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A SIMPLE ANALYSIS OF CUSTOMERS IMPATIENCE IN MULTISERVER QUEUES

Amit CHOUDHURY¹

PhD, Department of Statistics, Gauhati University, Assam, India

E-mail: achoudhury@rediffmail.com



Pallabi MEDHI²

Department of Statistics,
Gauhati University, Assam, India

E-mail: pmedhi16@gmail.com



Abstract: Balking and reneging are two ways through which customer impatience finds reflection. In the analysis of stochastic reneging, it has traditionally been assumed that the distribution of patience time is state independent. However in many queuing systems, the customer is aware of its position in the system state. Hence in this paper, we assume that a customer who arrives at the queuing system gets to know the state of the system. Consequently, both balking and reneging are taken as function of system state. Both types of reneging are considered. Explicit closed form expressions of a number of performance measures are presented. A numerical example is presented to demonstrate the results derived rounds off the paper.

Key words: Balking, Impatience, Queuing, Reneging.

1. Introduction

These days customers are busy entities. An assumption which is often attached to the analysis of many queuing model is the customers are willing to wait as long as it is necessary to obtain service. Our fast-paced life is often inconsistent with this assumption. In queuing terminology, two characteristics through which customer's impatience find reflection are balking and reneging. By balking, we mean the phenomenon of customers arriving for service into a non-empty queue and leaving without joining the queue. There is no balking from an empty queue. Haight (1957) has provided a rationale, which might influence a person to balk. It relates to the perception of the importance of being served which induces

an opinion somewhere in between urgency, so that a queue of certain length will not be joined, to indifference where a non-zero queue is also joined.

A customer will be said to have reneged if after joining the system it gets impatient and leave the system without receiving service. On joining the system, it has a patience time such that in case service is unavailable within this patience time the customers renege.

Reneging can be of two types- reneging till beginning of service (henceforth referred to as R_BOS) and reneging till end of service (henceforth referred to as R_EOS). A customer can renege only as long as it is in the queue and we call this as reneging of type R_BOS. It cannot renege once it begins receiving service. A common example is the barbershop. A customer can renege while he is waiting in queue. However once service get started i.e. hair cut begins, the customer cannot leave till hair cutting is over. On the other hand, if customers can renege not only while waiting in queue but also while receiving service, we call such behavior as reneging of type R_EOS. An example is processing or merchandising of perishable goods.

In the analysis of reneging phenomena, one approach is to assume that each customer has a Markovian patience time, the distribution of which is state independent. However, it is our common day observation that there are systems where the customer is aware of its state in the system. For example customers queuing at the O.P.D. (out patient department) clinic of a hospital would know of their position in the queue. This invariably causes waiting customers to have higher rates of reneging in case their position in the queue is towards the end. It is not unreasonable then to expect that such customers who are positioned at a distance from the service facility have reneging rates, which are higher than reneging rates of customers who are near the service facility. In other words, we assume that customers are "state aware" and in this paper we model the reneging phenomenon in such a manner that the Markovian reneging rate is a function of the state of the customer in the system. Customers at higher states will be assumed to have higher reneging rates.

The subsequent sections of this paper are structured as follows. Section 2 contains a brief review of the literature. Section 3 and section 4 contains the derivation of steady state probabilities and performance measures respectively. We perform sensitivity analysis in section 5. A numerical example is discussed in section 6. Concluding statements are presented in section 7. The appendix presented in section 8 contains some derivation.

2. Literature Survey

One of the earliest work on reneging was by Barrer (1957) where he considered deterministic reneging with single server markovian arrival and service rates. Customers were selected randomly for service. In his subsequent work, Barrer (1957) also considered deterministic reneging (of both R_BOS and R_EOS type) in a multiserver scenario with FCFS discipline. The general method of solution was extended to two related queuing problems. Another early work on reneging was by Haight (1959). Ancher and Gafarian (1963) carried out an early work on markovian reneging with markovian arrival and service pattern. Ghosal (1963) considered a D/G/1 model with deterministic reneging. Gavish and Schweitzer (1977) also considered a deterministic reneging model with the additional assumption that arrivals can be labeled by their service requirement before joining the queue and arriving customers are admitted only if their waiting plus service time do not exceed some fixed amount. This assumption is met in communication systems. Kok and Tijms (1985) considered a single server queuing system where a customer becomes a lost customer when its service has not begun within a fixed time. Haghighi et al (1986) considered a markovian multiserver queuing model with balking as well as reneging. Each customer had a balking probability which was independent of the state of the system. Reneging discipline considered by them was R_BOS. Liu et al (1987) considered an infinite server markovian queuing system with reneging of type R_BOS. Customers had a choice of individual service or batch service, batch service being preferred by the customer. Brandt et al (1999) considered a S-server system with two FCFS queues, where the arrival rates at the queues and the service may depend on number of customers 'n' being in service or in the first queue, but the service rate was

assumed to be constant for $n > s$. Customers in the first queue were assumed impatient customers with deterministic reneging. Boots and Tijms (1999) considered an M/M/C queue in which a customer leaves the system when its service has not begun within a fixed interval after its arrival. In this paper, they have given the probabilistic proof of 'loss probability', which was expressed in a simple formula involving the waiting time probabilities in the standard M/M/C queue. Wang et al (1999) considered the machine repair problem in which failed machines balk with probability $(1-b)$ and renege according to a negative exponential distribution. Bae et al. (2001) considered an M/G/1 queue with deterministic reneging. They derived the complete formula of the limiting distribution of the virtual waiting time explicitly. Choi et al. (2001) introduced a simple approach for the analysis of the M/M/C queue with a single class of customers and constant patience time by finding simple markov process. Applying this approach, they analyzed the M/M/1 queue with two classes of customer in which class 1 customer have impatience of constant duration and class 2 customers have no impatience and lower priority than class 1 customers. Performance measures of both M/M/C and M/M/1 queues were discussed. Zhang et al. (2005) considered an M/M/1/N framework with markovian reneging where they derived the steady state probabilities and formulated a cost model. Some performance measures were also discussed. A numerical example was discussed to demonstrate how the various parameters of the cost model influence the optimal service rates of the system. El- Paoumy (2008) also derived the analytical solution of $M^x/M/2/N$ queue for batch arrival system with markovian reneging. In this paper, the steady state probabilities and some performance measures of effectiveness were derived in explicit forms. Another paper on markovian reneging was by Yechiali and Altman (2008). They derived the probability generating function of number of customers present in the system and some performance measures were discussed. Choudhury (2009) considered a single server finite buffer queuing system (M/M/1/K) assuming reneging customers. Both rules of reneging were considered and various performance measures presented under both rules of reneging.

Other attempts at modeling reneging phenomenon include those by Baccelli et al (1984), Martin and Artalejo (1995), Shawky (1997), Choi, Kim and Zhu (2004), and Singh et al (2007), El- Sherbiny (2008) and El-Paoumy and Ismail (2009) etc.

An early work on balking was by Haight (1957). Another work using the concepts of balking and reneging in machine interference queue has been carried out by Al-Seedy and Al-Ibraheem (2001). There have been some papers in which both balking as well as reneging were considered. Here mention may be made of the work by Haghighi et al (1986), Shawky and El-Paoumy (2000), Zhang et al (2005), El- Paoumy (2008), El- Sherbiny (2008), Shawky and El-Paoumy (2008).

3. The Model and System State Probabilities

The model we deal with in this paper is the traditional M/M/k model with the restriction that customers may balk from a non-empty queue as well as may renege after they join the queue. We shall assume that the inter-arrival and service rates are λ and μ respectively. As for balking, we shall assume that each customer arriving at the system has a probability

' $1-p^{n-k+1}$ ' (for $n \geq k$ and 0 otherwise) of balking from a non-empty queue.

Customers joining the system are assumed to be of Markovian reneging type. We shall assume that on joining the system the customer is aware of its state in the system. Consequently, the reneging rate will be taken as a function of the customer's state in the system. In particular, a customer who is at state 'n' will be assumed to have random patience time following $\exp(v_n)$. Under R_BOS, we shall assume that

$$v_n = \begin{cases} 0 & \text{for } n = 0, 1, \dots, k \\ v + c^{n-k} & \text{for } n = k + 1, \dots \end{cases}$$

and under R_EOS,

$$v_n = \begin{cases} 0 & \text{for } n = 0. \\ v & \text{for } n = 1, 2, \dots, k \\ v + c^{n-k} & \text{for } n = k+1, k+2, \dots \end{cases}$$

where $c > 1$ is a constant.

Our aim behind this formulation is to ensure that higher the current state of a customer, higher is the reneging rate. Then it is clear that the constant c has to satisfy $c > 1$. This reneging formulation also requires that as a customer progresses in the queue from state n ($n \geq k+2$) to $(n-1)$, the reneging distribution would shift from $\exp(v + c^{n-k})$ to $(v + c^{n-k-1})$ under R_BOS . Similarly for R_EOS . In view of the memory less property, this shifting of reneging distribution is mathematically tractable, as we shall demonstrate in the subsequent sections. To the best of our knowledge, such a formulation of reneging distribution has not been attempted in literature. Advantages of the same are however obvious.

The steady state probabilities are derived by the Markov process method. We first analyze the case where customers renege only from the queue. Under R_BOS , let p_n denote the probability that there are ' n ' customers in the system. The steady state probabilities under R_BOS are given below,

$$\lambda p_0 = \mu p_1, \quad (3.1)$$

$$\lambda p_{n-1} + (n+1)\mu p_{n+1} = \lambda p_n + n\mu p_n; n=1, 2, \dots, k-1, \quad (3.2)$$

$$\lambda p^{n-k} p_{n-1} + \{k\mu + (n-k+1)v + c(c^{n-k+1} - 1)/(c-1)\} p_{n+1} = \lambda p^{n-k+1} p_n + \{k\mu + (n-k)v + c(c^{n-k} - 1)/(c-1)\} p_n$$

$$n = k, k+1, \dots \quad (3.3')$$

Solving recursively, we get (under R_BOS)

$$p_n = \{\lambda^n / (n! \mu^n)\} p_0; n=1, 2, \dots, k \quad (3.4)$$

$$p_n = \left[\lambda^n p^{\{(n-k)(n-k+1)\}/2} / \{k! \mu^k \prod_{r=k+1}^n (k\mu + r - kv + c(c^{r-k} - 1)/(c-1))\} \right] p_0; n=k+1, \quad (3.5)$$

where p_0 is obtained from the normalizing condition $\sum_{n=0}^{\infty} p_n = 1$ and is given as

$$p_0 = \left[\sum_{n=0}^k \lambda^n / (n! \mu^n) + \sum_{n=k+1}^{\infty} \lambda^n p^{\{(n-k)(n-k+1)\}/2} / \{k! \mu^k \prod_{r=k+1}^n (k\mu + r - kv + c(c^{r-k} - 1)/(c-1))\} \right]^{-1} \quad (3.6)$$

The steady state probabilities satisfy the recurrence relation. Under R_BOS

$$p_n = \{\lambda / (n\mu)\} p_{n-1}; n=1, 2, \dots, k$$

$$\text{and } p_n = \left[\lambda p^{n-k} / \{k\mu + n - kv + c(c^{n-k} - 1)/(c-1)\} \right] p_{n-1}; n = k+1, k+2, \dots$$

We shall denote by K_{R_BOS} the probability that an arriving unit has to wait on arrival (under R_BOS). Then

$$K_{R_BOS} = \Pr(N \geq k) = \sum_{n=k}^{\infty} p_n. \quad (3.7)$$

We may call K_{R_BOS} as Erlang's second (Erlang's delay probability) formula for state dependent balking and state dependent reneging (R_BOS) in line with similar nomenclature in Medhi (2003, page 87).

Under R_EOS where customers may renege from queue as well as while being served, let q_n denote the probability that there are n customers in the system. Applying the Markov theory, we obtain the following set of steady state equations.

$$\lambda q_0 = (\mu + v) q_1, \quad (3.8)$$

$$\lambda q_{n-1} + (n+1)(\mu + \nu)q_{n+1} = \lambda q_n + n(\mu + \nu)q_n ; n = 1, 2, \dots, k-1 \quad (3.9)$$

$$\lambda p^{n-k} q_{n-1} + \{k\mu + (n+1)\nu + c(c^{n-k+1} - 1)/(c-1)\}q_{n+1} = \lambda p^{n-k+1} q_n + \{k\mu + n\nu + c(c^{n-k} - 1)/(c-1)\}q_n$$

$$; n = k, k+1, \quad (3.10)$$

Solving recursively, we get (under R_EOS)

$$q_n = [\lambda^n / \{n!(\mu + \nu)^n\}]q_0 ; \quad n = 1, 2, \dots, k \quad (3.11)$$

$$q_n = \left[\lambda^n p^{\{(n-k)(n-k+1)\}/2} / \{k!(\mu + \nu)^k \prod_{r=k+1}^n (k\mu + r\nu + c\overline{c^{r-k} - 1/c - 1})\} \right] q_0 ; n = k+1, \dots$$

$$(3.12)$$

where q_0 is obtained from the normalizing condition $\sum_{n=0}^{\infty} q_n = 1$ and is given as

$$q_0 = \left[\sum_{n=0}^k \lambda^n / \{n!(\mu + \nu)^n\} + \sum_{n=k+1}^{\infty} \lambda^n p^{\{(n-k)(n-k+1)\}/2} / \{k!(\mu + \nu)^k \prod_{r=k+1}^n (k\mu + r\nu + c\overline{c^{r-k} - 1/c - 1})\} \right]^{-1}.$$

$$(3.13)$$

The recurrence relations under R_EOS are

$$q_n = [\lambda / \{n(\mu + \nu)\}]q_{n-1} ; \quad n = 1, 2, \dots, k.$$

$$\text{and } q_n = \{ \lambda p^{n-k} / (k\mu + n\nu + c\overline{c^{n-k} - 1/c - 1}) \} q_{n-1} ; \quad n = k+1, k+2, \dots$$

We shall denote by K_{R_EOS} the probability that an arriving unit has to wait on arrival (under R_EOS). Then

$$K_{R_EOS} = \Pr(N \geq k) = \sum_{n=k}^{\infty} q_n \quad (3.14)$$

which may be called Erlang's second (Erlang's delay probability) formula for state dependent balking and state dependent reneging (R_EOS).

4. Performance Measures

An important measure is 'L', which denotes the mean number of customers in the system. To obtain an expression for the same, we note that $L = P'(1)$ where

$$P'(1) = \frac{d}{ds} P(s) \big|_{s=1}.$$

Here $P(s)$ is the p.g.f. of the steady state probabilities. The derivation of $P'(1)$ is given in the appendix. From (8.1.17) and (8.2.3), the mean system size under two reneging rules are

$$L_{R_BOS} = (1/\nu)[\lambda(1 - K_{R_BOS}) + \{\lambda p_0 K_{R_BOS}(p\lambda, \mu, \nu)\} / \{p^{k-1} p_0(p\lambda, \mu, \nu)\} - (\mu - \nu) \sum_{n=1}^k n p_n - k(K_{R_BOS} - p_k)(\mu - \nu)$$

$$- \{p_0 K_{R_EOS}(c\lambda, \mu, \nu)\} / \{p_0(c\lambda, \mu, \nu)(c-1)c^{k-1}\} + (cK_{R_BOS})/(c-1)].$$

$$(4.1)$$

$$L_{R_EOS} = (1/\nu)[\lambda(1 - K_{R_EOS}) + \{\lambda q_0 K_{R_EOS}(p\lambda, \mu, \nu)\} / \{p^{k-1} q_0(p\lambda, \mu, \nu)\} - \mu \sum_{n=1}^k n q_n - k\mu(K_{R_EOS} - q_k)$$

$$- \{q_0 K_{R_EOS}(c\lambda, \mu, \nu)\} / \{q_0(c\lambda, \mu, \nu)(c-1)c^{k-1}\} + (cK_{R_EOS})/(c-1)].$$

$$(4.2)$$

Mean queue size can now be obtained and are given by

$$\begin{aligned}
 L_{q(R_BOS)} &= \sum_{n=k+1}^{\infty} (n-k)p_n \\
 &= L_{R_BOS} - \sum_{n=1}^k np_n - k(K_{R_BOS} - p_k) \\
 &= (1/\nu)[\lambda(1-K_{R_BOS}) + \{\lambda p_0 K_{R_BOS}(p\lambda, \mu, \nu)\} / \{p^{k-1} p_0(p\lambda, \mu, \nu)\} - \mu \sum_{n=1}^k np_n - k\mu(K_{R_BOS} - p_k) \\
 &\quad - \{p_0 K_{R_EOS}(c\lambda, \mu, \nu)\} / \{p_0(c\lambda, \mu, \nu)(c-1)c^{k-1}\} + (cK_{R_BOS})/(c-1)]. \\
 L_{q(R_EOS)} &= L_{R_EOS} - \sum_{n=1}^k nq_n - k(K_{R_EOS} - q_k) \\
 &= (1/\nu)[\lambda(1-K_{R_EOS}) + \{\lambda q_0 K_{R_EOS}(p\lambda, \mu, \nu)\} / \{p^{k-1} q_0(p\lambda, \mu, \nu)\} - (\mu+\nu) \sum_{n=1}^k nq_n - k(\mu+\nu)(K_{R_EOS} - q_k) \\
 &\quad - \{q_0 K_{R_EOS}(c\lambda, \mu, \nu)\} / \{q_0(c\lambda, \mu, \nu)(c-1)c^{k-1}\} + (cK_{R_EOS})/(c-1)].
 \end{aligned}$$

Using Little's formula, one can calculate the average waiting time in the system and average waiting time in queue from the above mean lengths both under R_BOS and R_EOS.

Customers arrive into the system at the rate λ . However all the customers who arrive do not join the system because of balking. The effective arrival rate into the system is thus different from the overall arrival rate and is given by

$$\begin{aligned}
 \lambda^e_{(R_BOS)} &= \lambda \sum_{n=0}^{k-1} p_n + \sum_{n=k}^{\infty} p^{n-k+1} p_n \\
 &= \lambda(1-K_{R_BOS}) + \{\lambda p_0 K_{R_BOS}(p\lambda, \mu, \nu)\} / \{p^{k-1} p_0(p\lambda, \mu, \nu)\}.
 \end{aligned}$$

Similarly in case of R_EOS

$$\lambda^e_{(R_EOS)} = \lambda(1-K_{R_EOS}) + \{\lambda q_0 K_{R_EOS}(p\lambda, \mu, \nu)\} / \{p^{k-1} q_0(p\lambda, \mu, \nu)\}.$$

We have assumed that each customer has a random patience time following $\exp(\nu)$. Clearly then, the reneging rate of the system would depend on the state of the system as well as the reneging rule. The average reneging rate under two reneging rules are given by

$$\begin{aligned}
 Avgrr_{(R_BOS)} &= \sum_{n=k+1}^{\infty} \{(n-k)\nu + c(c^{n-k} - 1)/(c-1)\} p_n \\
 &= \nu \left\{ p'(1) - \sum_{n=1}^k np_n \right\} - \nu k \left\{ 1 - \sum_{n=0}^k p_n \right\} + \{1/(c-1)c^{k-1}\} \sum_{n=k+1}^{\infty} c^n p_n - \{c/(c-1)\} \left\{ 1 - \sum_{n=0}^k p_n \right\} \\
 &= \lambda(1-K_{R_BOS}) + \{\lambda p_0 K_{R_BOS}(p\lambda, \mu, \nu)\} / \{p^{k-1} p_0(p\lambda, \mu, \nu)\} - \mu \sum_{n=1}^k np_n - k\mu(K_{R_BOS} - p_k). \\
 Avgrr_{(R_EOS)} &= \sum_{n=1}^k nq_n + \sum_{n=k+1}^{\infty} \{n\nu + c(c^{n-k} - 1)/(c-1)\} q_n \\
 &= \nu Q'(1) + \{1/(c-1)c^{k-1}\} \sum_{n=k+1}^{\infty} c^n p_n - \{c/(c-1)\} \left\{ 1 - \sum_{n=0}^k p_n \right\} \\
 &= \lambda(1-K_{R_EOS}) + \{\lambda q_0 K_{R_EOS}(p\lambda, \mu, \nu)\} / \{p^{k-1} q_0(p\lambda, \mu, \nu)\} - \mu \sum_{n=1}^k nq_n - k\mu(K_{R_EOS} - q_k).
 \end{aligned}$$

In a real life situation, customers who balk or renege represent the business lost. Customers are lost to the system in two ways, due to balking and due to reneging. Management would like to know the proportion of total customers lost in order to have an idea of total business lost.

Hence the mean rate at which customers are lost (under R_BOS) is

$$\begin{aligned} & \lambda - \lambda^e_{(R_BOS)} + \text{avgrr}_{(R_BOS)} \\ &= \lambda - \mu \sum_{n=1}^k np_n - k\mu(K_{R_BOS} - p_k). \end{aligned}$$

and the mean rate at which customers are lost (under R_EOS) is

$$\begin{aligned} & \lambda - \lambda^e_{(R_EOS)} + \text{avgrr}_{(R_EOS)} \\ &= \lambda - \mu \sum_{n=1}^k nq_n - k\mu(K_{R_EOS} - q_k). \end{aligned}$$

These rates helps in the determination of proportion of customers lost which is of interest to the system manager as also an important measure of business lost. This proportion (under R_BOS) is given by

$$\begin{aligned} & \{\lambda - \lambda^e_{(R_BOS)} + \text{avgrr}_{(R_BOS)}\} / \lambda \\ &= 1 - (1/\lambda) [\mu \sum_{n=1}^k np_n + k\mu(K_{R_BOS} - p_k)]. \end{aligned}$$

and the proportion (under R_EOS) is given by

$$\begin{aligned} & \{\lambda - \lambda^e_{(R_EOS)} + \text{avgrr}_{(R_EOS)}\} / \lambda \\ &= 1 - (1/\lambda) [\mu \sum_{n=1}^k nq_n + k\mu(K_{R_EOS} - q_k)]. \end{aligned}$$

The proportion of customer completing receipt of service can now be easily determined from the above proportion.

The customers who leave the system from the queue do not receive service. Consequently, only those customers who reach the service station constitute the actual load of the server. From the server's point of view, this provides a measure of the amount of work he has to do. Let us call the rate at which customers reach the service station as λ^s . Then under R_BOS

$\lambda^s_{(R_BOS)} = \lambda^e_{(R_BOS)} (1 - \text{proportion of customers lost due to reneging out of those joining the system})$

$$\begin{aligned} &= \lambda^e_{(R_BOS)} \left\{ 1 - \sum_{n=k+1}^{\infty} (n-k)p_n / \lambda^e_{(R_BOS)} \right\} \\ &= \lambda^e_{(R_BOS)} - \text{avgrr}_{(R_BOS)} \\ &= \mu \sum_{n=1}^k np_n + k\mu(K_{R_BOS} - p_k). \end{aligned}$$

In case of R_EOS, one needs to recall that customers may renege even while being served and only those customers who renege from the queue will not constitute any work for the server. Thus

$\lambda^s_{(R_EOS)} = \lambda^e_{(R_EOS)} (1 - \text{proportion of customers lost due to reneging from the queue out of those joining the system})$

$$= \lambda^e_{(R_EOS)} \left\{ 1 - \sum_{n=k+1}^{\infty} (n-k)q_n / \lambda^e_{(R_EOS)} \right\}$$

$$\begin{aligned}
 &= \lambda^e_{(R_EOS)} - \nu \left\{ Q'(1) - \sum_{n=1}^n n q_n \right\} + k \nu \left(1 - \sum_{n=0}^k q_n \right) \\
 &= \lambda^e_{(R_EOS)} - \nu Q'(1) + \nu \sum_{n=1}^n n q_n + k \nu (K_{R_EOS} - q_k) \\
 &= (\mu + \nu) \sum_{n=1}^k n q_n + k(\mu + \nu)(K_{R_EOS} - q_k) + \{q_0 K_{R_EOS}(c\lambda, \mu, \nu)\} / \{q_0(c\lambda, \mu, \nu)(c-1)c^{k-1}\} - (cK_{R_EOS})/(c-1).
 \end{aligned}$$

In order to ensure that the system is in steady state, it is necessary for the rate of customers reaching the service station to be less than the system capacity. This translates to $(\lambda^s/k\mu) < 1$.

5. Sensitivity Analysis.

It is interesting to examine and understand how server utilization varies in response to change in system parameters. The four system parameters of interest are λ, μ, ν . We place below the effect of change in these system parameters on server utilization. For this purpose, we shall follow the following notational convention in the rest of this section.

$p_n(\lambda, \mu, \nu)$ and $q_n(\lambda, \mu, \nu)$ will denote the probability that there are 'n' customers in a system with parameters λ, μ, ν in steady state under R_BOS and R_EOS respectively.

i) If $\lambda_1 > \lambda_0$, then

$$\begin{aligned}
 &\frac{p_0(\lambda_1, \mu, \nu)}{p_0(\lambda_0, \mu, \nu)} < 1 \\
 &\Rightarrow \frac{(\lambda_0 - \lambda_1)}{\mu} + \dots + \frac{(\lambda_0^k - \lambda_1^k)}{k! \mu^k} + \frac{p(\lambda_0^{k+1} - \lambda_1^{k+1})}{k! \mu^k (k\mu + \nu + c)} + \dots < 0
 \end{aligned}$$

which is true. Hence $p_0 \downarrow$ as $\lambda \uparrow$.

ii) If $\mu_1 > \mu_0$, then

$$\begin{aligned}
 &\frac{p_0(\lambda, \mu_1, \nu)}{p_0(\lambda, \mu_0, \nu)} > 1 \\
 &\Rightarrow \lambda \left(\frac{1}{\mu_0} - \frac{1}{\mu_1} \right) + \dots + \frac{\lambda^k}{k!} \left(\frac{1}{\mu_0^k} - \frac{1}{\mu_1^k} \right) + \frac{\lambda^{k+1} p}{k!} \left\{ \frac{1}{\mu_0^k (k\mu_0 + \nu + c)} - \frac{1}{\mu_1^k (k\mu_1 + \nu + c)} \right\} + \dots > 0
 \end{aligned}$$

which is true. Hence $p_0 \uparrow$ as $\mu \uparrow$.

iii) If $\nu_1 > \nu_0$, then

$$\begin{aligned}
 &\frac{p_0(\lambda, \mu, \nu_1)}{p_0(\lambda, \mu, \nu_0)} > 1 \\
 &\Rightarrow \frac{\lambda^{k+1} p}{k! \mu^k} \left\{ \frac{1}{(k\mu + \nu_0 + c)} - \frac{1}{(k\mu + \nu_1 + c)} \right\} \\
 &+ \frac{\lambda^{k+2} p^3}{k! \mu^k} \left\{ \frac{1}{(k\mu + \nu_0 + c) \{k\mu + 2\nu_0 + c(c^2 - 1)/(c-1)\}} - \frac{1}{\{(k\mu + \nu_1 + c)(k\mu + 2\nu_1 + c(c^2 - 1)/(c-1)\}} \right\} + \dots > 0
 \end{aligned}$$

which is true. Hence $p_0 \uparrow$ as $\nu \uparrow$.

The following can similarly be shown.

v) $q_0 \downarrow$ as $\lambda \uparrow$

vi) $q_0 \uparrow$ as $\mu \uparrow$

vii) $q_0 \uparrow$ as $\nu \uparrow$

The managerial implications of the above results are obvious.

6. Numerical Example

To illustrate the use of our results, we apply them to a queuing problem. We quote below an example from Allen (2005, page 352).

'Customers arrive randomly (during the evening hours) at the Kitten house, the local house of questionable services, at an average rate of five per hour. Service time is exponential with a mean of 20 minutes per customer. There are two servers on duty.

So many queuing theory students visits the Kitten house to collect data for this book that proprietress, Kitty Callay (also known as the Cheshire Cat) make some changes. She trains her kittens to provide more exotic but still exponentially distributed service and add three more servers, for a total of five. Her captivated, customers still complain that the queue is too long. Kitty commissions her most favoured customer Gralre K. Renga to make a study of her establishment. He is to determine the...., the number of servers she should provide so that....

the probability that an arriving customers must wait for service will not exceed 0.25.'

This is a design problem. Here $\lambda = 5/\text{hr}$ and $\mu = 3/\text{hr}$. As required by the owner of the Kitten house, we examine the minimum number of servers with different choices of k . Though not explicitly mentioned, it is necessary to assume reneging and balking.

We shall assume that reneging distribution is state dependent following $\exp(\nu_n)$ where ν_n is as described in section 3. Specifically, we shall assume $\nu = 0.5/\text{hr}$ and considered the scenario with $c = 1.1$. We further assume that balking rate is dependent of state and is 0.1.

Various performance measures of interest computed are given in the following table. These measures were arrived at using a FORTRAN 77 program coded by the authors. Different choices of k were considered. Results relevant with regard to the requirement that the Kitten house should provide servers so that the probability that an arriving customers will find all servers busy should be < 0.25 are presented in the following table. (All rates in the table as per hour rates)

Table 1: Table of Performance Measures (with $\lambda=5$, $\mu=3$, $\nu=0.5$, $p=0.9$ and $c=1.1$)

Performance Measure	Number of servers		
	k=2	k=3	k=4
$\sum_{n=k+1}^{\infty} p_n$	0.50513	0.23437	0.08755
λ^s (i.e. arrival rate of customers reaching service station)	3.95868	4.62644	4.88363
Effective mean arrival rate(λ^e)	4.58476	4.82917	4.94089
Fraction of time server is idle (p_0)	0.18557	0.18879	0.18902
Average length of queue	0.37868	0.12356	0.03507
Average length of system	1.69824	1.66570	1.66295
Mean reneging rate	0.62608	0.20273	0.05726
Average balking rate	0.08305	0.03417	0.01182
Mean rate of customers lost	1.04132	0.37356	0.11637
Proportion of customers lost due to reneging, and balking.	0.20826	0.07471	0.02327

From the above table it is clear that an ideal choice of k could be k=3 with

$\sum_{n=k+1}^{\infty} p_n = 0.23437$. It may be noted here that k=4 would also satisfy the design criteria. However, that would necessitate an additional server. Considering cost implications the idea would be to attain the design criteria with minimal number of servers. Under the assumption of balking and reneging, it appears that the proprietress need not increase the number of servers to five. Her design requirement would be met with three servers. She may therefore increase the number of servers by one.

7. Conclusion

The analysis of a multiserver Markovian queuing system with state-dependent balking and state dependent reneging has been presented. Even though balking and reneging have been discussed by others, explicit expression are not available. This paper makes a contribution here. Closed form expressions of number of performance measures have been derived. To study the change in the system corresponding to change in system parameters, sensitivity analysis has also been presented. A numerical example has been discussed to demonstrate results derived. The numerical example is of indicative nature meant to illustrate the benefits of our theoretical results in a design context.

8. Appendix

8.1. Derivation of P'(1) under R_BOS.

Let P(s) denote the probability generating function, defined by $P(s) = \sum_{n=0}^{\infty} p_n s^n$

From equation (3.2) we have

$$\lambda p_{n-1} + (n+1)\mu p_{n+1} = \lambda p_n + n\mu p_n; \quad n = 1, 2, \dots, k-1.$$

Multiplying both sides of the equation by s^n and summing over n, we get

$$\lambda s \sum_{n=1}^{k-1} p_{n-1} s^{n-1} - \lambda \sum_{n=1}^{k-1} p_n s^n = \mu \sum_{n=1}^{k-1} n p_n s^n - \frac{1}{s} \mu \sum_{n=1}^{k-1} (n+1) p_{n+1} s^{n+1} \quad (8.1.1)$$

From (3.3) we have

$$\lambda p^{n-k} p_{n-1} + \{k\mu + (n-k+1)\nu + c(c^{n-k+1} - 1)/(c-1)\} p_{n+1} = \lambda p^{n-k+1} p_n + \{k\mu + (n-k)\nu + c(c^{n-k} - 1)/(c-1)\} p_n$$

; $n = k, k+1, \dots$

Similarly multiplying both sides of the equation by s^n and summing over n

$$\begin{aligned} \lambda s \sum_{n=k}^{\infty} p^{n-k} p_{n-1} s^{n-1} - \lambda \sum_{n=k}^{\infty} p^{n-k+1} p_n s^n &= \sum_{n=k}^{\infty} \{k\mu + (n-k)\nu + (c^{n-k+1} - c)/(c-1)\} p_n s^n \\ &\quad - \frac{1}{s} \sum_{n=k}^{\infty} \{k\mu + (n-k+1)\nu + (c^{n-k+2} - c)/(c-1)\} p_{n+1} s^{n+1} \end{aligned} \quad (8.1.2)$$

Adding (8.1.1) and (8.1.2)

$$\begin{aligned} &\Rightarrow \lambda s \left[\sum_{n=1}^{k-1} p_{n-1} s^{n-1} + \sum_{n=k}^{\infty} p^{n-k} p_{n-1} s^{n-1} \right] - \lambda \left[\sum_{n=1}^{k-1} p_n s^n + \sum_{n=k}^{\infty} p^{n-k+1} p_n s^n \right] \\ &= \mu \sum_{n=1}^{k-1} n p_n s^n + \sum_{n=k}^{\infty} \{k\mu + (n-k)\nu + (c^{n-k+1} - c)/(c-1)\} p_n s^n - \frac{1}{s} \left[\mu \sum_{n=1}^{k-1} (n+1) p_{n+1} s^{n+1} + \sum_{n=k}^{\infty} \{k\mu + (n-k+1)\nu + (c^{n-k+2} - c)/(c-1)\} p_{n+1} s^{n+1} \right] \end{aligned}$$

$$\begin{aligned}
 &\Rightarrow \lambda s \left\{ P(s) - \sum_{n=k}^{\infty} p_n s^n \right\} + p p_k s^k + p^2 p_{k+1} s^{k+1} + \dots - \lambda \left\{ P(s) - p_0 - \sum_{n=k}^{\infty} p_n s^n \right\} - \lambda p p_k s^k - \lambda p^2 p_{k+1} s^{k+1} - \dots \\
 &= \mu s \left\{ P(s) - \sum_{n=k+1}^{\infty} n p_n s^{n-1} \right\} + k \mu \left\{ P(s) - \sum_{n=0}^k p_n s^n \right\} + v s \left\{ (k+1) p_{k+1} s^k + \dots \right\} - k v \left\{ p_{k+1} s^{k+1} + \dots \right\} + \{1/(c-1)c^{k-1}\} \{c^k p_k s^k + c^{k+1} p_{k+1} s^{k+1} + \dots\} \\
 &\quad - \{c/(c-1)\} \left\{ P(s) - \sum_{n=0}^{k-1} p_n s^n \right\} - \frac{1}{s} \left[\mu s \left\{ P(s) - p_1 - \sum_{n=k+1}^{\infty} n p_n s^{n-1} \right\} + k \mu \left\{ P(s) - \sum_{n=0}^k p_n s^n \right\} + v s \left\{ (k+1) p_{k+1} s^k + \dots \right\} - k v \left\{ p_{k+1} s^{k+1} + \dots \right\} \right. \\
 &\quad \left. + \{1/(c-1)c^{k-1}\} \{c^{k+1} p_{k+1} s^{k+1} + \dots\} - \{c/(c-1)\} \left\{ P(s) - \sum_{n=0}^k p_n s^n \right\} \right] \\
 &\Rightarrow \lambda s \left\{ P(s) - \sum_{n=k}^{\infty} p_n s^n \right\} - \lambda \left\{ P(s) - p_0 - \sum_{n=k}^{\infty} p_n s^n \right\} + [\{\lambda(s-1)\}/p^{k-1}] \{p^k p_k s^k + p^{k+1} p_{k+1} s^{k+1} + \dots\} \\
 &= \mu s \left\{ P'(s) - \sum_{n=k+1}^{\infty} n p_n s^{n-1} \right\} + k \mu \left\{ P(s) - \sum_{n=0}^k p_n s^n \right\} + v s \left\{ P'(s) - \sum_{n=1}^k n p_n s^{n-1} \right\} - k v \left\{ P(s) - \sum_{n=0}^k p_n s^n \right\} \\
 &\quad + \{1/(c-1)c^{k-1}\} \left\{ P(cs) - \sum_{n=0}^k p_n (cs)^n \right\} - \{c/(c-1)\} \left\{ P(s) - \sum_{n=0}^{k-1} p_n s^n \right\} - \mu \left\{ P'(s) - p_1 - \sum_{n=k+1}^{\infty} n p_n s^{n-1} \right\} \\
 &\quad - \frac{k \mu}{s} \left\{ P(s) - \sum_{n=0}^k p_n s^n \right\} - v \left\{ P'(s) - \sum_{n=1}^k n p_n s^{n-1} \right\} + \frac{k v}{s} \left\{ P(s) - \sum_{n=0}^k p_n s^n \right\} - \\
 &\quad \{1/(c-1)c^{k-1}s\} \left\{ P(cs) - \sum_{n=0}^k p_n (cs)^n \right\} + \{c/(c-1)s\} \left\{ P(s) - \sum_{n=0}^k p_n s^n \right\} \\
 &\Rightarrow \lambda s \left\{ P(s) - \sum_{n=k}^{\infty} p_n s^n \right\} - \lambda \left\{ P(s) - p_0 - \sum_{n=k}^{\infty} p_n s^n \right\} + [\{\lambda(s-1)\}/p^{k-1}] \{P(ps) - \sum_{n=0}^{k-1} p_n (ps)^n\} \\
 &= \mu s \left\{ P'(s) - \sum_{n=k+1}^{\infty} n p_n s^{n-1} \right\} + k \mu \left\{ P(s) - \sum_{n=0}^k p_n s^n \right\} + v s \left\{ P'(s) - \sum_{n=1}^k n p_n s^{n-1} \right\} - k v \left\{ P(s) - \sum_{n=0}^k p_n s^n \right\} \\
 &\quad + \{1/(c-1)c^{k-1}\} \left\{ P(cs) - \sum_{n=0}^k p_n (cs)^n \right\} - \{c/(c-1)\} \left\{ P(s) - \sum_{n=0}^{k-1} p_n s^n \right\} - \mu \left\{ P'(s) - p_1 - \sum_{n=k+1}^{\infty} n p_n s^{n-1} \right\} \\
 &\quad - \frac{k \mu}{s} \left\{ P(s) - \sum_{n=0}^k p_n s^n \right\} - v \left\{ P'(s) - \sum_{n=1}^k n p_n s^{n-1} \right\} + \frac{k v}{s} \left\{ P(s) - \sum_{n=0}^k p_n s^n \right\} - \\
 &\quad \{1/(c-1)c^{k-1}s\} \left\{ P(cs) - \sum_{n=0}^k p_n (cs)^n \right\} + \{c/(c-1)s\} \left\{ P(s) - \sum_{n=0}^k p_n s^n \right\}
 \end{aligned}$$

$$\begin{aligned}
 &\Rightarrow \lambda s P(s) - \lambda s \sum_{n=k}^{\infty} p_n s^n - \lambda P(s) + \lambda p_0 + \lambda \sum_{n=k}^{\infty} p_n s^n + \{ \lambda(1-s) \} / p^{k-1} P(ps) - \{ \lambda(1-s) \} / p^{k-1} \sum_{n=0}^{k-1} p_n (ps)^n \\
 &= \mu s P'(s) - \mu s \sum_{n=k+1}^{\infty} n p_n s^{n-1} + k \mu P(s) - k \mu \sum_{n=0}^k p_n s^n + \nu s P'(s) - \nu s \sum_{n=1}^k n p_n s^{n-1} - k \nu P(s) + k \nu \sum_{n=0}^k p_n s^n \\
 &+ \{ 1/(c-1)c^{k-1} \} P(cs) - \{ 1/(c-1)c^{k-1} \} \sum_{n=0}^k p_n (cs)^n - \{ c/(c-1) \} P(s) + \{ c/(c-1) \} \sum_{n=0}^{k-1} p_n s^n \\
 &- \mu P'(s) + \mu \frac{\lambda}{\mu} p_0 + \mu \sum_{n=k+1}^{\infty} n p_n s^{n-1} - \frac{k \mu}{s} P(s) + \frac{k \mu}{s} \sum_{n=0}^k p_n s^n - \nu P'(s) + \nu \sum_{n=1}^k n p_n s^{n-1} + \frac{k \nu}{s} P(s) - \frac{k \nu}{s} \sum_{n=0}^k p_n s^n - \\
 &\{ 1/(c-1)c^{k-1}s \} P(cs) + \{ 1/(c-1)c^{k-1}s \} \sum_{n=0}^k p_n (cs)^n + \{ c/(c-1)s \} P(s) - \{ c/(c-1)s \} \sum_{n=0}^k p_n s^n \\
 &\Rightarrow P'(s)(\mu + \nu) = \lambda P(s) - \lambda \sum_{n=k}^{\infty} p_n s^n + \{ \lambda P(ps) \} / p^{k-1} - \{ \lambda \sum_{n=0}^{k-1} p_n s^n \} / p^{k-1} + \mu \sum_{n=k+1}^{\infty} n p_n s^{n-1} - \frac{k \mu}{s} P(s) + \frac{k \mu}{s} \sum_{n=0}^k p_n s^n \\
 &\quad + \nu \sum_{n=1}^k n p_n s^{n-1} + \frac{k \nu}{s} P(s) - \frac{k \nu}{s} \sum_{n=0}^k p_n s^n - \frac{P(cs)}{(c-1)c^{k-1}s} + \frac{\sum_{n=0}^k p_n (cs)^n}{(c-1)c^{k-1}s} + \frac{c P(s)}{(c-1)s} - \frac{c \sum_{n=0}^k p_n s^n}{(c-1)s}
 \end{aligned}$$

Now

$$\lim_{s \rightarrow 1-} P'(s) = \lim_{s \rightarrow 1-} \frac{1}{(\mu + \nu)} \left[\begin{aligned} &\lambda P(s) - \lambda \sum_{n=k}^{\infty} p_n s^n + \{ \lambda P(ps) \} / p^{k-1} - \{ \lambda \sum_{n=0}^{k-1} p_n s^n \} / p^{k-1} + \mu \sum_{n=k+1}^{\infty} n p_n s^{n-1} - \frac{k \mu}{s} P(s) + \frac{k \mu}{s} \sum_{n=0}^k p_n s^n \\ &+ \nu \sum_{n=1}^k n p_n s^{n-1} + \frac{k \nu}{s} P(s) - \frac{k \nu}{s} \sum_{n=0}^k p_n s^n - \frac{P(cs)}{(c-1)c^{k-1}s} + \frac{\sum_{n=0}^k p_n (cs)^n}{(c-1)c^{k-1}s} + \frac{c P(s)}{(c-1)s} - \frac{c \sum_{n=0}^k p_n s^n}{(c-1)s} \end{aligned} \right]$$

$s \rightarrow 1- \quad s \rightarrow 1-$

$$P'(1) = \frac{1}{(\mu + \nu)} \left[\begin{aligned} &\lambda(1 - \sum_{n=k}^{\infty} p_n) + (\lambda / p^{k-1})(P(p) - \sum_{n=0}^{k-1} p_n p^n) + \mu \{ P'(1) - \sum_{n=1}^k n p_n \} - k \mu(1 - \sum_{n=0}^k p_n) \\ &+ \nu \sum_{n=1}^k n p_n + k \nu(1 - \sum_{n=0}^k p_n) - \frac{P(c)}{(c-1)c^{k-1}} + \frac{\sum_{n=0}^k p_n c^n}{(c-1)c^{k-1}} + \frac{c}{(c-1)}(1 - \sum_{n=0}^k p_n) \end{aligned} \right]$$

$$P'(1) = \frac{1}{\nu} \left[\begin{aligned} &\lambda(1 - \sum_{n=k}^{\infty} p_n) + (\lambda / p^{k-1})(P(p) - \sum_{n=0}^{k-1} p_n p^n) - \mu \sum_{n=1}^k n p_n - k \mu(1 - \sum_{n=0}^k p_n) \\ &+ \nu \sum_{n=1}^k n p_n + k \nu(1 - \sum_{n=0}^k p_n) - \frac{P(c)}{(c-1)c^{k-1}} + \frac{\sum_{n=0}^k p_n c^n}{(c-1)c^{k-1}} + \frac{c}{(c-1)}(1 - \sum_{n=0}^k p_n) \end{aligned} \right]$$

(8.1.3)

Here $P(c) = \sum_{n=0}^{\infty} p_n(\lambda, \mu, \nu) c^n$ and $P(p) = \sum_{n=0}^{\infty} p_n(\lambda, \mu, \nu) p^n$ where the symbol

$p_n(\lambda, \mu, \nu)$ is as described in section 7. We use p_n and $p_n(\lambda, \mu, \nu)$ interchangeably. However should any of the parameters λ, μ, ν change, it is explicitly stated. To obtain a closed form expression for $P(c)$ and $P(p)$, let us for the time being, consider two another queuing systems with parameters and assumptions similar to the queuing system we are presently considering except that the arrival rate is ' $c\lambda$ ' and ' $p\lambda$ ' respectively. For these new systems, the steady state equations are same as (4.1), (4.2) and (4.3) with ' λ ' replaced by ' $c\lambda$ ' and ' $p\lambda$ ' respectively. Denoting the steady state probabilities of these new systems by $p_n(c\lambda, \mu, \nu)$ and $p_n(p\lambda, \mu, \nu)$ respectively, we can obtain under R_BOS

$$p_n(c\lambda, \mu, \nu) = \{(c\lambda)^n / n! \mu^n\} p_0(c\lambda, \mu, \nu); n=1, 2, \dots, k \quad (8.1.4)$$

$$p_n(c\lambda, \mu, \nu) = \left[(c\lambda)^n p^{\{(n-k)(n-k+1)\}/2} / \{k! \mu^k \prod_{r=k+1}^n (k\mu + r - k\nu + cc^{r-k} - 1/c - 1)\} \right] p_0(c\lambda, \mu, \nu) \quad ; n = k+1, \dots \quad (8.1.5)$$

where

$$p_0(c\lambda, \mu, \nu) = \left[\sum_{n=0}^k (c\lambda)^n / (n! \mu^n) + \sum_{n=k+1}^{\infty} (c\lambda)^n p^{\{(n-k)(n-k+1)\}/2} / \{k! \mu^k \prod_{r=k+1}^n (k\mu + r - k\nu + cc^{r-k} - 1/c - 1)\} \right]^{-1}. \quad (8.1.6)$$

$$\text{and } p_n(p\lambda, \mu, \nu) = \{(p\lambda)^n / n! \mu^n\} p_0(p\lambda, \mu, \nu); n=1, 2, \dots, k-1$$

$$p_n(p\lambda, \mu, \nu) = \left[(p\lambda)^n p^{\{(n-k)(n-k+1)\}/2} / \{k! \mu^k \prod_{r=k+1}^n (k\mu + r - k\nu + cc^{r-k} - 1/c - 1)\} \right] p_0(p\lambda, \mu, \nu) \quad n=k+1, \dots \quad (8.1.7)$$

where

$$p_0(p\lambda, \mu, \nu) = \left[\sum_{n=0}^k (p\lambda)^n / (n! \mu^n) + \sum_{n=k+1}^{\infty} (p\lambda)^n p^{\{(n-k)(n-k+1)\}/2} / \{k! \mu^k \prod_{r=k+1}^n (k\mu + r - k\nu + cc^{r-k} - 1/c - 1)\} \right]^{-1}. \quad (8.1.8)$$

Similarly, we can derive the steady state probabilities of these queuing systems under R_EOS .

Let $P(S; c\lambda, \mu, \nu)$ denotes the probability generating function of this new queuing system so that

$$P(S; p\lambda, \mu, \nu) = \sum_{n=0}^{\infty} p_n(p\lambda, \mu, \nu) s^n.$$

Now

$$\begin{aligned} P(p) &= \sum_{n=0}^{\infty} p_n(\lambda, \mu, \nu) p^n \\ &= p_0 + \sum_{n=1}^k (p\lambda)^n p_0 / n! \mu^n + \sum_{n=k+1}^{\infty} \left[(p\lambda)^n p^{\{(n-k)(n-k+1)\}/2} / \{k! \mu^k \prod_{r=k+1}^n (k\mu + r - k\nu + cc^{r-k} - 1/c - 1)\} \right] p_0 \\ \Rightarrow [P(p) - p_0] / p_0 &= \sum_{n=1}^k (p\lambda)^n / n! \mu^n + \sum_{n=k+1}^{\infty} \left[(p\lambda)^n p^{\{(n-k)(n-k+1)\}/2} / \{k! \mu^k \prod_{r=k+1}^n (k\mu + r - k\nu + cc^{r-k} - 1/c - 1)\} \right] p_0 \end{aligned} \quad (8.1.9)$$

Now putting $S=1$ in $P(S; c\lambda, \mu, \nu)$ we get

$$\begin{aligned}
 P(1; p\lambda, \mu, \nu) &= p_0(p\lambda, \mu, \nu) + \sum_{n=1}^{\infty} p_n(p\lambda, \mu, \nu) \\
 \Rightarrow 1 &= p_0(p\lambda, \mu, \nu) + \sum_{n=1}^k (p\lambda)^n / n! \mu^n + \\
 &\sum_{n=k+1}^{\infty} \left[(p\lambda)^n p^{\{(n-k)(n-k+1)\}/2} / \{k! \mu^k \prod_{r=k+1}^n (k\mu + \overline{r-k} \nu + c\overline{c^{r-k} - 1 / c - 1})\} \right] p_0(p\lambda, \mu, \nu) \quad \text{using (8.1.7)} \\
 \Rightarrow 1 &= p_0(c\lambda, \mu, \nu) + \{(P(c) - p_0) / p_0\} p_0(c\lambda, \mu, \nu) \quad \text{using (8.9)} \\
 \Rightarrow P(p) &= p_0 / p_0(c\lambda, \mu, \nu).
 \end{aligned}$$

Similarly under R_EOS ,

$$Q(p) = q_0 / q_0(p\lambda, \mu, \nu). \quad (8.1.10)$$

Using the same procedure, we can obtain,

$$\text{under } R_BOS \quad P(c) = p_0 / p_0(p\lambda, \mu, \nu) \quad (8.1.11)$$

$$\text{and under } R_EOS \quad Q(c) = q_0 / q_0(c\lambda, \mu, \nu). \quad (8.1.12)$$

$$\text{Again let } K_{R_BOS}(p\lambda, \mu, \nu) = \sum_{n=k}^{\infty} p_n(p\lambda, \mu, \nu)$$

$$\begin{aligned}
 &= \sum_{n=k}^{\infty} [(p\lambda)^n p^{\{(n-k)(n-k+1)\}} / \{k! \mu^k \prod_{r=k+1}^n (k\mu + \overline{r-k} \nu + c\overline{c^{r-k} - 1 / c - 1})\}] p_0(p\lambda, \mu, \nu) \\
 &= \sum_{n=k}^{\infty} p_n p^n \{p_0(c\lambda, \mu, \nu) / p_0\}.
 \end{aligned}$$

Therefore,

$$\sum_{n=k}^{\infty} p_n p^n = p_0 K_{R_BOS}(p\lambda, \mu, \nu) / p_0(p\lambda, \mu, \nu) \quad (8.1.13)$$

and under R_EOS we have

$$\sum_{n=k}^{\infty} q_n p^n = q_0 K_{R_BOS}(p\lambda, \mu, \nu) / q_0(p\lambda, \mu, \nu). \quad (8.1.14)$$

similarly under R_BOS ,

$$\sum_{n=k}^{\infty} p_n c^n = p_0 K_{R_BOS}(c\lambda, \mu, \nu) / p_0(c\lambda, \mu, \nu) \quad (8.1.15)$$

and under R_EOS ,

$$\sum_{n=k}^{\infty} q_n c^n = q_0 K_{R_BOS}(c\lambda, \mu, \nu) / q_0(c\lambda, \mu, \nu). \quad (8.1.16)$$

Using these in (8.1.3), we

$$P(1) = \frac{1}{\nu} \left[\frac{\lambda(1 - K_{R_BOS}) + \{\lambda p_0 K_{R_BOS}(p\lambda, \mu, \nu)\} / \{p^{k-1} p_0(p\lambda, \mu, \nu)\} - (\mu - \nu) \sum_{n=1}^k n p_n - k(\mu - \nu)(K_{R_BOS} - p_k)}{-\{p_0 K_{R_BOS}(c\lambda, \mu, \nu)\} / \{(c-1)c^{k-1} p_0(c\lambda, \mu, \nu)\} + (c K_{R_BOS}) / (c-1)} \right]$$

$$\{\text{using (3.7)}\} \quad (8.1.17)$$

where $p_0(c\lambda, \mu, \nu)$ and $p_0(p\lambda, \mu, \nu)$ are given in (8.1.6) and (8.1.8) respectively.

8.2. Derivation of Q' (1) under R_EOS

From equation (3.9) we have,

$$\lambda q_{n-1} + (n+1)(\mu + \nu)q_{n+1} = \lambda q_n + n(\mu + \nu)q_n; \quad n=1,2,\dots,k-1.$$

Multiplying both sides of this equation by s^n and summing over n from we get

$$\lambda s \sum_{n=1}^{k-1} q_{n-1} s^{n-1} - \lambda \sum_{n=1}^{k-1} q_n s^n = (\mu + \nu) \sum_{n=1}^{k-1} n q_n s^n - \frac{1}{s} (\mu + \nu) \sum_{n=1}^{k-1} (n+1) q_{n+1} s^{n+1} \quad (8.2.1)$$

From equation (3.10)

$$\lambda p^{n-k} q_{n-1} + \{k\mu + (n+1)\nu + c(c^{n-k+1} - 1)/(c-1)\} q_{n+1} = \lambda p^{n-k+1} q_n + \{k\mu + n\nu + c(c^{n-k} - 1)/(c-1)\} q_n$$

; $n = k+1, k+2, \dots$

Multiplying both sides of this equation by s^n and summing over n from we get

$$\begin{aligned} \lambda s \sum_{n=k+1}^{\infty} p^{n-k} q_{n-1} s^{n-1} - \lambda \sum_{n=k+1}^{\infty} p^{n-k+1} q_n s^n &= \sum_{n=k+1}^{\infty} \{k\mu + n\nu + (c^{n-k+1} - c)/(c-1)\} q_n s^n \\ &\quad - \frac{1}{s} \sum_{n=k+1}^{\infty} \{k\mu + (n+1)\nu + (c^{n-k+2} - c)/(c-1)\} q_{n+1} s^{n+1} \end{aligned} \quad (8.2.2)$$

Adding (8.2.1) and (8.2.2) and proceeding in a manner similar to section (8.1), we obtain,

$$Q(1) = \frac{1}{\nu} \left[\begin{aligned} &\lambda(1 - K_{R_EOS}) + \{\lambda q_0 K_{R_EOS}(p\lambda, \mu, \nu)\} / \{p^{k-1} q_0(p\lambda, \mu, \nu)\} - \mu \sum_{n=1}^k n q_n - k\mu(K_{R_BOS} - q_k) \\ &- \{q_0 K_{R_EOS}(c\lambda, \mu, \nu)\} / \{(c-1)c^{k-1} q_0(c\lambda, \mu, \nu)\} + (cK_{R_EOS})/(c-1) \end{aligned} \right]$$

{using (3.14), (8.1.14) and (8.1.15)} (8.2.3)

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¹ Amit Choudhury (born 1968) is a faculty member of the Dept. of Statistics, Gauhati University, India. He completed his Ph.D. from the same university and M.Stat from Indian Statistical Institute, Kolkata, India. He has about fifteen years of teaching experience in Statistics/Operations Research. His research interest is in modeling Queuing scenarios as well as in inferential aspects of Queuing theory. In addition, he has interests in statistical applications in health and education. He has more than fifteen publications in peer reviewed journals of repute in topics ranging from Statistics, Operations Research to those in Public Health, Ophthalmology and Education. He is the life member of a number of professional bodies - Operational Research Society of India, Indian Statistical Association, Indian Society for Medical Statistics, Indian Bayesian Society. Prior to joining academics, he worked for more than three years in SIDBI (Small Industries Development Bank of India).

² Pallabi Medhi (born 1984) is a research scholar of the Dept. of Statistics, Gauhati University, India. She completed her M.Sc. from the same university in 2007. Her research interest is in modeling Queuing scenarios and in data analysis.

SPAM FILTERING FOR OPTIMIZATION IN INTERNET PROMOTIONS USING BAYESIAN ANALYSIS

Ion SMEUREANU¹

PhD Univ. Professor
Academy of Economic Studies
Bucharest, Pta. Romana, no. 2-5, district 1, Romania

E-mail: smeurean@ase.ro



Madalina ZURINI²

Academy of Economic Studies
Bucharest, Pta. Romana, no. 2-5, district 1, Romania

E-mail: madalina.zurini@gmail.com



Abstract: The main characteristics of an e-business and its promoting are presented. It contains ways of promoting an e-business, examined in depth the e-mail marketing principle along with advantages and disadvantages of the implementation. E-mail marketing metrics are defined for analyzing the impact on customers. A model for optimization the promoting process via email is created for reaching the threshold of profitability for electronic business. The model implements Bayesian spam filtering and applies an internal classification, using the principles of conditioned probabilities.

Key words: Bayesian spam filters; e-business; email marketing; knowledge society; metrics; Bayesian conditional probabilities

1. E-business in knowledge-based society

The knowledge society represents a new stage of human evolution, a superior quality lifestyle that involves intensive use of IT in all spheres of human activity, with major social and economic changes. Democracy, communication, understanding and cooperation are the main characteristics of this society, which makes knowledge society to be based on the multitude resources offered by Internet access.

The business environment has changed with the evolution of Internet. A web page with contact details and the activity field have almost every small firm. As a business, for most of the times, represents keeping contact with other persons, knowledge society based on the Internet brings advantages such as: presence, communication, updated information, well served clients, interfering with new techniques of propagation such as: images, sound and video.

The term of e-business is used for the first time by IBM, being defined as a "secure, flexible and integrated access for running various businesses with the combining of processes and systems that execute the basic operations of the business with those that

makes possible finding information over the Internet". E-business is more a strategy in terms of business, the main component being electronic commerce, „e-commerce“.

However, there is little information available on the Internet to help small businesses to explore the online space. Most studies simplify the structure of markets that practice e-economy by treating e-commerce companies as a generic group. The economic market encountered on the Internet can be divided into 3 structures: portal sites, market players and products and services distributors, each of which can be of 2 types: B2C, business oriented customers, or B2B, business oriented business.

Within this informatics and economic framework, the information is the one that makes the difference, and promoting an e-business is the key that opens the connections that any company needs to survive in the market.

In promoting products and services or for promotions, there are two approaches that are used in practice: mass marketing and direct marketing. Mass marketing involves media that disseminate information regarding the products through TV, radio, shops or newspapers. This type of marketing has a target group consisting in a variety of clients, there is no discrimination between them and the information is uniform transmitted. Direct marketing, on the other hand, is different from the mass one because the individual is the target group. Direct Marketing Association's (DMA) defines this type of marketing as "communication where information are systematically used to achieve quantitative objectives and where direct contact between companies and existing customers or potential ones is realized". From this definition it can be concluded that direct marketing classifies customers so that personalized advertisements and promotional activities can be targeted to a particular class of customers.

Over the past years, direct marketing increased in importance. Using the Internet has reduced operational costs of marketing and, even under conditions of low response from customers, it is sufficient to ensure success of the advertising campaigns.

Internet technologies offer more opportunities for online advertising. Most ads are generated by banners and sponsored content on web pages. However, in terms of online marketing, for a promotional campaign, email remains the most effective way of online promoting virtue of the precise destination, immediate response and incredibly low cost. Beyond this, it enables private communication and, properly used, helps building a long-term customers' confidence.

2. Differences between email marketing and spam

For an effective e-marketing strategy, all actions must be linked to the business's objectives. The main purpose of such a strategy is to have long term benefits of a competition, based on performance. This can be achieved through research, planning, experience, analysis and news in the online industry.

Site to the principles of offline marketing that are the 4P: product, price, promotion and placement, email marketing has the main characteristics: personalization, privacy, better customers' service, community, digital content, promotion sales, and security.

Promoting business means an expanded commercial see, public relationships, so that it brings the activity major advantages. In e-business, promotion is done with methods like:

- *Domain name* – as the company's web address is a suggestive name, the more quickly persons can retain it; visit more often, one that can be easily remembered. Most areas from .com or .co.uk have been sold already, so it takes ingenuity to be able to match the name, the business field of the firm with the web page's name. To increase the degree of familiarity, firms tend to find new names, not yet purchased, or use areas with other suffixes as .bizz.

- *Search engines* – Another way to encourage people to access the website of an electronic business is done through a registration on various search engines like Google, Lycos, HotBot or Yahoo. There are search engines that automatically index these websites, with links from other pages to this address. The higher the number of links, the better is the

position returned by the search engine, allowing people that seek information in a particular area to display the Internet pages most often used and spread. If you don't want this to be done manually, there are specialized firms that promote web addresses using search engines.

- *Sites for price comparison* – Web sites such as Kelkoo can be an effective way to promote an e-business so that, for every customer directed to the business, a fee is charged for the promotion site.

- *Electronic markets* – Sites such as Amazon, with external businesses presented, with stocks and prices, where you can buy directly, are useful to any small or large businesses for increasing the coverage of clients. A commission is charged for each sale through these markets, commission that does not diminish the e-business profit.

- *Banners, pop-ups and sponsored links* – Online ads from other web addresses are a good way to promote, with a click a person is directed to the target page. A good example is Google, who wins major revenue through those sponsored links that appear in the top right of the page where search results are returned.

- *Viral marketing* – Represents the spread of marketing messages via email to existing customers or a target group which are then sent by those users to their acquaintance. The advantage is given by the speed and low cost.

Commercial email marketing, also called viral, is one of the most advanced communication technologies in business, changing the way vendors interact with customers. FTC, Federal Trade Commission, in [1], defines a commercial message type email, CEMM, as any email that has as main purpose the promotion of a product or service, including the content of web pages commercially organized. For many organizations, commercial email has become a medium of communications for sales.

The power of modern sales has become impressively based on email. Many sellers can't imagine working without this tool. A research conducted by Electronic Commerce News shows that 85% of businesses use email for their communication with existing customers, 67% test the potential customers' market, 60% use email for an internal communication and 29% of the users dispose email for online orders.

The popularity of email marketing is given by advantages such as:

- lower cost than other means of communication such as direct mail marketing;

- possibility of information's distributing to a wide audience of potential customers using a list of emails;

- dispatch time of an email is very short, order of seconds or minutes, compared to a message sent by mail, which can last even several days;

- ads can be sent to a large number of subscribers to receive communicated via email with subjects that are part of the scope of their interests;

- more than half of Internet users check their emails in a normal working day;

- a precise analysis of the impact can be followed and proved to be high when done properly;

- tracking and metrics based on responses can optimize the transmission channel of emails through a continuous process of testing, using a statistical approach of the finding results;

- email marketing is green.

Email marketing offers the possibility of intercompany communication and their clients but, there are companies that send unsolicited messages, also called spam. The priority that email marketing analyses are focused is finding the threshold beyond which existing clients treat the received advertising emails as ham, legal emails, which they wish to receive, read and respond positive in line with the target set by the companies.

The difficulty of treating emails is given by the subjectivism of the customers that receive these messages.

What is practically a spam message? To answer this question is required both definitions of this type of email.

Spam, objectively and lawfully seen, is an advertisement that was not requested and not given the consent to be received. The subjective definition, however, is more restrictive and, thus, more difficult to respect. Spam is all that recipient considers to be spam.

The consequences of sending messages that may be considered spam by customers who receive them have a negative effect on electronic business.

Legally, in Romania, if you send spam you can't be punished with anything, maybe a fine if there are people with patience, time and will for claiming an email. But, technically, things get complicated.

Yahoo Mail, Gmail, Hotmail, and other such services offer the possibility of reporting those unsolicited messages, considered spam. Thus, if more people proceed in this way, future emails received by the same type as those reported will be automatically targeted spam, but not only for those users who have complained, but for all people using that mail service.

Another consequence is the possibility to block your account by the Internet provider. The blocking may occur through "black lists" which consist of spam senders and are interpreted by anti-spam filters offered default by mail servers or installed by the owners of email addresses, blocking the access of email that are considered spam.

To avoid this situation where email marketing is considered spam, it is important to take account of some recommendations such as:

- messages are sent only to those customers who have given their express consent;
- messages must be well formulated and relevant. Each mail must be unique, as each client has preferences that may differ from the others. It is important to take into account the preferences of each client as that person to handle that email as something personal;
- the address from which emails are sent must be visible, conclusive and, preferably, always the same;
- frequency of messaging is also important. Messages sent from a distance too great when the customer has an account and when he receives the first email advertising are more likely to be considered spam. Or in other circumstances, when the client is bombarded daily with messages, is not a desired approach;
- the use of specializing firm in sending messages, a company that has made a reputation with anti-spam policies formulated and implemented well. In this way, it can be avoided the blocking of these emails to different ISPs, and the rate of messages arriving in inbox, from junk mail, is higher.

3. The quality features of an e-business

Most studies conducted on the critical factors that ensure success in e-commerce focused on limited aspects of electronic business's operations.

On the other hand, Sung, in [2], has made an identification of 16 quality characteristics by applying a Likert scale on 5 levels for evaluation. The sample consisted of large firms in Korea, Japan and USA, who were questioned about the importance of each factor on the success of electronic business. This analysis did not include, however, SMEs (Small and Medium-sized Enterprises). Feidt, in 2002, also introduced these small businesses in research, [3], small innovative companies but with a high potential for development, suggesting that e-commerce is more a strategy than a technology. Table 1 shows, comparing the 2 analysis above, the influencing factors for small and medium companies, on the one hand, and large ones, on the other.

Table 1. Comparative analysis of influence factors

Nr.	Factors for SME (Feindt)	Factors for large companies (Sung)
1	Competitive profile	E-commerce strategy
2	Commitment	E-commerce expertise
3	Content	Variety of goods/services Plenty of information
4	Convenience	Easy to use
5	Control	Delivery of goods/services Payment process
6	Interaction	Customer services Customer privacy
7	Community	Customer relationship
8	Price sensibility	Competitive pricing
9	Brand image	No comparable factor
10	Partnership	No comparable factor
11	Process improvement	Evaluation of operations
12	Integration	Low cost operations
13	Facilitations	System stability, security and speed

Commitment. Studies such as [3] and [4] have identified a commitment to business growth and capacity of firms as a key factor for ensuring success. The growth motivation of the business can be seen by their willingness to adapt to market changes and by their strategies to constantly reinvent. They must be flexible and open to new ideas, technologies and opportunities. When necessary, businesses are prepared to seek strategic partners to provide appropriate technologies, financing, management expertise and contracts for a rapidly evolving of the business.

Content. The site serves as a central point for collecting and disseminating information related to marketing, sales and other functions of the organization, [5]. Web content must be accurate, informed, updated, understandable and related to users' needs. In addition, online sellers should provide an adequate range of products, accurate and relevant information about the products and services held to facilitate comparison of prices with other competitors by the customers.

Convenience. Convenience factor, similar to the degree of use, refer to the development of dedicated user interfaces, and also designs and models presented on the site, [5]. E-commerce sites can implement navigation tools, such as a menu, site map, search engines, audio, video, animation and games, tools for comparing products and services to improve communication. The problem with this is finding that balance between attractive design and fast loading time because "the system response time is inversely proportional to user's satisfaction", [5].

Interaction. Consumers expect high quality interaction with online merchants, mostly the customer service to be prompt and responsive, a non-stop help line, mailing lists and loyalty programs, and an effective system that realizes every step of the ordering process transparent to the users. Properly carried out these activities, it is proven to be very efficient to build confidence and customer loyalty, as it is presented in [3].

Control. The control factor evaluates the systems that e-business have integrated, defined in [5], to track and respond to questions and needs of users, the feedback and products returned, the control of the ordering process and credit risks controlling. Online sellers must audit the operations regular to measure and correct any dissatisfaction of the customers.

Community. This factor emphasizes the interaction between people and online organizations that allow participants to exchange information, such as forums, blogs. In [3],

it is shown that an online active community offers great opportunities for increasing the sales and productivity.

Price sensitivity. Customers expect lower prices of products and services on the Internet. A study by BizRate, [8], showed that the competitive environment forced electronic businesses to provide a reduction in transport cost, with cases in which it really is free.

Brand image. On the Internet, businesses are only one click away and customers are defined only within the language used in sites and geographical regions in which orders can be shipped concerning the economical view. Defining a brand name is essential to differentiate from other competitors in the industry. Lee and Kozar, in [7], have shown that users are willing to pay more for a company with prestige, reputation and a positive image of the brand.

Partnership. For small businesses, partnership is extremely important because of lower resources which force to work mostly closely with customers, suppliers and business partners for brand awareness, improve marketing, launching customized products. It is better to join in business community, because the position of state may attract new businesses and potential partners, being highlighted in the work [6].

Process improvement. Business process uses information technology to automate the organizational processes. This principle is easily applied to large companies, but in SMEs, financial constraints block the process.

Integration. Integrated electronic data exchange, EDI, is a system of services of quality, performance and productivity to support a large volume of business, [3]. Most small businesses don't have the capacity for full integration only if there is support from other partners in technologically advanced.

Following the studies made by Tan, [10], a correlation matrix is resulted, table 2, having 11 influencing factors previously defined. The main diagonal is equal cu 1, state that explains that any variable is all linked, 100% correlated with itself.

We define " \cap " as the correlation between two variables as:

$$x_i \cap x_i = 1$$

$$x_i \cap x_j \in [0;1]$$

For example, the strongest correlation is between Interaction and Content, with value 0.628, which means that the interaction is explained with a rate of 62.8% on web page content.

Table 2. Pearson correlation matrix

	CM	CT	CV	IR	CL	CY	PS	BI	PA	PI	SI
CM	1.000										
CT	0.540	1.000									
CV	0.191	0.494	1.000								
IR	0.422	0.628	0.299	1.000							
CL	0.137	0.339	0.478	0.454	1.000						
CY	0.494	0.350	0.464	0.359	0.361	1.000					
PS	0.192	0.182	0.175	0.219	0.087	0.310	1.000				
BI	0.286	0.251	0.261	0.254	0.187	0.336	0.233	1.000			
PA	0.207	0.269	0.471	0.159	0.470	0.510	0.236	0.179	1.000		
PI	0.220	0.420	0.549	0.334	0.388	0.462	0.243	0.319	0.423	1.000	
SI	0.227	0.356	0.554	0.394	0.537	0.432	0.167	0.273	0.435	0.378	1.000

CM: Commitment; CT: Content; CV: Convenience; IR: Interaction; CL: Control;

CY: Community; PS: Price Sensitivity; BI: Brand Image; PA: Partnership;

PI: Process Improvement; SI: Integration.

The results indicated by this analysis reveals that the 11 factors listed have a correlation between them, some more than others, correlations of which it must be taken into account in a proper marketing analysis. All parts are interconnected, not only improving a process to ensure success, it is important to find that way of comparative and correlate analysis between the main quality factors of an electronic business.

4. Metrics for analyzing email marketing's impact on customers

A good method of analyzing the email marketing's impact on customers is given by the interpretation of metrics showing in detail the main characteristics of electronic business and the level at which they stand.

Thus, research on email marketing and spam are going to 2 lines. The first is mainly focused on reducing spam from a perspective as broad as possible, and the second includes marketing studies to determine the response rate within a promotional campaign through email. For this analysis, the definitions of metrics follow, metrics that are calculated based on the results of a campaign through email but, prior to this campaign, for a further optimization at all stages of implementation.

Sent rate (SR) is the percentage of emails that were successfully sent in the total number of emails sent.

$$SR = \frac{\text{Number of emails successfully sent}}{\text{Number of total emails sent}}$$

, with $SR \in [0; 1]$.

If the emails are not sent for the first time to a list of addresses, this indicator should be included in the interval $[0.9; 1]$. If this indicator has values less than 0.9, this should be further examined to find the causes of this problem because such messages sent do not reach the recipients.

Opening rate (OPR) is the percentage of emails read by the customers of all emails sent successfully. This analysis can be performed on the total number of single openings and also on the multiple ones.

$$OPR = \frac{\text{Number of opens}}{\text{Total number of emails successfully sent}}$$

, with $OPR \in [0; 1]$.

Access rate (AR) is a very important metric, expressing how interested the customers were of those messages sent. It is calculated as the percentage of emails accessed, representing the following of the links in the message, to the total number of emails sent successfully.

$$AR = \frac{\text{Number of accessed email}}{\text{Total number of emails successfully sent}}$$

, with $AR \in [0; 1]$.

Unsubscribe rate (UR) is the percentage formed by the number of client that requested to unsubscribe from the electronic services provided through email to the total number of emails sent successfully.

$$UR = \frac{\text{Number of unsubscribes}}{\text{Total number of emails successfully sent}}$$

, with $UR \in [0; 1]$.

Forwarding rate (FR) is calculated if emails sent have a section that allows forwarding that message to another destination. It is calculated by the number of re-sends divided by the total number of emails successfully sent.

$$FR = \frac{\text{Number of forwarding}}{\text{Total number of emails successfully sent}}$$

, with $FR \in [0; 1]$.

Spam rate (SPR) is defined by the percentage of spam complains to the total number of email successfully sent.

$$SPR = \frac{\text{Number of spam complains}}{\text{Total number of emails successfully sent}}$$

, with $SPR \in [0; 1]$.

Response rate (RPR) is a metric that analyzes the last step, practical purpose of email marketing, namely the interaction between customers and e-business that is completed with the purchase of products or services.

$$RPR = \frac{\text{Number of completed orders}}{\text{Total number of emails successfully sent}}$$

, with $RPR \in [0; 1]$.

Spam probability (PSPAM) represents the weight given to the entire campaign sent emails to be classified as spam.

The main cause that occurs when an email sent is treated as spam by the recipient is explained by spam filters. In addition to the concepts defined in the preceding chapters of black lists and white lists, spam filters also have other technologies for classifying the received messages. Bayesian filtering contains an additional module which focuses on the principle of bayesian classification. This technique gives two probabilities for each word that appears in the email using the formula for conditional probabilities defined by Bayes, namely:

$$P(H/D) = \frac{P(D/H) * P(H)}{P(D)}$$

, where:

- H is the hypothesis and D data;
- $P(H)$ is the prior probability of H, the probability that H is true before the data D to be seen;
- $P(D/H)$ means the conditional probability assigned to the data D and the hypothesis H being satisfied;
- $P(D)$ represents the probability of data D to be realized;
- $P(H/D)$ is the probability that hypothesis H is satisfied, data D being given in reference to the earlier assumption.

From this formula, it was calculated the Bayesian formula of probabilities that a word must belong to a spam mail as:

$$P(S/W) = \frac{P(W/S)}{P(W/M)}$$

, where:

- $P(S/W)$ is the probability that the email containing the word W to be spam;
- $P(W/S)$ represents the probability of occurrence of the word W in spam emails;
- $P(W/M)$ is the probability of occurrence of the word W in all emails held, spam and ham.

Complementary probability above formulated to total $P(T) = 1$ is described by the formula:

$$P(H/W) = \frac{P(W/H)}{P(W/M)}$$

, where:

- $P(H/W)$ is the probability that the email containing the word W is ham;
- $P(W/H)$ stands for the probability of occurrence of the word W in ham emails.

As $H + S = M$ implies that $P\left(\frac{(H + S)}{M}\right) = P\left(\frac{M}{M}\right) = P(1) = 1$.

But, $P\left(\frac{(H + S)}{M}\right) = 1 = P\left(\frac{H}{M} + \frac{S}{M}\right) = P\left(\frac{H}{M}\right) + P\left(\frac{S}{M}\right)$. It is

enough to calculate the probability of an email containing a specific word W to be spam, $P(S/W)$ as the complementary probability, $P(H/W) = 1 - P(S/W)$.

Having defined the formula for a word appeared in an email, to extend to the entire set of words belonging to an email, the composition al probabilities was applied, due to the independent events:

$$P(SPAM/MAIL) = P(S/W_1) * P(S/W_2) * ... * P(S/W_n)$$

, where W_i , with $i \in \{1, 2, ..., n\}$, represents the n words belonging to the tested email. The value of n can be fixed or variable, depending on the desired implementation and optimization of the final test results.

For $PSPAM$ of all promotional campaigns, we aggregate the individual probabilities of each mail sent as:

$$PSPAM = P(SPAM/MAIL_1) * P(SPAM/MAIL_2) * ... * P(SPAM/MAIL_m)$$

, where:

- $MAIL_i$, with $i \in \{1, 2, ..., m\}$, represents the promotional emails and m the number of emails sent.

5. Defining an optimization model in an email campaign

To implement the principles presented in the previous chapters and of the analysis of the indicators defined, a progressive model is realized, consisted of steps to follow to improve quality results in an ad campaign run by email communication channel.

For this, there is need of a previous collection of data with reference to existing customers database containing information on name, email, address, age, gender, occupation and categories of interest of those customers as being buyers of products and services owned and marketed by the e-business. This information will be collected from the customers when they create accounts through the web site and integrated into a well structured database.

The following is a breakdown of each step containing figure 1, along with summarizing the statistical principles for a better understanding of the concept of statistical survey.

Step1. Verification of customers' database. As stated, the database must be constantly updated, for existing no differences between this information and the reality..

Step2. Determination of survey's dimension. To determine the sample size is required to conduct a statistical analysis of the customers. Next, we consider the statistical population as the entire database of the customers.

Suppose we want to find the dispersion of the population when it is divided by gender. For a simple random selection of non-renewable, and a limit error chosen with a probability guarantee of results, sample size, n , is directly proportional to the dispersion of population and inversely proportional to the size of the limit error. Proposed sample size will be equal to:

$$n = \frac{z^2 w(1-w)}{\Delta_w^2}$$

, where:

- z^2 is the probability indicator that is chosen from statistical tables for a 1- α confidence level chosen;
- $w(1-w)$ is the dispersion of the population with the gender classification, variable with two states, 1 or 0, w means the weight of the population that is 0, female;
- Δ_w refers to the limit error chosen, expressed as a percentage.

Knowing the sample size, it remains to choose those n customers from the total population to join the next survey. Choice may be left to the program, choosing randomly, or can be generated in a different way of sorting. To simplify the process, will choose the random method.

Step3. Training the spam filter. Bayesian spam filter is a type of artificial intelligence method, one of the initiators in this field being Paul Graham in [12], [13] and [14], a method involving previously training. Training is conducted on the basis of emails collected from various sources in the company, to be able to simulate as conclusive as possible a spam filter that a customer might use it. After all emails are loaded, the filter will calculate a spam weight for each word encountered in the emails. This weight will be used in the step where PSPAM probability will be calculated.

Step4. Formulation of advertising emails. This phase includes the formulation, analysis in terms of social, psychological, economic and visual future advertising emails from the promotional campaign. It will be taken in account of previous results made, for a better understanding of the customers, their way of thinking and decision making.

Step5. PSPAM calculation. Before sending the emails, PSPAM indicator is calculated as defined in the previous chapter in order to make an earlier analysis of the sent messages. The purpose is to allow internal simulation of the messages' interpretations by the customers.

Step6. Probability's analysis. The metric PSPAM is compared with a threshold given, suppose the value of 0.1. If the metric's value is below this threshold means that the emails have the probability to be considered spam less than 1%, and thus, step 7 will be preceded. Otherwise, the messages do not pass the first stage of analysis and will be redrafted and returned to step 4.

Step7. Emails' sending to the sample set. The emails are sent to the customers selected in step 2.

Step8. Analysis of the sample's results. Based on the results received and the feedback provided by sending the n emails, the metrics defined in chapter 4 are calculated.

Step9. Decision the extend the sample. The metrics' results are economical analyzed and a decision is being taken whether it can continue to expand the sample throughout the population. If the results are favorable, step 10 is preceded. In the other case, the process returns to step 4, for an reassessment of the existing situation.

Step10. Sending the remaining emails. At this stage, the advertising emails are considered to be as close to that level desired by the company, having a high probability of success for the entire set of customers. The remaining emails are sent to the rest of the customers.

Step11. Final analysis. The process being finished, the last conclusions are drawn reanalyzing the defined metrics, optimizing the conclusions for a future campaign correction.

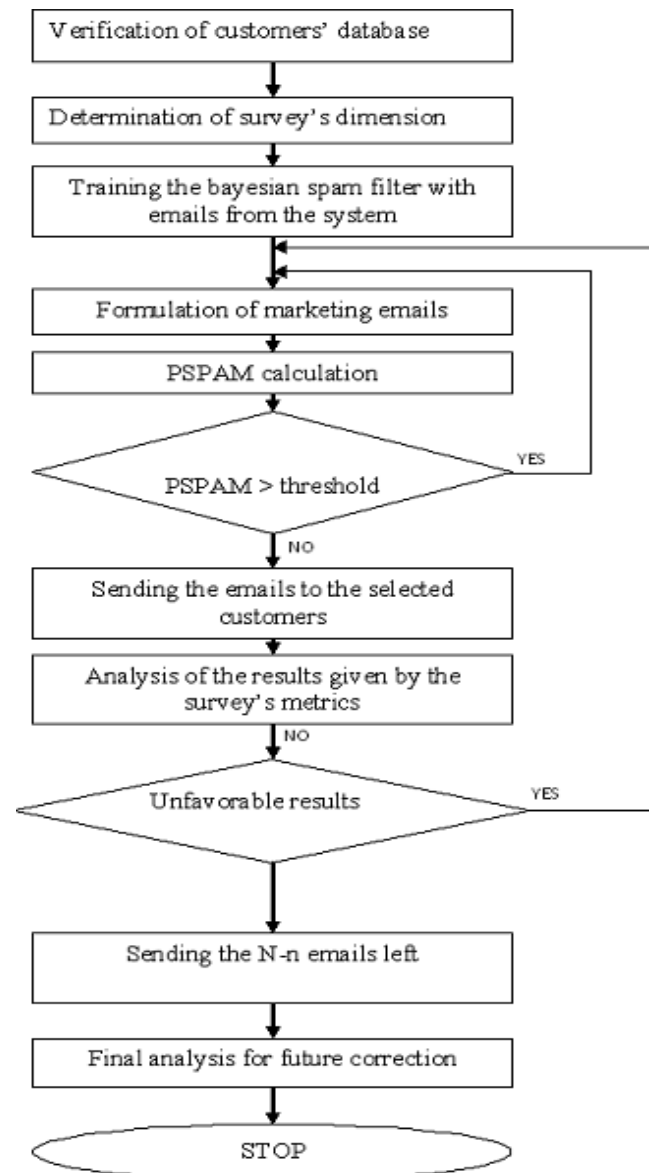


Fig 1. Steps of email promoting model's implementation

The principle of survey was chosen because it is more convenient for an electronic business to test an ad campaign on a more limited range of customers than to send a message that will be interpreted negatively to a much larger population. This theory is based on the trust that given by the customers to online businesses and the level of acceptance of entering into their privacy.

6. Conclusions

The knowledge society, along with the Internet communication channel, particularly the email, has made the promoting ways of businesses to get a different structure. Electronic businesses are seen by grown demanding of the customers who expect accurate information, concrete, rapid information to be made in a honest, convenient, interactive, to meet all the needs that a customer has.

Marketing's new paradigm, described in [15], paradigm which is introduced by the web, makes a change from product to customer, this including a new personalized micro-level and a management of customer relationship. The problems that arise in such a

framework are given for how such information should be submitted, to comply with customer's privacy, without diminishing the intensity of products and online services' promoting

The model presented in chapter 5 optimizes the flow of information and focuses the tests made for market evaluation on a limited number of customers. The major benefit is reducing the negative effects of an advertising campaign that hasn't the correct target group chosen, doesn't transmit the desired and expected information to the customers and doesn't consider the preference generally provided by the users when registering in the internal e-business system.

The best way to see the difference between email marketing and spam is in terms of the customers, those who receive these messages, because, as it was presented, spam is something subjective, in addition to legal and objective vision given by various laws, research and interpretations.

Spam is all that the recipient considers to be spam, so all we have to do left is to model the email sent to customer requirements, to personalize and optimize, to transform it from an advertising email to a personal one, applying sociological, mental, statistical, artificial intelligence and marketing theories.

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¹ **Ion SMEUREANU** has graduated the Faculty of Planning and Economic Cybernetics in 1980, as promotion leader. He holds a PhD diploma in "Economic Cybernetics" from 1992 and has a remarkable didactic activity since 1984, when he joined the staff of Bucharest Academy of Economic Studies. Currently, he is a full Professor of Economic Informatics within the Department of Economic Informatics and the dean of the Faculty of Cybernetics, Statistics and Economic Informatics from the Bucharest Academy of Economic Studies. He is the author of more than 16 books and an impressive number of articles. He was also project coordinator or member in many important research projects. He was awarded the Nicolae Georgescu-Roegen diploma, the award for the entire research activity offered by the Romanian Statistics Society in 2007 and many others.

² **Madalina ZURINI** has graduated the Faculty of Cybernetics, Statistics and Economic Informatics in 2008 and she is currently conducting a master in computer science researching the *implications of Bayesian classifications for optimizing spam filters*. She is also engaged in a Pedagogical Program as part of the Department of Pedagogical Studies. Her fields of interest are data classification, artificial intelligence, algorithm optimisations, mathematical and econometrical analysis of factors' influences. She wants to pursue a pedagogical career and also to give her PhD in the same area of research

E-BANKING IN DEVELOPING ECONOMY: EMPIRICAL EVIDENCE FROM NIGERIA

Elisha Menson AUTA

PhD Candidate, Centre for Policy and Economic Research,
University of Abuja, Abuja, Nigeria



E-mail: donelly2007@yahoo.com

Abstract: This paper empirically examines the impact of e-banking in Nigeria's economy using Kaiser-Meyer-Olkin (KMO) approach and Barlett's Test of Sphericity which support the use of factor analysis in order to extract independent variables associated with e-banking. The paper explores the major factors responsible for internet banking based on respondents' perception on various e-banking applications. It also provides a framework of the factors which are taken to assess the e-banking perception. Due to emergence of global economy, e-business has increasingly become a necessary component of business strategy and a strong catalyst for economic development. E-banking has become popular because of its convenience and flexibility, and also transaction related benefits like speed, efficiency, accessibility, etc. The results of this study shows that e-banking serves several advantages to Nigerian banking sector. The customers (respondents) perception is that e-banking provides convenience and flexible advantages. It also provides transaction related benefits like easy transfer, speedy transaction, less cost and time saving. However, the study shows that the Nigerian customers have security, access, and no enough knowledge regarding e-banking services rendering by banking sector in Nigeria. The study suggest that critical infrastructure like power and telecommunication should be provided and with high level of stability to ensure the application of e-banking in Nigeria. Also, the relative skewed nature of banks location mostly in urban area should be addressed to ensure spread and accessibility by rural dwellers.

Key words: E-banking, Developing Economy, Empirical Evidence, Nigeria

1. Introduction

Financial services industry over time has opened to historic transformation that can be termed as e-developments which is advancing rapidly in all areas of financial intermediation and financial markets such as e-finance, e-money, electronic banking (e-banking), e-brokering, e-insurance, e-exchanges, and even e-supervision. The new information technology (IT) is turning into the most important factor in the future development of banking, influencing banks' marketing and business strategies. In recent years, the adoption of e-banking began to occur quite extensively as a channel of distribution for financial services due to rapid advances in IT and intensive competitive banking markets (Mahdi and Mehrdad, 2010; Dube, et. al., 2009). The driving forces behind the rapid transformation of banks are influential changes in the economic environment include among others innovations in information technology, innovations in financial products, liberalization and consolidation of financial markets, deregulation of financial inter-mediation. These factors make it complicated to design a bank's strategy, which

process is threatened by unforeseen developments and changes in the economic environment and therefore, strategies must be flexible to adjust to these changes.

The e-banking is transforming the banking and financial industry in terms of the nature of core products /services and the way these are packaged, proposed, delivered and consumed. It is an invaluable and powerful tool driving development, supporting growth, promoting innovation and enhancing competitiveness (Gupta, 2008; Kamel, 2005). Banks and other businesses alike are turning to IT to improve business efficiency, service quality and attract new customers (Kannabiran and Narayan, 2005). Technological innovations have been identified to contribute to the distribution channels of banks and these electronic delivery channels are collectively referred to as electronic banking, (Goi, 2005). The evolution of banking technology has been driven by changes in distribution channels as evidenced by automated teller machine (ATM), Phone- banking, Tele-banking, PC-banking and most recently internet banking (Chang, 2003; Gallup Consulting, 2008).

E-banking is the term used for new age banking system. E-banking is also called online banking and it is an outgrowth of PC banking. E-banking uses the internet as the delivery channel by which to conduct banking activity, for example, transferring funds, paying bills, viewing checking and savings account balances, paying mortgages and purchasing financial instruments and certificates of deposits (Mohammed, et. al., 2009). It is difficult to infer whether the internet tool has been applied for convenience of bankers or for the customers' convenience. But ultimately it contributes in increasing the efficiency of the banking operation as well providing more convenience to customers. Without even interacting with the bankers, customers transact from one corner of the country to another corner.

Electronic banking has experienced explosive growth and has transformed traditional practices in banking (Gonzalez, 2008). As per prediction of Maholtra and Singh, (2007) the e- banking is leading to a paradigm shift in marketing practices resulting in high performance in the banking industry. Delivery of service in banking can be provided efficiently only when the background operations are efficient. An efficient background operation can be conducted only when it is integrated by an electronic system. The components like data, hardware, software, network and people are the essential elements of the system. Banking customers get satisfied with the system when it provides them maximum convenience and comfort while transacting with the bank. Internet enabled electronic system facilitate the operation to fetch these result.

According to Christopher, et. al., (2006), E banking has become an important channel to sell the products and services and is perceived to be necessity in order to stay profitable in successful. There is a growing interest in understanding the users' experience (Pyun, 2002), as e-banking is observed to be a larger concept than user satisfaction. From this perspective, assessing the user experience is essential for many technology products and services (Salehi, et. al., 2008). Customers have started perceiving the services of bank through internet as a prime attractive feature than any other prime product features of the bank. Customers have started evaluating the banks based on the convenience and comforts it provides to them.

This study aims to explore the major factors responsible for e-banking in Nigeria based on respondents' perception on various internet applications, participants perception about e-banking and whether the user and non-user perception differs. The three critical factors of interest are convenience and flexibility, transaction related benefits, and demographic variables (gender, location, etc.)

2. Review of Literature

The concept of e-banking is a delivery channel for banking services. Banks have used electronic channels for years to communicate and transact business with both domestic and international corporate customers. With the development of the Internet and the World Wide Web (WWW) in the latter half of the 1990s, banks are increasingly using electronic channels for receiving instructions and delivering their products and services to their

customers. This form of banking is generally referred to as e-banking or Internet banking, although the range of products and services provided by banks over the electronic channel vary widely in content, capability and sophistication. E-banking is defined as the automated delivery of new and traditional banking products and services directly to customers through electronic, interactive communication channels. The definition of e-banking varies amongst researches partially because electronic banking refers to several types of services through which bank customers can request information and carry out most retail banking services via computer, television or mobile phone (Daniel, 1999; Sathye, 1999). Salehi and Zhila, (2008), describes e-banking as an electronic connection between bank and customer in order to prepare, manage and control financial transactions. Electronic banking can also be defined as a variety of following platforms: (i) Internet banking (or online banking), (ii) telephone banking, (iii) TV-based banking, (iv) mobile phone banking, and e-banking (or offline banking).

E-banking includes the systems that enable financial institution customers, individuals or businesses, to access accounts, transact business, or obtain information on financial products and services through a public or private network, including the Internet or mobile phone. Customers access e-banking services using an intelligent electronic device, such as a personal computer (PC), personal digital assistant (PDA), automated teller machine (ATM), kiosk, or Touch Tone telephone. While some literature restricts the use of the term to internet banking (Daniel 1999), elsewhere the term is limited to retail banking (Aladwani 2001) or both retail and corporate banking (Simpson 2002). The common definition for e-banking, and the one used in this paper, comes from the Basel Committee Report on Banking Supervision (1998), "e-banking refers to the provision of retail and small value banking products and services through electronic channels. Such products and services can include deposit-taking, lending, account management, the provision of financial advice, electronic bill payment, and the provision of other electronic payment products and services such as electronic money".

Karjaluoto, et. al., (2002) indicated that banks have the choice to offer their banking services through various electronic distribution channels technologies such as Internet technology, video banking technology, telephone banking technology, and WAP technology. They also indicated that Internet technology is the main electronic distribution channel in the banking industry. In other words, e-banking as an online banking that involves the provision of banking services such as accessing accounts, transferring funds between accounts, and offering an online financial service.

Wang, et. al., (2003) claims that in the 1990s e-banking was under-utilised as business organisations used it only to market their products and services. Thornton and White (2001), examined customer orientations and usage of financial distribution channels in the Australian financial industry, found that more recently most financial institutions, faced with competitive pressure after the introduction of deregulation in 1983, have rethought their strategies to take full advantage of IT. Rafiu (2007) opines that the challenge to expand and maintain banking market share has influenced many banks to invest more in making better use of the Internet. The emergence of e-banking had made many banks rethink their IT strategies in competitive markets. This findings suggest that the banks that fail to respond to the emergence of e-banking in the market are likely to lose customers and that the cost of offering e-banking services is less than the cost of keeping branch banking.

This notion was also confirmed in a study conducted by Jasimuddin (2004) examined the role of e-banking in Saudi Arabia. He indicated that the majority of Saudi banks had taken advantage of Internet technology to establish web sites but few offered e-banking services. He suggested that if the Saudi Arabian banking industry wished to be successful in the global economy it would need to integrate Internet technology into its banking strategy.

Ayo (2006) investigated the prospects of e-commerce based on ability, motivation and opportunities (AMO) model and observed that virtually all companies have online presence. The paper reported the motivation and opportunities for e-commerce as low based on lack of e-Payment infrastructure and access to information and communication technology (ICT) facilities. Also, in an empirical assessment of customer acceptance of e-

Commerce carried out in Germany, Buse and Tiwari (2006) observed that: the highest mobile users are top management, followed by self employed, salaried class, students and others. Government employees were found not to patronize mobile banking; the most favoured reason for carrying out mobile banking is ubiquity, next is overview of bank account, followed by immediacy; and the highest fear of customers about mobile banking is that of insecurity, next is cost, and uncomfortably.

Mahdi and Mehrdad (2010) used chi-square to determine the impact of e-banking in Iran and there findings from the view points of customers is that, e-banking cause higher advantages to Iranians. In other words, Iran banks provide services that the customers are deriving satisfaction with particular reference to the use of e-banking. In a similar study, Jayawardhena and Foley (2000) explore e-banking as a new delivery channel arguing that e-banking may help to overcome the inherent disadvantages of traditional banks; it is very clear that if e-banking conducted successfully it leads to big volume of transactions. Further, Birch and Young (1997) argue that the internet may be exploited as a new delivery channel by the financial services industry to completely reorganize the structure of banks. It means that conducting e-banking in Iran leads more usage of ATM in Iran. The authors came to conclusion that the active ATM in banking sectors will cause cash circulation decreases, the efficiency of banking sector will increase, as: a. client banking costs decreases (less cash fees to pay), b. shop keeper / service provider costs will decrease, and c. bank costs decrease (cash storage, less checking and processing costs), costumers have not enough knowledge related to e-banking in Iran. Accordingly the null hypothesis is rejected also. The authors believe that the lack of enough information on e-banking in Iran may cause less efficiency of Iranian banks. To achieving high efficiency both bankers as well as Iranian legislators should introduce e-banking services at mass level.

Chiememeke et al. (2006) conducted an empirical investigation on adoption of e-banking in Nigeria. The study identified the major inhibiting factors to Internet banking adoption in Nigeria such as, insecurity, inadequate operational facilities including telecommunications facilities and electricity supply, and made recommendations on how Nigeria banks can narrow the digital divide. Also, the report revealed that Internet banking is being offered at the basic level of interactivity with most of the banks having mainly information sites and providing little Internet transactional services.

Similarly, Agboola (2006) investigated electronic payment systems and tele-banking services in Nigeria. The findings revealed that there has been a very modest move away from cash. Payments are now being automated and absolute volumes of cash transactions have declined. The result of the study revealed that tele-banking is capable of broadening the customer relationship, retain customer's loyalty and enable banks to gain commanding height of market share if their attendant problems such as, ineffectiveness of telecommunications services, epileptic supply of power, high cost, fear of fraudulent practices and lack of facilities necessary for their operation were taken care of.

Thus, going by the findings of most studies, we can argue that the literature on the impact of e-banking is inconclusive especially in developing economies and serve as an open ground for more research in the area of e-banking.

Framework of the factors

We mention that this research survey focuses on finding the customers' perception on various internet application related with e-banking. Various factors which contribute to the customers' perception such as convenience, flexible virtual banking system, reliability, time factor, real time access to information, saving transaction cost, on-line bill payments, digital signature for security, faster transfer, easy to use, user friendly, low transaction fees, any time and anywhere banking facility, access to current and historical transaction data, facility of fund transfer to third party, etc, (see Divya and Padhmanabhan, 2008) and speed, operational efficiency, better cash management, expanded financial reach (Mahrdr and Mehrdad, 2010). Some of the factors are discussed below.

1. Digital signature for security: Security is rated as the most important issue of online banking. There is a dual requirement to protect customers' privacy and protect

against fraud. Digital signature is a precautionary measure to prevent malpractices and tampering the information. It is a form of enhanced authentication (Williamson, 2006).

2. Convenience way of operating banking transactions: Online banking is a highly profitable channel for financial institutions. It provides customers convenience and flexibility and can be provided at a lower cost than traditional branch banking (Beer, 2006).

3. Faster transfer: The fundamental advantage of the e-banking is the transfer of the information about the money's worth to any place at any time with a mouse click's distance (Dube, et. al., 2009)

4. Reliability: Kamel (2005) identified one of the very important service quality dimensions of e-banking service quality is reliability. The online banking environment has grown tremendously over the past several years and will continue to grow as financial institutions continue to strive to allow customers to complete money transfers, pay bills, and access critical information online. Authenticating customers logging onto their online banking service has become a crucial concern of financial institutions (Williamson, 2006)

5. Time factor: Liu and Arnett in their study identified time factor as one of the prime factor that in e-banking service quality feature for the customers. Saving time is an importance factor which influences the customers prefers to use e-banking. (Beer, 2006). Banks can make the information of products and services available on their site, which is, an advantageous proposition.

6. Real time access to information: The banks started e-banking with simple functions such as real time access to information about interest rates, checking account balances and computing loan eligibility. Then, the services are extended to online bill payment, transfer of funds between accounts and cash management services for corporate organizations (Mohammed, 2009).

7. Queue management: One of the important dimensions of e-banking service quality is queue management (Agboola, 2006).

8. Saving transaction cost: Improving customer service, increasing market reach and reducing costs are now basic expectations of Internet banking services. If consumers are to use new technologies, the technologies must be reasonably priced relative to alternatives. Otherwise, the acceptance of the new technology may not be viable from the standpoint of the consumer (Al-Sukhar, 2005). Internet banking model offers advantages for both banks and customers. The Internet provides the banks with the ability to deliver products and services to customers at a cost that is lower than any existing mode of delivery. Another factor that would stand in the way of consumer adoption of e-banking is the cost factor.

9. Easy to use and user friendliness: Ease of use is an important determinant for the customer preferring the internet banking (Beer, 2006). In a study conducted by Karjaluoto, et. al., (2002); reported that ease of use of innovative product or service as one of the three important characteristics for adoption from the customer's perspective. The user friendliness of domain names as well as the navigation tools available in the web-sites is an important determinant for ease of use.

10. Any time and anywhere banking facility: Online banking users say that convenience is the most important factor, online banking lets them access their accounts from anywhere and at any time (Maholtra and Singh, 2007)

11. Access to current and historical transaction data: A customer can check balance by logging into banks website through a user name and password. In this way he can enquire balance, status of cheques, perform funds transfers, order drafts, request issue of cheque books etc (Gupta, 2008). Customers prefer to view account balances, transaction history and updates get e-statements, credit card and debit card transaction history and updates, checking the status of their credit card accounts, viewing information regarding their account, information on their fixed deposits on line.

3. Research Methodology

Research design

An exploratory research design was considered the most suitable approach in view of the nature of the problem being investigated. A structured questionnaire adapted and modified as used by Divya and Padhmanabhan, (2009) was used as the main data-gathering instrument. The questionnaire was divided into four sections. Section A captured basic demographic information regarding the banks such as age of the bank, capital base and the number of branches nation wide. Section B captured information about the adoption and usage of e- banking services. Section C sought to determine the perceived benefits of e-banking and while section D captured information about the nature of the challenges faced in the adoption and usage of e-banking. The last two sections used a five point Likert Scale battery where the respondents were asked to indicate the extent to which they agree/disagree with various statements. The Five-Point Likert's scale having the ratings of "strongly disagree" (1) and "strongly agree" (5) were used. Due to commercial confidentiality and sensitivity of the banking information the questionnaire was designed in a manner that did not require the respondents to reveal their names nor their banking institutions.

Data collection

The study sample consisted all the 25 commercial banks in Nigeria. All the commercial banks in Nigeria are head quartered in either Abuja (the Capital city) or Lagos, hence it was imperative to focus on these branches as they generally reflect technologies by sister branches. All the twenty-five (25) banks filled and returned the questionnaires. Data was collected over a period of three months commencing from the second week of January 2010 to the fourth week of March 2010. Statistical Package for Social Sciences (SPSS) version 10 was used as the statistical analysis tool while descriptive statistics were computed and used in the interpretation of findings.

A total of one thousand (1000) questionnaires were randomly administered to customers from diverse employment background and bankers. Seven hundred and fifty (750) were returned, which represents 75.00% of the total respondents or participants.

4. Empirical Analysis and Interpretation of Results

Factor Analysis was performed with 20 statements related with e-banking features. The Kaiser-Meyar-Olkin (KMO) for was .754 and significant Barlett's Test of Sphericity supported the use of factor analysis in order to extract independent variables associated with e-banking. The degree of common variance among the twenty variables is "mediocre" which reflects if a factor analysis is conducted, the factors extracted will account for fare amount of variance but not a substantial amount.

An exploratory principal component factor analysis was done using SPSS 16.0. Varimax rotation was used to identify the underlying factors for e-banking features. Items with eigen values greater than one were extracted and all factor loading greater than 0.5 were retained. Twenty items yielded 6 factors explaining 88.05 percent of variance as shown in table 1.

Table 1. Rotated Component Matrix

Factors	Items	Factor Loadings	Variance Explained	Reliability (Cronbach's Alpha)
1. Time factor			15.345	.812
2. Security			17.560	.765
3. Queue management			12.565	.764
4. User friendly			14.420	.910
5. Fund transfer			15.2085	.650
6. Accessibility			13.075	.673
			88.05	

The variables above are considered as fundamental consideration associated with e-banking. General perception about e-banking was gauged by 20 items. Out of which six items were related with convenience and flexibility and 9 items were related with transaction related benefits, and 5 items were related to access or location. All items were measured on a scale of 1 to 5. From the questionnaire convenience and flexibility related items clubbed together and average score taken to gauge the respondents' perception about convenience factor. Out of total respondents 84 % respondents felt that e-banking is very convenient and flexible. And same percentage that is, 84% from total users agrees or strongly agrees that e-banking is convenient. They felt that it gives benefits like no queuing in bank and one can do anytime and anywhere banking. Approx 79% of total respondent agreed that internet banking has transaction related benefits. These benefits include efficient and speedy transfer of funds with lower transaction cost. And, with e-banking can check transaction details regularly without any hassle. This result found support in Mahrddi and Mehrdad, (2010); Divya and Padhmanabhan, (2009) but contrary to Buse and Tiwani (2006).

Generally, most banks in Nigeria are located within the city or urban centres where infrastructure is available. Therefore, examining e-banking with respect to location became insignificant since banking services are rural-biased. Thus, the study has location challenge by rural and peri-urban dwellers in Nigeria who are compelled to move to the city to access banking services.

Table 2: Testing of factors which contribute to the customers' perception

Factors	Items	Cronbach Alpha
Convenience and flexibility	<ul style="list-style-type: none"> • -A convenient way of operating banking transactions - A - Avery flexible virtual banking system • - Save time as compared to conventional banking • - No queuing in bank branches • - Easy to use/ user friendly • - any time /any where banking facility • 	0.840
Transaction related benefits	<ul style="list-style-type: none"> -saves a lot of transaction cost -transaction is efficient - it has lower transaction fees - it allows speedy transfer - it allows easy access to transaction data both recent and historical -it can check transaction details and statement regularly - it gives facility to fund transfer to third party - Speed - Better cash management 	0.793

Source: Variables (factors and items) source from Divya and Padhmanabhan (2008)

The bio-data of participants is summarized in Appendix 1:

The bio-data in appendix 1 shows that out of 750 participants 411 (54.80%) are male and 339 are female. It means that the majority of participants are male. Majority of the respondents were between the ages 31 to 40 years, which represent 51.90% followed by ages 41 to 50 years with 29.60%. Relating to job position, most of the respondents were from the private sector 37.60% followed by public sector 32.14%. While 43 (5.73%) were clerks and others (teacher, student and housewife) 55(7.33%).

Regarding to educational background, out of 750 respondents 326(43.47%) had bachelor degree or higher national diploma (HND) followed by Diploma and national certificate of education (NCE) 120(16.00). Master degree had 195(26.00) participants. The least number of participants had PhD degree and Professor 109(14.53%). The location of the participants shows that most of the respondents resides in urban area 509(67.87), peri-urban 204(27.20), and rural area has the least with 37(4.93). The ratio of participants with respect to customer and bankers shows 448(59.73%) and 302(40.27%) respectively.

5. Conclusion

The banking industry play a significant role in supporting economic development through efficient financial services (Dube, et. al., 2009; Salehi and Azary, 2008). Nigerian banks have embraced innovative banking technologies and e-banking services in recent years. Almost all banks have invested in expanding and improving the Information Technology systems and a number of new e-banking services have been developed. All the 25 commercial banks operating in Nigeria have declared e-business as one of the core strategies for the future development. At the same time, e-banking acceptance depends probably on bank service quality, customer preferences and satisfaction.

The analysis done with the help of statistical tools clearly indicate the factors responsible for e-banking. Factor analysis results indicate that security, user friendly, queue management, accessibility, time factor and fund transfer are major factors. Out of total respondents' about 88% agreed that e-banking is convenient and flexible way of banking and it also has various transaction related benefits. Thus, Providing e-banking is increasingly

becoming a matter of need to banks to continue to compete in a globalized work and gain market competitive advantage.

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Appendix 1: Bio-data of participants or respondents

Variable	Category	Frequency	Percentage
Gender	Male	411	54.80
	Female	339	45.20
Age	20 – 30	54	7.20
	31 - 40	389	51.90
	41 – 50	222	29.60
	51 & above	85	11.30
Job Position	Clerk	43	3.73
	Private sector	282	37.60
	Public sector	241	32.14
	Public servant	129	17.20
	Others	55	7.33
Education	Diploma/NCE	120	16.00
	Degree/HND	326	43.47
	Master	195	26.00
	PhD/Professor	109	14.53
Location	Rural	37	4.93
	Peri-rural	204	27.20
	Urban	509	67.87
Participants	Customer	448	59.73
	Banker	302	40.27

Source: Field Survey 2010 (January – March)

WEB SERVICES INTEGRATION WITH DISTRIBUTED APPLICATIONS

Marilena DUMITRACHE

PhD Student, Faculty of Cybernetics Statistics and Economic Informatics, Academy of Economic Studies, Bucharest, Romania

E-mail: dm.marilena@gmail.com



Stelian DUMITRA

PhD Student, Faculty of Cybernetics Statistics and Economic Informatics, Academy of Economic Studies, Bucharest, Romania

E-mail: stelu20d@yahoo.com



Mircea BACIU

PhD Student, Faculty of Cybernetics Statistics and Economic Informatics, Academy of Economic Studies, Bucharest, Romania

E-mail: mircea.baciu@gmail.com



Abstract: A Web service is a distributed application component. Web services distributed computing model allows application-to-application communication. There is nothing fundamentally new about the basic concept and the related technologies. The innovative thing about this is the reach of Web services and its ubiquitous support by literally all major vendors. Most likely, heterogeneity will at the end no longer be an obstruction for distributed applications. This paper describes the concept of service-oriented architecture (SOA) in conjunction with the Web services technology and covers the core Web services specifications which form a powerful and robust foundation for building distributed systems. It is presented a case study regarding the integration of the Web services with the SAP system for handling interoperability issues.

The conclusions and the future proposed developments are presented in the end of the paper.

Key words: distributed application; web service; sap; protocols; security

1. Introduction

Because of the level of the application's integration, the Web services have grown in popularity and are beginning to improve the business processes. In fact, the Web services are being called the next evolution of the Web [1].

Web services provide a promising framework for development, integration, and interoperability of distributed software applications. Wide-scale adoption of the web services technology in critical business applications will depend on the feasibility of building highly dependable services. Web services technology enables interaction of software components across organizational boundaries. In such distributed environment, it is critical to eliminate errors at the design stage, before the services are deployed. Web services provide a promising framework for development, integration, and interoperability of distributed software applications. Wide-scale adoption of the web services technology in critical business applications will depend on the feasibility of building highly dependable services.

The remainder of the paper is structured as follows: section 2 provides information about service-oriented architecture (SOA) in conjunction with the Web services technology and the core Web services specifications; section 3 describes a case study on web services and associated key technologies. The application shows how to integrate Web Services on different platforms and how they allow the interoperability between applications running on these platforms, using the specific web services protocol stack presented in section 2. Section 4 concludes the paper and presents future proposed developments.

2. Service Oriented Architecture vs. Web Services

Distributed software systems, and the interactions between components within these systems, can exhibit a high level of complexity and lead to difficulty in the assessment of what system behavior is possible in multiple scenarios [2].

2.1. SOA

Nowadays, service-oriented architecture (SOA) and web services that enable flexible and loose integration of applications within and across enterprises have become one of the most phenomenal subjects both in academia and in industry.

SOA is a software architectural concept that defines the use of services to support the requirements of software users. It is a system for linking resources on demand. In an SOA, resources are made available to other participants in the network as independent services that are accessed in a standardized way. This provides for more flexible loose coupling of resources than in traditional systems architectures [3].

The SOA model treats three main elements that act as a find-bind/invoke-execute cycle as shown in Figure 1. The service provider offers a given service and publishes service description in a service registry. The service requester queries the registry to find a certain service. If found, it retrieves the location of the service and binds to the service endpoint, where the requester can finally invoke the operations of the service [4].

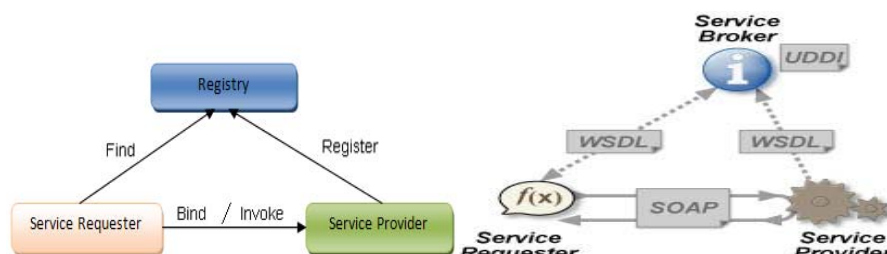


Figure 1. SOA triangle
Services framework

Figure 2. Architecture of the Web

SOA-based applications are distributed multi-tier applications that have presentation, business logic, and persistence layers. Services are the building blocks of SOA applications. While any functionality can be made into a service, the challenge is to define a service interface that is at the right level of abstraction. Services should provide coarse-grained functionality [5].

Within SOA, web service providers describe their services in WSDL to designate what they are and how to invoke them, and then publish these descriptions via a public UDDI registry. On the other hand, a service requester subscribes those WSDL descriptions and selects such services that satisfy an integration need. The requester often has to compose several services to accomplish complex tasks. Then, the requester invokes selected web services using XML/SOAP messages. All these can be seen in figure 2 [6].

2.2. Web Services

The client's basic needs over time don't really change, but the essential tools that are required in order to fulfill these needs are constantly evolving. Not long ago in nodes, which were present in distributed systems, existed the need to control the applications in a distributed manner. [7] Basically if an application that was running in a node happened to go down, that application was suppose to be restarted at another node. The creation of these types of distributed applications was nearly impossible. These days, it's routine and in fact there are many choices. The essential problem is not if it is possible for the components of distributed applications to communicate between them, but to choose the best technology in order to hold them together.

For example the .NET Framework, introduces good support for the two ways to architect a distributed application. Remoting is the architectural descendant of DCOM, allowing an object on one computer to make proxy-based calls to the methods of an object on another computer. Web services use a completely different technique, based on open XML and SOAP protocols, WSDL and UDDI, which are used to invoke methods on a remote machine. XML is used to tag the data, SOAP is used to transfer the data, WSDL is used for describing the services available and UDDI is used for listing available services.

"Web services are dynamic programs that enable data and applications to interact with each other on the Web through ad hoc connections—without any human intervention what so ever", said Sidharth, technical product manager for identity management at Sun. A Web service is normally intended to be a distributed application component. Its clients are other applications, not human beings. A Web Service is any piece of code that can communicate with other pieces of code via common Internet technology. A Web Service is a "virtual component" that hides "middleware idiosyncrasies" like the underlying component model, invocation protocol as far as possible.

The main advantages of Web services are flexibility and versatility: they support a lot of architecture and are independent of platforms and designs. Web services are built on several technologies that work in conjunction with emerging standards to ensure security and manageability, and to ensure that Web services can be combined to work independent of a vendor. Also Web services win on ease of development and interoperability.

Web services distributed computing model allows application-to-application communication. For example, one purchase-and-ordering application could communicate to an inventory application that specifies the items that need to be reordered or a Web service from a credit bureau which requests the credit history from the loan services, for prospective borrowers. In both cases, the data interaction must be protected to preserve its confidentiality.

Web Services are considered to be the future of the Internet. They are independent of the platform and also of the technology, but in reality they are XML/SML collections of standards which allow the interaction between systems (programs). Heather Kreger, one of the IBM's lead architects for SOA Standards which developed the standards for Web services, thought that Web Services are like an interface which describes a collection of operations, network accessible throughout the XML standard messages. Web Services have the main role to access different services and different data from different machines, and so they offer to the clients a single public interface.

2.3. The Web Services Architecture

Figure No. 3 describes the architecture of the Web Services. The architecture of the Web Services resembling with the TCP/IP reference model is presented in five levels: Network, Transport, Packaging, Description and Discovery. Each level is represented by different basic protocols. The network level concurs with the network level from the TCP/IP [8] reference model, offering basic communication, addressing and routing. Above the network level there is the transport level that offers the opportunity of direct communication between the existing applications from the network. The most important protocols are TCP/IP, UDP, FTP, HTTP, SMTP, Jabber [9].



Figure 3. Web Services Architecture

The web services can be implemented above any of the other protocols. The Packing level, which is above the Transport level, "packs" the data in the XML format – a format known by all the other parties involved in communication. XML and SOAP – Simple Object Access Protocol, are basic protocols of the Packing Level and are produced by the W3C standard.

2.4. SOAP – Simple Object Access Protocol

The role of SOAP is to encode the data in XML format and to make the message exchange possible between the applications in XML format. It uses the model request-answer, where the request is placed by the SOAP client, and the answer is given by the service provider, named SOAP server. Everything is shown in the below situated Figure 4.

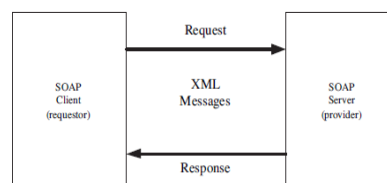


Figure 4. SOAP's basic request-response model

The protocol is used both to send and to receive messages from the Web Service. One advantage is to encapsulate the functionality of the RPC (Remote Procedure Call) using the extensibility and the functionality of the XML. SOAP defines a format for both messages, and a model for their processing by the receiver. In addition, SOAP – may also define a framework for protocol links, so that the SOAP messages can be transferred using the protocol stack from the transport level.

A SOAP message consists of a SOAP envelope [10], the root of the message, which in turn contains an optional header, and, necessarily, a body, independent of each other. SOAP message passes on its way from sender to receiver through many SOAP nodes, which can change the message. All the SOAP nodes form SOAP message path.

The Header contains general information about security – authentication and session, and about the message processing by the intermediary nodes. The data regarding the authentication usually is encrypted using WS-Security standard. The tag - "body" never misses from a SOAP message. Most of the times it is the last child of the "Envelope" node and it contains the information that is going to be transferred between applications (Web service input or output).

In Figure 5 it is shown an example of a SOAP message.

```
<?xml version="1.0" encoding="utf-8"?>
<soap:Envelope xmlns:xsi="http://www.w3.org/2001/XMLSchema-instance"
  xmlns:xsd="http://www.w3.org/2001/XMLSchema"
  xmlns:soap="http://www.w3.org/2003/05/soap-envelope">
  <soap:Body>
    <GetMaterialsResponse xmlns="http://tempuri.org/">
      <GetMaterialsResult>xmlxml</GetMaterialsResult>
    </GetMaterialsResponse>
  </soap:Body>
</soap:Envelope>
```

Figure 5. SOAP Message

In the above example, in the SOAP envelope the XML namespace and the type of the used message encoding are not specified. The Header node is missing in this example, and the body node contains the result of a "GetMaterials" method, which is a serialized Data Table type object in XML format.

SOAP was originally an acronym for Simple Object Access Protocol, But since SOAP Version 1.2 (SOAP 1.2 Part 0, 2003; SOAP 1.2 Part 1, 2003) it is technically no longer an acronym.

2.5. WSDL – Web Services Description Language

The description level, located above the packing level, is represented by the WSDL protocol, being based on the XML standard.

WSDL is a language written in XML [11], used as a model for describing Web services. WSDL reached version 2.0, but in version 1.1 the **D** stood for Definition. Version 1.2 of WSDL was renamed WSDL 2.0 because of the major differences between the two versions.

2.5.1. New features in WSDL 2.0

Nowadays, W3C recommends using WSDL 2.0 [11], but the problem is that it is not fully supported in all developing environments. The main differences between the two versions are:

- in WSDL 2.0 there's binding to all the HTTP request methods, whereas in WSDL 1.1 only the GET and POST methods;
- in WSDL 2.0 further semantics were added to the description language;
- WSDL 2.0 offers better support for RESTful web services;
- renaming of PortTypes (WSDL 1.1) into Interfaces (WSDL 2.0);
- renaming of Ports (WSDL 1.1) into Endpoints (WSDL 2.0);
- WSDL 2.0 can be implemented in a much simpler way.

WSDL is used in combination with SOAP and the XML schema representing the web service description. The main purpose of WSDL is that it leverages the connection between a client program and a web service, by determining the server available operations.

2.5.2. WSDL Components

Port/Endpoint – defines the address or connection to a web service; usually, it is represented by a simple URL.

Service – consists of a set of ports/endpoints, meaning the system functions exposed to the web based protocols.

Binding – defines a concrete message format and transmission protocol which may be used to define a port/endpoint.

PortType/Interface – defines a web service, all the operations that can be performed, and the messages used to perform the operation.

Operation – is an interaction with the service (a method) formed by a set of messages exchanged between the service and the other programs involved in the interaction.

Type – describes the data type definitions that are relevant for the exchanged messages.

Components 1-3 represent the concrete section of a WSDL, and components 4-6 represent the abstract section.

In Figure 6 it is shown an example of a WSDL message.

```
<definitions name="ServiceSAP" targetNamespace="http://192.168.1.12/ServiceSAP/ServiceSAP.asmx?WSDL">
  <wsdl:message name="GetMaterialsSoapIn">
    <wsdl:part name="parameters" element="tns:GetMaterials"/>
  </wsdl:message>
  <wsdl:message name="GetMaterialsSoapOut">
    <wsdl:part name="parameters" element="tns:GetMaterialsResponse"/>
  </wsdl:message>
  <wsdl:portType name="ServiceSAPSoap">
    <wsdl:operation name="GetMaterials">
      <wsdl:input message="tns:GetMaterialsSoapIn"/>
      <wsdl:output message="tns:GetMaterialsSoapOut"/>
    </wsdl:operation>
  </wsdl:portType>
  <wsdl:binding name="ServiceSAPSoap" type="tns:ServiceSAPSoap">
    <soap:binding transport="http://schemas.xmlsoap.org/soap/http"/>
    <wsdl:operation name="GetMaterials">
      <soap:operation soapAction="http://tempuri.org/GetMaterials" style="document"/>
      <wsdl:input>
        <soap:body use="literal"/>
      </wsdl:input>
      <wsdl:output>
        <soap:body use="literal"/>
      </wsdl:output>
    </wsdl:operation>
  </wsdl:binding>
  <wsdl:service name="ServiceSAP">
    <wsdl:port name="ServiceSAPSoap" binding="tns:ServiceSAPSoap">
      <soap:address location="http://192.168.1.12/ServiceSAP/ServiceSAP.asmx"/>
    </wsdl:port>
  </wsdl:service>
</definitions>
```

Figure 6. Example of a WSDL

2.6. UDDI – Universal Description, Discovery and Integration

UDDI represents a platform-independent framework [10], based on Extensible Markup Language (XML), a directory service where businesses can register and search for web services. UDDI is meant to be open for businesses, enabling them to publish and discover services, and to discover the interaction between them over the Internet.

2.6.1. UDDI – Characteristics of UDDI [8]

UDDI stores information about web services, it consists of web services interfaces written in WSDL. UDDI can be interrogated via SOAP messages and provides access to WSDL documents describing certain protocol bindings and message formats to interact with the web services. UDDI terminology contains also the following:

- Nodes – servers which support UDDI specifications nodes belong to a registry;
- Registries – collections of one or more UDDI nodes.

2.6.2. Benefits of UDDI

All businesses can benefit of UDDI because it solves the following problems:

- Discovering the right business from millions of online businesses;
- Once the preferred business is discovered, UDDI enables how to enable commerce;
- New customers can be reached and access to current customers can be increased;
- Market reach and offerings can be expanded;
- Barriers are removed to allow rapid participation in the global Internet economy;
- Services and business processes are programatically described in a single, open, and secure environment.

In Figure 7 it is shown an example of a UDDI message.

```
<discovery>
  <contractRef ref="http://192.168.1.12/ServiceSAP/ServiceSAP.asmx?wsdl" docRef="http://192.168.1.12/ServiceSAP/ServiceSAP.asmx"/>
  <soap address="http://192.168.1.12/ServiceSAP/ServiceSAP.asmx" binding="q1:ServiceSAPSoap"/>
  <soap address="http://192.168.1.12/ServiceSAP/ServiceSAP.asmx" binding="q2:ServiceSAPSoap12"/>
</discovery>
```

Figure 7. Example of a UDDI

Although the basic specifications of Web Services: XML, SOAP, WSDL and UDDI provide an acceptable level of interoperability and integrity [11] a significant effort has made to increase the applications area of Web Services, and to address to higher various issues from the real world. Thus new specifications emerged for Web Services' reliability, security, metadata management, transactions and orchestration, all of which have extended the Web Services' architecture. Among the new specifications it is worth to be mentioned:

- Metadata Management: WS-Addressing, WS-Policy, WS-MetadataExchange;
- Reliable Messaging: WS-Reliability, WS-ReliableMessaging, WS-Eventing, WS-Notifications;
- Security: WS-Authorization, WS-SecurityPolicy, WS-Trust, WS-SecureConversation, WS-Federation, WS-Privacy, XML Encryption, XML Signature;
- Transactions: WS-Transactions family(WS-AtomicTransaction, WS-BusinessActivity, WS-Coordination), WS-Composite Application Framework(WS-CAF);
- Orchestration
- Choreography

3. Case study regarding the integration of Web Services on different platforms

This section describes a case study on web services and associated key technologies, offering practical tests in order to support the previous presented. The application shows how to integrate Web Services on different platforms and how they allow the interoperability between applications running on these platforms, using the specific web services protocol stack presented in previous sections.

The problem we have modeled is as follows: "A company that sells used hardware in the fields of retail and food industry wants to integrate some mobile terminals, which are going to be used in order to scan the bar codes of the store's hardware equipments with an ERP system, in our case SAP."

Using a mobile terminal [12], we want to make different types of storage-specific operations, such as: reception, delivery and material's inventory.

The solution we present for integrating the mobile devices with SAP - consists in developing a web service on a .NET platform, that can be used as a proxy between the two platforms.

Figure 8 presents the data flow between these distributed systems, where the interoperability occurs based on the Web services.

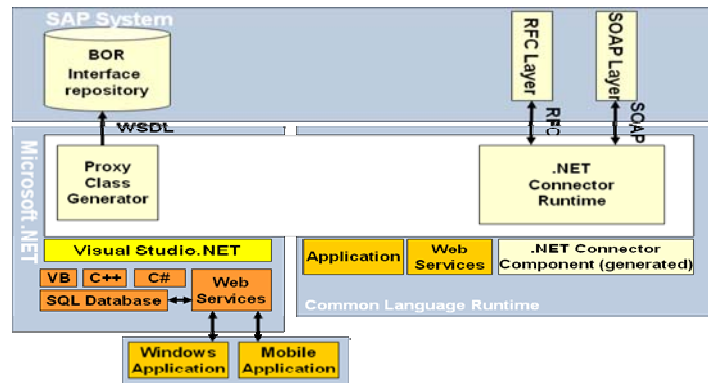


Figure 8. Interfacing .NET Applications with SAP via Web Services

Brief description of the scheme.

Integrated with SAP Business Object Repository (BOR) [13] is an object-oriented repository that contains SAP objects and SAP interfaces, and also their components, such as methods, attributes and events. In SAP Web Application Server, SOAP runtime provides us the mechanism of using SOAP protocol to call and access the RFC functions (Remote Function Call) [14] via HTTP. A web service in SAP can be seen as an RFC function.

The interoperability between the Microsoft .NET platform and the SAP system is done with the help of a component, the SAP Connector for Microsoft .NET [14]. It supports SAP Remote Function Call (RFC) and Web Services and it allows the development of various applications on the .Net platform. The Proxy class generation, which enables the calling for web service (RFC function) is based on WSDL. In this case, it can be said that SAP connector acts as a proxy between SAP and the .Net platform.

In order to integrate the mobile terminal with SAP, it has been developed a web service that becomes the wrapper class for the SAP connector and also it becomes a proxy between the application from the terminal and SAP.

In order to integrate the Web Service with both distributed systems some steps must be taken:

Stage 1: Developing and publishing the Web Services

Using the developing platform Visual Studio.NET, a series of appropriate methods for the data exchange is constructed. In this initial step the .Net connector is integrated to SAP. If the SAP authentication was successfully done, the description of the available RFC functions it is brought from SAP with the help of UDDI and WSDL. Since WSDL –is based on the XML standard the description of the functions is brought into a file with a .sapwsdl extension in XML format. Based on the WSDL and on the .NET connector, the proxy class, that contains the signature of the functions from the file, is being generated in order to become available for calling by other work methods.

In the example from below is shown a description of a RFC function from SAP on a proxy call basis, function that allows the over taking of the materials from the system:

```

/// <summary>
/// Remote Function Module Z_RFC_MATERIALE.
///
/// </summary>
/// <param name="Gv_Eroare"></param>
/// <param name="Gv_Mtart"></param>
/// <param name="Gt_Matnr"></param>
[RfcMethod(AbapName = "Z_RFC_MATERIALE")]
[SoapDocumentMethodAttribute("http://tempuri.org/Z_RFC_MATERIALE",
RequestNamespace = "urn:sap-com:document:sap:rfc:functions",

```



```

RequestElementName = "Z_RFC_MATERIALE",
ResponseNamespace = "urn:sap-com:document:sap:rfc:functions",
ResponseElementName = "Z_RFC_MATERIALE.Response"))]
public virtual void Z_Rfc_Materiale (
[RfcParameter(AbapName = "GV_MTART",RfcType=RFCTYPE.RFCTYPE_CHAR, Optional =
false, Direction = RFCINOUT.IN, Length = 4, Length2 = 8)]
[XmlElement("GV_MTART", IsNullable=false, Form=XmlSchemaForm.Unqualified)]
string Gv_Mtart,
[RfcParameter(AbapName = "GV_EROARE",RfcType=RFCTYPE.RFCTYPE_CHAR, Optional =
true, Direction = RFCINOUT.OUT, Length = 80, Length2 = 160)]
[XmlElement("GV_EROARE", IsNullable=false, Form=XmlSchemaForm.Unqualified)]
out string Gv_Eroare,
[RfcParameter(AbapName = "GT_MATNR",RfcType=RFCTYPE.RFCTYPE_ITAB, Optional = true,
Direction = RFCINOUT.INOUT)]
[XmlArray("GT_MATNR", IsNullable=false, Form=XmlSchemaForm.Unqualified)]
[XmlElement("item", IsNullable=false, Form=XmlSchemaForm.Unqualified)]
ref ZMATNRTable Gt_Matnr)
{
object[] results = null;
results = this.SAPInvoke("Z_Rfc_Materiale",new object[] {Gv_Mtart,Gt_Matnr });
Gv_Eroare = (string) results[0];
Gt_Matnr = (ZMATNRTable) results[1];
}

```

In the following example it is shown a method of web service "GetMaterials" used to call RFC function rendered above:

```

[WebMethod]
public DataTable GetMaterials()
{
try
{
string error = "";
SAPProxy proxy = new SAPProxy(this.BuildConnectionString());
ZMATNRTable materialsERSA = new ZMATNRTable();
proxy.Z_Rfc_Materiale("ERSA", out error, ref materialsERSA);
if (error != String.Empty)
throw new Exception(error);
Access access = new
Access(ConfigurationManager.ConnectionStrings["ConnectionString"].ConnectionString);
for (int i = 0; i < materialsERSA.Count; i++)
access.SynchronizeMaterials(materialsERSA[i], "ERSA");
return materialsERSA.ToADODDataTable();
}
catch (Exception exc)
{
throw (exc);
}
}

```

Stage 2: Rendering the Web Service

At this stage, after the construction of the web service, it will be rendered in the application from the mobile terminal which is developed on the .Net platform. If the service is available, similar to Stage 1, the description of the service in the application is brought with the help of UDDI and WSDL.

Stage 3: Calling methods of service

After rendering the contract and after the generation of the proxy class to call the service, in the mobile device application one can call the service methods.

Exchanging data between applications in the background, in the form of SOAP messages is done asynchronously and at runtime. If there is a large volume of data, due to the interoperability based on XML standard, the process may be slowed because of the need for parsing of the XML messages in different types of objects. This exchange of messages is done by following these steps:

Step 1: Request the mobile application service

While calling a method from the proxy class, on a SOAP framework basis, a request is developed, in XML format. This message is attached to the SOAP - and will on a HTTP protocol basis to the service. If the service is not available and if there are no different restrictions on the network, next will be followed step 2, otherwise the application ends.

Step 2: Request service from SAP

The request was received by the service and it will be forwarded to SAP through connector and the proxy class. Based on the SOAP Runtime protocol, which provides access and the calling to the RFC functions, all the messages are sent in XML format via HTTP, and they are parsed as objects and BOR specific structures.

Step 3: The SAP's answer to the service

The result of the request from Step 2 is sent in XML format by the SOAP Runtime and it is parsed by the .NET connector into the service's specific objects.

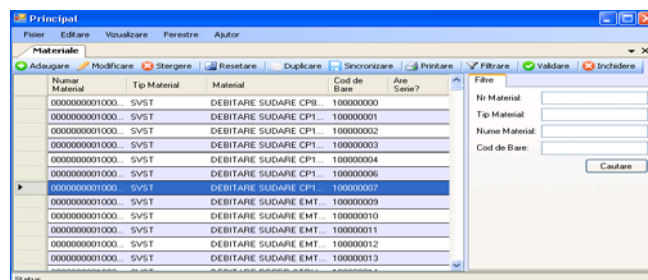


Figure 9. Simple call web service from Windows application

Step 4: The answer of the service to the mobile application

The result from Step 3, as a response from service to customer, it is sent like in Step 1. The XML message is transformed by SOAP's framework into specific objects from the development environment, in our case a DataTable object type which can be seen in Figure 9.

4. Conclusions

Web services play a similar role with older technologies such as Remote Procedure Call (RPC), Common Object Request Broker Architecture (CORBA), Distributed Component Object Model (DCOM), but also offers several advantages over these. The great advantage of Web Services is that they can integrate different platforms and allow the interoperability between different distributed systems. The Integrity and the interoperability can be addressed with the SOA's help in a two-stage process that involves publishing and orchestrating the web services. The Independence with respect to various platforms is due to common standard that they all have as a basis, the XML. Using Web services provides to the developers the opportunity to create high quality applications more quickly.

In this paper, the authors came up with a pattern of using and integrating the Web services, proofing the interoperability between multiple distributed applications running on different platforms.

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A STUDY ON WEIBULL DISTRIBUTION FOR ESTIMATING THE PARAMETERS

Paritosh BHATTACHARYA

CEM, Kolaghat, Midnapore, India

E-mail: p_bhattacharya2001@yahoo.co.in

Rakhi BHATTACHARJEE

BITM, Shantiniketan, India

Abstract: The wind resource varies with of the day and the season of the year and even some extent from year to year. Wind energy has inherent variances and hence it has been expressed by distribution functions. In this paper, we present some methods for estimating Weibull parameters, namely, shape parameter (k) and scale parameter (c). The Weibull distribution is an important distribution especially for reliability and maintainability analysis. The suitable values for both shape parameter and scale parameters of Weibull distribution are important for selecting locations of installing wind turbine generators. The scale parameter of Weibull distribution also important to determine whether a wind farm is good or not. The presented method is the analytical methods and computational experiments on the presented methods are reported.

Key words: Wind Speed; Probability Distribution; Weibull Distribution; Linear Least Square Method

1. Introduction

Today, most electrical energy is generated by burning huge fossil fuels and special weather conditions such as acid rain and snow, climate change, urban smog, regional haze, several tornados, etc., have happened around the whole world. It is now clear that the installation of a number of wind turbine generators can effectively reduce environmental pollution, fossil fuel consumption, and the costs of overall electricity generation. Although wind is only an intermittent source of energy, it represents a reliable energy resource from a long-term energy policy viewpoint. Among various renewable energy resources, wind power energy is one of the most popular and promising energy resources in the whole world today. At a specific wind farm, the available electricity generated by a wind power generation system depends on mean wind speed (MWS), standard deviation of wind speed, and the location of installation. Since year-to-year variation on annual MWS is hard to predict, wind speed variations during a year can be well characterized in terms of a probability distribution function (pdf). This paper also addresses the relations among MWS, its standard deviation, and two important parameters of Weibull distribution.

2. Weibull distribution

The Weibull distribution is characterized by two parameters, one is the shape parameter k (dimensionless) and the other is the scale parameter c (m/s)

The cumulative distribution function is given by

$$F(v) = 1 - \exp\left[-\left(\frac{v}{c}\right)^k\right] \dots\dots\dots(1)$$

And the probability function is given by

$$f(v) = \frac{dF(v)}{dv} = \frac{k}{c} \left(\frac{v}{c}\right)^{k-1} \exp\left[-\left(\frac{v}{c}\right)^k\right] \dots\dots\dots(2)$$

The average wind speed can be expressed as

$$\bar{v} = \int_0^{\infty} v f(v) dv = \int_0^{\infty} \frac{vk}{c} \left(\frac{v}{c}\right)^{k-1} \exp\left[-\left(\frac{v}{c}\right)^k\right] dv \dots\dots\dots(3)$$

$$\text{Let } x = \left(\frac{v}{c}\right)^k, \quad x^{\frac{1}{k}} = \frac{v}{c} \quad \text{and} \quad dx = \frac{k}{c} \left(\frac{v}{c}\right)^{k-1} dv$$

Equation (3) can be simplified as

$$\bar{v} = c \int_0^{\infty} x^{\frac{1}{k}} \exp(-x) dx \dots\dots\dots(4)$$

By substituting a Gamma Function

$$\Gamma(n) = \int_0^{\infty} e^{-x} x^{n-1} dx \quad \text{into (4) and let } y = 1 + \frac{1}{k} \quad \text{then we have}$$

$$\bar{v} = c \Gamma\left(1 + \frac{1}{k}\right) \dots\dots\dots(5)$$

$$\text{The standard deviation of wind speed } v \text{ is given by } \sigma = \sqrt{\int_0^{\infty} (v - \bar{v})^2 f(v) dv} \dots\dots\dots(6)$$

i.e.

$$\begin{aligned} \sigma &= \sqrt{\int_0^{\infty} (v^2 - 2v\bar{v} + \bar{v}^2) f(v) dv} \\ &= \sqrt{\int_0^{\infty} v^2 f(v) dv - 2\bar{v} \int_0^{\infty} v f(v) dv + \bar{v}^2} \\ &= \sqrt{\int_0^{\infty} v^2 f(v) dv - 2\bar{v} \cdot \bar{v} + \bar{v}^2} \dots\dots\dots(7) \end{aligned}$$

Use

$$\int_0^{\infty} v^2 f(v) dv = \int_0^{\infty} v^2 \frac{k}{c} \left(\frac{v}{c}\right)^{k-1} dv = \int_0^{\infty} c^2 x^{\frac{2}{k}} \frac{k}{c} \left(\frac{v}{c}\right)^{k-1} dv = \int_0^{\infty} c^2 x^{\frac{2}{k}} \exp(-x) dx \dots\dots\dots(8)$$

And put $y = 1 + \frac{2}{k}$, then the following equation can be obtained

$$\int_0^{\infty} v^2 f(v) dv = c^2 \Gamma\left(1 + \frac{2}{k}\right) \dots\dots\dots(9)$$

Hence we get

$$\sigma = \left[c^2 \Gamma \left(1 + \frac{2}{k} \right) - c^2 \Gamma^2 \left(1 + \frac{1}{k} \right) \right]^{\frac{1}{2}}$$

$$= c \sqrt{\Gamma \left(1 + \frac{2}{k} \right) - \Gamma^2 \left(1 + \frac{1}{k} \right)} \dots \dots \dots (10)$$

$$\text{i.e. } \frac{1}{1-F(v)} = \exp \left[\left(\frac{v}{c} \right)^k \right] \dots \dots \dots (13)$$

$$\text{i.e. } \ln \left\{ \frac{1}{1-F(v)} \right\} = \left[\left(\frac{v}{c} \right)^k \right] \dots \dots \dots (14)$$

But the cumulative Weibull distribution function is transformed to a linear function like below:

$$\text{Again } \ln \ln \left\{ \frac{1}{1-F(v)} \right\} = k \ln v - k \ln c \dots \dots \dots (15)$$

Equation (14) can be written as $Y = bX + a$

where $Y = \ln \ln \left\{ \frac{1}{1-F(v)} \right\}$, $X = \ln v$, $a = -k \ln c$, $b = k$

By Linear regression formula

Linear Least Square Method (LLSM)

Least square method is used to calculate the parameter(s) in a formula when modeling an experiment of a phenomenon and it can give an estimation of the parameters. When using least square method, the sum of the squares of the deviations S which is defined as below, should be minimized.

$$S = \sum_{i=1}^n w_i^2 [y_i - g(x_i)]^2 \dots \dots \dots (11)$$

In the equation, x_i is the wind speed, y_i is the probability of the wind speed rank, so (x_i, y_i) mean the data plot, w_i is a weight value of the plot and n is a number of the data plot.

The estimation technique we shall discuss is known as the Linear Least Square Method (LLSM). It is so commonly applied in engineering and mathematics problem that is often not thought of as an estimation problem. The linear least square method (LLSM) is a special case for the least square method with a formula which consists of some linear functions and it is easy to use. And in the more special case that the formula is line, the linear least square method is much easier. The Weibull distribution function is a non-linear function, which is

$$F(v) = 1 - \exp \left[- \left(\frac{v}{c} \right)^k \right] \dots \dots \dots (12)$$

$$b = \frac{N \sum_{i=1}^n X_i Y_i - \sum_{i=1}^n X_i \sum_{i=1}^n Y_i}{N \sum_{i=1}^n X_i^2 - (\sum_{i=1}^n X_i)^2} \dots\dots\dots (16)$$

$$a = \frac{\sum_{i=1}^n X_i^2 \sum_{i=1}^n Y_i - \sum_{i=1}^n X_i \sum_{i=1}^n X_i Y_i}{N \sum_{i=1}^n X_i^2 - (\sum_{i=1}^n X_i)^2} \dots\dots\dots (17)$$

4. Maximum Likelihood Estimator(MLE)

The method of maximum likelihood (Harter and Moore (1965a), Harter and Moore (1965b), and Cohen (1965)) is a commonly used procedure because it has very desirable properties.

Let x_1, x_2, \dots, x_n be a random sample of size n drawn from a probability density function $f(x, \theta)$ where θ is an unknown parameter. The likelihood function of this random sample is the joint density of the n random variables and is a function of the unknown parameter. Thus

$$L = \prod_{i=1}^n f_{X_i}(x_i, \theta) \dots\dots\dots (18)$$

is the Likelihood function. The Maximum Likelihood Estimator (MLE) of θ , say $\bar{\theta}$, is the value of θ , that maximizes L or, equivalently, the logarithm of L . Often, but not always, the MLE of θ is a solution of

$$\frac{d \log L}{d \theta} = 0 \dots\dots\dots (19)$$

Now, we apply the MLE to estimate the Weibull parameters, namely the shape parameter and the scale parameters. Consider the Weibull probability density function (pdf) given in (2), then likelihood function will be

$$L(x_1, x_2, \dots, x_n, k, c) = \prod_{i=1}^n \left(\frac{k}{c} \right) \left(\frac{x_i}{c} \right)^{k-1} e^{-\left(\frac{x_i}{c} \right)^k} \dots\dots\dots (20)$$

On taking the logarithms of (20), differentiating with respect to k and c in turn and equating to zero, we obtain the estimating equations

$$\frac{\partial \ln L}{\partial k} = \frac{n}{k} + \sum_{i=1}^n \ln x_i - \frac{1}{c} \sum_{i=1}^n x_i^k \ln x_i = 0 \dots\dots\dots (21)$$

$$\frac{\partial \ln L}{\partial c} = \frac{-n}{c} + \frac{1}{c^2} \sum_{i=1}^n x_i^k = 0 \dots\dots\dots (22)$$

On eliminating c between these two above equations and simplifying, we get

$$\frac{\sum_{i=1}^n x_i^k \ln x_i}{\sum_{i=1}^n x_i^k} - \frac{1}{k} - \frac{1}{n} \sum_{i=1}^n \ln x_i = 0 \dots\dots\dots (23)$$

which may be solved to get the estimate of k . This can be accomplished by Newton-Raphson method. Which can be written in the form

$$x_{n+1} = x_n - \frac{f(x_n)}{f'(x_n)} \dots\dots\dots (24)$$

Where

$$f(k) = \frac{\sum_{i=1}^n x_i^k \ln x_i}{\sum_{i=1}^n x_i^k} - \frac{1}{k} - \frac{1}{n} \sum_{i=1}^n \ln x_i \dots\dots\dots (25)$$

And

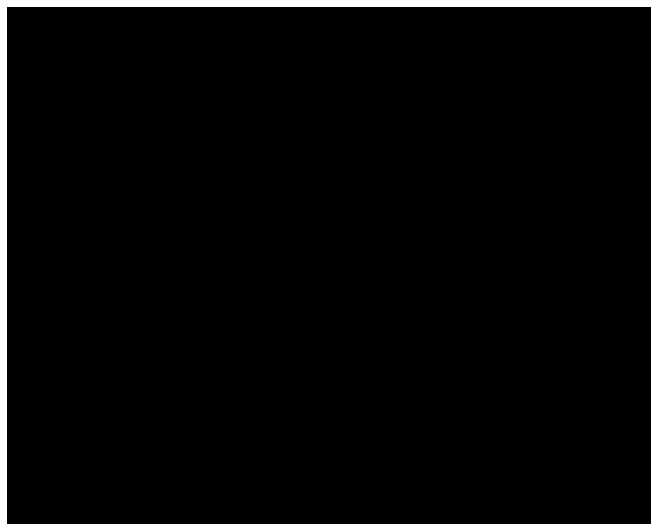
$$f'(k) = \sum_{i=1}^n x_i^k (\ln x_i)^2 - \frac{1}{k^2} \sum_{i=1}^n x_i^k (k \ln x_i - 1) - \left(\frac{1}{n} \sum_{i=1}^n \ln x_i \right) \left(\sum_{i=1}^n x_i^k \ln x_i \right) \dots\dots (26)$$

Once k is determined, c can be estimated using equation (22) as

$$c = \frac{\sum_{i=1}^n x_i^k}{n} \dots\dots\dots (27)$$

5. Results and Discussions

When a location has $c=6$ the pdf under various values of k are shown in Fig. 1. A higher value of k such as 2.5 or 4 indicates that the variation of Mean Wind speed is small. A lower value of k such as 1.5 or 2 indicates a greater deviation away from Mean Wind speed.



When a location has $k=3$ the pdf under various values of c are shown in Fig.2. A higher value of c such as 12 indicates a greater deviation away from Mean Wind speed.

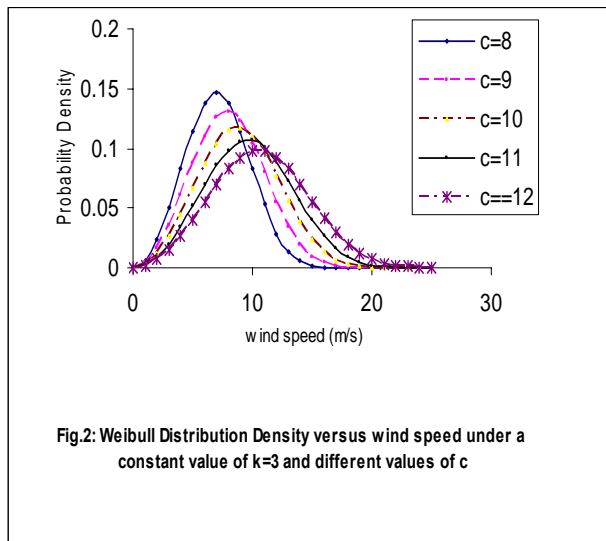


Fig. 3 represents the characteristic curve of $\Gamma\left(1 + \frac{1}{k}\right)$ versus shape parameter k .

The values of $\Gamma\left(1 + \frac{1}{k}\right)$ varies around .889 when k is between 1.9 to 2.6.

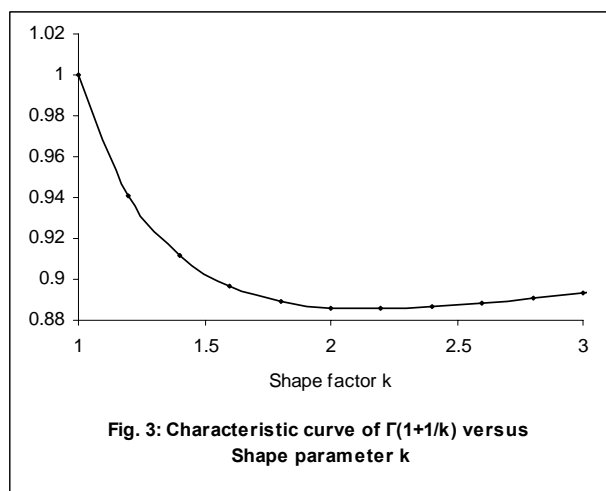
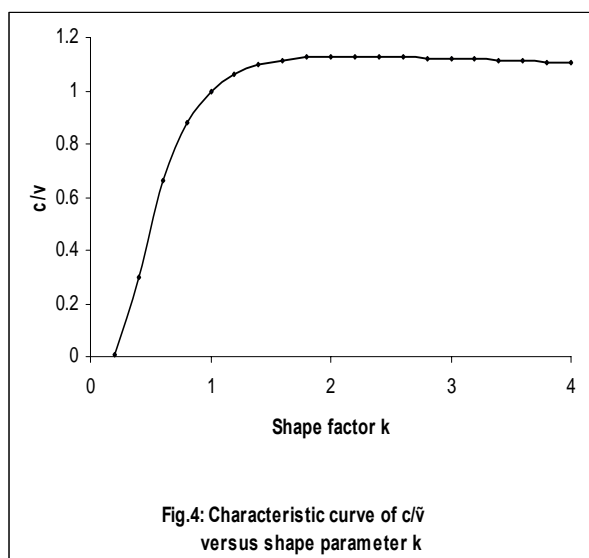


Fig.4 represents the characteristic curve of $\frac{c}{\bar{v}}$ versus shape parameter k . Normally the wind speed data collected at a specified location are used to calculate Mean Wind speed. A good estimate for parameter c can be obtained from Fig.4 as $c = 1.128\bar{v}$ where k ranges from 1.6 to 4. If the parameter k is less than unity, the ratio $\frac{c}{\bar{v}}$ decrease rapidly.

Hence c is directly proportional to Mean Wind speed for $1.6 \leq k \leq 4$ and Mean Wind speed is mainly affected by c . The most good wind farms have k in this specified range and estimation of c in terms of \bar{v} may have wide applications.



Example: Consider the following example where x_i represents the Average Monthly Wind Speed (m/s) at kolkata (from 1st March, 2009 to 31st March, 2009)

March2009	Wind Speed (m/s)	March, 2009	Wind Speed (m/s)
1	0.56	17	0.28
2	0.28	18	0.83
3	0.56	19	1.39
4	0.56	20	1.11
5	1.11	21	1.11
6	0.83	22	0.83
7	1.11	23	0.56
8	1.94	24	0.83
9	1.11	25	1.67
10	0.83	26	1.94
11	1.11	27	1.39
12	1.39	28	0.83
13	0.28	29	2.22
14	0.56	30	1.67
15	0.28	31	2.22
16	0.28		

Also let $F(x_i) = \frac{i}{n+1}$ and using equations (16) and (17) we get $k = 1.013658$

and $c = 29.9931$

But if we apply maximum Likelihood Method we get $k = 1.912128$ and $c = 1.335916$. There is a huge difference in value of c by the above two methods. This is due to the mean rank of $F(x_i)$ and k value is tends to unity.

6. Conclusions

In this paper, we have presented two analytical methods for estimating the Weibull distribution parameters. The above results will help the scientists and the technocrats to select the location for Wind Turbine Generators.

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MINIMIZATION, CONSTRAINTS AND COMPOSITE BÉZIER SURFACES¹

Arie JACOBI²

Department of Software Engineering,
Sami Shamoon Academic College of Engineering,
Beer Sheva, Israel

E-mail: ariej@mscc.huji.ac.il



Shilo LIFSCHUTZ³

Department of Accounting,
Academic Center of Law and Business,
Ramat-Gan, Israel

E-mail: shilo@clb.ac.il



Abstract: This paper presents a global method for approximation and/or construction of surfaces using constraints. The method is based on a min max problem which describes approximation and differential geometric characteristics, constrained in order to achieve desired geometrical effects. The numerical solution of the problem takes full advantage of the Finite-Elements method and of constrained optimization algorithms.

Key words: Bézier surface; Offset surface; Surface Fitting; Offset Surface; Approximate Conversion; surface approximation; Surface simplification; Variational Problem Formulation; Finite-Element Method (FEM); FEM's hp-method; Lagrange Multipliers Formulation

1. Introduction

Approximation of surfaces and construction of offset surfaces has a variety of applications. For example:

- Approximation to a set of scattered points in three-dimensional space originated from scientific experiments, earth terrain description, or data from satellites
- Exchanging format of formal data. It is required in geometric modeling systems for free form surfaces, as they use different mathematical representations and different polynomial bases for curves and surface representation
- Conversion between non-polynomial representations (such as rational surfaces) to polynomial ones.

Additional motivations for approximation are the ability of merging curves and surfaces in order to reduce information or the construction of offset curves and surfaces which are needed in tool paths planning for numerical control machines and in construction of a thick surface that is used as the outer (or inner) surface of objects such as, a car, an airplane, or a mold.

Finite element methods are an essential tool for the approximation of a solution of a partial differential equation and are based on the weak variational formulation of boundary and initial value problems. The importance of this property is twofold:

- It provides the proper setting for the existence of a very irregular solution to differential equations.
- The solution appears in the integral of a quantity over a domain, which can be broken up into the sum of integrals over an arbitrary collection of almost disjoint sub-domains whose union is the original domain.

These properties allow analysis to be done locally on a typical sufficiently small sub-domain, so that polynomial functions of various degrees are adequate for representing the local behavior of the solution (see [27]). In order to arrive at a global approximation of a solution of a partial differential equation in the finite element method, their contributions of local approximation over individual elements are assembled together in a systematic way. This leads to schemes which are robust in appropriate norms and insensitive to distortions and singularities of the mesh.

Desirable Properties of an Approximation Surface

We consider the following properties for curve approximation and construction:

- *End Points Interpolation* - The approximation surface's end points should interpolate the approximated surface's end points.
- *End Directions Preservation* - The approximation surface's boundary curves' end tangents should have the same direction as the approximated surface's boundary curves' end tangents.
- *Parametric and Geometric Continuity* - Creation of a smooth approximation surfaces. There are cases where a higher parametric continuity degree (C^1 or C^2) between the approximation surface patches is needed. It is also possible to ensure geometric continuity of first degree (GC^1) between the approximation surface patches.

2. Previous Work

There are several approaches for approximation of curves, surfaces, or points in three-dimensional space. Among the early important works in this field for the approximate conversion of curves and surfaces as well as the construction of offset curves and offset surfaces, we would like to signify the works of [11,26] and [16]-[24]. Among the recent works there is the work of Weiss et al. [30], that attempts to provide practical solutions to overcome problems of irregular distribution of data points which are over topologically irregular domains. The Weiss et al. method includes algorithms to compute a good initial parametrization, a procedure for handling weakly defined control points, a shape dependent knot refinement, and a fitting strategy to maintain tight tolerances and smoothness simultaneously. Their method achieves a high accuracy relative to the published 'standard' solutions.

Borges and Pastva [9] deal with the problem of fitting a single Bézier curve segment to a set of ordered data so that the error is minimized in the total least squares sense. They developed an algorithm for applying the Gauss-Newton method to this problem with a direct method for evaluating the Jacobian based on implicitly differentiating a pseudo-inverse. Chen Guo-Dong and Wang Guo-Jin [10] consider simultaneous fitting of multiple curves and surfaces to 3D measured data captured as part of a reverse engineering process, where constraints exist between the parameters of the curves or surfaces. Enforcing such constraints may be necessary

- to produce models of sufficiently accurate tolerances for import into a CAD system, and

- to produce models which successfully reproduce regularities and symmetries required by engineering applications.

There are several works in the CAGD field that use Bernstein-Bézier finite elements in the context of approximation. One of the earliest works on actual approximate conversion is Bercovier and Jacobi [3, 4] and Luscher [25]. Examples of later works that use FEM in CAGD are hierarchical methods for linear spline approximation and construction of surface triangulations or quadrangulations by adaptively subdividing a surface to a form of tree. It is used in an approximation to a set of scattered points in three-dimensional space using hierarchical spline and surface approximation methods such as in [8], or for approximation over irregular domain as introduced in [5, 6, 7, 31]. An implementation of cubic tetrahedral Bernstein-Bézier finite elements and their application in the context of facial surgery simulation is presented in [28]

3. Our Approach and its Strategy

Our approach uses the combination of the finite element method with the Bernstein-Bézier representation, introducing a valuable finite element due to the many advantages of the Bernstein-Bézier shape functions [14]. It introduces the construction of surfaces given by piecewise definitions of their parameter range and allows surface editing, both in the direction of several (lower order) surfaces approximating a single given one, or, conversely replacing several by a single one. This approach exploits the *p-method* and the *h-method* in FEM in order to improve approximation. This is done by using elements of higher degrees to overcome areas that are difficult to approximate and by using elements with lower degrees for approximating the rest of surface for implementing the *p-method*. For the implementation the *h-method* we use element mesh which is refined to increase accuracy. In order to achieve desired geometrical properties for the approximation surface, we incorporate parametric (C^0 - C^2) and geometric (GC^1) continuity constraints between the approximation surface patches, and other constraints, such as a constraint for the interpolation of the approximation surface end points with the end points of the given surface, or, a constraint to enforce the directions of the end edges of each boundary curve control polygon of the approximation surface to have the same directions as the tangents at the end points of the approximated surface boundary curves.

The strategy we use (see [3, 4]) is a global and continuous method for the approximation and construction of parametric surfaces. The method is based on a variational formulation which includes geometrical relations between surfaces and constraints upon the geometry and/or parameterization of the approximation surface. The variational formulation is based on the squared integrals of the zeroth, first, and second derivative (semi) norms of the approximation and approximated surfaces. We introduce the Lagrangian multiplier formulation for the constraints implementation. A weighting factor is related to each derivative (semi) norm. These weighting factors allow one to control the approximation of the related norm. The solution of this constrained variational problem is done by the Finite Element Method (FEM) over Bernstein basis functions.

Outline of this Paper

Description of the unconstrained problem, the constrained problem and its solution, is presented in sections 2-4. In section 5, a survey of the methods for estimating the approximation errors is given. Section 6, introduces the constraints we use to improve the approximation. A number of examples involving constraints, for surface degree reduction, surface merging and construction of offset surfaces are shown in section 7.

4. The Problem

Problem Statement

We will first define the problem without constraints. Given a parametric surface:
 $\mathbf{f}(u, v) = \mathbf{f}(f_1(u, v), f_2(u, v), f_3(u, v)), \quad u \in [a, b], \quad v \in [c, d],$

find the unknown vector function

$\mathbf{x}(u, v) = \mathbf{x}(x_1(u, v), x_2(u, v), x_3(u, v)), \quad u \in [a, b], \quad v \in [c, d],$

which is the solution by minimization, of one of the following three related problems we consider in this article:

$$\text{problem 1: } J^0(\mathbf{x}) = E(\mathbf{x}), \quad (1)$$

$$\text{problem 2: } J^1(\mathbf{x}) = E(\mathbf{x}) + \bar{E}(\mathbf{x}) \quad \text{and} \quad (2)$$

$$\text{problem 3: } J^2(\mathbf{x}) = E(\mathbf{x}) + \bar{E}(\mathbf{x}) + \hat{E}(\mathbf{x}), \quad (3)$$

where

$$E(x) = \alpha \iint_{\Omega} (\mathbf{x}(u, v) - \mathbf{f}(u, v))^2 \, dudv, \quad (4)$$

$$\bar{E}(x) = \beta \iint_{\Omega} \left(\frac{\partial}{\partial u} \mathbf{x}(u, v) - \frac{\partial}{\partial u} \mathbf{f}(u, v) \right)^2 + \left(\frac{\partial}{\partial v} \mathbf{x}(u, v) - \frac{\partial}{\partial v} \mathbf{f}(u, v) \right)^2 \, dudv, \quad (5)$$

and

$$\begin{aligned} \hat{E}(x) = \gamma \iint_{\Omega} & \left(\frac{\partial^2}{\partial u^2} \mathbf{x}(u, v) - \frac{\partial^2}{\partial u^2} \mathbf{f}(u, v) \right)^2 + \left(\frac{\partial^2}{\partial v^2} \mathbf{x}(u, v) - \frac{\partial^2}{\partial v^2} \mathbf{f}(u, v) \right)^2 + \\ & \left(\frac{\partial^2}{\partial u \partial v} \mathbf{x}(u, v) - \frac{\partial^2}{\partial u \partial v} \mathbf{f}(u, v) \right)^2 \, dudv, \end{aligned} \quad (6)$$

are the zeroth, first, and second error (semi) norms, respectively, and α , β , and γ positive moduli which are used as weighting factors.

Solution for the Problem Using the FEM Technique

In the following section we present the solution of the problem stated in section 2.1 using the FEM technique. The solution process includes: the partition of the problem's domain into two-dimensional elements, the calculation of a stiffness matrix \mathbf{M}_e and load vector \mathbf{m}_e for a given element e , the assembly of the elements' stiffness matrices and load vectors into the main stiffness matrix and load vector, and the calculation of the approximation error. The solution to one of the problems (1-3), follows the Galerkin-Ritz solution scheme (a computational example for the Rayleigh-Ritz and Galerkin methods, using the strong form of Poisson's equation can be seen in [29]).

The Approximation FEM Space

Given the partition:

$$a = u_0 < u_1 < u_2 \cdots < u_m = b$$

$$c = v_0 < v_1 < v_2 \cdots < v_n = d, \quad (7)$$

of the rectangular range

$$\bar{\Omega} = [a \leq u \leq b ; \quad c \leq v \leq d], \quad (8)$$

each sub-range

$$\Delta_e = [u_i \leq u \leq u_{i+1} ; \quad v_j \leq v \leq v_{j+1}] \quad (9)$$

for $i = 0, \dots, m-1$, $j = 0, \dots, n-1$ is the global parameter range of an element e , where $e = 0, \dots, L-1$ and L is the number of elements.

We use the following linear transformation to establish the relation between the global parameters $u, v \in \Delta_e$ and the local parameters $r, s \in [0, 1]$:

$$\begin{bmatrix} u \\ v \end{bmatrix} = \begin{bmatrix} u_i \\ v_j \end{bmatrix} + ra + sb, \quad (10)$$

where

$$\mathbf{a} = \begin{bmatrix} u_{i+1} \\ v_j \end{bmatrix} - \begin{bmatrix} u_i \\ v_j \end{bmatrix}, \quad \text{and} \quad \mathbf{b} = \begin{bmatrix} u_i \\ v_{j+1} \end{bmatrix} - \begin{bmatrix} u_i \\ v_j \end{bmatrix}. \quad (11)$$

We introduce an $n \times m$ dimensional approximation space $V^{l,m,n}$, consisting of functions which are piecewise C^0 Bézier patches over the range $\bar{\Omega}$. Let

$$V^{l,m,n} = \{ \mathbf{x}(u, v) : \begin{aligned} \mathbf{x}(a, c) &= \mathbf{f}(a, c), \quad \mathbf{x}(b, c) = \mathbf{f}(b, c), \\ \mathbf{x}(a, d) &= \mathbf{f}(a, d), \quad \mathbf{x}(b, d) = \mathbf{f}(b, d) \end{aligned} \}$$

such that there are

$$p \leq m \text{ and } q \leq n \text{ with}$$

$$\mathbf{x}(u, v) |_{\Delta_e} = \mathbf{S}_{p,q}(r(u), s(v))$$

a Bézier patch, for all $0 \leq e \leq L - 1$

$r(u), s(v)$ derived from equation (??),

$$\mathbf{x}(u, v) \in C^{l-1}([a, b]) \quad l = 0, 1, 2, 3, \text{ and}$$

C^{-1} is the space of functions

with discontinuities at u_e , or v_e only}. (12)

We define

$$V^l \equiv \bigcup_{m,n=1}^{\infty} V^{l,m,n}, \quad (13)$$

to be the minimization space, where $V^{l,m,n}$ is the finite-dimensional subspace (12) of V^l .

Problem Description for a given element e

Using partition (7) and given element e , where $e = 0, \dots, L - 1$ and L is the number of elements, we set:

$$E_e(\mathbf{x}) = \alpha \iint_{\Delta_e} (\mathbf{x}(u, v) - \mathbf{f}(u, v))^2 du dv, \quad (14)$$

$$\bar{E}_e(\mathbf{x}) = \beta \iint_{\Delta_e} \left(\frac{\partial}{\partial u} \mathbf{x}(u, v) - \frac{\partial}{\partial u} \mathbf{f}(u, v) \right)^2 + \left(\frac{\partial}{\partial v} \mathbf{x}(u, v) - \frac{\partial}{\partial v} \mathbf{f}(u, v) \right)^2 du dv, \quad (15)$$

and

$$\begin{aligned} \hat{E}_e(\mathbf{x}) = \gamma \iint_{\Delta_e} & \left(\frac{\partial^2}{\partial u^2} \mathbf{x}(u, v) - \frac{\partial^2}{\partial u^2} \mathbf{f}(u, v) \right)^2 + \left(\frac{\partial^2}{\partial v^2} \mathbf{x}(u, v) - \frac{\partial^2}{\partial v^2} \mathbf{f}(u, v) \right)^2 + \\ & \left(\frac{\partial^2}{\partial u \partial v} \mathbf{x}(u, v) - \frac{\partial^2}{\partial u \partial v} \mathbf{f}(u, v) \right)^2 du dv, \end{aligned} \quad (16)$$

to be the zeroth, first, and second error (semi) norms for the element e , and for the element sub-range Δ_e (as in (9)), respectively.

Let

$$J^l(\mathbf{x}(u, v)) = \sum_{e=0}^{L-1} J_e^l(\mathbf{x}(u, v)), \quad (17)$$

where

$$J_e^l(\mathbf{x}(u, v)) = J^l(\mathbf{x}(u, v))|_{u,v \in \Delta_e}, \quad (18)$$

or,

$$J_e^l(\mathbf{x}(u, v)) = \left\{ \frac{1}{2} \mathbf{b}_e^T \mathbf{M}_e \mathbf{b}_e - \mathbf{m}_e^T \mathbf{b}_e \right\} \quad \text{for } 0 \leq e \leq L-1, \quad (19)$$

where for the e -th element,

$$\mathbf{b}_e \equiv [\mathbf{b}_{e0,0}, \dots, \mathbf{b}_{e,m,n}]^T,$$

is the vector of the unknown Bézier points, \mathbf{m}_e is the **element load vector** and \mathbf{M}_e is the **element stiffness matrix**.

Our objective is to find for all m, n the function $\mathbf{x}(u, v)$ which is taken over the space $V^{l,m,n}$, in order to approximate $\mathbf{f}(u, v)$ in some sense to be defined later.

General Solution of the Problem

After integration element by element, we obtain:

$$\sum_{e=0}^{L-1} J_e^l(\mathbf{x}_e) = \left\{ \frac{1}{2} \mathbf{b}^T \mathbf{M}^l \mathbf{b} - \mathbf{m}^{lT} \mathbf{b} \right\} \quad \text{for } l=0,1,2, \quad (20)$$

or

$$J^l(\mathbf{x}) = \left\{ \frac{1}{2} \mathbf{b}^T \mathbf{M} \mathbf{b} - \mathbf{m}^T \mathbf{b} \right\} \quad \text{for } l=0,1,2. \quad (21)$$

The elements' stiffness matrices \mathbf{M}_e^l , and load vectors \mathbf{m}_e^l ($e = 0, \dots, L-1$), are assembled into the global stiffness matrix \mathbf{M} and load vector \mathbf{m} .

The minimum of each $J^l(\mathbf{x})$ in (21), is given by the approximation surface $\mathbf{x}(u, v)$, where \mathbf{b} is the solution of the system

$$\nabla_{\mathbf{b}} J^l(\mathbf{x}) = \mathbf{M} \mathbf{b} + \mathbf{m} = \vec{0}. \quad (22)$$

The system (22) is linear symmetric positive definite, and we use the LDL^T algorithm or the Conjugate-Gradient [32] to solve it. The coordinate components of $J^l(\mathbf{x})$ are decoupled, and the solution of the system refers to each coordinate component by itself.

Properties of the Global Stiffness Matrix

The global stiffness matrix is sparse, square banded, symmetric and positive definite. All of its elements are positive, the largest element per row or column is in the main diagonal, and the sum of all elements in a row (or in a column), is constant for each n . Its graph's shape is determined by the numbering of each of the degrees of freedom involved in the problem.

Let,

- $s = 3L(m+1)(n+1)$ be the sum of degrees of freedom for all the elements,
- ElementsAlongU, ElementsAlongV be the number of elements along u and v parameter lines respectively,
- and NodesAlongU, NodesAlongV be the number of nodes along u and v parameter lines respectively (every node contains 3 degrees of freedom).

For each element of degrees $m \times n$, there are $3(m+1)(n+1)$ degrees of freedom. Every boundary curve which shares two neighboring elements (patches) decreases the total number of degrees of freedom by:

$$s_b = \text{NodesAlong}U * (\text{ElementsAlong}V - 1) + \text{NodesAlong}V * (\text{ElementsAlong}U - 1).$$

Every cross section of such two boundary curves (meaning, at a cross section of four neighboring elements) decreases the total number of degrees of freedom by 9. The total sum of degrees of freedom at the cross sections is:

$$s_c = (\text{ElementsAlong}V - 1) * (\text{ElementsAlong}U - 1).$$

Therefore, the total number of the problem's degrees of freedom and the order of the global stiffness matrix is at most $s - s_b - s_c$. The creation of the global stiffness matrix is fast and efficient, since it involves only the assembly of the element stiffness matrices, which are pre-calculated and small.

The constrained Problem and its solution

Among the exiting numerous algorithms for solving a constrained optimization problem, some solve the constrained problem by replacing it with a family of unconstrained optimization problems. We use the Lagrangian multiplier formulation that converts the constrained minimization problem,

minimize $J^l(\mathbf{x})$ subject to the constraints $\varphi_i(\mathbf{x}) = 0, 1 \leq i \leq m$,
into the following unconstrained min max problem,

$$L_{\min \mathbf{x}, \max \lambda}^\Phi(\mathbf{x}, \lambda) = J^l(\mathbf{x}) + \sum_{i=1}^m \lambda_i \varphi_i(\mathbf{x}), \quad (23)$$

where

$$\lambda \equiv [\lambda_1, \dots, \lambda_m]^T. \quad (24)$$

λ_i is called the Lagrange multiplier for the constraint $\varphi_i(\mathbf{x}) = 0$. Solution of problem (23) with regard to the degrees of freedom in \mathbf{b} and λ yields the following necessary conditions:

$$\nabla L^\Phi(\mathbf{x}, \lambda) = \begin{cases} M\mathbf{b} - \mathbf{m} + \frac{\partial \sum_{i=1}^m \lambda_i \varphi_i(\mathbf{x})}{\partial \mathbf{b}} = \vec{0} \\ \sum_{i=1}^m \varphi_i(\mathbf{x}) = 0. \end{cases} \quad (25)$$

Error Estimation

Since the approximation depends on the parametrization of $\mathbf{x}(u, v)$ and $\mathbf{f}(u, v)$, it does not necessarily yield orthogonal error vectors between corresponding values of parameters. The absolute Euclidean minimum (or maximum) is at the point where two normals are collinear. Therefore a re-parametrization is needed so that the correction of the parametrization will direct the error vectors to be as orthogonal as possible to the tangent plane of \mathbf{x} at (u, v) . This will result in a better error estimation [16].

We use two types of discrete error estimators:

- the largest error Euclidean distance δ :
$$\delta = \max\{\|\mathbf{x}(u_i, v_j) - \mathbf{f}(u_i, v_j)\|, u_i \in [a, b], v_j \in [c, d]\}. \quad (26)$$

- the maximal angle deviation ϑ , between the normals:

$$\vartheta = \max \left\{ \arccos \left(\frac{\mathbf{n}_x(u_i, v_j) \cdot \mathbf{n}_f(u_i, v_j)}{\|\mathbf{n}_x(u_i, v_j)\| \|\mathbf{n}_f(u_i, v_j)\|} \right), \quad u_i \in [a, b], \quad v_j \in [c, d] \right\}. \quad (27)$$

In order to use error measurements which are not dependent on the parametrization, we use the zeroth, first and second derivative error (semi) norms (4-6) for error estimation. The zero derivative error norm (squared error integral) is the L^2 norm. The first and second derivative error (semi) norms are used to estimate the error in the first and second partial derivatives displacements.

We also use the error (semi) norms $E(x)/S^2$, $\bar{E}(x)/S^2$ and $\hat{E}(x)/S^2$, which measure the mean error displacement per unit area, where S is an approximated area of $f(u)$. For error in curvature, we use the Gaussian curvature L^2 error norm:

$$E^K = \int_a^b \int_c^d (\tilde{K} - K)^2 dudv, \quad (28)$$

where \tilde{K} is the Gaussian curvature of the approximated surface $\mathbf{x}(u, v)$, and the Gaussian curvature mean deviation error:

$$E_m^K = \frac{\int_a^b \int_c^d (\tilde{K} - K)^2 dudv}{\int_a^b \int_c^d (K)^2 dudv}. \quad (29)$$

We also use the mean curvature L^2 error norm:

$$E^H = \int_a^b \int_c^d (\tilde{H} - H)^2 dudv \quad (30)$$

where \tilde{H} is the mean curvature of the approximated surface $\mathbf{x}(u, v)$, and the mean curvature mean deviation error:

$$E_m^H = \frac{\int_a^b \int_c^d (\tilde{H} - H)^2 dudv}{\int_a^b \int_c^d (H)^2 dudv}. \quad (31)$$

Error estimations of all types presented in this section are presented in tables 1-2, for the approximation of a Bézier patch of degrees 5×3 , with sizes of $3 \times 28.5 \text{ mm}$ and approximated area of $S = 85.36 \text{ mm}^2$ (see Figure 1), by a Bézier patch with different degrees and continuity orders between elements.

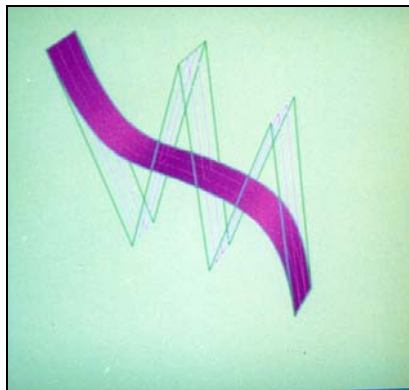
degrees	segments	δ	ϑ	$E(x)/S^2$	$\bar{E}(x)/S^2$	$\hat{E}(x)/S^2$	
2 x 3	1 x 1	1.479	0.348	2.348264e-04	2.523826e-05	5.107362e-05	
3 x 3	1 x 1	0.380	0.240	1.465209e-05	1.818131e-06	5.128979e-05	
4 x 3	1 x 1	0.365	0.220	1.395186e-05	1.595004e-06	1.254311e-05	
5 x 3	1 x 1	0.0	1.13e-07	4.447024e-19	2.911940e-18	1.115636e-04	
3 x 3	2 x 1	0.450	0.245	1.396280e-05	2.934349e-06	1.763063e-04	
4 x 3	2 x 1	0.041	0.039	1.395732e-07	7.531847e-08	1.712148e-04	
2 x 3	3 x 1	0.540	0.276	1.530292e-05	2.978221e-06	1.713536e-04	
3 x 3	3 x 1	0.051	0.058	1.651474e-07	1.487066e-07	1.769908e-04	

4 x 3	3 x 1	0.002	0.006	5.137242e-10	2.253008e-09	1.929809e-04	
4 x 3	3 x 1	0.002	0.005	4.885186e-10	1.619946e-09	1.930424e-04	
4 x 3	3 x 1	0.011	0.001	8.110637e-09	8.997212e-09	1.935975e-04	

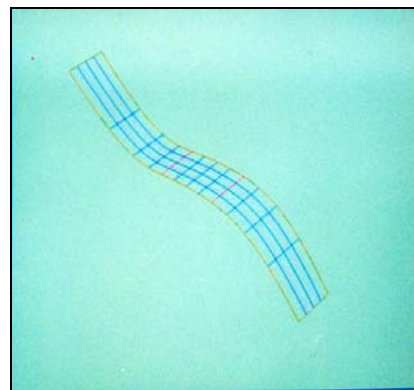
Table 1: Various error estimations by the derivative error (semi) norms, for the approximation of a Bézier patch of degrees 5×3 , with sizes of 3×28.5 mm and approximated area of $S = 85.36$ mm², by a Bézier patch with different degrees. See Figure: 1

degrees	E^K	E_m^K	E^H	E_m^H
2 x 3	3.339768e+00	9.764867e-01	5.592338e+00	8.735588e-01
3 x 3	2.059826e+00	5.233216e-01	3.087451e-01	5.079714e-02
4 x 3	6.194731e-01	1.573840e-01	2.562805e-01	4.216526e-02
5 x 3	2.012408e-14	5.084107e-15	1.561175e-14	2.442805e-15
3 x 3	8.055888e-01	3.363193e-01	3.795821e-01	5.976159e-02
4 x 3	8.473989e+00	3.537743e+00	8.030403e-02	1.264311e-02
2 x 3	4.354274e-01	5.014103e-01	4.724982e-01	7.439097e-02
3 x 3	1.259487e+01	1.449829e+01	1.270752e-01	1.998752e-02
4 x 3	8.319730e+01	9.577063e+01	6.819476e-01	1.072628e-01
4 x 3	8.309422e+01	9.565197e+01	6.816075e-01	1.072093e-01
4 x 3	8.281638e+01	9.533215e+01	6.837922e-01	1.075529e-01

Table 2: Various error estimations by the curvature error norms, for the same given Bézier patch and approximation surfaces as in Table 1.



(a) Given surface



(b) The approximation surface.

Figure 1: Reduction of a Bézier patch of degrees (5×3) , with sizes of 3×28.5 mm and approximated area of $S = 85.36$ mm², by an approximation (C^2, c^3) Bézier surface with three patches of degrees (4×3) .

5. Incorporation of Constraints

In this section we introduce the description and implementation of constraints which are imposed upon the approximation or the construction of surfaces in order to achieve the following desired geometrical effects:

Constraints for Surface's End Points Interpolation

In order to create a better approximation, the end points of the approximation surface should interpolate the end points of the given surface. Meaning, the conditions: $\mathbf{x}(a, c) = \mathbf{f}(a, c)$, $\mathbf{x}(b, c) = \mathbf{f}(b, c)$, $\mathbf{x}(a, c) = \mathbf{f}(a, c)$, and $\mathbf{x}(b, c) = \mathbf{f}(b, c)$, (32) must be satisfied.

First Derivative Interpolation at the End Points of Boundary Curves

The direction of the end edges of each boundary curve control polygon of the approximation surface, are constrained to have the same directions as the tangents at the end points of the approximated surface boundary curves. The method used for this constraint, is the same one used for curves, as described in [4].

Constraints for C^k , Continuity Between Elements

Higher continuity between elements will not necessarily improve the approximation but will create a smoother approximation surface, a feature which is desirable in many cases, such as, the design of mechanical parts which requires first- or second- order smoothness, or in the definition of a tool path for NC machine where we need the speed and the acceleration of the tool to be continuous. This can be achieved by C^1 and C^2 parametric continuity.

A Bézier surface may contain several patches joined together with a given continuity. Let $\mathbf{x}^{m,n}(u, v)$ and $\mathbf{y}^{m,n}(u, v)$ be two Bézier patches defined over $[u_{i-1}, u_i] \times [v_{j-1}, v_j]$ and $[u_i, u_i + 1] \times [v_{j-1}, v_j]$ respectively. $\mathbf{x}^{m,n}$ and $\mathbf{y}^{m,n}$ have C^q continuity along the parametric line $u = u_i$, if

$$\frac{\partial^q}{\partial u^q} \mathbf{x}^{m,n}(u_i, v) = \frac{\partial^q}{\partial u^q} \mathbf{y}^{m,n}(u_i, v). \quad (33)$$

For Bézier surfaces it is possible to reduce the surface problem to several curve problems. Using the *cross boundary derivative* (the derivative at the common boundary curve) with respect to the global parameters u, v we obtain:

$$\frac{1}{(\Delta_{i-1})^q} \sum_{j=0}^n \Delta^{q,0} \mathbf{b}_{m-q,j} B_j^n(v) = \frac{1}{(\Delta_i)^q} \sum_{j=0}^n \Delta^{q,0} \mathbf{b}_{m,j} B_j^n(v). \quad (34)$$

It follows that along the parametric line $u = u_i$,

$$\frac{1}{(\Delta_{i-1})^q} \Delta^{q,0} \mathbf{b}_{m-q,j} = \frac{1}{(\Delta_i)^q} \Delta^{q,0} \mathbf{b}_{m,j} \quad j = 0, \dots, n. \quad (35)$$

Equations (35) are conditions on rows of control points across the boundary curve which represents C^q Bézier curves (see [14,23]).

The continuity constraints can be imposed only on part of the patches. This is usually done on a boundary of complete two rows or two columns in the mesh of elements. This way we construct a surface the in some parts it has a high continuity degree (C^2 for example), and in other parts a C^0 continuity is left to produce a corner or a cusp.

Examples

Approximation of surfaces can be used in the applications of: degree reduction of surfaces, merging of surfaces with large number of patches, and construction of offset

surfaces. Reduction of degree of high order polynomial surfaces to polynomials of a lower order. The degree reduction approximation might reduce or increase the number of surface patches. It is sometimes needed to merge a surface which is constructed from many small patches into a surface with less patches of the same degrees, or even with higher degrees. The construction of a parametric offset surface, is done by approximating the offset surface $\mathbf{x}_d(u, v)$ to the given surface $\mathbf{f}(u, v)$.

$$\mathbf{x}_d(u, v) = \mathbf{f}(u, v) + d\mathbf{n}(u, v), \quad (36)$$

where $\mathbf{n}(u, v)$ is the principal normal vector, and d is the offset distance along $\mathbf{n}(u, v)$.

Among the surfaces that we used in the section, there are the following three surfaces:

1. The surface BRODE (Fig. 2(a)), which is a 9 x 9 Bézier patch with sizes of 100 x 140 mm.
2. The surface SEITE1 (Fig. 2(b)), which is a collection of 9 x 7 Bézier patches, each of degrees 3 x 3 and sizes about 500 x 2200 mm.
3. The surface SURFB (Fig. 2(c)), which is a collection of 17 x 63 Bézier patches, each of degrees 5 x 5 and sizes about 450 x 1800 mm.

These surfaces were used as test examples (bench-mark) for the comparison between spline conversion methods (See for example, [12, 13, 15, 24].

The bench-mark specifications were to convert these surfaces to a 3 x 3 or a 5 x 5 Bézier or B-Spline surface with a maximal error tolerance of 0.1 and 0.01 mm. For each of the surfaces involved in the bench-mark a table was presented, in which the degrees and the segment number of the result approximation surface are listed in the first two columns. In the third column the inner continuities in both parameter directions are entered. The fourth column contains the compression factor. The compression factor for the conversion to Bézier surface is given by the quotient:

$$\frac{\text{number of Bézier points of the given patches}}{\text{number of Bézier points of the converted patches}},$$

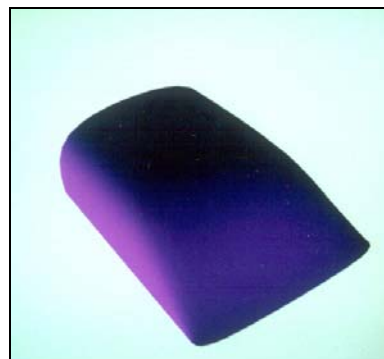
and the compression factor for the conversion to B-Spline surface is given by the quotient:

$$\frac{3(\text{number of Bézier points of the given patches})}{3(\text{number of B-Spline control points}) + \text{knots in the knot vectors}}.$$

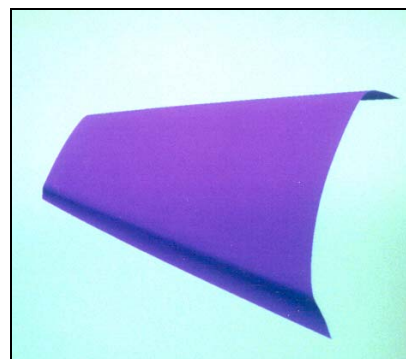
In the fifth column the approximation presents the prescribed approximation tolerance, and in the sixth column the largest error Euclidean distance

$$\delta = \max\{\|\mathbf{x}(u_i, v_j) - \mathbf{f}(u_i, v_j)\|, \quad u_i \in [a, b], \quad v_j \in [c, d]\},$$

is given. We will use the same format of table for our examples.



(a) The BRODE surface



(b) The SEITE1 surface



(c) The SURFB surface

Figure 2 : Benchmark Surfaces.

Constraints results

Examples of constrained approximations are introduced in our results of the benchmark surfaces BRODE, SEITE1 and SURFB in tables (3-5). These examples emphasize the *p*-method and *h*-method of FEM in the approximation. We see for example in table (5) that presents approximation results for the SURFB surface (Figure 2(c)), that each line in the table introduces an approximation with a different mesh size (signified by the *segments* column of the table) and patch degrees. One can see the different continuity degrees achieved by the imposing C^1 or C^2 continuity constraints between the approximation surface patches. The last two lines of table (3), represent an approximation of the BRODE surface (Figure 2(a)), that uses a mesh of elements of different degrees, attached with C^0 continuity. These results display how the combination of the *hp*-methods in FEM was used in the approximation scheme of this research.

Another example that displays an approximation of an offset surface is presented in figure (3). The approximation offset surface that approximated a surface of degrees (3 x 3) (Figure 3(a)), has an offset distance of -0.4 mm , two patches of degrees (3 x 3) and (C^1, C^3) continuity between its patches. The approximation integrates the end points and first derivative interpolation constraints.

All the above examples integrated the end points and first derivative interpolation constraints (see sections 6.1-6.2, respectively).

Table 3. surface: BRODE:

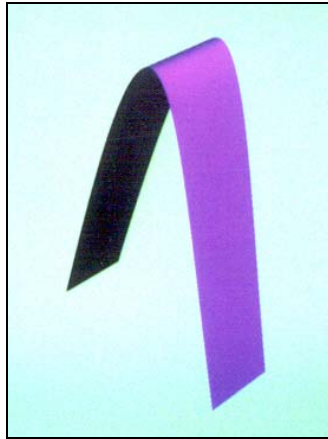
surface: BRODE: degree: 9 x 9, segments: 1 x 1					
degree	segments	minimal continuity	compression factor	error tolerance	measured error
3 x 3	5 x 3	C^2, C^2	0.42	0.01	0.008
	3 x 2	C^1, C^1	1.04	0.01	0.01
	2 x 2	C^1, C^1	1.56	0.1	0.042
	3 x 2	C^2, C^2	1.04	0.1	0.018
4 x 3	2 x 2	C^2, C^2	1.25	0.1	0.017
4 x 4	2 x 2	C^2, C^2	1.0	0.01	0.006
	2 x 1	C^2, C^2	2.0	0.1	0.026
4 x 5	2 x 1	C^2, C^2	1.66	0.1	0.016
5 x 3	1 x 2	C^2, C^2	2.08	0.1	0.023
5 x 5	2 x 2	C^2, C^2	0.69	0.01	0.002
	2 x 1	C^2, C^2	1.39	0.1	0.015
	1 x 1		2.78	0.1	0.022
7 x 6	1 x 1		1.79	0.1	0.004
7 x 7	1 x 1		1.56	0.1	0.001
3,5,3 x 3,4	3 x 2	C^0, C^0	1.04	0.01	0.008425
3,3,2,2,3,3 x 3,4	6 x 2	C^0, C^0	0.52	0.01	0.009057

Table 4. surface: SEIT1

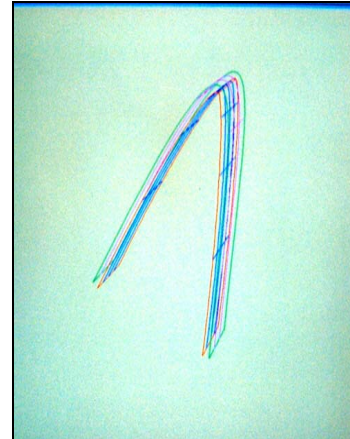
surface: SEITE1: degree: 3 x 3, segments: 9 x 7					
degree	segments	minimal continuity	compression factor	error tolerance	measured error
3 x 3	8 x 7	C^2, C^2	1.13	0.01	0.0088
	2 x 7	C^2, C^2	4.5	0.1	0.1
4 x 3	2 x 7	C^2, C^2	3.6	0.1	0.068960
5 x 3	3 x 7	C^2, C^2	2.0	0.01	0.003987
5 x 5	3 x 7	C^2, C^2	1.33	0.01	0.003985
	2 x 7	C^2, C^2	2.0	0.1	0.023714
	1 x 7	C^2, C^2	4.0	0.1	0.1

Table 5. surface: SURFB

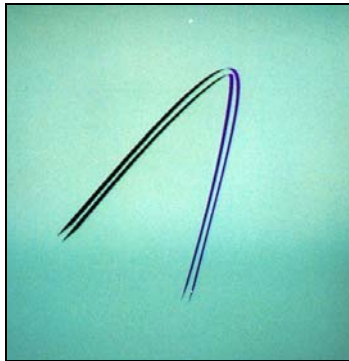
surface: SURFB: degree: 5 x 5, segments: 17 x 63					
degree	segments	minimal continuity	compression factor	error tolerance	measured error
3 x 3	17 x 29	C^2, C^2	4.89	0.01	0.008632
	17 x 16	C^2, C^2	8.86	0.1	0.076872
3 x 4	17 x 13	C^2, C^2	8.72	0.1	0.016416
3 x 5	17 x 13	C^2, C^2	7.27	0.01	0.004687
	17 x 8	C^2, C^2	11.81	0.1	0.041205
4 x 4	14 x 17	C^2, C^2	6.48	0.01	0.008888
	8 x 11	C^2, C^2	17.53	0.1	0.079889
5 x 4	6 x 11	C^2, C^2	19.47	0.1	0.095811
5 x 5	14 x 13	C^2, C^2	5.88	0.01	0.008055
	6 x 6	C^2, C^2	29.75	0.1	0.075385



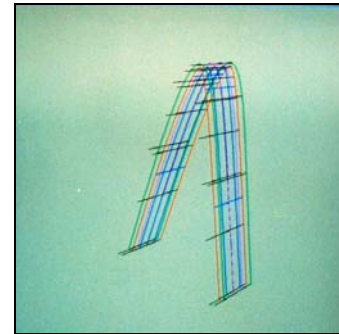
(a) Bézier patch of degrees



(b) The shaded surfaces



(c) The shaded surfaces



(d) The surfaces with drawing of the tangents along the v-parameter lines

Figure 3: An offset Bézier surface to the surface (a), with two patches of degrees 3×3 , (C^1, C^3) continuity between patches and offset length of -0.4.

6. Conclusions and Future Work

In our work we introduced a global and continuous method for approximation and/or construction of surfaces. It is based on a minimization of a functional which makes use of global and continuous criteria (4-6), for approximation and construction. Constraints were integrated in the approximation to obtain some specific characteristics. Some of the properties we would like to obtain are aimed at improving the approximation, such as, end points interpolation and end direction preservation. Others are designed to impose a desirable form upon the approximation surface, for instance, parametric continuity between the approximation surface's patches. The constraints used were linear, equality, and operates on a discrete parameter range. We used the Lagrangian multiplier formulation for the constrained problem.

The numerical solution of the functional uses FEM with the Bernstein-Bézier representation for the shape functions, and presents cardinal advantages:

- The approximation method operates globally on the given problem's domain.
- Segmentation of the approximation surface is natural to FEM, because of the subdivision of a FEM problem's domain into elements.

- Every element is treated separately, and its "influence" is added to the general stiffness matrix such that there is no limitation on the form of the general range combined from a collection of elements.
- It is possible to approximate, using different elements with different degrees of elements (Bézier patches).
- The system of equations is linear for any degrees of the elements and any order of parametric continuity C^n between the elements.
- The use of Bézier-Bernstein representation grants good properties for the stiffness matrix, and saves much of the approximation calculation using proper solution methods (such LDL^T). Most of the calculation of the element matrices is prepared in advance and can be used regardless of the subdivision of problem's domain.
- There is not a pre-requirement on the given surface's continuity.
- It is possible to approximate, using different elements with different degrees of parametric continuity, including C^0 .

As the next step in this research, we intend to include different types of linear and nonlinear constraints, of equality and/or inequality types, on a discrete or a continuous parameter range. These constraints are aimed at improving the approximation, or at imposing a desirable form upon the approximation surface. Examples of such constraints are, optimal construction using reparametrization, opening loops for given looped surfaces, or avoiding loops in the approximation of offset surfaces.

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² **Arie Jacobi** is Senior Lecturer in the Software Engineering Department, Sami Shamoon College of Engineering, Israel. He was born in Israel in 1956, and received his B.Sc., M.Sc. and Ph.D. degrees in Computer Science from the Hebrew University of Jerusalem, Israel. Dr. Jacobi's professional experience includes 28 years of engineering and management positions in Israeli software and high-tech industries. His current research interests are in Computer Aided Geometric Design and Approximation Theory, in which Dr. Jacobi investigate different numerical and geometrical methods for approximation and construction of curves and surfaces. Another field of research interest is Complexity in Social Networks. Dr. Jacobi is now investigating problems of trust for information, which is distributed over social networks whose growth and evolution are primarily driven by personal interactions. Dr. Jacobi has published in leading journals such as Computer Aided Geometric Design (CAGD) and Mathematical Modeling and Numerical Analysis (MAN).

³ **Shilo Lifschutz**, is the chairman of the accounting department and a faculty member of accounting and finance at the Academic Center of Law and Business, Ramat Gan, Israel. He holds a PhD from the Hebrew University of Jerusalem and is a CPA (Israel and Illinois, USA). His main research interests are quantitative methods in finance and risk management.

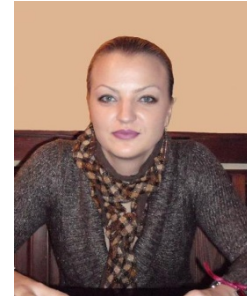
COMPARATIVE MULTIDIMENSIONAL ANALYSIS OF WHEAT PRODUCTION AT COUNTY LEVEL

Aura POPA

PhD Candidate, Department of Statistics and Econometrics,
University of Economics, Bucharest, Romania

Email: aura.popa@csie.ase.ro,

Web page: www.aurapopa.ase.ro



Abstract: *This article presents a series of multidimensional analysis methods on agricultural factors at county level, performing subsequently a comparison between the results obtained in 2004 and, respectively, in 2008. By applying factor analysis, principal components analysis and hierarchical classes analysis, this article will seek to highlight the main variables that influence the agricultural production of grain, by territorial grouping by county, and then forming some specific county typologies for Romanian wheat production.*

Key words: *agriculture; multidimensional data analysis; cluster analysis; factor analysis; county typologies for wheat production*

Introduction

Multivariate analysis used to study wheat cultivation in many articles, both in terms of its taxonomy, agronomy or agrometeorology, in the habitat assessment and the conduct of their parameters change scenarios at the limits considered normal. Such studies have been conducted in agriculture for decades, but in particular only inside experimental models, as well as those of **Waugh (1942)** which shows an example consisting of two sets of variables describing the type of spring wheat from Canada and the flour characteristics that resulted, or **Walton (1971)** who conducted studies on wheat genetics, or more recently Dixon (2006) on the same curriculum. In combination with engineering and cutting-edge technologies, it attempts to study production at the spatial level, by topographing and scanning the biomass, but with the latter it showed major disadvantages of data collection, yet unimproved. From a multidimensional perspective, **Moayedi et al. (2010)** investigated the effect of irrigation regimes during different phases of growth and development on yield and yield components, for some genotypes of wheat intended exclusively for bakery uses. Examples may continue, but it is to be remembered that the majority has as units of analysis the parcel, or small areas, homogeneous in terms of descriptors of natural and technical endowments utilised.

Several national and regional analysis, but also as derived from the literature, reveals that in the case of agriculture, matters must be considered more closely. The attempt to analyze at a certain level—even if it is at regional, macro-regional or national—makes many "players" lose their important role. Therefore, it is necessary to examine this issue at the county level, although the ideal, but nevertheless hypothetical, the most pertinent analysis is at the micro level, i.e. parcel level, of the area under cultivation. However, this would only be possible with the support of the implementation of a geographic information system software, a topographic database with detailed information about the soil type, meteorological situation type of crop on each parcel of land, the amount and the type of fertilizers that are used, agro-technical methods implemented throughout the cultivation

period, mapping the network of water etc. Because some of these data sources are not freely available, and others are not collected in an existing database, we must apply a multidimensional analysis at county level which is intended to outline for the whole group some areas for this specific analysis

The agricultural sector needs and uses more and more raw materials to improve its production and, therefore, it becomes even more dependent on agro-industrial complex than we might think. The farmer is subjected to a process of alienation with its means of production, which in turn affects later social and financial status (**Goodman et al., 1987**).

Farmers, especially those with less land, can not afford to lose even the smallest piece of means of income and therefore will not bypass inputs that will ensure a great deal of success (equipment, mechanical equipment, chemical fertilizer, better quality seeds etc.) (**Friedman, 1991**). Therefore, it should not surprise us that a Romanian farmer, from his little land available, will buy fertilizer or pesticide with the risk to cover costs at the limit rather than to live a mere subsistence life.

According to analysis conducted by the National Meteorological Administration from Romania, for the years 2004 and 2008, agrometeorological characteristics showed no extreme events, vegetative rhythms on both field crops and fruit-growing species have evolved, in a whole, normally overall across the country, the processes of growth and development until harvest did not register impediments, although sometimes there were some extreme phenomena locally, on small areas, but without causing major changes at the county, regional or national levels. These are some reasons why the author of this article chose these years for the analysis.

In order to observe major changes at territorial level and concerning wheat production, requires a distance of five years between the current and the base year of study. Usually, it is taken as reference years that are multiples of five. In this case, the most recent data were those for 2000 and 2005, but according to ECA&D¹ many extreme events, were recorded, particularly in April 2005, when floods affected thousands of homes in over 140 localities and 30,000 hectares of cultivated farmland were covered in water, and in May of the same year, very large quantities of rain fell over more than 130,000 hectares of arable land. Several areas were also affected in the same year, in June and July—2,300 households, and in August about 1,400 homes were destroyed with all the agricultural land affected within their surrounding area. For this reason, a comparison analysis with this year could lead to a distortion of the overall situation, and because in terms of extreme weather phenomena such events have never been applicable in Romania for more than 50 years.

Multivariate models used to analyze wheat production in agriculture

Hierarchical methods, non-hierarchical and statistical classification try to recover, as far as possible, the real structure and form of groups of factors that underly the successful cultivation of wheat. Because these methods are already known, the following part of the article will focus on the study of results and their comparison. It will be presented in a series of multivariate data analysis methods used to identify some characteristics of wheat farming in Romania.

In the first part, in the descriptive analysis, conducted using SAS² software, data that were used in this study are summarized, both for the year 2004 and for the year 2008, and then will be presented the results of the methods used to identify features of clustering at the county level.

Descriptive data analysis

In a typical problem of numerical classification, expression of data will be in the form of a matrix, with n individuals—in our case represented by counties, on which were evaluated p characteristics, as follows:

- FOM—civilian employed population in agriculture at NACE Rev. 2 section, at counties level (per thousand persons);
- PROD—wheat production per counties (tons);

- SUP—surface cultivated with wheat, per counties (hectares);
- Park Numbers of agricultural tractors (TR), of plows (PLU), of mechanical cultivators (CUL), of mechanical drills (SEM), of self-propelled combine harvester, respectively, straw and hay balers (PRE) for each county (number of pieces);
- Insecticides (INS), fungicides (FUN), herbicides (ERB) and pesticides (PES) in whatever form or substance or mixture of substances, including their mixtures with ingredients for use in agriculture, forestry, the storage areas and other activities in order to prevent, reduce, remove or destroy pests, pathogens, weeds and other forms of animal or plant life, including viruses, harmful to plants and domestic animals, insects and rodents carrying diseases transmissible to humans, and adjusting products affecting plant growth, defoliation or splitting them; they are reported for each county substance (kilograms);
- Chemical fertilizers (CHIM) presented as industrial products as their contents may be nitrogen (AZ), phosphate (FOS), potassium (POS); they are expressed in active substance used on arable land in the county in question (tonnes);
- NAT—natural fertilizers including manure from all species of animals and birds (fresh or fermented) and liquid slurry, which is expressed in gross weight used at the county level (tonnes).

All data were collected for year 2004 and 2008, being extracted from regional statistical yearbooks published by the Romanian National Institute of Statistics, and the newly created base was first validated to eliminate any errors caused by any incorrect data entry, followed by application of proposed methods, interpretation of results for each year separately and so then reveal significant changes at temporal comparisons.

In the literature, some authors recommend standardisation of variables as a previous step to calculate the gap between values that may be due to different metrics, in order to eliminate the effect of scale on the results of the final hierarchization.

The most common types of standardisation are those using standard deviation or with the help of ranking. Thereby, for example, **Milligan and Cooper (1988)** recommended the second option because it has a linear standardisation, being more robust in the presence of aberrant values.

For the simple descriptive analysis of data, resulted in large overall coefficients of variation within several variables taken into account, therefore, because the data that we have present no aberrant values, we express information in a standardized form applying as first proposed by standardized values of mean zero and dispersion equal to 1.

Even though in the 2008 data there have been increases in the case of some factors such as technical equipment—the fleet enriched in average with about 620 machines, the used quantities of fertilizers increased from 1,03 in the case of nitrates—the yield of this year was with 29,43% less than for the year 2004, this being due to the particular decrease of people employed in agriculture, an absolute number of 205 thousands persons, and the diminuation with 8,09 percentage points of agricultural area cultivated with wheat.

However, this change can be made on the behalf of the previous year, 2007, when there were recorded the highest annual average temperatures, causing drought in large areas, therefore the production was very low in almost all types of grains, and increasing wheat prices on international markets led farmers to be unable to buy enough seed for sowing in the autumn of that period.

As was expected, the top 10 are the counties of Timis, Calarasi, Teleorman, Olt and Dolj are found in both years chosen for analysis, and the bottom places are counties such as: Salaj, Hunedoara, Bistrita-Nasaud, Maramures, Harghita or Sibiu (we omitted from the analysis Bucharest which has the smallest agricultural area of all the administrative areas considered).

One obvious reason is the profile of agricultural land these leading counties hold, the total area of arable land in each county in part plays an important role, which have the highest weights of the total arable land in the country (for example, in 2004, in Constanta county have been cultivated 5,03% of arable land from the total farmland with wheat across the county, while in Sibiu county it was only 1,01%), but also because of the high altitudes, in many counties the presence of hills and mountains is a major influence.

Principal components analysis and factor analysis.

Factor analysis is a multivariate statistical technique aimed at extracting a small number of hidden factors (latent) responsible for correlations between the original variables. If these correlations are relevant, it is believed to be caused by the existence of one or more common „hidden“ factors for all variables. Factor analysis allows us to confirm the statistical results on the group of original variables. The variables are grouped together and, as such, they may be regarded as forming a homogeneous group and will be considered related to the same factor. Multivariate analysis techniques are increasingly used throughout many areas of research, and to characterize wheat production, in terms of quantity or quality, as certain aspects of the goal.

Based on descriptive information available about the total wheat production, the area under cultivation, the agricultural machinery, the amount of pesticides and the quantity of fertilizers used in every county in Romania, for 2004 and 2008, we can identify trends in certain areas not necessarily regionally, but links between factors that have been chosen.

Application of factorial analysis on data that we have, will result in a first correlation matrix and will conclude that:

- For 2004, there are strong direct links between the components of the available agricultural vehicles between the fleet and the production of wheat, between the fleet and fertilizer, but most importantly to note is that between the production of wheat and other factors considered in the analysis. As expected, the area under this kind of crop is a main „character“ ($r=0,875$)³, but also the sowers ($r=0,897$), the combines ($r=0,788$), and among the fertilizers, the nitrates are emphasized ($r=0,75$), between the machinery and fertilizers—the sowers and nitrates, with a total variation synthesizing in 90,56% between the latter variable.

- For 2008, we can say that this time, the correlation matrix have strong direct links between the same pairs of variables as in the previous year considered, but it is most important to note again the strong linkage between the production of wheat and other factors taken into account in the analysis.

- This year, the area for this kind of agriculture has the greatest impact ($r = 0,989$), and other variables are slowing down in effect. Tractors and plows, with a correlation of 0,983 (also confirmed the fact that, logically, as the number of tractors increased, the more there will be bought plows for them). The sowers and the combines are in a direct linkage, with a correlation of 0,929, which indicates that the number of drills once increased, ads to more drilling, and so the harvest will be bigger;

- Both for 2004 and for 2008, it is noted that the employment variables in agriculture, insecticides, fungicides and natural and potassic fertilizers are poorly correlated, so in the further analysis they will be not considered. Therefore, from the 17 factors that we considered at first, only 12 remain.

Linking fertilizers with the rest of variables considered, only in the case of nitrates, it was expected as the agro-chemical substance used in such crops is the main one, and therefore, as known from the literature, there is a quatitatively role for the final harvest.

Therefore, the new group of values associated with previous correlation matrix, will be determined by its own values. A value greater than 1 for a component indicates the fact that it has a larger contribution than that of the original value, as indicated from the extract at (Figure 1).

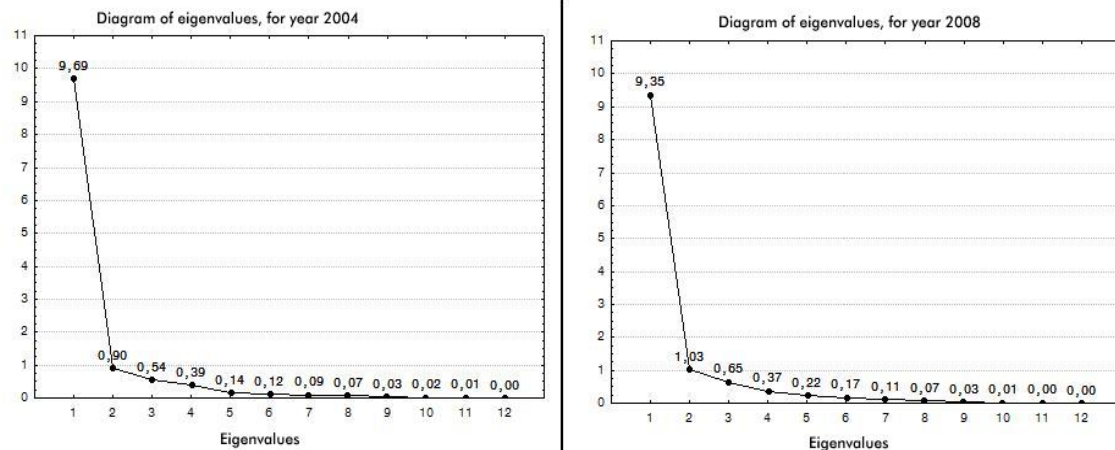


Figure 1.—Diagram of eigenvalues of correlation matrix (left for the year 2004, right for the year 2008).

From the obvious difficulty of viewing multi-dimensional space, having sizes larger than four, principal components analysis is used mostly to minimize the dimensionality of variables in at least two, but no more than three dimensions, as recommended in an example given by **Braun and Maindonald (2010)**.

This method summarizes initial data variability on all the 12 factors with which we started, in number of twelve, in several unrelated parts together, called principal components. Each principal component is extracted in the linear combination of original variables, respectively with the minimum loss of information. The first principal component extracted from this combination is one that takes the maximum possible change from the original data. Thus, in both years, the biggest eigenvalue, of 9,69, it is of wheat yield, with a variance of 80,71% for the year 2004 and, respectively, of 77,88%, with an absolute value of 9,35, for year 2008. The second principal component, the area planted with wheat, takes less variance of only 7,46% (0,89) for the base year, and slightly higher in 2008, i.e. 8,54% (1,02 in absolute value), and so on for the rest of their eigenvalues that corresponds to the following considered values.

As the first principal components take more than 88,71% from the variance of initial data for the first year, and, respectively, of 86,43% for the other year of our study, then we can say that the scope to reduce the dimensionality was achieved.

The notion of score observations can be understood if we interpret the observations as vectors in an n -dimensional space of variables, as determined by **Stewart and Muller (2006)**.

The last part of the principal components analysis involves the loadings's interpretation that are correlation coefficients between columns-score and original variables. Extremely important is the study of correlation coefficients between original variables and the first two principal components that emerged above. Strong coefficients show that corresponding variables can be considered responsible for the variance of the data.

Therefore, for the year 2004, we have strong correlations between the second factor and the wheat production ($r=0,84$) and the cultivated area ($r=0,9$), it also presents a powerful link between the second factor and the first variable ($r=0,91$) and the second one ($r=0,90$).

Normally, to obtain factors with small loads, that were not significant, and therefore to simplify the interpretation of factors, there is performed a rotation through the well-known varimax method.

For the remaining variables, it is not correlating with any other principal component, nor it is correlating with components that have small eigenvalues (for the year 2004, it is applicable to tractors ($r=0,755$) and plows ($r=0,713$) with the forth factor, the chemical fertilizers ($r=0,767$) and nitrates ($r=0,712$) with the third factor; for year 2008, in isolation, a correlation is revealed between the fourth main component and balers

($r=0,845$), and also, between the third principal component and the nitrates ($r=0,7515$))⁴, this suggests that those variables have a minor contribution to the variance of the dataset, so that the "unimportant" variables will be removed in order to simplify the overall analysis.

Eigenvectors, associated eigenvalues, will be those weights in the calculation of linear combinations. Communality is part of the variance of initial variables, expressed as a percentage, which is due to all factors found.

When we assume that all communalities of the variables are equal to 100% — when all original variables are completely "explained" by factors— the actually result of the analysis coincides with the one of the principal components analysis, the results obtained in this case are:

Table 1. *Communality*

Variable	From factor 1 (in year 2004)	From factor 2 (in year 2008)
PROD	0,851	0,881
SUPR	0,822	0,905

Note: Extracton method: Principal Components Rotation: Varimax.

Factor loading coefficients form a matrix of size $p \times k$, each element expressing the correlation between an original variable and a factor. We have p variables and latent factors k , and $k < p$. From the correlation matrix between original and new variables obtained, we can ascertain which variables are correlated with new factors.

Thus, for the both years considered, factor 1 is directly and strong correlated with the wheat production, and the cultivated area, but weaker with the plows, seeders, sowers; the second factor is strongly correlated with the wheat yield and the area on which was cultivated, but weaker in intensity compared to other factors, and so on.

Classification methods.

Cluster analysis are designed to identify homogenous group of individuals, with similar variables that characterize them, it is necessary to define and to measure the similarity (or its complement, i.e. the distance) between two individuals or between groups of individuals, known as the proximity index. When all variables are continuous, the most commonly used method for calculating the distance between two individuals (observations) is the Euclidean or the Manhattan one. Hierarchical analysis was applied to the 42 cases, according to the two factors of the principal components analysis, production and area cultivated with wheat.

In the Figure 2, left, it can be easily observed that the biggest similarity will be found between counties that have the largest surfaces cultivated with wheat that resulted from the very simple descriptive analysis of the data from the beginning, and the high dissimilarity between these two variables proved to be the utmost importance from the previous analysis.

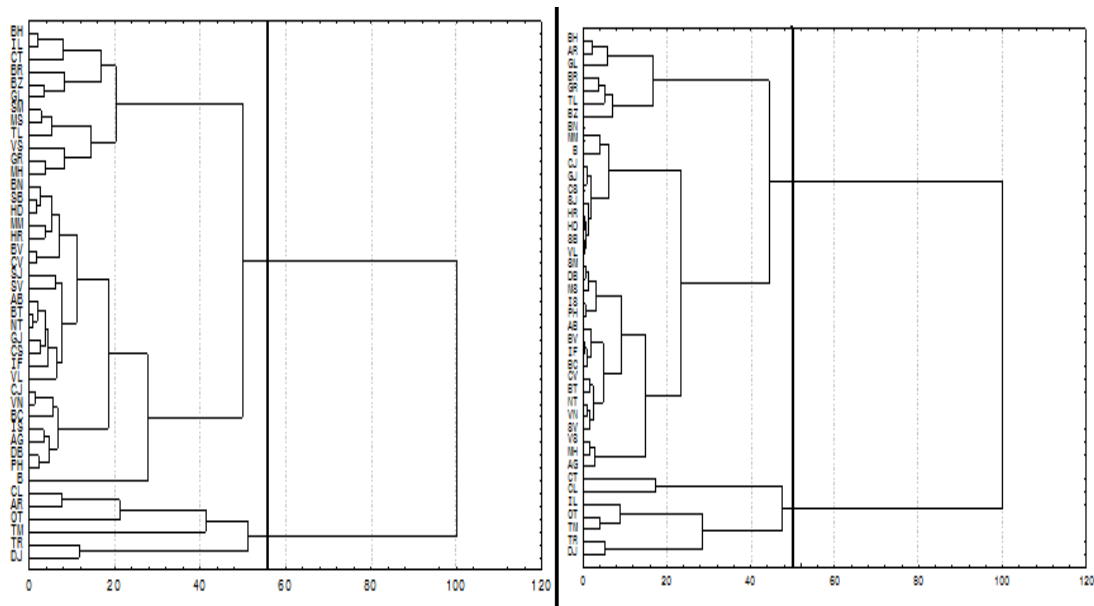


Figure 2.—Dendrogram of wheat production and growing surfaces, at counties level in Romania (left for the year 2004, and right for the year 2008).

However the penultimate stage is highlighted at the end of the concatenation, that all counties by grouping according to these variables will be divided into two major groups. The second cluster consists of six counties—Calarasi, Arad, Olt, Timis, Teleorman and Dolj being grouped by similar characteristics. We re-create a dendrogram for the year 2008 (Figure 2, right).

As analysts, we could cut the dendrogram at the level of also two major groups, prob could be worded better but this time the steps of concatenation is achieved much earlier, and we can observe that the structure of groups, members that are forming the counties groups, changes in distance between the two years from the leading counties that places a greater emphasis this year than the year taken as a base of comparison.

Among the members of the two groups with few structural changes, then we can say that Calarasi, Teleorman, Olt or Dolj county are on top or both rankings in different years, at a distance of five years. These counties are known for their specific inclination towards agriculture and in these areas for the farmers tendency for wheat cultivation.

From the final dendrograms which resulted, using cluster analysis, we can see that if we would have started analyzing at the macro-regional or at the regional level it would be lost in specificity and contribution for each county separately.

One reason for the loss of information, as much as is conducted through an overall analysis, it is due to the mismatch in territorial division that is purely administrative and not natural, within the specifics of agricultural land and crops found in each county separately.

Conclusions

The analysis in this study is based on data calculations for only two years, which have not been affected by any major influences from other factors, such as distribution of other types of crops on arable land available.

Instead, the great disadvantage of this comparison is the relatively small number of years and incomplete and unbalanced representation of all variables that are used in agricultural management, particularly the cultivation of wheat. If we could have access to a metadata base that holds all the necessary information at parcel level, this type of analysis would lead to more conclusive results. Such a database would be similar to that used by **Ziv and Goldman (1987)** for wheat cultures. In this case, the overall result indicates the

potential of all multidimensional data considered in order to identify predictors that lead to increased production, and therefore profits.

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Appendix1— County Codification

County	Code	County	Code	County	Code
Bihor	BH	Iasi	IS	Dambovita	DB
Bistrita-Nasaud	BN	Neamt	NT	Giurgiu	GR
Cluj	CJ	Suceava	SV	Ialomita	IL
Maramures	MM	Vaslui	VS	Prahova	PH
Satu-Mare	SM	Braila	BR	Teleorman	TR
Salaj	SJ	Buzău	BZ	Dolj	DJ
Alba	AB	Constanta	CT	Gorj	GJ
Brasov	BV	Galati	GL	Mehedinti	MH
Covasna	CV	Tulcea	TL	Olt	OT
Harghita	HR	Vrancea	VN	Valcea	VL
Mures	MS	Ilfov	IF	Arad	AR
Sibiu	SB	Bucuresti	B	Caras-Severin	CS
Bacau	BC	Arges	AG	Hunedoara	HD
Botosani	BT	Calarasi	CL	Timis	TM

Appendix 2—Dataset of agriculture at the regional level in year 2004
(source: Romanian Statistical Yearbook, 2005) (I)

Code	FOM04	PROD04	SUP04	TR04	PLU04	CUL04	SEM04	COM04
BH	92,3	881640	75141	9246	6918	1265	3089	1359
BN	45,9	169621	13302	2167	1665	143	552	269
CJ	78,1	426790	32051	5330	4452	593	1384	780
MM	77,4	133853	10796	3959	2724	81	433	235
SM	58,9	641429	46262	4535	3879	617	1735	829
SJ	37,2	241021	21598	4340	3654	498	1033	515
AB	51,6	377042	31524	3127	2373	720	1225	495
BV	33,7	151502	27518	4614	3283	839	994	443
CV	24,8	148469	24133	4002	2907	790	840	377
HR	43,9	88163	14553	3956	2400	1086	571	419
MS	74,8	601364	45007	5029	3950	654	1920	876
SB	29,5	204097	14478	2925	2135	444	813	309
BC	70,0	474330	22505	2424	1988	199	989	321
BT	79,0	367078	28383	2561	2289	361	1257	495
IS	99,7	461432	39341	3303	2922	696	1390	460
NT	87,1	355203	28541	2107	1840	239	987	402
SV	112,6	260088	33246	4497	3504	522	846	605
VS	73,0	689407	63888	2663	2413	249	1076	314
BR	41,7	960294	52962	3509	2827	830	1505	391
BZ	79,7	830255	45247	3549	2867	688	1247	323
CT	68,4	947655	86504	4938	3732	821	2150	660
GL	65,4	867146	49716	3243	2762	650	1464	346
TL	31,8	695579	43023	2521	1961	458	1102	448
VN	66,6	419478	29748	3191	2759	350	1119	316
Code	FOM04	PROD04	SUP04	TR04	PLU04	CUL04	SEM04	COM04
IF	36,1	307361	25628	1616	1243	404	829	177
B	8,8	2281	175	280	142	55	74	34
AG	74,1	504848	42400	3065	2426	406	1161	456
CL	51,2	1251109	113043	4504	3346	880	2278	851
DB	74,3	491899	34677	4771	3741	916	1685	378
GR	49,8	669364	81541	3756	2787	633	1704	478
IL	48,1	854255	75810	3147	2383	574	1678	514
PH	69,4	486930	30604	2162	1440	264	988	360
TR	93,0	1330201	216587	7058	6148	1556	3727	1243
DJ	115,9	1428540	234739	7422	6165	1731	4203	1952
GJ	41,3	344164	22232	2391	1925	214	837	304
MH	52,4	638239	75818	3592	3022	766	1730	622
OT	83,4	987795	145031	6041	5251	775	3509	1228
VL	60,1	323026	14770	2331	2569	96	651	229
AR	49,4	1148139	118651	7952	6132	1948	3464	1222
CS	43,0	379006	24208	5758	4099	566	869	396
HD	42,9	192911	17580	3969	2885	441	905	361
TM	76,0	1670001	142987	10260	8192	2578	5333	1861

Appendix 2—Dataset of agriculture at the regional level in year 2004
(source: Romanian Statistical Yearbook, 2005) (II)

PRE04	INS04	FUN04	ERB04	CHIM04	AZ04	FOS04	POS04	NAT04
279	31826	42391	226529	25373	15627	7402	2344	1127269
44	12808	27608	42966	5558	3293	1957	308	752073
134	30217	118985	144079	12631	7431	5155	45	1275402
50	7008	14547	19384	2992	1999	815	178	896211
255	16849	29140	174008	13288	8751	3753	784	503800
41	10272	36563	65767	4689	3486	1026	177	448545
63	13436	33859	77566	6816	4218	2252	346	740681
130	12656	46039	47477	4657	2862	1318	477	457430
170	18081	71243	42568	8469	4644	2566	1259	475351
137	7788	31508	25079	3621	2127	905	589	697423
182	24256	43526	149834	10151	6240	3425	486	893841
119	10005	22891	60493	3734	2786	741	207	752146
26	19039	31896	30991	8727	6399	2261	67	404995
112	13061	18522	23584	6190	3828	2189	173	1020449
77	86602	98484	47818	9678	6046	3042	590	416711
56	7675	14707	38309	6927	4354	2140	433	379700
154	20727	55296	51428	11407	7915	2594	898	837702
42	15157	22750	35280	6612	4211	2050	351	419890
130	54583	54693	166379	9026	6033	2966	27	6290
124	6821	33625	28530	3212	2514	563	135	207840
230	46511	82148	102492	5926	3606	1924	396	33956
147	42002	108035	98057	6431	4628	1580	223	124266
161	20250	34506	34355	3660	2837	777	46	10770
66	26270	389480	33533	4037	3181	811	45	179709
PRE04	INS04	FUN04	ERB04	CHIM04	AZ04	FOS04	POS04	NAT04
100	11721	10646	38320	3712	2416	1104	192	56970
15	140	241	40	32	22	7	3	361
68	41704	55166	94097	11864	8759	3059	46	315110
200	17578	20630	173666	9475	7133	2103	239	3936
110	37074	50534	70198	10926	7486	2561	879	524368
121	22371	24117	114163	10675	9076	1536	63	136160
110	14559	24435	94740	4651	3601	780	270	4805
79	14462	84185	30798	6526	5672	823	31	252470
131	15956	28115	157304	20796	15284	5456	56	302690
174	33100	127906	108298	18734	15736	2873	125	166856
17	4530	6611	12103	5093	4639	426	28	456690
22	1508	9855	55789	4950	4737	206	7	337281
90	31053	35421	77569	21327	17841	3303	183	128871
12	10267	29391	16838	7753	6855	862	36	459658
313	16777	33491	298335	16789	12718	3834	237	381568
43	1421	13323	84379	4640	3898	740	2	435901
50	2895	15559	47087	2841	1961	659	221	396645
537	45304	98807	530758	35408	23281	9529	2598	326036

¹ **Source:** European Climate Assessment & Dataset (ECA&D) site: <http://eca.knmi.nl/>

² through access to the SAS Centre of Excellence, partnership between Academy of Economics Studies Bucharest and SAS Romania

³ All data analyzed in this article were relevant statistically, with a significance level of 95%, and r is the correlation coefficient.

⁴ $\alpha=0,05$.

A GENETIC ALGORITHM BASED RELIABILITY REDUNDANCY OPTIMIZATION FOR INTERVAL VALUED RELIABILITIES OF COMPONENTS¹

L. SAHOO

Department of Mathematics, Raniganj Girls' College, Raniganj-713347, India

E-mail: lxsahoo@gmail.com

A.K. BHUNIA

Department of Mathematics, The University of Burdwan, Burdwan-713104, India

E-mail: math_akbhunia@buruniv.ac.in

D. ROY

Centre for Management Studies, The University of Burdwan, Burdwan-713104, India

E-mail: dr.diliproy@gmail.com

Abstract: The goal of this paper is to solve the constrained redundancy allocation problem of series-parallel/parallel-series/complex system with the interval valued reliability of each component. For maximizing the overall system reliability under limited resource constraints, the problem is formulated as an unconstrained integer programming problem with interval coefficients by two different penalty techniques. To solve the transformed problem, we have developed two different real coded GAs for integer variables with tournament selection, uniform crossover, uniform mutation and different fitness functions based on penalty techniques. As a special case, considering the lower and upper bounds of the initial valued reliabilities of the component as the same, the corresponding problem has been solved. To illustrate the model, some numerical examples have been solved by our developed GAs and the results have been compared. Finally, to study the stability of our developed GAs with respect to the different GA parameters (like, population size, crossover and mutation rates), sensitivity analyses have been shown graphically.

Key words: Redundancy allocation; Complex system; Reliability optimization; Genetic algorithm; Interval numbers; Order relations; Penalty technique

1. Introduction

While advanced technologies have raised the world to an unprecedented level of productivity, our modern society has become more delicate and vulnerable due to the increasing dependence on modern technological system that often require complicated operations and highly sophisticated management. From any respect, the system reliability is a crucial measure to be considered in system operation and risk management. When designing a highly reliable system, there arises an important question how to obtain a balance between the reliability and other resources e.g., cost, volume and weight. As a result, addition of redundant components or increase of component reliability leads to the increase of the system reliability. In the last few decades, optimization problems including redundancy allocation have been treated by many researchers as integer nonlinear

programming problems with one or several resource constraints [1]. During the last two decades, numerous reliability design techniques have been introduced. These techniques can be classified as implicit enumeration, dynamic programming, branch and bound technique, linear programming, Lagrangian multiplier method, heuristic methods and so on. To solve this type of problem, one may refer to the works of Ghare and Taylor[2], Tillman et al.[3], Nakagawa et al.[4], Tzafestas [5], Shantikumar and Yao[6], Misra and Sharama[7], Sundarajan[8], Chern[9], Ohtagaki et al. [10], Sun and Li[11], Ha and Kuo[12], Gen and Yun[13]. Tillman and Kuo et al.[14] have extensively reviewed the several optimization techniques for system reliability design in their books. In those works, the reliabilities of the system components are assumed to be known at a fixed positive level which lies in the open interval $(0,1)$. However, in real life situations, the reliabilities of these individual components may vary due to different reasons. Any technology cannot produce different components with exactly identical reliabilities. Moreover, the human factor, improper storage facilities and other environmental factors may affect the reliabilities of the individual components. Therefore it is sensible to treat the component reliabilities as positive imprecise numbers in open interval $(0,1)$, instead of fixed real numbers. To tackle the problem with imprecise numbers, generally stochastic, fuzzy and fuzzy-stochastic approaches are used. In stochastic approach, the parameters are assumed as random variables with known probability distribution. In fuzzy approach, the parameters, constraints and goals are considered as fuzzy sets with known membership functions or fuzzy numbers. On the other hand, in fuzzy-stochastic approach, some parameters are viewed as fuzzy sets and other as random variables. However, for a decision maker to specify the appropriate membership function for fuzzy approach and probability distribution for stochastic approach and both for fuzzy-stochastic approach is a formidable task. To overcome these difficulties for representation of imprecise numbers by different approaches, one may represent the same by interval number as it is the best representation among others. Due to this new representation, the objective function of the reduced redundancy allocation problem is interval valued, which is to be maximized under given constraints.

In this study, we have considered GA-based approaches for solving reliability optimization problems with interval objective. As objective function of the redundancy allocation problem is interval valued, to solve this type of problem by GA method, order relations of interval numbers are essential for selection/ reproduction operation as well as for finding the best chromosome in each generation. Recently, Mahato and Bhunia[15] proposed the modified definitions of order relations with respect to optimistic and pessimistic decision maker's point of view for maximization and minimization problems.

In this paper, we have considered the problem of constrained redundancy allocation in series system, hierarchical series-parallel system and complex or non-parallel-series system with interval valued reliability components. The problem is formulated as a non-linear constrained integer programming problem with interval coefficients for maximizing the overall system reliability under resource constraints. During the last few years, several techniques were proposed for solving constraints optimization problem with fixed coefficient with the help of GAs [16-19]. Among them, penalty function techniques are very popular in solving the same by GAs [13, 16, 19]. This method transforms the constrained optimization problem to an unconstrained optimization problem by penalizing the objective function corresponding to the infeasible solution. In this work, to solve the constrained optimization problem we have converted it into unconstrained one by two different type of penalty techniques and the resulting objective function would be interval valued. So, to solve this problem we have developed two different GAs for integer variables with same GA operators like tournament selection, uniform crossover, uniform mutation and elitism of size one but different fitness function depending on different penalty approaches. These methods have been illustrated with some numerical examples and to test the performance of these methods, results have also been compared. As a special case considering the lower and upper bounds of interval valued reliabilities of components as same, the resulting problem becomes identical with the existing problem available in the literature. This type of redundancy allocation problem with fixed valued of reliabilities of

components have been solved and illustrated with some existing numerical examples. Finally, to study the stability of our developed GAs for interval valued objective with respect to different GA parameters, sensitivity analyses have been performed and shown graphically.

2. Finite interval arithmetic

An interval number A is a closed interval defined by $A = [a_L, a_R] = \{x : a_L \leq x \leq a_R, x \in R\}$, where a_L and a_R are the left and right limits, respectively and R is the set of all real numbers. An interval A can also be expressed in terms of centre and radius as $A = \langle a_C, a_W \rangle = \{x : a_C - a_W \leq x \leq a_C + a_W, x \in R\}$, where a_C and a_W are, respectively, the centre and radius of the interval A i.e., $a_C = (a_L + a_R)/2$ and $a_W = (a_R - a_L)/2$. Actually, each real number can be regarded as an interval, such as for all $x \in R$, x can be written as an interval $[x, x]$ which has zero radius. The basic arithmetical operations like, addition, subtraction, multiplication, division and integral power of interval number are available in the book of interval analysis [20].

3. Order relations of interval numbers

In this paper, as we have considered some parameters as interval valued, to find the optimal solution of the optimization problem so we have to discuss the order relations of interval numbers for maximization problems. Let A and B be two closed intervals. These two intervals may be one of the following three types.

Type1. Two intervals A and B are disjoint.

Type2. Intervals are partially overlapping.

Type3. Either $A \subset B$ or $B \subset A$

In this case, we shall consider the definitions of order relations for maximization problems developed recently by Mahato and Bhunia [15] in the context of optimistic and pessimistic decision maker's point of view.

4. Assumptions and Notations

To develop the paper, the following assumptions and notations have been considered.

4.1 Assumptions

1. The component reliabilities are imprecise and interval valued.
2. The failure of any component is independent of that of the other components.
3. All redundancy is active redundancy without repair.

4.2 Notations

The following notations have been used in the entire paper.

m	Number of resource constraints
n	Number of stages of the system
x	(x_1, x_2, \dots, x_n)
j	Index for stage
x_j	Number of redundant components at stage j
l_j	Lower limit on x_j
u_j	Upper limit on x_j
r_j	Reliability of component at stage j which is interval valued

r_{jL}	Lower limit of r_j
r_{jR}	Upper limit of r_j
b_i	Total amount of i -th resource available
$R_j(x_i)$	$1 - (1 - r_j)^{x_i}$, the reliability of stage j
$R_{jL}(x)$	Lower bound of $R_j(x)$
$R_{jR}(x)$	Upper bound of $R_j(x)$
Q_j	$1 - R_j$
$R_S = [R_{SL}, R_{SR}]$	System reliability which is interval valued

5. Problem formulation

Let us consider a system consisting of n components. Assume that the system and its components have two states, viz. "operating state" and "failure state". To represent the state of the component j ($j = 1, 2, 3, \dots, n$), let us define a binary variable y_j as follows:

$$y_j = \begin{cases} 1, & \text{if the component } j \text{ is in operating state,} \\ 0, & \text{otherwise,} \end{cases}$$

In this case, we get a set of 2^n numbers of binary vectors with n components each. Similarly to represent the state of the system, we define another binary variable ϕ given by

$$\phi = \begin{cases} 1, & \text{if the system is in operating state} \\ 0, & \text{otherwise} \end{cases}$$

Clearly the value of ϕ is 1 for some possibilities of y and 0 for other possibilities. Therefore, in most of the reliability systems, we can explicitly define ϕ as a function of y [say $\phi(y)$]. So that the system state ϕ corresponding to the vector y is given by $\phi = \phi(y)$. The function $\phi(y)$ is called the structure function of the system.

Now we consider a system with structure function $\phi(y)$. A component j is said to be relevant to function $\phi(y)$ either

$$\phi(y_1, \dots, y_{j-1}, 1, y_{j+1}, \dots, y_n) = 1 \quad \forall y_r, r \neq j \text{ and } r = 1, 2, \dots, n$$

$$\text{or, } \phi(y_1, \dots, y_{j-1}, 0, y_{j+1}, \dots, y_n) = 0 \quad \forall y_r, r \neq j \text{ and } r = 1, 2, \dots, n$$

Definition 5.1 A system is called coherent if and only if the structure function $\phi(y)$ satisfies the following conditions:

- (i) $\phi(y)$ is a non-decreasing function for each y_j ,
- (ii) Each component is relevant to $\phi(y)$.

Now, we consider a coherent system consisting of n subsystems and m constraints, in which each subsystem has a number of redundancies and all components, are stochastically independent. The objective of the redundancy allocation problem is to find the number of redundancies in each subsystem in order to maximize the overall system reliability subject to the given constraints. The corresponding problem formulated as an Integer Non-Linear Programming (INLP) is as follows:

$$\begin{aligned} &\text{Maximize } R_S \\ &\text{subject to the constraints} \\ &g_i(x) \leq b_i, \quad \text{for } i = 1, 2, \dots, m \end{aligned}$$

$$l_j \leq x_j \leq u_j, \text{ for } j=1,2,\dots,n$$

Now if we consider the reliability of each component as interval valued, the earlier mentioned problem is transformed to the non-linear constrained integer programming problem with interval objective as follows:

$$\text{Maximize } [R_{SL}, R_{SR}]$$

subject to the constraints

$$g_i(x) \leq b_i, \text{ for } i=1,2,\dots,m$$

$$l_j \leq x_j \leq u_j, \text{ for } j=1,2,\dots,n$$

Now, we shall discuss the different types of redundancy allocation problems.

5.2 Constrained redundancy optimization problem for a series system

It is well known that a series system (ref. Fig.1) with n independent components must be operating only if all the components are functioning. In order to improve the overall reliability of the system; one can use more reliable components. However, the expenditure and more often the technological limits may prohibit an adoption of this strategy. An alternative technique is to add redundant components as shown in Fig. 2. The goal of the problem is to determine an optimal redundancy allocation so as to maximize the overall system reliability under limited resource constraints. These constraints may arise out of the size, cost and quantities of the resources. Mathematically, the constrained redundancy optimization problem for such a system for fixed values of reliability can be formulated as follows:

IN
OUT

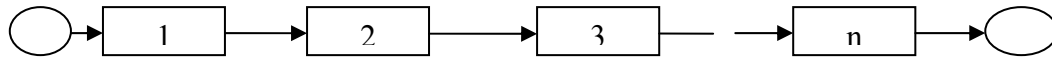


Figure 1

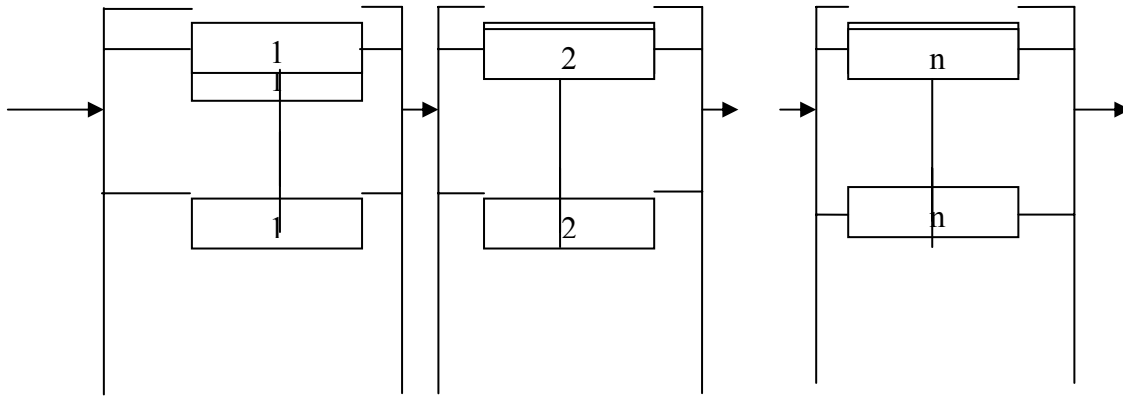


Figure 2

$$\text{Maximize } R_S = \prod_{j=1}^n [1 - (1 - r_j)^{x_j}]$$

$$\text{subject to } g_i(x) \leq b_i, \quad i=1,2,\dots,m$$

$$\text{and } l_j \leq x_j \leq u_j, \text{ for } j=1,2,\dots,n$$

$$\text{where } r_j \in (0,1)$$

Now if we consider component reliabilities as interval valued, i.e., $r_j = [r_{jL}, r_{jR}]$ then the above problem reduces to

$$\text{Maximize } [R_{SL}, R_{SR}] = \prod_{j=1}^n [\{1 - (1 - r_{jL})^{x_j}\}, \{1 - (1 - r_{jR})^{x_j}\}]$$

$$\text{subject to } g_i(x) \leq b_i, \quad i = 1, 2, \dots, m$$

$$\text{and } l_j \leq x_j \leq u_j, \text{ for } j = 1, 2, \dots, n$$

This is an INLP with interval valued objective.

5. 3. Hierarchical series-parallel system

A reliability system is called a hierarchical series parallel system (HSP) if the system can be viewed as a set of subsystems arranged in a series parallel; each subsystem has a similar configuration; subsystems of each subsystem have a similar configuration and so on. For example let us consider a HSP system ($n=10, m=2$) shown in the Fig. 3. This system has a nonlinear and non separable structure and consists of nested parallel and series system. The system reliability of HSP is given by

$$R_S = \{1 - \langle 1 - [1 - Q_3(1 - R_1R_2)]R_4 \rangle (1 - R_5R_6) \} (1 - Q_7Q_8Q_9)R_{10}$$

Mathematically, the constrained redundancy optimization problem for this system for fixed values of reliability can be formulated as follows:

$$\text{Maximize } R_S = \{1 - \langle 1 - [1 - Q_3(1 - R_1R_2)]R_4 \rangle (1 - R_5R_6) \} (1 - Q_7Q_8Q_9)R_{10}$$

$$\text{subject to } g_i(x) \leq b_i, \quad i = 1, 2, \dots, m$$

$$\text{and } l_j \leq x_j \leq u_j, \text{ for } j = 1, 2, \dots, n$$

$$\text{where } r_j \in (0, 1)$$

Now if we consider the component reliabilities as interval valued, i.e., $r_j = [r_{jL}, r_{jR}]$ then the above problem reduces to the following form:

$$\text{Maximize } [R_{SL}, R_{SR}] = \{1 - \langle 1 - (1 - [Q_{3L}, Q_{3R}](1 - [R_{1L}, R_{1R}][R_{2L}, R_{2R}]))[R_{4L}, R_{4R}] \rangle (1 - [R_{5L}, R_{5R}][R_{6L}, R_{6R}]) \} (1 - [Q_{7L}, Q_{7R}][Q_{8L}, Q_{8R}][Q_{9L}, Q_{9R}])[R_{10L}, R_{10R}]$$

$$\text{subject to } g_i(x) \leq b_i, \quad i = 1, 2, \dots, m$$

$$\text{and } l_j \leq x_j \leq u_j, \text{ for } j = 1, 2, \dots, n$$

This is an INLP with interval valued objective.

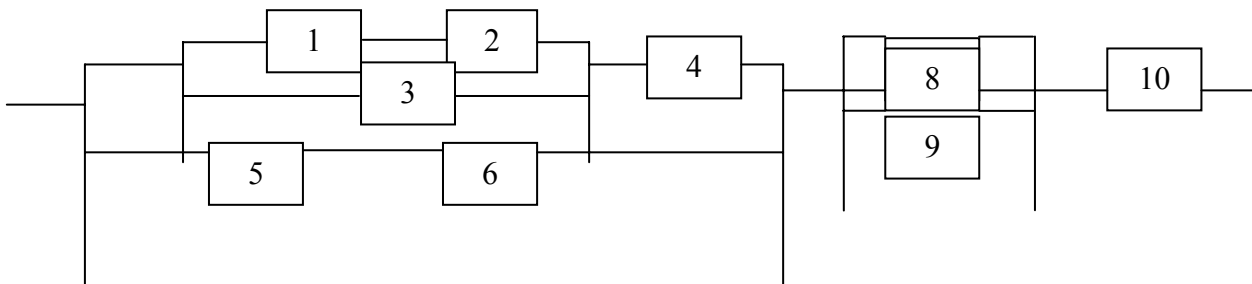


Figure 3

5.4 Complex System

When a reliability system can be reduced to series and parallel configurations, there exist combinations of components which are connected neither in a series nor in parallel. Such systems are called complex or non parallel series systems. This system is also called the bridge system.

For example, let us consider a bridge system ($n=5, m=3$) shown in Fig.4. This system consists of five subsystems and three nonlinear and non-separable constraints. The overall system reliability R_S is given by the expression as follows:

$$R_S = R_5(1 - Q_1 Q_3)(1 - Q_2 Q_4) + Q_5[1 - (1 - R_1 R_2)(1 - R_3 R_4)],$$

where $R_j = R_j(x_j)$ and $Q_j = 1 - R_j$ for all $j = 1, \dots, 5$.

Mathematically, the constrained redundancy optimization problem for such complex system for fixed values of reliability can be formulated as follows:

$$\text{Maximize } R_S = R_5(1-Q_1Q_3)(1-Q_2Q_4) + Q_5[1-(1-R_1R_2)(1-R_3R_4)]$$

$$\text{subject to } g_i(x) \leq b_i, \quad i=1,2,\dots,m$$

and $l_j \leq x_j \leq u_j$, for $j = 1, 2, \dots, n$

where $r_i \in (0, 1)$

Since the overall system has a complex structure, the objective function is also non-linear and non-separable.

Now if we consider the component reliabilities as interval valued, i.e., $r_i = [r_{iL}, r_{iR}]$ then the above problem reduces to the following form as follows:

$$\begin{aligned} \text{Maximize}[R_{SL}, R_{SR}] = & [R_{5L}, R_{5R}](1 - [Q_{1L}, Q_{1R}][Q_{3L}, Q_{3R}](1 - [Q_{2L}, Q_{2R}][Q_{4L}, Q_{4R}]) \\ & + [Q_{5L}, Q_{5R}]\{1 - (1 - [R_{1L}, R_{1R}][R_{2L}, R_{2R}]) (1 - [R_{3L}, R_{3R}][R_{4L}, R_{4R}])\} \end{aligned}$$

subject to $g_i(x) \leq b_i, \quad i=1,2,\dots,m$

and $l_j \leq x_j \leq u_j$, for $j = 1, 2, \dots, n$

This is an INLP with interval valued objective.

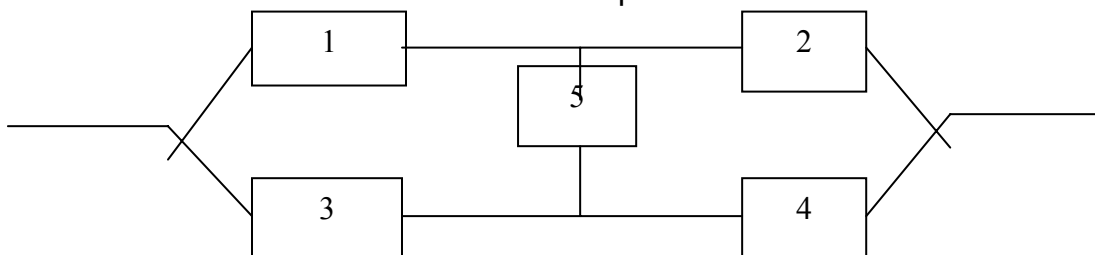


Figure 4

5.5 k -out-of- n System

A k -out-of- n system is an n -component system which functions when at least k of its n components function. This redundant system is sometimes used in the place of pure parallel system. It is also referred to as k -out-of- $n:G$ system. An n -component series system is a n -out-of- $n:G$ system whereas a parallel system with n -components is a 1 -out-of- $n:G$ system. When all of the components are independent

and identical, the reliability of k -out-of- n system can be written

as $R_S = \sum_{j=k}^n \binom{n}{j} r^j (1-r)^{n-j}$, where r is the component reliability.

6. GA based constraint handling technique

In the application of Genetic Algorithm for solving the reliability optimization problems with interval objective [21-23], there arises an important question for handling the constraints relating to the problem. During the past, several approaches viz. penalty-based method, methods that preserve the feasibility of solutions, methods that clearly distinguish between feasible and unfeasible solutions and hybrid methods have been proposed to handle the constraints in evolutionary algorithms [16-19] for solving the problems with fixed objective [24]. Among these approaches, penalty-based technique is very well known and widely applicable. In this approach, constraints are added /subtracted to the objective function in different ways. To avoid the infeasible solution, the fitness function is increased /decreased with a penalty term multiplied by a so called penalty coefficient. However, there arises a difficulty to select the initial value and updating strategy for the penalty coefficient. To avoid this situation, Deb [24] proposed a GA based penalty technique called Parameter Free Penalty (PFP) technique replacing the objective function value of GA by worst objective value of feasible solution of previous generation and without multiplying the penalty coefficient. This means that the fitness function values of infeasible solution do not depend on the objective function value. As the mentioned constrained optimization problem is of interval valued, therefore the problem to be solved with a new fitness function is of the following interval form:

Maximize

$$[\hat{R}_{SL}(x), \hat{R}_{SR}(x)] = [R_{SL}(x), R_{SR}(x)] - \left[\sum_{i=1}^m \max\{0, g_i(x) - b_i\}, \sum_{i=1}^m \max\{0, g_i(x) - b_i\} \right] + \theta(x) \quad (1)$$

$$\text{where } \theta(x) = \begin{cases} [0, 0] & \text{if } x \in S \\ -[R_{SL}(x), R_{SR}(x)] + \min[R_{SL}, R_{SR}] & \text{if } x \notin S \end{cases}$$

$$\text{and } S = \{x : g_i(x) \leq b_i, i = 1, 2, \dots, m \text{ and } l \leq x \leq u\}$$

Here the parameter $\min[R_{SL}, R_{SR}]$ is the value of interval valued objective function of the worst feasible solution in the population. Alternatively, the problem may be solved with another fitness function by penalizing a large positive number (say M which can be written in interval form as $[M, M]$) [25]. We denote this penalty as Big-M penalty and its form is as follows:

$$\text{Maximize } [\hat{R}_{SL}(x), \hat{R}_{SR}(x)] = [R_{SL}(x), R_{SR}(x)] + \theta(x)$$

(2)

$$\text{where } \theta(x) = \begin{cases} [0, 0] & \text{if } x \in S \\ -[R_{SL}(x), R_{SR}(x)] + [-M, -M] & \text{if } x \notin S \end{cases}$$

$$\text{and } S = \{x : g_i(x) \leq b_i, i = 1, 2, \dots, m \text{ and } l \leq x \leq u\}$$

The above problems (1) and (2) are non-linear unconstrained integer programming problem with interval coefficient.

7. Solution procedure

Now we have to solve the above nonlinear maximization problems (1) and (2) with the help of GA. For this purpose, we have developed two types of GA with different fitness function based on PFP and Big-M penalty techniques. We call these GAs as PFP-GA and Big-M-GA respectively.

The different steps of this algorithm are described as follows:

7.1 Algorithm

Step-1: Initialize the parameters of genetic algorithm, bounds of variables and different parameters of the problem.

Step-2: $t = 0$ [t represents the number of current generation].

Step-3: Initialize the chromosome of the population $P(t)$ [$P(t)$ represents the population at t -th generation].

Step-4: Evaluate the fitness function of each chromosome of $P(t)$ considering the objective function of either (1) or (2) as fitness function.

Step-5: Find the best chromosome from the population $P(t)$.

Step-6: t is increased by unity.

Step-7: If the termination criterion is satisfied go to step-14, otherwise, go to next step.

Step-8: Select the population $P(t)$ from the population $P(t-1)$ of earlier generation by tournament selection process.

Step-9: Alter the population $P(t)$ by crossover, mutation and elitism process.

Step-10: Evaluate the fitness function value of each chromosome of $P(t)$.

Step-11: Find the best chromosome from $P(t)$.

Step-12: Compare the best chromosome of $P(t)$ and $P(t-1)$ and store better one.

Step-13: Go to step-6

Step-14: Print the last found best chromosome (which is the solution of the optimization problem).

Step-15: End.

For implementing the above GA in solving the problems (1) and (2) the following basic components are to be considered.

- GA Parameters
- Chromosome representation
- Initialization of population
- Evaluation of fitness function
- Selection process
- Genetic operators (crossover, mutation and elitism)
- Termination criteria

7.2 GA Parameters

There are different parameters used in the genetic algorithm, viz. population size (p_size), maximum number of generations (m_gen), crossover rate/probability of crossover (p_cross) and mutation rate/probability of mutation (p_mute). There is no hard and fast rule for choosing the population size for GA. However, if the population size is considered to be large, storing of the data in the intermediate steps of GA may create some difficulties at the time of computation with the help of computer. On the other hand, for very small population size, some genetic operations can not be implemented. Particularly, mutation operator does not work properly as the mutation rate is very low. Regarding the maximum number of generations, there is no indication for considering this value. Generally, it is problem

dependent on problem to problem. Particularly, it depends upon the number of genes (variables) of a chromosome (solution) in artificial genetics. Again, from the natural genetics, it is obvious that the crossover rate is always greater than that of mutation rate. Usually, the crossover rate varies from 0.8 to 0.95 whereas the mutation rate varies from 0.05 to 0.2. Sometimes, it is considered as $1/n$ where n be the number of genes (variables) of the chromosomes (solutions).

7.3 Chromosome representation and initialization

In the applications of GA, the appropriate chromosome (individual) representation of solution for the given problems is an important task to the users. There are different types of representations, viz. binary, real, octal, hexadecimal coding, available in the existing literature. Among these representations, real coding representation is very popular as this type of chromosome representation looks like a vector. In this representation, each component (gene) of the chromosome is the values of decision variables of the optimization problems which are to be solved by GA.

7.4 Initialization of population

After the selection of chromosome representation, the next step is to initialize the chromosomes that will take part in the artificial genetic operations like natural genetics. This procedure produces population size number of chromosomes in which every component for each chromosome is randomly generated within the bounds of the corresponding decision variable. There are different procedures for selecting a random number for each component of the chromosomes. Here for each component of the chromosome, a random value is selected from the discrete set of values within its bounds.

7.5 Evaluation of fitness function

After getting a population of potential solutions, we need to check how good they are. So we have to calculate the fitness value for each chromosome. In this evaluation, the value of objective function corresponding to the chromosome is taken as the fitness value of that chromosome. Here, the transformed unconstrained objective function due to different penalty technique is considered as the fitness function.

7.6 Selection

In artificial genetics, the selection operator plays an important role. Usually, it is the first operator applied to the population. The primary objective of this operator is to emphasize on the above average solutions and eliminate below average solutions from the population for the next generation under the well-known evolutionary principle "survival of the fittest". In this work, we have used tournament selection scheme of size two with replacement as the selection operator. This operator selects the better chromosome/individual from randomly selected two chromosomes/individuals. This selection procedure is based on the following assumptions:

- (i) When both the chromosomes / individuals are feasible then the one with better fitness value is selected.
- (ii) When one chromosome/individual is feasible and another is infeasible then the feasible one is selected.
- (iii) When both the chromosomes/individuals are infeasible with unequal constraint violations, then the chromosome with less constraint violation is selected.
- (iv) When both the chromosomes/individuals are infeasible with equal constraint violations, then any one chromosome/individual is selected.

7.7 Crossover

After the selection process, other genetic operators like crossover and mutation are applied to the resulting chromosomes (those which have survived). Crossover is an operation that really empowers the GA. It operates on two or more parent solutions at a time and

generates offspring by recombining the feature of the parent solutions. In this operation, expected $\lceil p_cross * p_size \rceil$ (* denotes the product and $\lceil \rceil$ denotes the integral value) number of chromosomes will take part. Hence in order to perform the crossover operation, select $\lceil p_cross * p_size \rceil$ numbers of chromosomes are selected. In our work, the operation is done in the following manner.

Step-1: Find the integral value of $p_cross * p_size$ and store it in N .

Step-2: Select the chromosomes v_k and v_i randomly from the population for crossover.

Step-3: The components v'_{kj} and v'_{ij} ($j=1,2,\dots,n$) of two offspring will be created by either $v'_{kj} = v_{kj} - g$ and $v'_{ij} = v_{ij} + g$ if $v_{kj} > v_{ij}$

Or, $v'_{kj} = v_{kj} + g$ and $v'_{ij} = v_{ij} - g$, where g is a random integer number between 0 and $|v_{kj} - v_{ij}|$.

Step-4: Repeat step-2 and step-3 for $\frac{N}{2}$ times.

7.8 Mutation

The aim of mutation operation is to introduce the random variations into the population. Sometimes, it helps to regain the information lost in earlier generations. Mainly, this operator is responsible for fine tuning capabilities of the system. This operator is applied to a single chromosome only. Usually, its rate is very low; otherwise it would defeat the order building generated through selection and crossover operations. Mutation attempts to bump the population gently into a slightly better way, i.e., the mutation changes single or all the genes of a randomly selected chromosome slightly. In this work, we have used uniform mutation. If the gene v_{ik} of chromosome v_i is selected for mutation and domain of v_{ik} is $[l_{ik}, u_{ik}]$, then the reduced value of v_{ik} is given by

$$v'_{ik} = \begin{cases} v_{ik} + \Delta(u_{ik} - v_{ik}), & \text{if random digit is 0.} \\ v_{ik} - \Delta(v_{ik} - l_{ik}), & \text{if random digit is 1.} \end{cases}$$

where $k \in \{1, 2, \dots, n\}$ and $\Delta(y)$ returns a value in the range $[0, y]$.

In our work, we have taken

$\Delta(y) =$ A random integer between $[0, y]$.

7.9 Elitism

Sometimes, in any generation, there is a chance that the best chromosome may be lost when a new population is created by crossover and mutation operations. To remove this situation the worst individual/chromosome is replaced by that best individual/chromosome in the current generation. Instead of single chromosome one or more chromosomes may take part in this operation. This process is called as elitism.

7.10 Termination criteria

The termination condition is to stop the algorithm when either of the following three conditions is satisfied:

- (i) the best individual does not improve over specified generations.
- (ii) the total improvement of the last certain number of best solutions is less than a pre-assigned small positive number or
- (iii) The number of generations reaches maximum number of generation i.e., max_gen .

In this work we have used first condition and we take 10 generations as a specified generation.

8. Numerical Examples

To illustrate the proposed GAs (viz. PFP-GA and Big-M-GA) for solving constrained redundancy allocation problems with interval valued reliabilities of components, we have solved four numerical examples. It is to be noted that for solving the said problem with fixed valued reliabilities of components, the reliability of each component are taken as interval with the same lower and upper bounds of interval. In first three examples, the reliabilities of the components are interval valued whereas in the last example (taken from Ha and Kuo [12]), it is fixed. For each example, 20 independent runs have been performed by both the GAs, of which the following measurements have been collected to compare the performance of PFP-GA and Big-M-GA.

- (i) Best found system reliability
- (ii) Average generations
- (iii) Average CPU times

The proposed Genetic Algorithms are coded in C programming language and run in Linux environment. The computation work has been done on the PC which has Intel core-2 duo processor with 2 GHz. In this computation, different population size has been taken for different problems. However, the crossover and mutation rates are taken as 0.95 and 0.15 respectively.

Example-8.1 (Ref. section 5.2)

$$\text{Maximize } [R_{SL}, R_{SR}] = \prod_{j=1}^5 [\{1 - (1 - r_{jL})^{x_j}\}, \{1 - (1 - r_{jR})^{x_j}\}]$$

subject to

$$\sum_{j=1}^5 p_j x_j^2 - P \leq 0,$$

$$\sum_{j=1}^5 c_j [x_j + \exp(\frac{x_j}{4}) - C \leq 0,$$

$$\sum_{j=1}^5 w_j x_j \exp(\frac{x_j}{4}) - W \leq 0,$$

Table 1: Parameter used in Example -8.1

i	r_i	p_i	P	c_i	C	w_i	W
1	[0.76,0.83]	1	110	7	175	7	200
2	[0.82,0.87]	2		7		8	
3	[0.88,0.93]	3		5		8	
4	[0.61,0.67]	4		9		6	
5	[0.70,0.80]	2		4		9	

Example-8.2 (Ref. section 5.3)

$$\text{Maximize } [R_{SL}, R_{SR}] = \{1 - (1 - (1 - [Q_{3L}, Q_{3R}](1 - [R_{1L}, R_{1R}][R_{2L}, R_{2R}]))[R_{4L}, R_{4R}](1 - [R_{5L}, R_{5R}][R_{6L}, R_{6R}]))\} \\ (1 - [Q_{7L}, Q_{7R}][Q_{8L}, Q_{8R}][Q_{9L}, Q_{9R}][R_{10L}, R_{10R}])$$

subject to

$$c_1 \exp\left(\frac{x_1}{2}\right)x_2 + c_2 \exp\left(\frac{x_3}{2}\right) + c_3x_4 + c_4[x_5 + \exp\left(\frac{x_5}{4}\right)] + c_5x_6^2x_7 + c_6x_8 + c_7x_9^3 \exp\left(\frac{x_{10}}{2}\right) - 120 \leq 0,$$

$$w_1x_1^2x_2 + w_2 \exp\left(\frac{x_3x_4}{2}\right) + w_3x_5 \exp\left(\frac{x_6}{4}\right) + w_4x_7x_8^3 + w_5[x_9 + \exp\left(\frac{x_9}{2}\right)] + w_6x_{10} \exp\left(\frac{x_{10}}{4}\right) - 130 \leq 0,$$

Table 2: Parameter used in Example-8.2

j	1	2	3	4	5	6	7	8	9	10
r_j	[.80,.84]	[.87,.90]	[.89,.93]	[.84,.86]	[.88,.90]	[.9,.95]	[.8,.85]	[.91,.95]	[.8,.83]	[.88,.92]
c_j	8	4	2	2	1	6	2	8	-	-
w_j	16	6	7	12	7	1	9	-	-	-
l_j	1	1	1	1	1	1	1	1	1	1
u_j	4	5	6	7	5	5	3	3	4	6

Example-8.3 (Ref. section 5.4)

$$\text{Maximize } [R_{SL}, R_{SR}] = [R_{5L}, R_{5R}](1 - [Q_{1L}, Q_{1R}][Q_{3L}, Q_{3R}](1 - [Q_{2L}, Q_{2R}][Q_{4L}, Q_{4R}]) \\ + [Q_{5L}, Q_{5R}]\{1 - (1 - [R_{1L}, R_{1R}][R_{2L}, R_{2R}](1 - [R_{3L}, R_{3R}][R_{4L}, R_{4R}])\}$$

subject to

$$10 \exp\left(\frac{x_1}{2}\right)x_2 + 20x_3 + 3x_4^2 + 8x_5 - 200 \leq 0,$$

$$10 \exp\left(\frac{x_1}{2}\right) + 4 \exp(x_2) + 2x_3^3 + 6[x_4^2 + \exp\left(\frac{x_4}{4}\right)] + 7 \exp\left(\frac{x_5}{4}\right) - 310 \leq 0,$$

$$12[x_2^2 + \exp(x_2)] + 5x_3 \exp\left(\frac{x_3}{4}\right) + 3x_1x_4^2 + 2x_5^3 - 520 \leq 0,$$

$$(1, 1, 1, 1, 1) \leq (x_1, x_2, x_3, x_4, x_5) \leq (6, 3, 5, 6, 6),$$

where

$$R_1(x_1) = \{[0.78, 0.82], [0.83, 0.88], [0.89, 0.91], [0.915, 0.935], [0.94, 0.96], [0.965, 0.985]\};$$

$$R_2(x_2) = 1 - (1 - [0.73, 0.77])^{x_2};$$

$$R_3(x_3) = \sum_{k=2}^{x_3+1} \binom{x_3+1}{k} ([0.87, 0.89])^k ([0.11, 0.13])^{x_3+1-k};$$

$$R_4(x_4) = 1 - (1 - [0.68, 0.72])^{x_4};$$

$$R_5(x_5) = 1 - (1 - [0.83, 0.86])^{x_5};$$

Example-8.4 (Ref. section 5.4)

$$\text{Maximize}[R_{SL}, R_{SR}] = [R_{5L}, R_{5R}](1 - [Q_{1L}, Q_{1R}][Q_{3L}, Q_{3R}](1 - [Q_{2L}, Q_{2R}][Q_{4L}, Q_{4R}]) \\ + [Q_{5L}, Q_{5R}]\{1 - (1 - [R_{1L}, R_{1R}][R_{2L}, R_{2R}](1 - [R_{3L}, R_{3R}][R_{4L}, R_{4R}])\}$$

subject to

$$10 \exp\left(\frac{x_1}{2}\right)x_2 + 20x_3 + 3x_4^2 + 8x_5 - 200 \leq 0,$$

$$10 \exp\left(\frac{x_1}{2}\right) + 4 \exp(x_2) + 2x_3^3 + 6[x_4^2 + \exp\left(\frac{x_4}{4}\right)] + 7 \exp\left(\frac{x_5}{4}\right) - 310 \leq 0,$$

$$12[x_2^2 + \exp(x_2)] + 5x_3 \exp\left(\frac{x_3}{4}\right) + 3x_1x_4^2 + 2x_5^3 - 520 \leq 0,$$

$$(1, 1, 1, 1, 1) \leq (x_1, x_2, x_3, x_4, x_5) \leq (6, 3, 5, 6, 6), \quad x \in \mathbb{R}_n^+$$

where

$$R_1(x_1) = \{[.8, .8], [.85, .85], [.9, .9], [.925, .925], [.95, .95], [.975, .975]\};$$

$$R_2(x_2) = 1 - (1 - [0.75, 0.75])^{x_2};$$

$$R_3(x_3) = \sum_{k=2}^{x_3+1} \binom{x_3+1}{k} ([0.88, 0.88])^k ([0.12, 0.12])^{x_3+1-k};$$

$$R_4(x_4) = 1 - (1 - [0.7, 0.7])^{x_4};$$

$$R_5(x_5) = 1 - (1 - [0.85, 0.85])^{x_5};$$

Table 3: Numerical results for Example 8.1-8.4

Method	Example	x	Best found system reliability R_s	Average CPU time (sec.)	Average Generation	Population Size
PFP-GA	8.1	(3,2,2,3,3)	[0.860808,0.930985]	0.0001	12.10	50
	8.2	(1,2,2,5,4,4,2,2,1,5)	[0.999909,0.999987]	0.0105	17.55	100
	8.3	(5,1,2,4,4)	[0.991225,0.999872]	0.0200	11.20	200
	8.4	(3,2,4,4,2)	[0.999382,0.999382]	0.0100	12.40	100
Big-M-GA	8.1	(3,2,2,3,3)	[0.860808,0.930985]	0.0001	12.80	50
	8.2	(1,2,2,5,4,4,2,2,1,5)	[0.999909,0.999987]	0.0110	17.75	100
	8.3	(5,1,2,4,4)	[0.991225,0.999872]	0.0200	10.90	200
	8.4	(3,2,4,4,2)	[0.999382,0.999382]	0.0100	12.55	100

9. Sensitivity Analysis

To study the performance of our proposed GAs like PFP-GA and Big-M-GA based on two different types of penalty techniques, sensitivity analyses have been carried out graphically on the centre of the interval valued system reliability with respect to GA parameters like, population size, crossover and mutation rate separately keeping the other parameters at their original values. These are shown in Fig.5-Fig.7.

From Fig.5, it is evident that in case of PFP-GA, smaller population size gives the better system reliability. However, both the GAs are stable when population size exceeds the number 30.

From Fig.6, it is observed that the system reliability is stable if we consider the crossover rate between the interval (0.65, 0.95) in case of PFP-GA. In both GAs, it is stable when crossover rate is greater than 0.8.

In Fig.7, sensitivity analyses have been done with respect to mutation rate. In both GAs, the value of system reliability be the same.

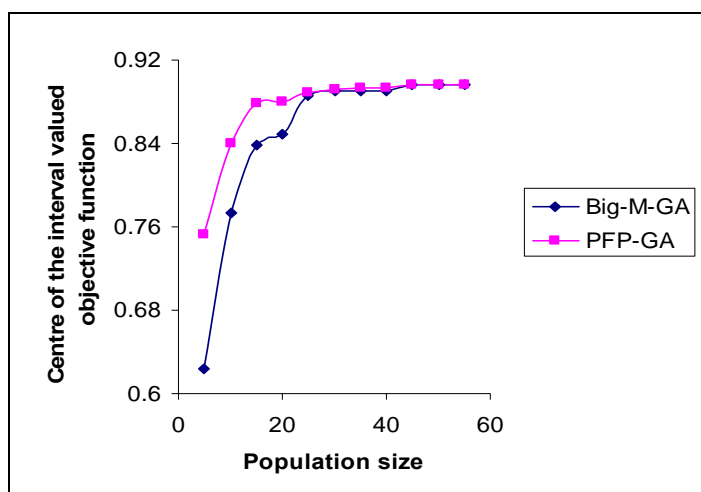


Figure 5. Population size vs. centre of the objective function value

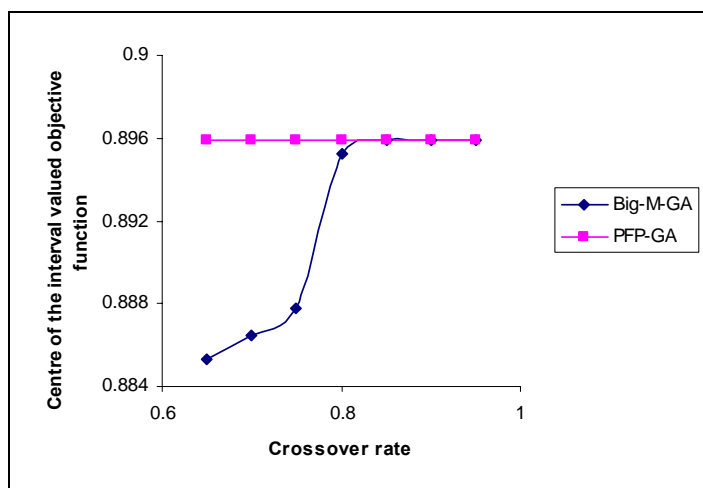


Figure 6. Crossover rate vs. centre of the objective function value

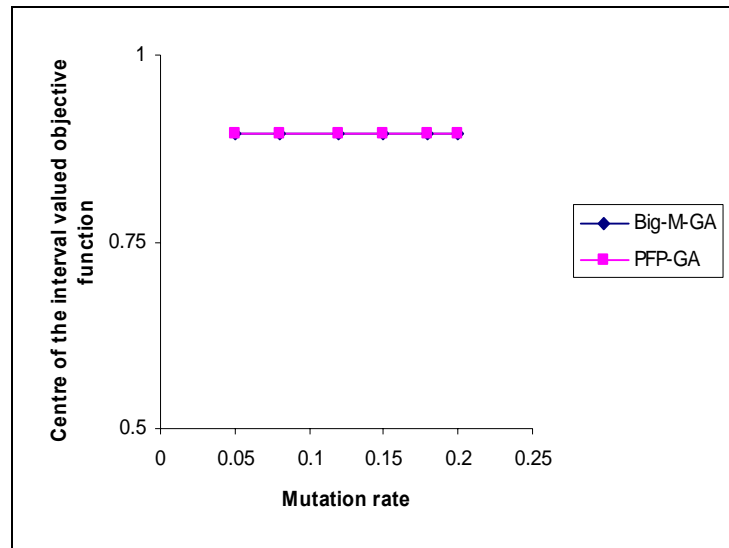


Figure 7. Mutation rate vs. centre of the objective function value

10. Conclusions

In this paper, reliability redundancy allocation problems of series-parallel/parallel-series/complex system with some resource constraints have been solved. In those systems, reliability of each component has been considered as imprecise number and this imprecise number has been represented by an interval number which is more appropriate representation among other representations like, random variable representation with known probability distribution, fuzzy set with known fuzzy membership function or fuzzy number. For handling of resource constraints, the corresponding problem has been converted into unconstrained optimization problem with the help of two different parameter free penalty techniques. Therefore, the transformed problem is of unconstrained interval valued optimization problem with integer variables. To solve the transformed problem, we have developed real coded GA for integer variables with interval valued fitness function, tournament selection, uniform crossover, uniform mutation and elitism of size one. In tournament selection and elitism operation, recently developed definitions of interval ranking have been used. In the existing penalty function technique, tuning of penalty parameter is a formidable task. From the performance of GAs, it is observed that both the GAs with both fitness function due to different penalty techniques take very lesser CPU time with very small generations to solve the problems. It is clear from the expression of the system reliability, that the system reliability is a monotonically increasing function with respect to the individual reliabilities of the components. Therefore, there is one optimum setup irrespective of the choice of the upper bound or lower bound of the component reliabilities. As a result, the optimum setup obtained from the upper bound/lower bound will provide both the upper bound and the lower bound of the optimum system reliability. However, the interval approach presented in this paper has a wider applicability. For future research, one may use the proposed GAs in solving other reliability optimization problems like Chance- constrained reliability optimization problems, Network reliability optimization problems, etc.

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MULTINOMIAL LOGISTIC REGRESSION: USAGE AND APPLICATION IN RISK ANALYSIS

Anass BAYAGA

School of Initial Teacher Education (SITE), Faculty of Education,
University of Fort Hare, South Africa



E-mail: abayaga@ufh.ac.za

Abstract: The objective of the article was to explore the usage of multinomial logistic regression (MLR) in risk analysis. In this regard, performing MLR on risk analysis data corrected for the non-linear nature of binary response and did address the violation of equal variance and normality assumptions. Additionally, use of maximum likelihood (-2log) estimation provided a means of working with binary response data. The relationship of independent and dependent variables was also addressed.

The data used included a cohort of hundred risk analyst of a historically black South African University. In this analysis, the findings revealed that the probability of the model chi-square (17.142) was 0.005, less than the level of significance of 0.05 (i.e. $p < 0.05$). Suggesting that there was a statistically significant relationship between the independent variable-risk planning (Rp) and the dependent variable-control mechanism (control mecs) ($p < 0.05$). Also, there was a statistically significant relationship between key risks assigned (KSA) and time spent on risk mitigation. For each unit increase in confidence in control mecs, the odds of being in the group of survey respondents who thought institution spend too little time on Rp decreased by 74.7%. Moreover, the findings revealed that survey respondents who had less confidence in control mecs were less likely to be in the group of survey respondents who thought institution spent about the right amount of time on risk planning.

Key words: Binary variable; Log odds ratio; Logistic regression model; (log) Likelihood ratio statistic; -2 Log Q; Wald statistic; Model fit; -2 Log L. Quantitative risk analysis

1. Context of study

Modeling of risk processes such as risk awareness, risk identification, monitoring and reporting, planning and mitigation etc is among rather difficult subjects tackled by risk analyst especially in applying multinomial logistic regression in dynamic (social) setting. Invariably though, social science research (Yu, Lai & Wang, 2008; Fan & Xiao, 2006) problems somewhat call for analysis and prediction of a dichotomous¹ outcomes. Traditionally, such research outcomes were addressed by either ordinary least squares (OLS) regression or linear discriminant function analysis (Hosmer & Lemeshow, 2000). However, both techniques, as a result of their nature, depend on strict statistical assumptions, thus, normality of independent variables, linearity of relationships, multicollinearity among independent variables, equal dispersion matrices for discriminant analysis (Tabachnick et al., 2001). These assumptions which are not easily observed in a dynamic setting are part of multiple² regression.

Introduction of multinomial logistic regression was an alternative regression analysis to cater for conditions that do not necessarily obey the assumptions listed above with the exception of multicollinearity (Hosmer & Lemeshow, 2000). In the last decade, the technique, like other univariate and multivariate data analysis methods, started to find a

prominent place in the medicine, engineering and the manufacturing industries. This development led researchers in risk analysis to build more accurate and useful statistical models by applying it in risk analysis (Liebenberg & Hoyt, 2003; Hosmer & Lemeshow, 2000). Meanwhile, despite being in use in general statistical analysis for many years, it has received rather little attention in the risk analysis literature compared to other regression applications regarding modelling of explanatory and response variable (Liebenberg & Hoyt, 2003; Hosmer & Lemeshow, 2000).

Recent studies (Crane, Gibbons, Jolley, Van Belle, 2007; Hedeker, 2003; Menard, 2002; Tabachnick, Fidell & Osterlind, 2001; Harrell, 2001; Hosmer & Lemeshow, 2000) have noted that modelling relationship between explanatory (predictor) and response variables is a fundamental activity encountered in risk analysis. The accounts of these studies suggest that simple linear regression is often used to investigate the relationship between a single predictor variable and a single response (dependent) variable. But, when there are several explanatory variables though, multinomial logistic regression is used.

However, often the response (dependent variable) as some of the authors argued (Menard, 2002; Tabachnick et al., 2001; Harrell, 2001; Hosmer & Lemeshow, 2000) is not a numerical value. Instead, the response is simply a designation of one of two possible outcomes (a binary response); example, alive or dead, success or failure, yes or no. Data involving relationship between explanatory variables and binary responses proliferate in just about every discipline from engineering to the natural sciences, to medicine etc. Invariably though, what remains a matter of concern for many practitioners and theorists of risk analysis in University is the questions of how to model relationship between explanatory variables and a binary response variable (Liebenberg & Hoyt, 2003; Tabachnick et al., 2001; Harrell, 2001; Hosmer & Lemeshow, 2000).

Scholars (Hamilton, 2003; Hendrickx, 2000; McCullagh, 1980) have argued that the difficulty for both practitioners and theorists in modeling of risk processes steams from the social setting within which risk parameters are applied. Additionally, the authors suggested that little scholarly literature has delved into application of multinomial logistic regression in analysis of risks parameters particularly in a University context.

Whiles, there remain little studies being conducted in risk analysis with regards to MLR in the context of a University, many recent studies (Van Gelderen, Thurik & Bosma, 2006; Jalilvand & Switzer, 2005; McNeil, Frey & Embrechts, 2005; Mishra & El-Osta, 2002) have encouraged its usage due to its relevance to the field of risk analysis. Following the above, the paper explored the application of multinomial logistic regression via University-wide risk analysis. It does this by using concepts from simple and multiple linear regressions which are carried over to MLR. Additionally, ideas of maximum likelihood estimation are central to the modelling of the MLR data.

1.1. Multinomial Logistic Regression

The multinomial (polytomous) logistic regression model is a simple extension of the binomial logistic regression model. It is used when the dependent variable has more than two nominal or unordered categories, in which dummy coding³ of independent variables is quite common. In using multinomial logistic regression in risk analysis, the *dependent* (response) variable is dummy coded into multiple 1/0 variables (cf. sections 3 for details). This means that there is a variable for all categories but one, so if there are M categories, there will be M-1 dummy variables. All but one category has its own dummy variable. Each category's dummy variable has a value of 1 for its category and a 0 for all others. One category, the reference category, does not need its own dummy variable, as it is uniquely identified by all the other variables being 0.

With regards to the above, risk analyst using multinomial logistic regression can then estimate a separate binary logistic regression model for each of those dummy variables. The result is M-1 binary logistic regression models. The most significant factor to consider here is that each one tells the effect of the predictors of risk on the probability of success in that category, in comparison to the reference category. Noting though that each model has its own intercept and regression coefficients- the reason being that predictors of risk analysis processes could affect each category differently (cf. sections 2 & 3 for details).

1.2. Why Multinomial Logistic Regression instead of other Techniques?

Most of multivariate analysis techniques require the basic assumptions of normality and continuous data, involving independent and/or dependent variables as aforementioned. This necessity manifests itself also in the application of MLR data collection and measurement steps in risk analysis, but to varying degree. Thus, whereas much stronger interval and ratio scales provide a good basis for a more comprehensive multivariate analysis, commonly used risk measurement scales such as five-point likert, ordinal, and nominal scales are usually considered unsuitable for multivariate analysis techniques, due to various assumption as listed above. For this reason, multinomial logistic regression was used where the above assumptions tend to be violated. This is evident in one main way in MLR analysis. Thus, it has alternative data distribution assumptions, suggesting that it generates more appropriate and correct findings in terms of model fit and correctness of the analysis regardless of any assumption (cf. section 3.1 for details).

A multinomial logistic regression model is a form of regression where the outcome variable (risk factor-dependent variable) is binary or dichotomous and the independents are continuous variables, categorical variables, or both. The application of multinomial logistic regression in risk analysis arises when an analyst analyses relationships between a non-metric dependent variable and metric or dichotomous independent variables. It compares multiple groups of risk processes such as risk mitigation, planning, monitoring identification, through a combination of binary logistic regressions (cf. section 3). The comparisons are equivalent to the comparisons for a dummy-coded dependent variable, with the group with the highest numeric score used as the reference group. Additionally, it provides a set of coefficients for each of the two comparisons. Multinomial logistic regression exists to handle the case of dependents with more classes. This is referred to as the multivariate case.

Thus, it is expected that multinomial logistic regression approach would do better when there is evidence of substantial departures from multivariate normality as is the case where there are some dichotomous or zero/one variables or where distributions are highly skewed or heavy-tailed especially in dynamic settings. In MLR however, hypotheses on significance of explanatory variables cannot be tested in quite the same way as say linear regression. Recall that in linear regression, where the response variables are normally distributed, one can use t- or F-test statistics for testing significance of explanatory variables. But in logistic regression, the response variables are Bernoulli distributed⁴, meaning that a risk analyst has to use different test statistics, which exact distributions are unknown. In this paper though, the researcher would not go into any technical details about test statistics, but focus on interpreting the results of a MLR analysis (cf. methodology, results and discussion of results). To this effect, two different types of test statistics, the (log) likelihood ratio statistic (often referred to as the -2log or deviance) and the Wald statistic would be used. The model is written somewhat differently in some software (cf. SPSS & SAS) than usual mathematical approach. In some software (cf. SAS), the sign is a plus, suggesting that increases in predictor values leads to an increase of probability in the lower-numbered response categories. The converse is true for software such as SPSS with a minus sign between the intercept and all the regression coefficients (cf. section 3 for details). This is a convention ensuring that for positive beta coefficients, increases in predictor values leads to an increase of probability in the higher-numbered response categories. It is recommended risk analyst make sure they understand how the model is set up in any statistical package before interpreting results. In general, the likelihood statistic is superior to the Wald statistic (in the sense that it gives more reliable results), so the paper would mainly concentrate on the likelihood ratio statistic (the reason for considering the Wald statistic too, is that it is computationally easy and is given automatically in the output of most statistical computer packages).

Tabachnick et al. (2001) argued that multinomial logistic regression technique has number of major advantages as a summary to the discussion above: (1) it is more robust to violations of assumptions of multivariate normality and equal variance-covariance matrices across groups; and (2) it is similar to linear regression, but more easily interpretable diagnostic statistics. Further, advantages of the analysis that raise its popularity come from

the following assumptions: (3) most importantly, MLR does not assume a linear relationship between the dependent and independent variables; (4) independent variables need not to be interval (5) MLR does not require that the independents be unbounded and lastly (6) normally distributed error terms are not assumed.

Widely use of MLR as a problem solving tool, particularly in the fields of medicine, psychology, mathematical finance and engineering are as a result of the above advantages listed. This listed relevance attracted the present author's attention in the case for University-wide risk analysis. Standing on the advantages, the purpose and extent of the usage of MLR in risk analysis research deserves a well-prepared review and application of the method (cf. sections 2 & 3). Such an application would provide valuable clues for future University-wide risk analyst about the scope of MLR, what would be expected from it, and how it would work in various risk problems. Following the above contestations and underlying advantages of MLR in risk analysis, the intent of this paper are as seen below (cf. 1.3).

1.3. Research Objectives

The aims of the paper are to:

- a) Demonstrate the use of multinomial logistic regression in risk analysis
- b) Evaluate the use of maximum likelihood methods to perform multinomial logistic regression in risk analysis.
- c) Demonstrate how to assess the fit of a multinomial logistic regression model in risk analysis with the presence of independent variables.
- d) Determine the significance of explanatory variables.

Following third research aim, the hypothesis posed is:

H₀ = there is no difference between model without independent variables (WINV) and the model with independent variables (INV)

H_a = there is a difference between model without independent variables and the model with independent variables.

2. Method

The study was undertaken in a historically black South African University in the greater Eastern Cape Province. The University has a high but relatively heterogonous population density of risk analyst in various committees mandated to risk manage the university. This means that the University is equipped with various (academic, finance, human resource, information systems/infrastructure) directors and or managers who are largely in the knowledge of risk analysis. Historically, the risk/quality unit in University is the overseer of risks and quality checks of the University, which is headed by a director. The population for this study was hundred (100) risk analysts in various committees which were stratified. Information was collected from respondents with the aid of a structured and validated interview schedule, consisting of closed ended questions, based on the objectives/questions of the study. Data analysis was with the aid of both descriptive and inferential analysis (cf. section 3). Independent variables for this study were variables of risk analysis, in this case grouped into two (i) institutional risk planning and (ii) institutional risk mitigation. The dependent variable was characteristics of risk variables, which was measured in terms of six elements, viz (i) institution embedded risk management into its planning and operational processes to a sufficient extent (ii) institutional policy documents deal with risk management issues; iinternal auditors conduct audits as part of statutory regulation (iii) institution has control mechanisms to mitigate risk (iv) responsibility for the oversight of individual key risks are assigned to appropriate managers (v) the institution's overall approach to risk management, as assessed for one-academic year is adequate for its strategic objectives (vi) the issues arising from audits are brought to the attention of the executive management team as appropriate. Since the independent variable was ordinal, the following cautions were considered, thus assumptions. Firstly, MLRA does not make any assumptions of normality, linearity, and homogeneity of variance for the independent variables (cf. sections 1.1 & 1.3). Secondly, because it does not impose these requirements, it

is preferred to other class of analysis (e.g discriminant analysis) when the data does not satisfy these assumption

The overall test of relationship among the independent variables and groups defined by the dependent was based on the reduction in the likelihood values for a model which does not contain any independent variables and the model that contains the independent variables (cf. sections 3.1 & 3.2). This difference in likelihood followed a chi-square distribution χ^2 , and was referred to as the model chi-square (cf. section 3). The significance test for the final model chi-square was the researcher's statistical evidence of the presence of a relationship between dependent and independent variables (cf. section 3).

3. Results: Description of the data

This section describes results of the study. This included the description of the data. Firstly, consideration was given to overall test of relationship, this described the overall test of relationship. Secondly, strength of MLR relationship was tested, this was done to establish the strength of MLR relationship. Thirdly, evaluating for the usefulness of logistic model and relationship between the independent and dependent variables.

3.1. Overall test of Relationship

The first thing in MLR for any risk analyst is to describe the overall test of relationship, in this case a relationship between the dependent and independent variables (cf. section 2-method). The presence of a relationship between the dependent and combination of independent variables is based on the statistical significance of the final model chi-square in the table 1; termed model fitting information. In this analysis, the distribution reveals that the probability of the model chi-square (17.142) was 0.005, less than the level of significance of 0.05 (i.e. $p < 0.05$). The null hypothesis that there was no difference between the model without independent variables and the model with independent variables was rejected (cf. section 1.4). As evidenced in table 3.1 this suggested that the existence of a relationship between the independent variables and the dependent variable was supported, hence accepting the alternate (H_a) hypothesis.

Table 3.1. Model fitting information

Model	-2logLikelihood	Chi-Square	df	Sig
Intercept Only	272.024	.	.	.
Final	225.334	17.142	6	.005

3.2. Strength of Multinomial Logistic Regression Relationship

Once the relationship is established, the next important thing to do is to establish the strength of multinomial logistic regression relationship. While, MLR does compute correlation measures to estimate the strength of the relationship (pseudo R square measures, such as Nagelkerke's R^2), these correlation measures do not really tell an analyst much about the accuracy or errors associated with the model. A more useful measure to assess the utility of a multinomial logistic regression model was classification accuracy, which compares predicted group membership based on the logistic model to the actual, known group membership, which is the value for the dependent variable (cf. section 3.2.1). To assess the strength of multinomial logistic regression relationship, however, the evaluation of the usefulness for logistic models was considered (cf. sections 3.2.1. & 3.2.2).

In this case, using Cox & Snell R Square and the Nagelkerke R square value, they provide an indication of the amount of variation in the dependent variable. These are described as pseudo R square. The distribution in table 3.2 below reveals that the values are 0.181 and 0.322 respectively, suggesting that between 18.1% percent and 32.2% percent of the variability is explained by this set of variables used in the model.

Table 3.2. Pseudo R-Square

Step	Cox & Snell R ²	Nagelkerke R ²
	.181	.322

3.2.1. Evaluating Usefulness for Logistic Models

The proportional by chance accuracy rate was computed by calculating the proportion of cases for each group based on the number of cases in each group in and then squaring and summing the proportion of cases in each group ($0.371^2 + 0.557^2 + 0.072^2 = 0.453$). The proportional by chance accuracy criteria however was 56.6% ($1.25 \times 45.3\% = 56.6\%$). This warrants comparing accuracy rates. To characterise the model as useful, the study compared the overall percentage accuracy rate produced as 25% more than the proportional by chance accuracy. The classification accuracy rate was 60.5% (cf. table 3.3 classification) which was greater than the proportional by chance accuracy criteria of 56.6%, suggesting that the model was useful.

Table 3.3. Classification

Observed	Predicted			% Correct
	1	2	3	
1	15	47	0	24
2	7	86	0	92.5
3	5	7	0	.0
Overall %	16.2%	83.8%	0%	60.5

3.2.2. Relationship of Independent and Dependent Variables

Once the above sections are clarified, it warrants a further scrutiny of the relationship of independent and dependent variables. There are two types of tests for individual independent variables (cf. 1.3 for details). The likelihood ratio test evaluates the overall relationship between an independent variable and dependent variables. While, the Wald test evaluates whether or not the independent variable is statistically significant in differentiating between two groups in each of embedded binary logistic comparisons (cf. table 3.4). Risk analyst need to be cautions though that if an independent variable has an overall relationship to the dependent variable, it does not necessarily suggest statistical significance. In fact, it might or might not be statistically significant in differentiating between pairs of groups defined by the dependent variable.

Table 3.4. Likelihood Ratio Tests⁵

Effect	-2log likelihood of Reduced Model	Chi-Square	df	Sig
Intercept	248.323	2.000	2	.010
i-Emb risk	248.625	2.150	2	.020
ii-Policy doc	260.395	3.423	2	.010
iii-Control mecs	265.195	8.200	2	.010

Following the argument above and referring to table 3.4, there is a statistically significant relationship between the independent variable risk planning (Rp) and the dependent variable ($0.010 < 0.05$). As well, the independent variable Rp is significant in distinguishing both category 1 of the dependent variable from category 3 of the dependent variable. ($0.027 < 0.05$) see table 3.5 for this case.

Table 3.5. Parameters Estimates

Risk planning a		B	S.E	Wald	df	Sig.	Exp(B)	95.0% C.I for EXP(B)	
								Lower	Upper
1	Intercept	2.240	1.478	1.000	1	.001	.	.	.
	Emb risk	.019	.020	.906	1	.000	1.019	4.237	24.000
	Policy D.	.071	.106	.427	1	.011	1.073	8.696	302.304
	Control M	-1.373	.620	4.913	1	.027	.253	.000	
2	Intercept	3.639	2.456	2.195	1	.008	.	.245	1.000
	Emb risk	.003	.020	.017	1	.020	1.003	.655	1.200
	Policy D.	.002	.110	2.463	1	.117	1.188	.779	1.000
	Control M	.540	.401	4.392	1	.007	.191	.	.

The reference category is: 3

In addition, the independent variable Rp is significant in distinguishing category 2 (cf. table 3.5) of the dependent variable from category 3 of the dependent variable ($0.007 < 0.05$). What does this imply?

The above suggest that survey respondents who had less confidence in were less likely to be in the group of survey who thought the institution spent too much time on Rp (DV category 3). For each unit increase in confidence in control mecs, the odds of being in the group of survey respondents who thought the institution spent too little time on Rp decreased by 74.7%. ($0.253 - 1.0 = -0.747$).

Also, an assessment of table 3.6 revealed that there is a statistically significant relationship between key risks assigned (KSA) and the dependent variable, spending on risk mitigation (RisMit) (cf. table 3.2). Moreover, key risks assigned plays a statistically significant role in differentiating the too little group from the too much (reference) group ($0.007 < 0.5$). However, key risks assigned does not differentiate the about right group from the too much (reference) group ($0.51 > 0.5$). Survey respondents who were managers (code 1 for key risks assigned) were less likely to be in the group of survey respondents who thought institution spent too little time on RisMit (DV category 1), rather than the group of survey respondents who thought institution spent too much time on RisMit (DV category 3). Survey respondents who were managers were 88.5% less likely ($0.115 - 1.0 = -0.885$) to be in the group of survey respondents who thought institution spent too little time on RisMit.

Table 3.6. Parameter Estimates⁶

RisMit		B	S.E	Wald	df	Sig.	Exp(B)	95.0% C.I for EXP(B)	
								Lower	Upper
Too Little	Intercept	8.434	2.233	14.261	1	.000	.	1.214	1.442
	audits kn	-.023	.017	1.756	1	.005	.677		
	APPRM	-.066	.102	.414	1	.000	.631		
	audits E	-.575	.251	5.234	1	.021	.563		
	KSA=1	-2.167	.805	7.242	1	.007	.115	.119	1.910
	KSA=2	0 ^b	.	.	0	.		.143	1.303
About Right	Intercept	4.485	2.255	3.955	1	.0004	.	.245	1.102
	audits kn	-.001	.018	.003	1	.000	.999	.655	1.268
	APPRM	.011	.104	.011	1	.002	1.011	.779	1.300
	audits E	-.397	.257	2.375	1	.003	.673		
	KSA=1	-1.606	.824	3.800	1	.003	.201		
	SA=2								

K 0
b

4. Discussion of findings

The variables emb risk, policy documents, control mecs were useful predictors for distinguishing between groups on risk planning. These predictors differentiate survey respondents who thought institution spent too little time on Rp from survey respondents who thought institution spend too much time on Rp. It also differentiated survey respondents who thought institution spent about the right amount of time on Rp from survey respondents who thought institution spent too much time on Rp.

Among this set of predictors, confidence in control mecs was helpful in distinguishing among the groups defined by responses about spending on Rp. Survey respondents who had less confidence in control mecs were less likely to be in the group of survey respondents who thought institution spent too little time on Rp, rather than the group of survey respondents who thought institution spent too much time on Rp. For each unit increase in confidence in control mecs, the odds of being in the group of survey respondents who thought we spend too little time on Rp decreased by 74.7%. Survey respondents who had less confidence in control mecs were less likely to be in the group of survey respondents who thought institution spent about the right amount of time on Rp. For each unit increase in confidence in control mecs, the odds of being in the group of survey respondents who thought institution spent about the right amount of time on Rp decreased by 88.5%.

Implication MLR for risk analyst: Just as with ordinary least squares regression analyst need some means of determining the significance of the estimates of the model parameters (cf. section 3). The analyses also need a means of assessing the fit, or lack of fit, of the logistic model (Hedeker, 2003; Menard, 2002; Tabachnick et al., 2001). The deviance is twice the log-likelihood ratio statistic. The deviance for a logistic model can be likened to the residual sum of squares in ordinary least squares regression for the linear model. The smaller the deviance the better the fit of the logistic model. A small value for the deviance is an indication that there is a significant fit for the logistic model and some other model may be more appropriate. Asymptotically, the deviance has a χ^2 distribution, therefore, to perform tests of hypotheses regarding the fit of the model the deviance is compared to the percentiles of a χ^2 distribution.

Numerical problems: The maximum likelihood method used to calculate multinomial logistic regression is an iterative fitting process that attempts to cycle through repetitions to find an answer. Sometimes, the method will break down and not be able to converge or find an answer. Sometimes the method will produce wildly improbable results, reporting that a one-unit change in an independent variable increases the odds of the modeled event by hundreds of thousands or millions. These implausible results can be produced by multicollinearity, categories of predictors having no cases or zero cells, and complete separation whereby the two groups are perfectly separated by the scores on one or more independent variables.

If an independent variable has an overall relationship to the dependent variable, it might or might not be statistically significant in differentiating between pairs of groups defined by the dependent variable. The interpretation for an independent variable focuses on its ability to distinguish between pairs of groups and the contribution which it makes to change the odds of being in one dependent variable group rather than the other. Analyst should not interpret the significance of an independent variable's role in distinguishing between pairs of groups unless the independent variable also has an overall relationship to the dependent variable in the likelihood ratio test. The interpretation of an independent variable's role in differentiating dependent variable groups is the same as the researchers used in binary logistic regression. The difference in multinomial logistic regression is that analyst could have multiple interpretations for an independent variable in relation to different pairs of groups.

5. Conclusion

With reference to the research objectives, firstly, consideration was given to situations where response variables are binary random variables, taking the values 1 and 0, for 'success' and 'failure', risk analysis processes respectively. The parameters in the model were estimated using the method of maximum likelihood (cf. research objectives a-d). Odds ratios for the response variables were calculated from the parameters of the fitted model. In order to test hypotheses in logistic regression, the study used the likelihood ratio test and the Wald test. In order to evaluating usefulness for logistic models, the benchmark that was used to characterise the MLR model as useful was a 25% improvement over the rate of accuracy achievable by chance alone. This was referred to as by chance accuracy. The estimate of by chance accuracy that we will use is the proportional by chance accuracy rate, computed by summing the squared percentage of cases in each group. The odds of the response variable being success, for given values of the explanatory variables, are the ratio between the probability that the response is a success and the probability that the response is failure, given the values of the explanatory variables (cf. section 3). The odds ratio compares the odds of the response variable being success for two different sets of values of the explanatory variables.

Secondly, the finding also revealed that the probability of the model chi-square (17.142) was 0.005, less than the level of significance of 0.05 (i.e. $p < 0.05$). Suggesting a statistically significant relationship between the independent variable risk planning (Rp) and the dependent variable (0.010 < 0.05). Also, an assessment of table 3.6 revealed that there is a statistically significant relationship between key risks assigned (KSA) and the dependent variable, spending on RisMit. Survey respondents who were managers (code 1 for key risks assigned) were less likely to be in the group of survey respondents who thought we spend too little time on RisMit, rather than the group of survey respondents who thought we spend too much time on RisMit. For each unit increase in confidence in control mecs, the odds of being in the group of survey respondents who thought institution spent too little time on Rp decreased by 74.7%. Survey respondents who had less confidence in control mecs were less likely to be in the group of survey respondents who thought we spend about the right amount of time on Rp. Implying that both risk planning and mitigation are integral component of risk analysis. There play important role in variables such as control mechanism policy formulation and audit. Hence, it is recommended that attention be given to such components of risk analysis in effort to quality assure an institution.

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¹ (Whether an outcome would succeed or fail, whether a respondent should be classified as a user or nonuser, whether a respondent is prone to engage in risky attitude or not, male/female, yes/no, user/nonuser, satisfied/unsatisfied, etc.)

² Note that logistic regression is a variation of multiple regression, which does not necessarily obey the assumptions of the latter.

³ Dummy coding provides one way of using categorical predictor variables in various kinds of estimation models. Dummy coding uses only ones and zeros to convey all of the necessary information on group membership.

⁴ Bernoulli distribution, is a discrete probability distribution, which takes value 1 with success probability p and value 0 with failure probability $q = 1 - p$. It is a good model for any random experiment with two possible outcomes, for example, yes/no answer (of a respondent in an opinion poll), died/survived (in a drug trial) etc. A random variable x has a Bernoulli distribution with parameter $0 < p < 1$ if $P(A)$ is the probability of outcome A . The parameter p is often called the "probability of success". For example, a single toss of a coin has a Bernoulli distribution with $p=0.5$ (where 0 = "head" and 1 = "tail").

⁵ (i) institution embeds risk management into its planning and operational processes to a sufficient extent (embed risk) (ii) institutional policy documents deal with risk management issues; internal auditors conduct audits as part of statutory regulation (policy documents) (iii) institution has control mechanisms to mitigate risk (control mechanisms)

⁶ RiskMit: (iv) responsibility for the oversight of individual key risks are assigned to appropriate managers (key risks assigned-KSA) (v) the institution's overall approach to risk management, as assessed for one-academic year is adequate for its strategic objectives; (approach to risk management strategic objectives-APRMSOBJ) (vi) the issues arising from audits are brought to the attention of the executive management team as appropriate (audits know to EMT)

THE COIN CHANGING PROBLEM AS A MATHEMATICAL MODEL

Mike C. PATTERSON

Dillard College of Business, Midwestern State University,
Wichita Falls, TX, USA

E-mail: mike.patterson@mwsu.edu

Bob HARMEL

Dillard College of Business, Midwestern State University,
Wichita Falls, TX, USA

Abstract: The coin changing problem is a well-known problem to the general public and to operations research specialists and computer scientists. It is a popular topic for both discussion and programming assignments (Ghosh, 2008). The problem is a good example of the recursive nature of many problem solving techniques. Dynamic programming and greedy heuristics are perhaps the most common approaches to the coin changing problem (Johnsonbaugh and Schaefer, 2004).

Key words: Mathematical Model; Coin Changing Problem; Excel Solver Function

Statement of the Coin Changing Problem

In simple terms, the coin changing problem is stated below. Given a finite set of monetary value coins (i.e. 1¢, 5¢, 10¢, 25¢, 50¢ and \$1), make change for any amount utilizing the minimum number of coins possible. The recursive nature of the problem can be illustrated in a model where one chooses the largest denomination coin and multiplies the value of that coin by the maximum number which can be divided into remaining change to be made until the final change remaining is zero. Figure 1 is a model diagram of the problem where the change to be made is \$2.83. All change is to be made in coins of values 1¢, 5¢, 10¢, 25¢, 50¢ and \$1.

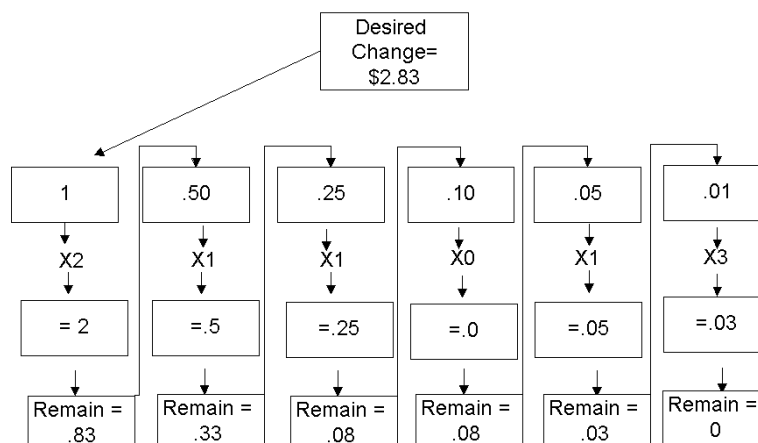


Figure 1. Schematic Model of Coin Changing Problem Desired Change = \$2.83

Excel Spreadsheet Formulation

Microsoft Excel possesses a powerful add-in tool in solving optimization problems. The tool is called Solver. This Excel add-in was developed by Frontline Systems, which has also developed more powerful versions of the optimization tool for more complex problems such as non-linear, quadratic programming and discontinuous problems. The Frontline web site (Frontline Systems, n.d.) provides an in-depth discussion of the various optimization tools the company markets as well as of optimization in general. The basic Solver, which is a part of Excel, was utilized for this model. The initial spreadsheet formulation is displayed in Figure 2. Cells B2:G2 hold the values of each of the six coins. Cells B4:G4 are the cells which will be changed during the optimization process and will hold the number of each coin to be used to make the desired change. Cell I5 is the cell where the desired change should be entered. Cell I3 will hold the calculated change upon optimization which must be equal to the value entered into I5. Cell I7 is the cell which will be optimized to minimize the total number of coins used in making change and will equal the sum of cells B4:G4. Figure 3 displays the formula view of the model.

1/A	B	C	D	E	F	G	H	I
2	1	0.5	0.25	0.1	0.05	0.01		Change
3								0
4	0	0	0	0	0	0		Desired
5								0
6								Total Coins
7								0

Figure 2. Initial Spreadsheet Formulation of Coin Changing Problem

1/A	B	C	D	E	F	G	H	I
2	1	0.5	0.25	0.1	0.05	0.01		Change
3								=SUMPRODUCT(B4:G4,B2:G2)
4	0	0	0	0	0	0		Desired
5								0
6								Total Coins
7								=SUM(B4:G4)

Figure 3. Formula View Spreadsheet Formulation of Coin Changing Problem

A mathematical statement of the problem is displayed below.

Minimize: $A + B + C + D + E + F$

Where: A = number of \$1 coins

B = number of 50¢ coins

C = number of 25¢ coins

D = number of 10¢ coins

E = number of 5¢ coins

F = number of 1¢ coins

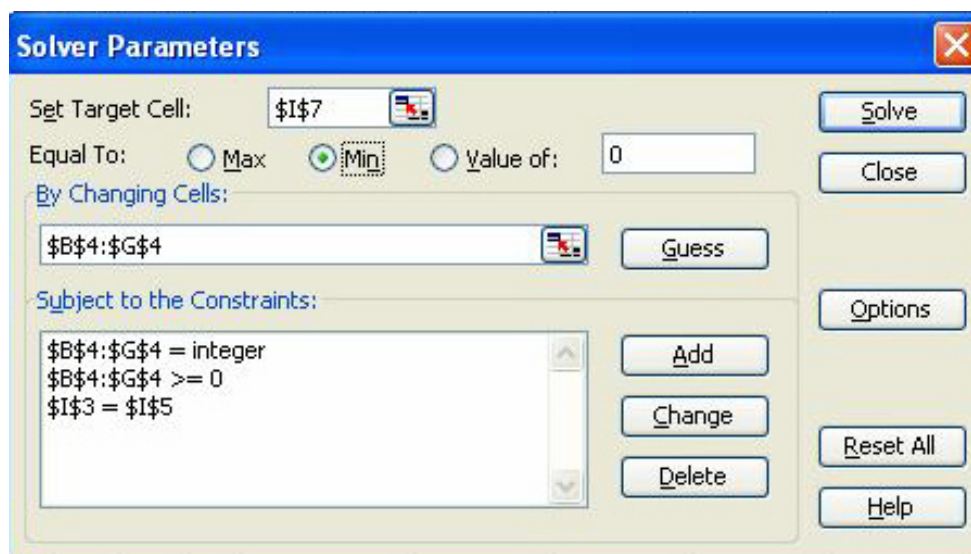
Subject to the following constraints:

$$1.00A + .50B + .25C + .10D + .05E + .01F = \text{Desired Change}$$


A, B, C, D, E, F = Integers

A, B, C, D, E, F ≥ 0


The Excel Solver parameters which utilize the mathematical statement of the problem are displayed in Figure 4.



Solver Parameters

Set Target Cell: 

Equal To: ☐ Max ☒ Min ☐ Value of:

By Changing Cells: 

Subject to the Constraints:

-
-
-

Buttons: Solve, Close, Options, Add, Change, Delete, Reset All, Help

Figure 4. Solver Parameters for Coin Changing Problem

Figure 5 displays the model output solution to the previously discussed desired change amount of \$2.83. Figure 6 and 7 displays output for other change amounts

1/A	B	C	D	E	F	G	H	I
2	1	0.5	0.25	0.1	0.05	0.01		Change
3								2.83
4	2	1	1	0	1	3		Desired
5								2.83
6								Total Coins
7								8

Figure 5. Solution View of Coin Changing Problem Desired Change = \$2.83

1/A	B	C	D	E	F	G	H	I
2	1	0.5	0.25	0.1	0.05	0.01		Change
3								0.92
4	0	1	1	1	1	2		Desired
5								0.92
6								Total Coins
7								6

Figure 6. Solution View of Coin Changing Problem Desired Change = \$.92

1/A	B	C	D	E	F	G	H	I
2	1	0.5	0.25	0.1	0.05	0.01		Change
3								3.54
4	3	1	0	0	0	4		Desired
5								3.54
6								Total Coins
7								8

Figure 7. Solution View of Coin Changing Problem Desired Change = \$3.54

Summary

The well-known coin changing problem is a widely used exercise in recursive logic exercises in business and computer science classes. Most of the examples observed by the authors involved heuristic type logic, dynamic programming, greedy algorithms and backtracking techniques. The spreadsheet model in this paper is designed to illustrate what the authors believe to be a more concise and intuitive approach which will be interesting and accessible to most instructors, students, and individuals.

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A MUTIVARIATE STRATEGIC PERSPECTIVE FOR THE EVALUATION OF CUSTOMER SATISFACTION IN GREAT DISTRIBUTION

Gabriella MILONE

University of Naples - Federico II, Naples, Italy

E-mail: gamilone@unina.it



Abstract: The author proposes a strategy for the analysis of data from Customer Satisfaction (CS) in Great Distribution. The aim of this paper is to evaluate CS through a comparison of multivariate statistical methodologies.

In this paper the author compares different estimations of Structural Equation Model (SEM) in a case study: evaluation of customer satisfaction in a supermarket. Overall satisfaction is determined by reference to three departments: "Salami and Cheese", "Butchery", "Fruit and Vegetables". Each department is assessed through three aspects: "Assortment", "Staff" and "Offer".

Initially, the links between the different variables are verified through factor analysis and subsequently inserted into a structural equation model. To estimate the model the approach of "maximum likelihood" was used, with LISREL software. Finally, the "Partial Least Squares (PLS)" approach was used to confirm the results.

Key words: customer satisfaction; multivariate analysis; great distribution; LISREL

1. Introduction

In the social sciences [14] the study of the evolution of customers' buying behavior and style plays a particularly delicate role.

Companies must adapt to rapid changes. They must take action to meet and even to anticipate expectations. In particular, their aim must be to impress customers with a high level of service in order to strengthen the bond of trust.

For this reason, companies must be aware of the customer's habits and must increase potential loyalty. This can lead to an immediate economic return and a future increase in the total value of the company. In fact, it can survive by relying on the turnover and profitability provided by its customers.

However, the satisfaction-loyalty connection is not always automatic, because of the consumption of goods considered luxuries. There are fringes of customers who, while happy with the product-service received, for new purchases tend to turn to other producers. Thus, satisfaction and loyalty, although closely related, are two different concepts.

In fact, we talk about satisfaction only in cases where those who give an opinion have actually experienced the use of the goods or services purchased. They must be able to give their opinions both in brief and by evaluating tangible and intangible aspects.

Therefore, typical customer satisfaction is real. The potential client can only express an opinion on image, reputation, or repute of the producer of goods or services.

Moreover, particular attention must be paid to customer expectations. These are influenced by a number of factors whose importance varies from individual to individual.

Consider the difficulty of assessing the impact of personal needs, in turn influenced by past experiences, both direct and indirect.

2. Customer Satisfaction in GD

In recent years, Great Distribution (GD) has a dominant role in the chain transferring products/services to market. In most sectors, the supply of products has increased demand. The critical success of the company tends to depend on the ability to reach the end customer through a more efficient distribution format.

Companies are forced to innovate with new competitive strategies in an attempt to stand out and to gain new market niches [12].

These dynamics are due to continuous economic change, intensifying competition, complexity and articulation of the offer. This explains the growing importance of evaluation of customer satisfaction (CS). In particular, in GD issues of quality are increasingly important, in the sense of the ability to meet customers' implicit and explicit needs.

The evaluation ratings of customers are affected by:

- cultural elements and character traits, such as occupation, educational qualifications, etc.

- psychological factors, such as cognitive and emotional elements

- other factors, such as customer's knowledge of competing firms.

The GD is characterized, therefore, by the presence of an asymmetry of information between the customer receiving the service and the structure that it provides.

For these reasons, in order to improve the quality of services provided, with a view to satisfying customers, GD should pay attention to those factors that will influence the customer's decision to purchase (special promotions, advertising, product placement on shelves, availability of sales staff, etc.).

3. Evaluation of CS through a process-oriented approach

The study of the international context (D'Ambra, L. and Gallo M., 2006) shows that the measurement of satisfaction essentially follows two approaches:

- application-oriented approach of quality models and evaluation of the size characteristics of quality

- oriented approach to evaluate the customer experience in the process of service delivery.

To monitor CS it is necessary to follow the customer through a system that uses the tools dictated by the literature and strictly follow the steps preceding and following data collection (UNI 11098th: 2003).

The author proposes stages of an integrated approach for the evaluation of Customer satisfaction in GD (Table 1).

Table 1. Stages of an integrated approach to process

Step 1	Definition of model, questionnaire, rating scales
Step 2	Universe of reference and sampling plan
Step 3	Data Collection
Step 4	Data quality and pre-treatment
Step 5	Data Analysis
Step 6	Decision support based on results of CS

3.1. Sampling plan

The sampling adopted is based on two different units of detection:

- *Loyal customers.* These are customers who make the pilgrimage to the point of sale, fidelity card holders, giving a probabilistic sampling, since the distribution structure has the list of all the loyal customers and can then allocate to each individual the probability of becoming part of the sample. In this case the design may be simple random sampling, stratified by sex or age groups, etc.;

- *Occasional customers.* These are customers who can be identified by non-probability sampling, through "face to face" interviews, trying to stratify customers by time and relative abundance (through a survey of presence). An alternative is random sampling in clusters (cluster sampling). The sample is formed by randomly selecting some clusters (bands). In order for the cluster sampling to be effective it is necessary for the clusters not to be too large (for all customers to be interviewed), that their size be as uniform as possible and that the units of which they are composed are the most heterogeneous in terms of type.

3.2. Data quality and pre-treatment

The statistical techniques adopted to validate the questionnaires include:

- Rasch Analysis to verify the scalability of the questionnaire
- Factor Analysis, to evaluate the one-dimensionality
- Use of the coefficient of stability (test-retest), which consists of applying the same tool to the same subject in two subsequent cases, calculating the correlation coefficient between the two sets of scores
- Use of the equivalence factors, which stem from the calculation of the correlation between "parallel forms" of a set of items (a scale or test), administered in a single application (or shortly after). Two or more items (or two or more tests) are called parallel when in measuring the same construct, they are similar in content and/or difficulty.

The coefficients of equivalence most often used are the following:

- Split-half technique. It consists in dividing the questions related to the same concept (of a business process) into two parts and calculating the correlation between the two sets of scores thus obtained. The ratio thus obtained indicates the equivalence of the two halves and the loyalty of the middle test (or scale). This is then corrected using the formula of Spearman-Brown
- Calculation of Cronbach's coefficient alpha. The reliability of Cronbach's coefficient alpha is measured in terms of internal consistency, "it reflects the degree of agreement among multiple measurements of the same theoretical concept obtained at the same time of administration with the same method.

Before analyzing the data, it was decided to pre-treat the data, so as to ensure the quality of information that will be extracted, relating to:

- assessing the quality of the data, responding to the requirements defined by Eurostat documentation in evaluating the quality of statistics produced by the member countries of the European Community concerning the following dimensions: relevance, accuracy, timeliness, transparency, comparability, consistency, completeness
- the treatment of missing data, using deterministic techniques (deductive imputation, imputation by medium)
- quantification of Thurstone which allows for an analysis respecting the ordinal nature of the data through a transformation from ordinal data to linear data (D'Ambra et al, 2001, 2002).

4. Structural Equation Model

For the detection of CS, following a process-oriented approach, a flexible tool must be used to assess the satisfaction of customers who access the service through different pathways.

The questionnaire should be structured so that:

- it describes all processes and activities related to services
- it stratifies the population into subgroups and items of interest.

The conceptual model can be achieved through the theoretical construct of the structural equation [4] (Structural Equation Model - SEM) method commonly used in the scientific community.

The objective of this methodology is to provide a simplified representation of real processes, taking into account not only the multiplicity of causes that act on a dependent variable, but also the connections between different causes. One of the main reasons is due to the increasing use of software capable of performing statistical analysis based on both the covariance (LISREL, EQS, Amos, etc.) and the search for components (PLS) [3] [6].

Typically, the study of CS is made through the evaluation of customer-perceived satisfaction with aspects of the service received.

These aspects, latent dimensions of satisfaction (latent variables) are quantified through manifest variables, usually expressed on an ordinal scale of scores. The relationship between manifest variables and latent variables can be formalized through patterns that ensure rigour, firstly, in the process of defining the concept of CS, and subsequently in its measurement. The SEM include a first linear system of equations with unknown coefficients which links a set of (endogenous) variables, not observed by each other, with a second set of (exogenous) variables, that are unobservable and linked to expectations.

This structure is complemented by two other sets of equations linking the endogenous and exogenous variables but others have observed. Each set of equations is disturbed by the presence of accidental errors.

The degree of overall satisfaction is identified with one of the latent variables.

$$\eta = B\eta + \Gamma\xi + \zeta$$

$$X = \Delta^X\xi + \delta \text{ (b)}$$

$$Y = \Delta^Y\eta + \varepsilon \text{ (c)}$$

where:

η = vector of m endogenous variables

ξ = vector of n exogenous variables

ζ = vector of m error

B, Γ = matrices of structural coefficients (the first, linkages between endogenous variables, and others, linkages between endogenous and exogenous)

X, δ = vectors to exogenous variables and errors observed

Δ^X = matrix of structural coefficients between observed variables and latent variables

Y, ε = vectors of endogenous variables and the errors observed

Δ^Y = matrix of structural coefficients between the observed variables and latent variables

4.1. Factor Analysis

The general concept of factor analysis is responsible for a series of statistical techniques whose ultimate goal is to deliver a set of observed variables in terms of a smaller number of hypothetical variables (latent, in LISREL terminology) called factors [13].

Factor analysis provides two approaches: a confirmatory type and an exploratory one, which will be referred to in the case-study considered.

The confirmatory-type approach assumes that the researcher already has a theoretical model of reference, which plans to submit empirical data to verification, while the in exploratory approach there are no assumptions about the number of factors, the identity of the factors and relationships between factors and manifest variables, so it is necessary to estimate all the parameters λ , but a model of this kind (in which all the possible links between manifest and latent are activated) is not identified, and should any of these be bound (usually zero).

In a model with correlated factors, as in our case, in the model thus identified the student must have at least one variable for each factor "saturated" solely by that factor, as is clear from the matrix Λ^x .

$$\Lambda^x = \begin{bmatrix} \lambda_{1,1}^x & 0 & 0 & 0 \\ 0 & \lambda_{2,2}^x & 0 & 0 \\ 0 & 0 & \lambda_{3,3}^x & 0 \\ 0 & 0 & 0 & \lambda_{4,4}^x \\ \lambda_{5,1}^x & \lambda_{5,2}^x & \lambda_{5,3}^x & \lambda_{5,4}^x \\ \lambda_{6,1}^x & \lambda_{6,2}^x & \lambda_{6,3}^x & \lambda_{6,4}^x \\ \lambda_{7,1}^x & \lambda_{7,2}^x & \lambda_{7,3}^x & \lambda_{7,4}^x \\ \lambda_{8,1}^x & \lambda_{8,2}^x & \lambda_{8,3}^x & \lambda_{8,4}^x \\ \lambda_{9,1}^x & \lambda_{9,2}^x & \lambda_{9,3}^x & \lambda_{9,4}^x \\ \lambda_{10,1}^x & \lambda_{10,2}^x & \lambda_{10,3}^x & \lambda_{10,4}^x \\ \lambda_{11,1}^x & \lambda_{11,2}^x & \lambda_{11,3}^x & \lambda_{11,4}^x \\ \lambda_{12,1}^x & \lambda_{12,2}^x & \lambda_{12,3}^x & \lambda_{12,4}^x \end{bmatrix}$$

4.2. Partial Least Squares

The estimation method Partial Least Squares [15] is an exploratory non-parametric approach, it is an not inferential instrument, so the results are valid only for the sample. This definition shows that one cannot make global statistical tests (the only practicable one, the "bootstrap test", is in fact a resampling test that can measure the significance of a link), the PLS [5] approach is optimized by the variance and covariance structures, and there are no errors (the errors are diagonal matrices). The difference between the LISREL and the PLS methods lies in the estimation of the parameters, which happens through the LISREL maximum likelihood method, optimizing a global function to define a single measure of goodness of fit, while the PLS approach is algorithm based on "fixed points", seeking the points of local minimum, or minimum points referring to each of the latent variables.

5. Evaluation of Customer Satisfaction with an Italian supermarket

The data used was collected in an Italian supermarket, in order to be able to develop, test and validate a system to monitor customer satisfaction and the quality offered, designed to assess the level of overall satisfaction perceived by customers.

5.1. Definition of the variables

The variables considered [1] [7] [9] are broadly sixteen, nine of which are manifest exogenous variables (X), corresponding to the three factors considered (range, staff, offerings) collected for each department, three latent exogenous variables (ξ), corresponding to satisfaction with the meats and cheeses, butchery, fruit and vegetables departments, and a latent endogenous variable (η) corresponding to the overall satisfaction score with the supermarket, which in turn is measured by three manifest endogenous variables (Y) corresponding to the overall satisfaction towards the set of staff and goods offered, taken as a whole (Table 2).

Table 2. Definition of Variables

Latent Variables		Manifest Variables	
Salami and cheese	ξ_1	Assortment	X_1
		Staff	X_2
		Goods Offered	X_3
Butchery	ξ_2	Assortment	X_4
		Staff	X_5
		Goods Offered	X_6
Greengrocery	ξ_3	Assortment	X_7
		Staff	X_8
		Goods Offered	X_9
Customer Satisfaction	η_1	Assortment	Y_1
		Staff	Y_2
		Goods Offered	Y_3

The data collected is qualitative. Therefore, it was transformed to be treated as quantities by the Thurstone procedure, following these steps:

- calculation of the absolute frequencies
- calculation of the relative frequencies
- calculation of the cumulative relative frequencies
- calculation of the inverse of the standard normal distribution function
- quantification of the new scale.

After having assessed the data is to load the new data quantified in the program LISREL, after defining the correlation matrix, obtained by the data-centered and standardized.

5.2 Cronbach'alfa and Correlation of Item-Scale

The construction of a questionnaire drawn up by the questions, then administered to a sample.

Subsequently, it assesses the internal consistency of the scale.

The internal consistency is used to verify the existence of elements of the scale that are not consistent with the others. The instruments used are the Item-Scale Correlation and the Cronbach' alfa coefficient (Table 3).

Table 3. Internal consistency of the scales

Latent Variables		Cronbach'alfa	Manifest Variables		Item-Scale Correlation
Slami and cheese	ξ_1	0,7	Assortment	X_1	0,54
			Staff	X_2	0,55
			Goods Offered	X_3	0,46
Butchery	ξ_2	0,75	Assortment	X_4	0,64
			Staff	X_5	0,53
			Goods Offered	X_6	0,56
Greengrocery	ξ_3	0,82	Assortment	X_7	0,69
			Staff	X_8	0,63
			Goods Offered	X_9	0,74
Customer Satisfaction	η_1	0,76	Assortment	Y_1	0,59
			Staff	Y_2	0,59
			Goods Offered	Y_3	0,59

The α coefficients measured lead to the scale being accepted. In fact, they exceeded the acceptance limit of 0.7. The values obtained, all tending to 1, showing a good degree of correlation. Therefore, it was not necessary to remove any items of the scale.

Using the matrix above and the data loaded into LISREL, the measurement model was developed.

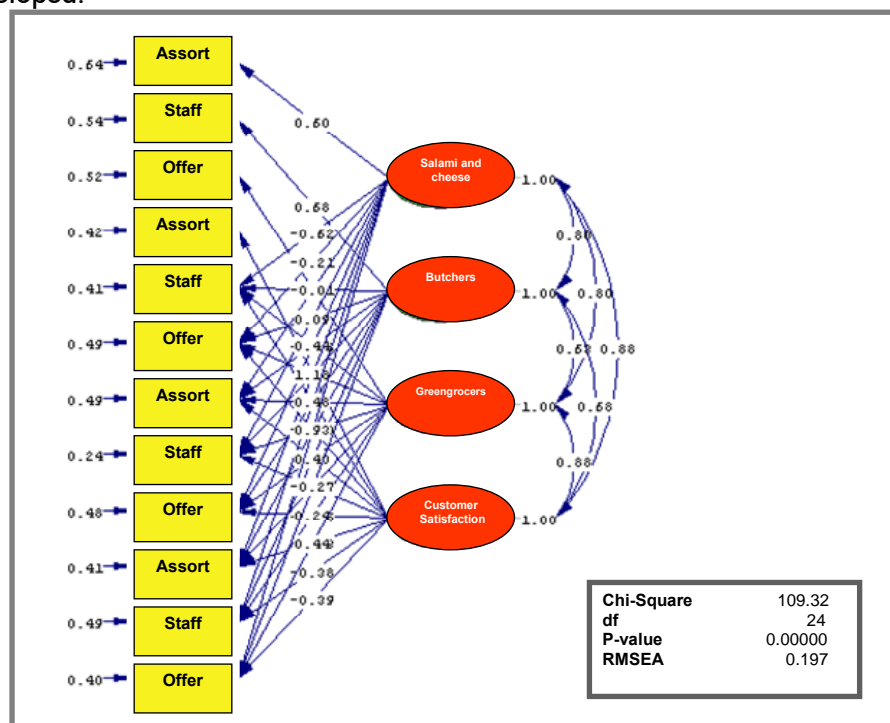


Figure 1. Initial Path Diagram of Factor Analysis

Figure 1 shows that there are many non-significant links, highlighted by the values of the T-value. Consequently, those links were removed, equaling zero so that LISREL does not estimate them. This was followed by the activation of the links suggested by the software through the indices of change and on the basis of "common sense" (by placing the parameters that are to be released in the matrix corresponding to 1).

These operations led us to the definition of the following model, depicted in the second path diagram of Figure 2.

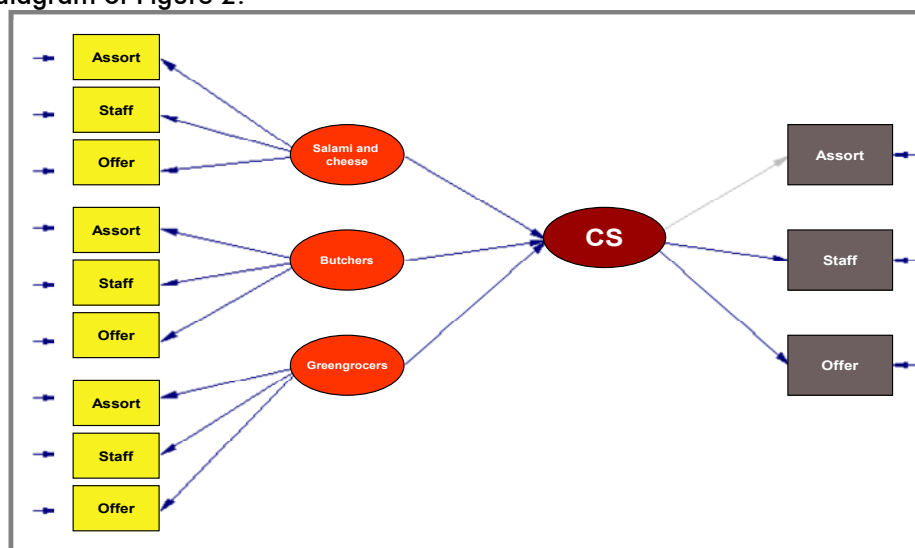


Figure 2. Final Path Diagram of Factorial Analysis

The model has a Chi-square equal to 105.17 with 45 degrees of freedom, the P-value equal to 0.00000 and does not accept the null hypothesis ($H_0: \Sigma-S=0$), since it falls in the rejection area.

Therefore, we reject the model and nothing can be done to improve it.

The Factor analysis presented does not give useful indications about the possible relationships between the manifest variables and factors.

5.3. Definition of the Model

Progressing in the analysis of the model [8] [10], not being able to use the results obtained from factor analysis, we assumed a theoretical model as presented in Figure 3:

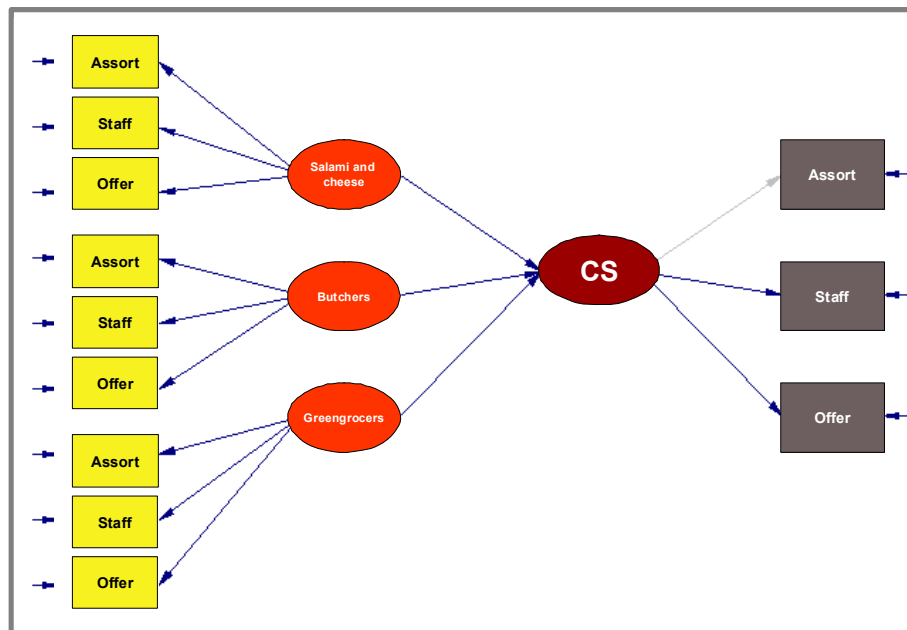


Figure 3. Conceptual Diagram

The model shown has nine manifest exogenous variables, X_n (assortment, staff, offer) underlying three latent exogenous variables:

- ξ_1 , satisfaction department "Salami and cheese"
- ξ_2 , satisfaction department "Butchery"
- ξ_3 , satisfaction department "Greengrocery".

Exogenous variables, through the links of causation, express a latent endogenous variable η (Customer Satisfaction) measured by three manifest endogenous variables Y_n .

The starting point is that this model starts from the idea that the overall satisfaction in a supermarket may depend on the satisfaction with the three departments that comprise it.

The LISREL model is summarized by three basic equations, which for the model in question are expressed by:

• *Structural Model*, for the causal relationships between endogenous and exogenous variables:

$$\eta_{(1 \times 1)} = B_{(1 \times 1)} \eta_{(1 \times 1)} + \Gamma_{(1 \times 3)} \xi_{(3 \times 1)} + \zeta_{(1 \times 1)} \quad (1)$$

which in matrix formulation:

$$[\eta_1] = [\gamma_{1,1} \quad \gamma_{1,2} \quad \gamma_{1,3}] \begin{bmatrix} \xi_1 \\ \xi_2 \\ \xi_3 \end{bmatrix} + [\zeta_1]$$

- Endogenous measurement model:

$$Y_{(3 \times 1)} = \Lambda_{(3 \times 1)}^x \eta_{(1 \times 1)} + \varepsilon_{(3 \times 1)} \quad (2)$$

which in matrix formulation:

$$\begin{bmatrix} y_1 \\ y_2 \\ y_3 \end{bmatrix} = \begin{bmatrix} \lambda_{1,1}^y \\ \lambda_{2,1}^y \\ \lambda_{3,1}^y \end{bmatrix} [\eta_1] + \begin{bmatrix} \varepsilon_1 \\ \varepsilon_2 \\ \varepsilon_3 \end{bmatrix}$$

- Exogenous measurement model:

$$X_{(9 \times 1)} = \Lambda_{(9 \times 3)}^x \xi_{(3 \times 1)} + \delta_{(9 \times 1)} \quad (3)$$

which in matrix formulation:

$$\begin{bmatrix} x_1 \\ x_2 \\ x_3 \\ x_4 \\ x_5 \\ x_6 \\ x_7 \\ x_8 \\ x_9 \end{bmatrix} = \begin{bmatrix} \lambda_{1,1}^x & 0 & 0 \\ \lambda_{2,1}^x & 0 & 0 \\ \lambda_{3,1}^x & 0 & 0 \\ 0 & \lambda_{4,2}^x & 0 \\ 0 & \lambda_{5,2}^x & 0 \\ 0 & \lambda_{6,2}^x & 0 \\ 0 & 0 & \lambda_{7,3}^x \\ 0 & 0 & \lambda_{8,3}^x \\ 0 & 0 & \lambda_{9,3}^x \end{bmatrix} \begin{bmatrix} \xi_1 \\ \xi_2 \\ \xi_3 \end{bmatrix} + \begin{bmatrix} \delta_1 \\ \delta_2 \\ \delta_3 \\ \delta_4 \\ \delta_5 \\ \delta_6 \\ \delta_7 \\ \delta_8 \\ \delta_9 \end{bmatrix}$$

To complete the formulation of the model, the other four matrices must be specified:

- Φ , which defines the correlation between the latent exogenous variables, the matrix
- Ψ , which defines the correlation of the errors of the endogenous latent variables
- Θ^ε , which defines the correlation between the errors of measurement of the endogenous model
- Θ^δ , which defines the correlation of the errors of the model exogenously.

The four matrices above are all square and symmetrical and the diagonal is the variances of the corresponding variables.

$$\Phi = \begin{bmatrix} \phi_{1,1} & & \\ \phi_{2,1} & \phi_{2,2} & \\ \phi_{3,1} & \phi_{3,2} & \phi_{3,3} \end{bmatrix} \quad \Psi = [\psi_1] \quad \Theta^{\epsilon} = \begin{bmatrix} \theta^{\epsilon}_{1,1} & & \\ 0 & \theta^{\epsilon}_{2,2} & \\ 0 & 0 & \theta^{\epsilon}_{3,3} \end{bmatrix}$$

$$\Theta^{\delta} = \begin{bmatrix} \theta^{\delta}_{1,1} & & & & & & & & \\ 0 & \theta^{\delta}_{2,2} & & & & & & & \\ 0 & 0 & \theta^{\delta}_{3,3} & & & & & & \\ 0 & 0 & 0 & \theta^{\delta}_{4,4} & & & & & \\ 0 & 0 & 0 & 0 & \theta^{\delta}_{5,5} & & & & \\ 0 & 0 & 0 & 0 & 0 & \theta^{\delta}_{6,6} & & & \\ 0 & 0 & 0 & 0 & 0 & 0 & \theta^{\delta}_{7,7} & & \\ 0 & 0 & 0 & 0 & 0 & 0 & 0 & \theta^{\delta}_{8,8} & \\ 0 & 0 & 0 & 0 & 0 & 0 & 0 & 0 & \theta^{\delta}_{9,9} \end{bmatrix}$$

5.4. Results with LISREL

Having identified the eight matrices of the model included in the LISREL software, you get the Path Diagram (Figure 4).

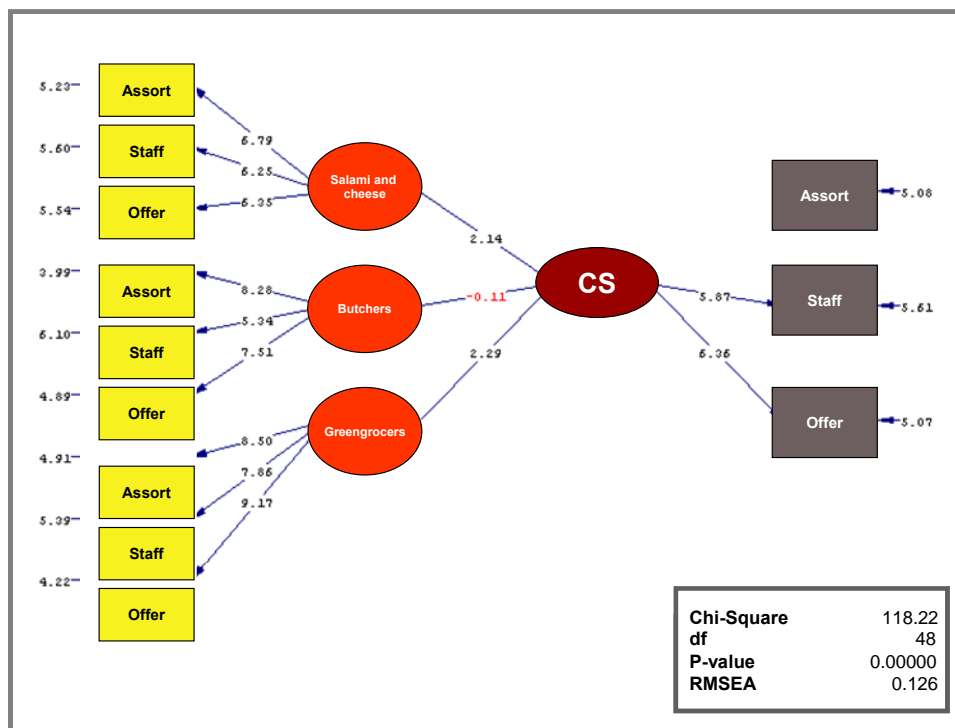


Figure 4. Path Diagram with T-value

The model has a value of chi-square equal to 118.23 with 48 degrees of freedom, a P-value equal to 0.00000, and does not accept the null hypothesis ($H_0: \Sigma-S=0$), since it falls in the rejection area.

Therefore, we reject the model and nothing can be done to improve it.

LISREL through the value of the T-value indicates the existence of a parameter not significantly different from zero, corresponding to the link $Y_{2,1}$.

Removing this link, it is made equal to zero in the corresponding matrix so that the software does not make its estimate. From a conceptual point of view this implies that the latent variable ξ_2 (Butchery) does not directly affect endogenous latent variable η_1 (CS).

Changes made by the new model are presented in Figure 5:

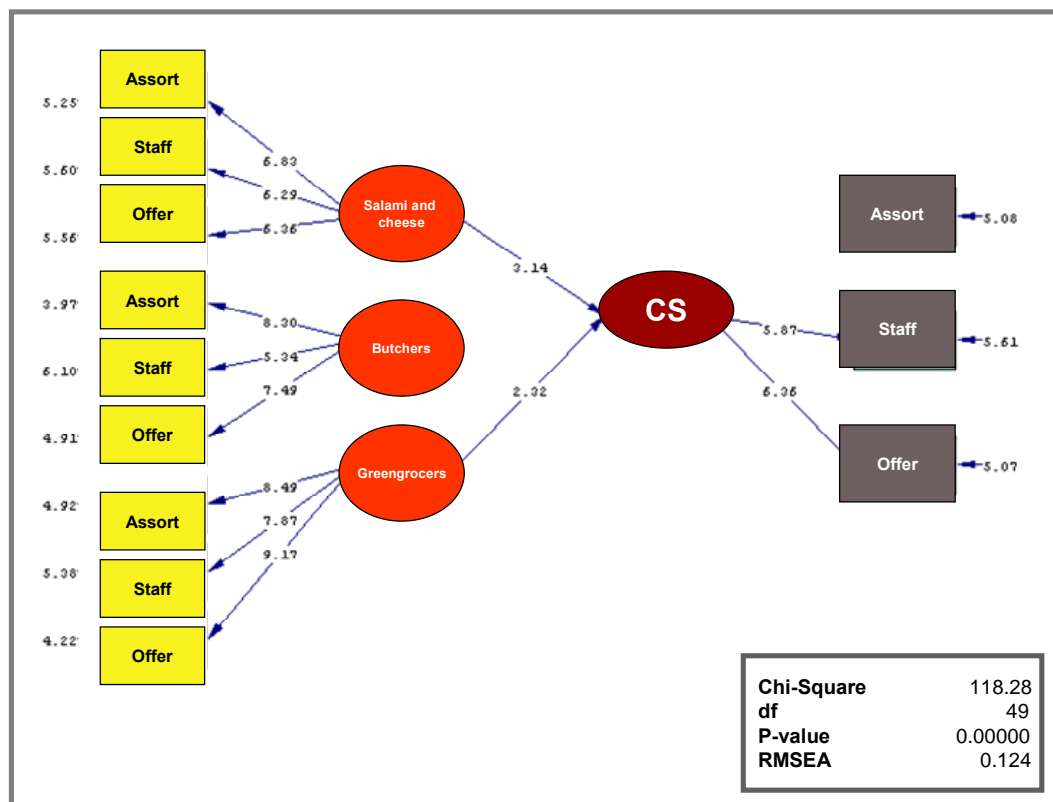


Figure 5. Path Diagram with estimated parameter

The model at this point though slightly better than the previous one (with values of $\chi^2 = 118.28$ and $df = 49$), is absolutely still to be rejected due to the low P-value (less than 0.10).

Moreover, a conceptual evaluation was done to eliminate the correlations among the latent exogenous variables $\phi_{3,1}$ and $\phi_{3,2}$, keeping only the correlation between satisfaction with "Salami and Cheese" department and the satisfaction with the "Butchery" department ($\phi_{2,1}$). However the resulting model had a value of χ^2 significantly higher than the previous model, which led us to maintain the initial correlations.

Similarly, an improvement to the model analyzed for possible links suggested by LISREL using the indices of change, was not activated because the links are not consistent with an assessment based on "common sense".

Therefore, no other ways of improving the model can be seen.

5.5. Results with PLS

The model analyzed in this case involves measuring the overall satisfaction with a supermarket (latent variable, CS) using the satisfaction with three other latent variables, "Salami and cheese", "Butchers", "Greengrocers" (Figure 6).

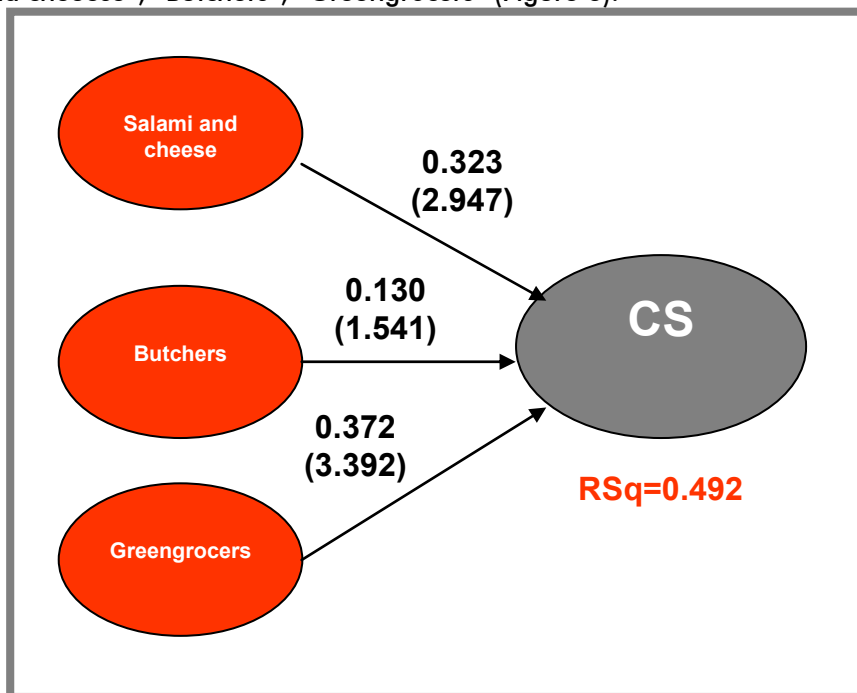


Figure 6. Initial PLS Model

The bootstrap [11] test shows a significant link is not relevant to the relationship between satisfaction with Butchery and customer satisfaction, giving a value of 1.541 (<2). We proceeded to the elimination of this bond and consequently the elimination of the latent variable "Butchery" (Figure 7, Table 4).

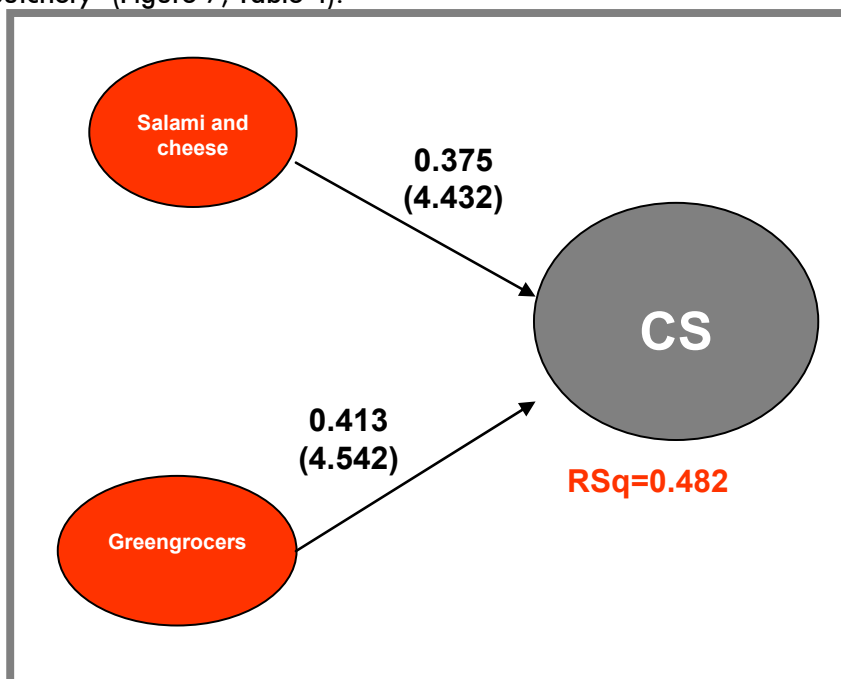


Figure 7. Final PLS Model

Table 4. Results with Bootstrap

Measurement Mode (weight)					
		Entire Sample Estimate	Mean of Subsamples	Standard Error	T-Statistic
Salami and cheese	Assortment	0.4561	0.4585	0.0890	5.1227
	Staff	0.4009	0.3969	0.0824	4.8625
	Goods Offered	0.4044	0.3981	0.0789	5.1269
Greengrocery	Assortment	0.3571	0.3524	0.0454	7.8719
	Staff	0.4114	0.4181	0.0469	8.7654
	Goods Offered	0.3910	0.3896	0.0500	7.8133
Customer Satisfaction	Assortment	0.4253	0.4272	0.0493	8.6348
	Staff	0.3743	0.3730	0.0419	8.9420
	Goods Offered	0.4130	0.4157	0.0505	8.1794

Table 5. Structure Model

Structure Model				
	Entire Sample Estimate	Mean of Subsamples	Standard Error	T-Statistic
Salami and cheese ->CS	0.3750	0.3724	0.0846	4.4319
Greengrocery ->CS	0.4130	0.4241	0.0909	4.5422

5.6. Comparison of results according to the LISREL and PLS approaches

The model presented a report giving "reflective", which can be expressed by the covariances between the latent variables and manifest variables.

This differs from the reports giving "instructive" in which the variables are used as manifest indicators of latent variables.

Regarding the validation of the model that characterizes the PLS [2] approach is the existence of individual indices of goodness of fit (R^2) is not inferential. Therefore, the results are valid only for the sample. Equally, with the LISREL model there is a global parametric index. Therefore, it is an inferential index and extended to the entire population.

The case study observed a value of R^2 between 0 and 1, which makes the model acceptable.

Bootstrap analysis of the text referring to this model reveal the values of T-Statistic are all greater than 2, which means that the links are all significant, because their values fall in the rejection of the null hypothesis ($H_0: \mu = 0$).

Finally, conceptually, analysis of the model shows that the overall satisfaction with a supermarket is not directly dependent on the variable of satisfaction with the "Butchery" department. This confirms the final evaluations obtained with LISREL.

6. Final remarks

The verification of the conceptual model in LISREL showed that Customer Satisfaction with a supermarket is not adequately measured by the variables considered. This means that the satisfaction with one department is not considered a serious effect on overall satisfaction. In particular, the satisfaction with the "butchery" department has no significant impact on "overall satisfaction". For this reason, the corresponding relationship is eliminated.

What led to the falsification of the model is the magnitude of the residual as a whole. It showed that a wide discrepancy between the observed matrix (S) and the matrix Hold (Σ) is not attributable to simple stochastic fluctuations. For this purpose the following were analyzed:

- the *Steamleaf Plot* for the distribution of residuals, which has a bell shaped curve, considering the "probability sample"
- the *Q-plot* for the dispersion of the standardized residuals, which shows the straight line interpolating the residuals (near 45° line), making it seem that there is a good model-to-data fit.

The sensitivity of the Chi-square to the sample size (though not particularly large), granted to the analysis of alternative measures of overall adaptation of the model:

- the GFI (goodness of fit index), where the value of the t-statistic is standardized with the maximum value it can reach (it should be between 0 and 1). The reference model is equal to 0.82 (good fit). However, this measure takes no account of degrees of freedom, so that the model is parsimonious
- the AGFI (adjusted goodness of fit index), which is a modified version of the previous year. The value obtained was 0.72 (also between 0 and 1)
- the RMR (root mean squared residuals), which represents a pure average of the squared residuals, which shows a value of 0.073. However, these measures have the disadvantage of not having a statistical distribution. For this reason, we cannot perform significance tests of the model, being valid only for the sample considered.

Having verified the proper fit of the model to data, show that the falsification of the model may be due to an incorrect formulation of the questionnaire, or the fact that other possible variables were not considered.

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INFORMATION THEORETIC ESTIMATION IMPROVEMENT TO THE NONLINEAR GOMPERTZ'S MODEL BASED ON RANKED SET SAMPLING

Enrico CIAVOLINO

Università Salento, Salento, Italy

E-mail: enrico.ciavolino@ateneo.unile.it

Amajd D. AL-NASSER

Yarmouk University, Irbid, Jordan

E-mail: amjadyu@yahoo.com

Abstract: The aim of this paper is to apply both Generalized Maximum Entropy (GME) estimation method and Ranked Set Sampling (RSS) technique to improve the estimations of the Gompertz's Model. The Gompertz's model is a simple formula which expresses the geometrical relationship between the force of mortality and age.

The choice of evaluating the RSS is due to the fact that in many practical applications of the Gompertz's model, as in biological or environmental sciences, the variable of interest is more costly to measure but is associated with several other easily obtainable.

In this paper, we have used Monte Carlo experiments to illustrate the performance of the GME estimator based on two different sampling techniques: the Simple Random Sample (SRS) and RSS. Moreover, the results are compared with the traditional Maximum Likelihood Estimates (MLE).

Key words: Generalized Maximum Entropy; Ranked Set Sampling; Gompertz's Model; Maximum Likelihood Estimates; Monte Carlo experiments

1. Introduction

Gompertz's law (Gompertz, 1865) has a very important role in modelling the human mortality rate and providing the actuarial tables. Moreover, in recent years, it has been applied to several other fields, such as in the fertility rate model (Booth, 1984), in biological medical data (Ricklefs et al., 2002), in environmental data, or in the study of reliability (Yamada et al., 1985).

In this paper, we propose the GME estimator as an alternative to Maximum Likelihood Estimation (MLE), since it is widely used in literature in cases of non linear function (Golan et al., 2001). In particular we can highlight some points:

- The GME approach uses all the data points and does not require restrictive moment or distributional error assumptions.
- Thus, unlike the MLE estimator, the GME is robust for a general class of error distributions.
- The GME estimator can be used when the sample is small, when there are many covariates, and when the covariates are highly correlated.

It means that the GME was proposed because it is particularly useful when there are no restrictive sampling assumptions or in the case of ill-posed problems or

underdetermined ones, and also because in these cases MLE is unattractive and not robust to the underlying (unknown) distribution (Golan, 2008).

Moreover, RSS (McIntyre, 1952) could be used to increase the precision of estimates and to reduce costs, by using the researchers' experiences or inexpensive measurements.

The format of this paper is as follows. In Section 2 the Gompertz's model is introduced in its characteristics and analytical formulation by considering the MLE method. In Section 3, firstly the classic GME formulation is specified and then the Gompertz's model is expressed in the framework of the GME. In Section 4, the RSS method is presented and a discussion on how it can be utilized for applications in which the Gompertz's model is used. In Section 5, Monte Carlo experiments show the numerical performance for the proposed estimators, based on the RSS and SRS sampling schemes. Section 6 contains concluding remarks and discussion of future works.

2. The Gompertz's Model

Benjamin Gompertz, in 1825, showed that the mortality rate increases in a geometrical progression, defining one of the most informative actuarial functions for investigating the ageing process. He observed a law of geometrical progression in death rates by analysing a sample of people, aged between 20 and 60 years, in England, Sweden, and France. Gompertz's law has become the most successful law to model the dying out process of living organisms (Willemse et al., 2000). The relationships of the force of mortality, such as the power function of age, is also called the Weibull model (Carey, 2001) and it can be expressed by the following formula:

$$y = B \cdot C^x + \varepsilon \quad (1)$$

Where $B > 0$ reflects the general level of mortality in the population, and $C > 1$ is Gompertz's constant that reflects the rate at which the force of mortality increases with age; $x > 0$ is the age, and y is the force of mortality at age x .

The estimates of the parameters B and C can be obtained numerically using different methods, however, the MLE is the most commonly used for this model and it is discussed by Garg et al. (1970).

The MLE estimation method, presents the assumption that the random variables D_x (number of deaths at age x before reaching age $x+1$) follows a Binomial distribution: $B(n_x, q_x)$. The p_x , reported in the following formula (2), is the conditional probability function of the Gompertz's model given in (1):

$$p_x = e^{-\int_x^{x+1} \mu_t dt} = e^{-\int_x^{x+1} BC^t dt} = e^{-BC^x(C-1)/\log C} \quad (2)$$

Or considering the log function:

$$\log p_x = -B \cdot C^x \cdot (C-1)/\log C \text{ and } \log p_{x+1}/\log p_x = C \quad (3)$$

The other parameters are defined as follows: $q_x = 1 - p_x$, n_x is the number of people of age x and d_x is the value of the random variable D_x . Given this assumption, the logarithm of the joint likelihood function is the following:

$$\log L(B, C) = \sum_x (n_x - d_x) \cdot \log p_x + d_x \cdot \log(1 - p_x) \quad (4)$$

The general way of estimating the parameters is to use Nonlinear Maximization (NM), so as to find the minimum of the following function and to plug into (2) for the p_x :

$$\min_{B,C} \sum_x \left[(n_x - d_x) \cdot B \cdot C^x \cdot (C-1) / \log C - d_x \cdot \log \left(1 - e^{-B \cdot C^x \cdot (C-1) / \log C} \right) \right] \quad (5)$$

The NM problem can be solved via numerical methods such as the Newton-Raphson iteration or the Simplex algorithm.

3. Generalized Maximum Entropy

In the framework of the Information Theoretic we propose the use of the Generalized Maximum Entropy (GME) to estimate the non linear relationship between age and the force of mortality. In this section, we start by briefly describing the traditional maximum entropy (ME) estimation method, then we introduce the GME formulation as a method for recovering information from the Gompertz's model.

3.1. The GME estimation method

The *Entropy of Information* was first introduced by Claude Shannon in 1948 as a propriety associated to any probability distribution, defining an axiomatic method of measuring the uncertainty (state of knowledge) of a collection of events.

Letting X be a random variable with possible outcomes $\{x_1, x_2, \dots, x_s\}$, with the corresponding probability $P = \{p_1, p_2, \dots, p_s\}$ such that $\sum p_i = 1$, Shannon defined the *entropy of information* of probabilities distribution function as:

$$H(P) = -k \cdot \sum_{i=1}^s p_i \cdot \log(p_i) \quad (6)$$

Where k is a constant usually equal to 1; and $0 \cdot \ln(0) = 0$. The quantity $\{-\log(p_i)\}$ is called self information of the x_i event. The average on the self information is defined as the Entropy. The function $H(P)$ is called *Entropy*, *Shannon's Entropy* or *Information Entropy*.

Edwin Thompson Jaynes (1957a, 1957b) extended the entropy of information by defining the *Maximum Entropy Principle* (MEP). The MEP estimates an unknown probability distribution from given moment constraints and adding up normalization constraints on the probabilities.

The frequency that maximizes entropy is an intuitively reasonable estimate of the true distribution when we lack any other information. If we have information about an experiment, such as the sample moments, or non-sample information about the random variable, such as restrictions from economic theory, we can to alter our "intuitively reasonable" estimate. The method of maximum entropy proceeds by choosing the distribution that maximizes entropy, subject to the sample and non-sample information.

Letting X be a random variable with possible outcomes $\{x_1, x_2, \dots, x_s\}$, the objective of the MEP is to recover the unknown probability distribution $P = \{p_1, p_2, \dots, p_s\}$, taking into account: *consistency constraints*, by defining the functions $\{f_1(x_i), f_2(x_i), \dots, f_T(x_i)\}$, which represents the constraints (or information) generated by the data; *measurable values* $\{y_1, y_2, \dots, y_T\}$, obtained by a priori knowledge on the phenomenon, or by training dataset.

The constraints generated from the data and the measurable values, are expressed by the following equation:

$$\sum_{i=1}^s f_t(x_i) \cdot p_i = y_t \quad t = 1, 2, \dots, T \quad (7)$$

The adding up *normalization constraints*, which means $\sum p_i = 1$, are the following:

$$\sum_{i=1}^s p_i = 1 \quad (8)$$

The MEP estimates the probability distribution by the maximization of the entropy function, equation (6), based on the consistency and normalization constraints, equation (7, 8), where the consistency constraints are expressed in form of expectation values of a discrete random variable¹:

Amos Golan *et al.* proposed in 1996 an alternative method to solve many standard and ill-posed econometric problems in the framework of the MEP developed by Jaynes,

This information theoretic method, which is called Generalized Maximum Entropy (GME), is based on the *re-parameterization* and *re-formulation* of a general linear model. Considering a regression model $\mathbf{y} = \mathbf{X}\boldsymbol{\beta} + \boldsymbol{\varepsilon}$ with n units and m variables, the coefficients and the error terms can be re-parameterized as a convex combination of expected value of a discrete random variable, as in the following equation:

$$\mathbf{y}_{n,1} = \mathbf{X}_{n,m} \cdot \boldsymbol{\beta}_{m,1} + \boldsymbol{\varepsilon}_{n,1} = \mathbf{X}_{n,m} \cdot \mathbf{Z}_{m,M} \cdot \mathbf{p}_{m \cdot M,1} + \mathbf{V}_{n,N} \cdot \mathbf{w}_{n \cdot N,1} \quad (9)$$

The matrix inner products $(\mathbf{Z} \cdot \mathbf{p})$ and $(\mathbf{V} \cdot \mathbf{w})$, represent respectively the re-parameterization of the regression coefficients and the error terms as in form of expected value of a discrete random variable.

The matrices \mathbf{Z} and \mathbf{V} are diagonal and the generic matrix element is represented respectively by the vectors $\mathbf{z}'_k = [-c \quad -c/2 \quad 0 \quad c/2 \quad c]$ with $\{k = 1, \dots, m\}$ and $\mathbf{v}'_h = [-b \quad -b/2 \quad 0 \quad b/2 \quad b]$ with $\{h = 1, \dots, n\}$.

These vectors (\mathbf{z}_k and \mathbf{v}_h) define the support values, called *fixed points*, usually with five elements ($M=N=5$) with a given constants ' c ' and ' b ', uniformly and symmetrically chosen around zero with equally spaced distance discrete points, or as Golan suggested, for the error terms, to follow the six-sigma rule (Pukelshiem, 1994), where the number of fixed points is 3, and the constant c is equal to \widehat{s}_y .

The super vectors \mathbf{p} and \mathbf{w} associated are probabilities and have to be estimated by maximization of the Shannon entropy function:

$$H(p, w) = -\mathbf{p}'_{1,m \cdot M} \cdot \ln \mathbf{p}_{m \cdot M,1} - \mathbf{w}'_{1,n \cdot N} \cdot \ln \mathbf{w}_{n \cdot N,1}$$

subjected to some *normalization* and *consistency constraints*. The steps for the GME algorithm are shown in the following table 1.

Table 1. The Generalized Maximum Entropy Algorithm

1. **Re-parameterize** the unknown parameters and the disturbance terms as a convex combination of expected value of a discrete random variable;
2. **Re-formulate the model** with the new re-parameterization as the data constraint;
3. **Define the GME problem as non-linear programming** problem in the following form:

Objective Function = Shannon's Entropy Function

1. The consistency constraints, which represents the new formulation of the model;
2. The normalization Constraints.
4. **Solve the non-linear programming** by using numerical method

The constraints defined for estimating the unknown parameters refer to *consistency* and *normalization constraints*. The first one represents the information generated from the

data, that means a part of the model defined in the equation (9); the second one identifies the conditions: $0 \leq p_{ki} \leq 1$, $\{i=1, \dots, M; k=1, \dots, m\}$, $\sum p_{ki}=1$ $\{k=1, \dots, m\}$ and $0 \leq w_{hi} \leq 1$, $\{i=1, \dots, N; h=1, \dots, n\}$, $\sum w_{hi}=1$ $\{h=1, \dots, n\}$.

The main advantages of using GME estimation method, as above defined (*supra* § 1), are its desirable properties which can be briefly summarized: does not require restrictive moments or distributional error assumptions; it's robust for a general class of error distributions; may be used with small samples, with many highly correlated covariates; moreover, using the GME method, it is easy to impose nonlinear and inequality constraints.

Therefore the GME works well in case of ill-behaved data and the above listed cases, where the MLE estimator cannot proceed.

In the following section the GME formulation for the Gompertz's model is explained, moreover the definition of the optimization function and both consistency and normalization constraints will be discussed.

3.2. GME for the Gompertz's Model

The GME formulation as a method for recovering information from the Gompertz's model, starts from the relationship between the age and the force of mortality, expressed by:

$$y_{n,1} = B \cdot C^{x_{n,1}} + \varepsilon_{n,1} \quad (9)$$

The starting point, following the algorithm in table 1, is the *re-parameterization* of the unknown parameters and error terms, as a convex combination of expected values. For the Gompertz's model, the number of predictor variables is just one, age, which means, as in the above general formulation, $m=1$ and the matrix \mathbf{Z} is just a vector $\mathbf{z}'_k = [-c \quad -c/2 \quad 0 \quad c/2 \quad c]$, considering $M=5$ fixed points. The error term is obtained by considering the diagonal matrix with generic element $\mathbf{v}'_k = [-3 \cdot \hat{s}_y \quad 0 \quad 3 \cdot \hat{s}_y]$, based on the three-sigma rule, with $N=3$ fixed points.

The *re-parameterization* as convex combination of expected values is expressed in the following equations:

$$B = \mathbf{z}'_{1,M} \cdot \mathbf{p}_{M,1} \quad (10)$$

$$C = \mathbf{c}'_{1,M} \cdot \mathbf{q}_{M,1} \quad (11)$$

$$\varepsilon_{n,1} = \mathbf{V}_{n,n \cdot N} \cdot \mathbf{w}_{n \cdot N,1} \quad (12)$$

Moreover, *normalization constraints* are necessary, because for each probability vector of the coefficients and error terms (ex., $\mathbf{p}_{M,1}$), the sum of probabilities estimated have to be equal to 1, which means $\sum p_i = 1$, $\sum q_i = 1$ and $\sum w_{hi} = 1$ $\{h=1, \dots, n\}$.

These constraints are formalized by the following expressions:

$$\mathbf{p}'_{1,M} \cdot \mathbf{1}_{M,1} = 1 \quad (13)$$

$$\mathbf{q}'_{1,M} \cdot \mathbf{1}_{M,1} = 1 \quad (14)$$

$$\mathbf{J}^*_{n,n \cdot N} \cdot \mathbf{w}_{n \cdot N,1} = \mathbf{1}_{n,1} \quad (15)$$

The matrix \mathbf{J}^* is the Kronecker product between the identity matrices $\mathbf{I}_{n,n}$ and the vector of one $\mathbf{1}_{M,1}$, which means $\mathbf{J}^*_{n,n \cdot N} = (\mathbf{I}_{n,n} \otimes \mathbf{1}'_{1,N})$:

$$\mathbf{J}_{n,n \cdot N}^* \cdot \mathbf{w}_{n \cdot N,1} = \mathbf{1}_{n,1} \Rightarrow \begin{bmatrix} 111 & 0 & \dots & 0 \\ 0 & 111 & \dots & 0 \\ \vdots & \vdots & \ddots & \vdots \\ 0 & 0 & \dots & 111_n \end{bmatrix} \cdot \begin{bmatrix} \mathbf{w}_1 \\ \mathbf{w}_2 \\ \vdots \\ \mathbf{w}_n \end{bmatrix} = \begin{bmatrix} 1 \\ 1 \\ \vdots \\ 1_n \end{bmatrix} \quad (16)$$

Given the re-parameterization, the Gompertz's model can be re-formulated in the following way, where all the parameters, unknowns and error terms, are expressed as expected values:

$$\mathbf{y}_{n,1} = (\mathbf{z}_{1,M}' \cdot \mathbf{p}_{M,1}) \cdot (\mathbf{c}_{1,M}' \cdot \mathbf{q}_{M,1})^{\mathbf{x}_{n,1}} + \mathbf{V}_{n,n \cdot N} \cdot \mathbf{w}_{n \cdot N,1} \quad (17)$$

The definition of GME problem as non linear programming problem is the following:

$$\textbf{Objective function:} \quad \text{Max } H(P, Q, W) = -\mathbf{p}' \cdot \ln \mathbf{p} - \mathbf{q}' \cdot \ln \mathbf{q} - \mathbf{w}' \cdot \ln \mathbf{w}$$

$$\textbf{Consistency Constraints:} \quad \mathbf{y}_{n,1} = (\mathbf{z}_{1,M}' \cdot \mathbf{p}_{M,1}) \cdot (\mathbf{c}_{1,M}' \cdot \mathbf{q}_{M,1})^{\mathbf{x}_{n,1}} + \mathbf{V}_{n,n \cdot S} \cdot \mathbf{w}_{n \cdot S,1}$$

$$\textbf{Normalization Constraints:} \quad \mathbf{p}_{1,M}' \cdot \mathbf{1}_{M,1} = 1 ; \mathbf{q}_{1,M}' \cdot \mathbf{1}_{M,1} = 1 ; \mathbf{J}_{n,n \cdot N}^* \cdot \mathbf{w}_{n \cdot N,1} = \mathbf{1}_{n,1}$$

The non-linear programming system is solved by the formulation of the Lagrangian function and the first order conditions which provides the basis for the solution. The Lagrangian function is expressed by the following formulation:

$$L = -\mathbf{p}' \cdot \ln \mathbf{p} - \mathbf{q}' \cdot \ln \mathbf{q} - \mathbf{w}' \cdot \ln \mathbf{w} + \lambda' \cdot \left[\mathbf{y} - (\mathbf{z}' \cdot \mathbf{p}) \cdot (\mathbf{c}' \cdot \mathbf{q})^{\mathbf{x}} - \mathbf{V} \cdot \mathbf{w} \right] + \theta \cdot [1 - \mathbf{p}' \cdot \mathbf{1}] + \nu \cdot [1 - \mathbf{q}' \cdot \mathbf{1}] + \tau' \cdot [1 - \mathbf{J}^* \cdot \mathbf{w}] \quad (18)$$

where θ , ν , λ , τ , are respectively the scalars and the vectors of the Lagrangian multipliers. By taking the gradient of L it is possible to derive the first-order-condition. However, the equations system will not be in closed form and to get the final values, a numerical optimization technique (successive quadratic programming method) should be used to compute probabilities (Ciavolino, 2007).

The estimations can be expressed by:

$$\hat{B} = \mathbf{z}_{1,M}' \cdot \hat{\mathbf{p}}_{M,1} \quad (19)$$

$$\hat{C} = \mathbf{c}_{1,M}' \cdot \hat{\mathbf{q}}_{M,1} \quad (20)$$

$$\hat{\mathcal{E}} = \mathbf{V}_{n,n \cdot S} \cdot \hat{\mathbf{w}}_{n \cdot S,1} \quad (21)$$

Respectively for the general level of mortality (19), the force of mortality (20) and the error term (21).

4. Ranked Set Sampling Method

The concept of Ranked Set Sampling is a recent development that enables more structure to be provided to the collected sample items, although the name is a bit of a misnomer as it is not as much a sampling technique as a data measurement technique.

This approach to data collection was first proposed by McIntyre in 1952, for situations where taking the actual measurements for sample observations is difficult (e.g., costly, destructive, time-consuming), but mechanisms for either informally or formally ranking a set of sample units is relatively easy and reliable.

In particular, McIntyre was interested in improving the precision in estimation of average yield from large plots of arable crops without a substantial increase in the number of fields from which detailed, expensive and tedious measurements needed to be collected. For discussions of some of the settings where ranked set sampling techniques have found application, see Chen *et al.* (2004).

The idea of using the RSS for the Gompertz's model, is in some practical applications, where the response variable is too expensive to measure, destructive or time-consuming, but the predictor variable can be measured easily with relatively negligible cost.

Some examples can be found in studies in the ecological or biology fields, where, for instance, the death rate of plants or insects caused by the contamination in hazardous waste sites are analyzed using the Gompertz's law. The analysis can be improved by inspections for ranking areas of soil based on visual inspection or other expert opinion about the sample units (Chen *et al.*, 2004).

The scheme we used in this study is the balanced RSS, which involves drawing m sets of Simple Random Samples, each of size m from a population, and ranking each set with respect to the variable of interest, for instance *the age*. Then, from the first set, the element with the smallest rank is chosen for the actual measurement. From the second set, the element with the second smallest rank is chosen. The process is continued until we have selected the largest unit from the last simple random sample.

The procedure can be formalized as the following steps:

1. Select m^2 units randomly from the population.
2. Randomly allocate the m^2 units into m subsets, each of size m .
3. Order the units within each subset, based on the perception of the interest variable.
4. In the RSS, the smallest unit in the first subset is selected for actual measurement, the second smallest unit in the second subset is selected for actual measurement, we continue in this process until the largest ranked unit is selected from the m^{th} subset.
5. These four steps are called *cycle* and can be repeated r times until the desired sample size is reached, equal to $n = r \cdot m$.

If just three ranks and one cycle are considered, the selected RSS is denoted by: $\{X_{[1]}, X_{[2]}, X_{[3]}\}$. In order to select a RSS with a sample size equal to $n = r \cdot m$, the cycle is repeated r independent times, yielding the following sample:

$$\{X_{[1],j}, X_{[2],j}, \dots, X_{[m],j}\}, \text{ for } j = 1, \dots, r \quad (22)$$

It can be noted that the selected elements are mutually independent order statistics but not identically distributed. In practice, the set size m is kept small to ease the visual ranking, RSS literature suggested that $m = 2, 3, 4, 5$ or 6 .

Table 2. Ranked Set Sampling Design

Cycle ($r=4$)	Rank ($m=3$)		
	Young	Adult	Elderly
1	$X_{[1],1}$	-	-
	-	$X_{[2],1}$	-
	-	-	$X_{[3],1}$
2	$X_{[1],2}$	-	-
	-	$X_{[2],2}$	-
	-	-	$X_{[3],2}$
3	$X_{[1],3}$	-	-
	-	$X_{[2],3}$	-

	-	-	$X_{[3],3}$
4	$X_{[1],4}$	-	-
	-	$X_{[2],4}$	-
	-	-	$X_{[3],4}$

Table 2 shows an illustrative example, where the set size m is equal to 3 (the number of ranks) and the cycles $r = 4$. In the table, we assume that the units are selected according to their age, therefore there are three ranks: *Young*, *Adult*, *Elderly*, and the rows represent the ordered sample within each cycle, where $X_{[i],i}$ is the sample unit included in the RSS, which represents the i^{th} order unit in the i^{th} cycle.

The number of units randomly selected are 36, $m^2=3^2=9$, for 4 cycles, but only 12 are included in the RSS.

5. Simulation and Results

We perform different simulation studies in order to draw conclusions about the performance of GME and MLE estimation methods for the Gompertz's model by using the RSS technique. The simulation experiments start with the analysis of the fixed points to evaluate the sensitivity of the GME for the Gompertz's model. The analysis of the sensitivity is made by changing the value of the constant ' c ' and the number of the fixed points. This analysis is recommended and useful (Golan *et al.*, 1996), to verify the support spaces on the coefficients and error terms to measure the sensitivity of results across support space specifications.

Four simulation experiments are considered to measure the sensitivity of the GME estimator: the first two simulations, to determine the variance and the bias of the Gompertz's coefficients; the last two simulations, to examine the variance and the bias of the error terms.

The sensitivity analysis is measured in term of Bias and Mean Squared Error (MSE), as in the following equations for the B coefficient:

$$\text{Bias}(B) = n^{-1} \cdot \sum_{i=1}^n (\hat{B}_i) - B$$

$$\text{MSE}(B) = n^{-1} \cdot \sum_{i=1}^n (\hat{B}_i - B_i)^2$$

After the sensitivity simulation studies, that allow to choose the best support space specification for the GME estimator, we performed two simulation studies for the evaluation GME and MLE performance based on the SRS and RSS.

The measures Bias and MSE are used to evaluate the performance of the estimator applied to both sampling techniques, computing also relative efficiency (*eff*); where the *eff* of B is defined as:

$$\text{eff}(B) = \frac{\text{MSE}(B_{\text{GME}})}{\text{MSE}(B_{\text{MLE}})}$$

To perform all the simulation studies, the parameters of the Gompertz's function B and C were first initialized to 0,3 and 2. Using these values as initial values, a simulation study was carried out by generating 1000 samples according to the following relationship:

$$y_i = 0,3 \cdot (2)^{x_i} + \varepsilon_i, \quad i = 1, 2, \dots, n \quad (23)$$

where $\varepsilon \sim N(0,1)$ and $X \sim \text{Exp}(1)$. The sensitivity simulation studies for the GME parameters choice were conducted by fixing the sample size to be 20.

All the pseudo distributions varieties are generated from a build in subroutines of International Mathematical and Statistical Library (IMSL). The solution of GME system were solved and generated by using a successive quadratic programming method to solve a nonlinear programming problems based on the NCONF and NLPQL subroutines. The basic FORTRAN codes for computational purposes were developed by Schittkowski (1986) and necessary modifications were made by the authors.

5.1. The Sensitivity Analysis of the GME Estimates

Tables 3, 4, 5 and 6 show the results of the sensitivity analysis for the *fixed points*. It has been considered four experiments: the first two to evaluate the support value 'c' and the number of fixed points for the parameters B and C; the last two for the error term (*supra* § 3).

Experiment 1. This experiment is performed to select the support parameter bounds of B and C. we start by using only three support values for each of these parameters in the interval $[-c, 0, c]$, where $c = 1, 5, 10, 50$ or 100 . The support bounds of the error term are fixed to be three data points selected according to the three sigma rule. The results of this simulation study are given in Table.3. The results indicated that the GME estimates were more accurate and more efficient than the MLE estimates for all the selected support bound. However, the best support bound should be selected in the interval $[-5, 0, 5]$ for both parameters, which; almost, gives the minimum MSE and bias for the GME estimates.

Table 3. Selecting the 'c' value of the parameters (B e C) support points

METHODS		GME: different 'c' VALUES of the fixed points					MLE
PARAMETERS		[-1, 1]	[-5, 5]	[-10, 10]	[-50, 50]	[-100, 100]	
B	Bias	-0,0098	-0,0107	-0,0097	-0,0114	-0,012	0,1715
	MSE	0,0108	0,011	0,0115	0,0153	0,0131	0,0394
C	Bias	-0,0159	-0,0029	-0,0092	0,0171	0,0141	0,0504
	MSE	0,0028	0,0022	0,0023	0,0026	0,0028	0,0034

Experiment 2. This experiment is repeated under the same assumptions of experiment two, and by fixing the support bounds of the parameters in the interval $[-5, 5]$. The aim now is to determine the number of support points within the suggested interval that leads to better results. Consequently, we start to increase the numbers of support points and allocate them in an equidistant fashion. The results of this experiment in Table. 4 suggested that there is no statistical improvement of the estimators if we increase the number of support points from 3 to 7 data points. Therefore, according to this simulation experiment, we suggest to fix the number of support points to be three in the interval $[-5, 5]$.

Table 4. Fixing the number of support points for the parameters (B e C)

METHODS		GME: Different NUMBERS of fixed points in [-5 5]					MLE
PARAMETERS		3	4	5	6	7	
B	Bias	-0,0107	-0,0071	-0,0104	-0,0077	-0,0098	0,1715
	MSE	0,011	0,0128	0,0118	0,014	0,0128	0,0394
C	Bias	-0,0029	-0,0136	-0,0112	-0,008	-0,0077	0,0504
	MSE	0,0022	0,0024	0,0023	0,0024	0,0021	0,0034

Experiment 3. We conducted a sampling experiment based on the experimental design outlined above and based on the results of experiment 1 and experiment 2. Hereafter, the support bounds of the error term is selected in the interval $[-c S_y, 0, c S_y]$, then we start changing the value of 'c' to be equal to 1, 2, 3, 4 and 5. The results of this experiment in Table 5 indicate that the best value for 'c' is 3, this result is consistent with the results of Golan et al (1996), which suggest in using the three sigma rules when we setup the support bonds of the error terms.

Table 5. Selecting the 'c' value for the error term

		GME: different 'c' VALUES of the fixed points					MLE
METHODS		[-1S, 1S]	[-2S, 2S]	[-3S, 3S]	[-4S, 4S]	[-5S, 5S]	
PARAMETERS							
B	Bias	-0,0067	-0,0107	-0,0107	-0,0117	-0,0139	0,1715
	MSE	0,0107	0,0114	0,011	0,0129	0,014	0,0394
C	Bias	-0,0144	-0,0143	-0,0029	-0,0098	-0,0111	0,0504
	MSE	0,0023	0,0021	0,0022	0,0019	0,0018	0,0034

Experiment 4. Likewise experiment 2, the simulation trial were repeated by shifting the support points of the error term to be equally spaced in the interval $[-3 S_y, 0, 3 S_y]$. The results of this experiment are in Table 6 suggested that we should fix three support values of the error term according to the three sigma rule.

Table 6. Fixing the number of support points for the error term

		GME: different 'c' VALUES of the fixed points					MLE
METHODS		3	4	5	6	7	
PARAMETERS							
B	Bias	-0,0107	-0,0003	-0,0032	-0,0005	-0,0023	0,1715
	MSE	0,011	0,0169	0,0191	0,0168	0,0165	0,0394
C	Bias	-0,0029	-0,0138	-0,0092	-0,0127	-0,0161	0,0504
	MSE	0,0022	0,0021	0,0028	0,0022	0,0021	0,0034

5.2. Performance Analysis of the RSS technique

In this section, based on the main simulation conditions and the results of the last four experiments, we are comparing between two the estimation methods from the view of point of the sampling techniques. Two experiments were performed, the first one based on SRS and the other one based on RSS. Noting that, in the RSS scheme the ranking done based on the X variable. Other ranking schemes could be used such as Double RSS (Al-Saleh and Al-Kadiri, 2000), which is not of our interest in this paper, to be an alternative to the classical ranking scheme.

Experiment 5. Based on the results of experiment 1 to experiment 4, by selecting three support values for each of the parameters in the $[-5, 5]$ interval and three support values the error term according to the three sigma rule, this experiment, under the simulation assumption outlined above, is conducted by increasing the sample size n , i.e., $n = 20, 25, 30, 40$ and 50 . The results in Table 7 indicate that the GME is more accurate and more efficient than the MLE estimates for all sample sizes.

Table 7. Comparisons between GME and MLE using SRS

N	Method	C			B		
		Bias	MSE	eff	Bias	MSE	eff
20	GME	-0,0029	0,0022	1,54	-0,0107	0,0110	3,58
	MLE	0,0504	0,0034		0,1715	0,0394	
25	GME	-0,0150	0,0029	1,14	0,0043	0,0138	2,44
	MLE	0,0496	0,0033		0,1597	0,0337	
30	GME	-0,0085	0,0029	1,21	-0,0005	0,0096	4,11
	MLE	0,0510	0,0035		0,1726	0,0395	
40	GME	-0,0046	0,0018	1,67	-0,0006	0,0090	4,09
	MLE	0,0474	0,0030		0,1663	0,0368	
50	GME	-0,0013	0,0012	2,08	0,0007	0,0086	3,36
	LE	0,0429	0,0025		0,1482	0,0289	

Experiment 6. The aim of this experiment is to improve the simulation results by using the simulated RSS. For the RSS sampling scheme we used a set size $r = 4$ or 5 and the number of cycles $m = 5, 6, 8$ and 10; to achieve the desired sample size. The results of this experiment are given in Table 8.

Table 8. Comparisons between GME and MLE using RSS

r	m	Method	C			B		
			Bias	MSE	Eff	Bias	MSE	eff
5	4	GME	0,0005	0,0056	1,76	0,0039	0,0059	8,55
		MLE	0,0532	0,0099		0,1938	0,0505	
5	5	GME	-0,0302	0,0053	1,85	0,0154	0,0144	3,55
		MLE	0,0543	0,0098		0,1953	0,0511	
6	5	GME	-0,0101	0,0025	1,64	-0,0003	0,0128	3,89
		MLE	0,0551	0,0041		0,1917	0,0499	
8	5	GME	-0,0179	0,0032	1,22	-0,0161	0,0107	3,78
		MLE	0,0506	0,0039		0,1726	0,0405	
10	5	GME	0,0078	0,0033	1,03	-0,0003	0,0084	6,12
		LE	0,0535	0,0034		0,1929	0,0514	

We conclude by summarizing our findings:

1. In this study we used simulated ranked data with GME and MLE estimation methods to estimate the parameters of the Gompertz's model.
2. Under the simulation assumptions, GME based on both sampling techniques SRS and RSS gives a better estimate than MLE from the MSE point of view. The simulated efficiency score is more than 1 in both experiments for all parameters and under different sample sizes.

6. Conclusions and Discussion

To sum up, we consider fitting the Gompertz's model by two different estimation methods MLE and GME and based on two different sampling techniques: SRS and RSS. Despite the sampling technique, the simulation results demonstrate that the GME estimates are superior and often more efficient than MLE estimates in terms of MSE.

Table 9. Simulation results GME (SRS vs RSS)

N	Method	B			C		
		Bias	MSE	Eff	Bias	MSE	eff
20	SRS	-0,0107	0,0110	1,86	-0,0029	0,0022	0,39
	RSS	0,0039	0,0059		0,0005	0,0056	
25	SRS	0,0043	0,0138	0,96	-0,015	0,0029	0,55
	RSS	0,0154	0,0144		-0,0302	0,0053	
30	SRS	-0,0005	0,0096	0,75	-0,0085	0,0029	1,16
	RSS	-0,0003	0,0128		-0,0101	0,0025	
40	SRS	-0,0006	0,0090	0,84	-0,0046	0,0018	0,56
	RSS	-0,0161	0,0107		-0,0179	0,0032	
50	SRS	0,0007	0,0086	1,02	-0,0013	0,0012	0,36
	RSS	-0,0003	0,0084		0,0078	0,0033	

For all situations used in the simulation study considering different sample size, the GME estimates are more efficient and more accurate than the MLE estimates. However, there is no improvement in parameter estimation by MLE rely on RSS.

However, Table 9 shows the results obtained by comparing the SRS and RSS sampling schemes in the context of GME. We find out that they are comparable and no method is better than any other method, therefore, we can used GME as a robust estimation method to fit Gompertz's model whatever the sample technique is used.

Moreover it is widely shown in the literature that in empirical circumstances, the RSS can be employed to gain more information than SRS while keeping the cost of, or the time constraint on, the sampling about the same.

Consequently, the GME estimator based on any sampling scheme can be recommended for estimating the parameter of Gompertz's model as an alternative estimation method to the classical MLE.

Moreover, other statistical procedures and new methodologies in the context of RSS are proposed, like extreme ranked set sampling (ERSS), median ranked set sampling (MRSS) or L ranked set sampling (LRSS) by Al-Nasser (2007). It may be useful to investigate these different methods for the Gompertz's model, considering the GME estimator.

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¹ The estimates will not be in closed form and to get the final values, a numerical optimization technique (successive quadratic programming method) should be used to compute probabilities.

THE RASCH MODEL FOR EVALUATING ITALIAN STUDENT PERFORMANCE¹

Ida CAMMINATIELLO

University of Naples Federico II, Naples, Italy

E-mail: camminat@unina.it



Michele GALLO

University of Naples L'Orientale, Naples, Italy

E-mail: mgallo@unior.it



Tullio MENINI

University of Naples L'Orientale, Naples, Italy

E-mail: menini@unior.it



Abstract: In 1997 the Organisation for Economic Co-operation and Development (OECD) launched the OECD Programme for International Student Assessment (PISA) for collecting information about 15-year-old students in participating countries.

Our study analyse the PISA 2006 cognitive test for evaluating the Italian student performance in mathematics, reading and science comparing the results of different local governments. For this purpose the most proper statistic methodology is Item Response Theory - IRT that collects several models, the simplest is Rasch Model – MR (1960). As the items used in the analysis are both dichotomous that polytomous, we apply Partial Credit Model (PCM).

Key words: Programme for International Student Assessment; Italian student performance; Partial Credit Model

1. Introduction

The prosperity of countries now derives to a large extent from their human capital. This consciousness urges many countries to monitor students' learning. Comparative international assessments can extend and enrich the national picture by providing a larger context within which to interpret national performance. In response to this need, the Organisation for Economic Co-operation and Development (OECD) launched the OECD Programme for International Student Assessment (PISA) in 1997.

The PISA surveys have been designed to collect information about 15-year-old students in participating countries. PISA examines how well students are prepared to meet the challenges of the future, rather than how well they master particular curricula. PISA

surveys take place every three years. The first survey took place in 2000, the second in 2003 and the third in 2006. For each assessment, one of the three areas (science, reading and mathematics) is chosen as the major domain and given greater emphasis. The remaining two areas, the minor domains, are assessed less thoroughly. In 2000 the major domain was reading; in 2003 it was mathematics and in 2006 it was science. The results of these surveys have been published in a series of reports (OECD, 2001, 2003, 2004, 2007) and a wide range of thematic and technical reports.

In this paper we focus on the PISA 2006 survey. In all countries the survey includes:

- a cognitive test for evaluating the student performance
- a student questionnaire to collect information from students on various aspects of their home, family and school background
- a school questionnaire to collect information from schools about various aspects of organisation and educational provision in schools.

As in previous surveys, additional questionnaire material was developed, which was offered as international options to participating countries. In PISA 2006, two international options were available, the Information Communication Technology (ICT) familiarity and the parent questionnaire.

The PISA 2006 results show wide differences in the performance of countries that participated to the survey. Also the Italian results show performance differences within the country, in particular between local governments and between different schools (INVALSI, 2007).

Our study analyses the cognitive test for evaluating the Italian student performance in reading, mathematics and science, comparing the results of different local governments. Several papers show the measures obtained by students, we are going to focus on measurement instrument for studying:

- the abilities required by PISA 2006 test to which the Italian students are or not able to answer;
- if the students of a local government are scoring better than the students of another local government on an item (Differential Item Functioning, DIF).

For this purpose the most proper statistic methodology is Item Response Theory - IRT (Baker & Kim 2004), that collects several models, the simplest is Rasch Model – MR (1960). As the items used in the analysis are both dichotomous and polytomous, we apply Partial Credit Model (PCM).

2. Rasch model

The aim of the IRT is to test people. Hence, their primary interest is focused on establishing the position of the individual along some not directly observable dimension called latent trait. Because of the many educational applications the *latent trait* is often called *ability*.

The IRT derives the probability of each response as a function of the latent trait and some *item parameters*. The same model is then used to obtain the likelihood of ability as a function of the actually observed responses and, again, the item parameters. The ability value that has the highest likelihood becomes the ability estimate. For this purpose IRT makes the important assumption of *local independence*. This means that the responses given to the separate items in a test are mutually independent *given ability*.

The objective of each IRT model is to predict the probability that a person will give a certain response to a certain item. People can have different levels of ability, and items can have different levels of ability. To keep track of this, we denote the probability of a correct response with $P_{n,s}$: the index s refers to the item, and the index n refers to the person. When an item allows for more than two options, we denote the probability with $P_{n,s,x}$ where the index x refers to the options.

The simplest IRT model is the RM. Rasch's basic idea is that the Models for Measurement make it possible to *measure properly*, and, equally importantly, to validate

which data conform to measurement and which does not: Rasch has specified demands for a social sciences measurement to be of the same quality as measurements in the natural sciences and he has then found out exactly which kind of statistical models conform to these specified requirements, namely the Models for Measurement (Rasch, 1968). The conclusion therefore is that a given data set yield measurements in Rasch's well-defined meaning of the word, if, and only if, the data conform to one of the Models for Measurement. So, if the Models for Measurement did not describe the data, then, in certain situations, it is considered better to discard the data than the model.

This view of Rasch's upon data is indeed controversial and quite a contrast to the traditional approach where the statistical model is expanded to fit the data. Closely connected to the Models for Measurement is the concept of specific objectivity, which by and large is the name Rasch chose for his requirements for measurements.

For a dichotomous item the RM has only one item parameter. The probability of a correct response given l'item parameter δ_s , and the person parameter β_n , is

$$P_{n,s} = \frac{\exp(\beta_n - \delta_s)}{1 + \exp(\beta_n - \delta_s)} \quad (1)$$

where δ_s characterizes the difficulty of item s , and β_n characterizes the ability of examinee n .

The literature offers a number of alternative procedures for estimating parameters, including Joint maximum likelihood, Conditional maximum likelihood (CML) and Marginal maximum likelihood (MML). Under appropriate assumptions these solutions are asymptotically equivalent, consistent and multivariate normal (Haberman, 1977; de Leeuw & Verhelst, 1986).

When the items are polytomous with a different number of categories which have not the same distance, the most proper version of the IRT is the Partial Credit Model (PCM) proposed by Wright & Masters (1982). The probability that a subject n answers to a item s through the category x ($x = 1, 2, \dots, w, \dots, M_s$) is calculated by tht formula:

$$P_{nsx} = \frac{\exp \sum_{j=0}^x [\beta_n - (\delta_{sj} + \tau_j)]}{\sum_{w=0}^{M_s} \exp \sum_{j=0}^w [\beta_n - (\delta_{sj} + \tau_j)]} \quad (2)$$

δ_{sj} characterizes the difficulty of item s , for the threshold j and τ_j are category thresholds.

2.1. Rasch diagnostics

In literature there are different tools to evaluate the goodness of fit of the model to observed data. One of the most used is based on the residuals analysis for each individual (or item). The residual can be standardized as follows:

$$z_{ns} = \frac{x_{ns} - E_{ns}}{\sqrt{w_{ns}}}$$

where w_{ns} is the estimated variance of responses reproduced by model, x_{ns} is the response of the individual n to the item s and E_{ns} is the expected value of the response.

The interpretation of standardised residuals is simple but too analytic because it is referred to each individual or item. For obtaining a synthetic information, the mean value of squared standardised residuals z_{ns}^2 can be calculated: $U_n = \frac{1}{K} \sum_{s=1}^K z_{ns}^2$ for each individual

where K is the number of items and $U_s = \frac{1}{H} \sum_{n=1}^H z_{ns}^2$ for each item where H the number of individuals.

The expected value of U_n and U_s (outfit or Unweighted Mean Square statistic) is equal to 1. However Linacre proposes different ranges around 1 according to the origin of observed data: for small samples and/or tests with few items, there is a good fit if the statistics is in the range [0.6; 1.4]; otherwise the values should be in [0.8; 1.2]. Anyway values greater than 2 are bad for the measurement.

It can be demonstrated that the outfit statistics is sensitive to big differences between β e δ ; for balancing this characteristic it is possible to weigh the squared residuals with the variance, obtaining another synthetic statistics defined INFIT (or Weighted

Mean Square statistic): $V_n = \frac{\sum_{s=1}^K (w_{ns} z_{ns}^2)}{\sum_{s=1}^K (w_{ns})}$ for each individual where K is the number of

items and $V_s = \frac{\sum_{n=1}^H (w_{ns} z_{ns}^2)}{\sum_{n=1}^H (w_{ns})}$ for each item where H the number of individuals

The infit statistic is sensitive to unexpected behaviour affecting responses to items near the person ability level and the outfit statistic is outlier sensitive, so it is useful to calculate both the statistics.

With reference to the estimations of parameters of RM $\hat{\beta}_n$ and $\hat{\delta}_s$ it is possible to

calculate the Standard Error (SE): $SE(\hat{\beta}_n) = \left[\frac{1}{\sum_{s=1}^K (w_{ns})} \right]^{\frac{1}{2}}$, and $SE(\hat{\delta}_s) = \left[\frac{1}{\sum_{n=1}^H (w_{ns})} \right]^{\frac{1}{2}}$

Producing a synthesis with respect to the SE of estimations $\hat{\delta}_s$ it is possible to calculate the mean square error: $ME_\delta^2 = \frac{1}{K} \sum_{s=1}^K [SE(\hat{\delta}_s)]^2$ the squared root of which supplies the mean error of item calibration ME_δ .

The ratio between such value and the squared root of unbiased variance SA_δ gives the separation index: $SI_\delta = \frac{SA_\delta}{ME_\delta}$ where $SA_\delta = \sqrt{(S_\delta^2 - ME_\delta^2)}$ and $S_\delta^2 = \frac{1}{K} \sum_{s=1}^K \delta_s^2$ is the variance of estimations $\hat{\delta}_s$. If the index is far from one, the item are well separated.

In terms of the separation index, reliability index can be expressed as follow: $RE_\delta = \frac{SI_\delta^2}{1 + SI_\delta^2} = \frac{SA_\delta^2}{S_\delta^2} = 1 - \frac{ME_\delta^2}{S_\delta^2}$. It has the property that $RE_\delta = 0$ if there is no reproducibility of the measures, $RE_\delta = 1$ if there is perfect reproducibility of the measures, otherwise, $0 \leq RE_\delta \leq 1$

The goodness of fit can be evaluated graphically by the analysis of Item Characteristic Curves (ICC) and Category Probability Curves (CPC). The ICC of i -th item represents the probability of achieving a given score for the item, depending on the parameter value β . The misfit of s -th item is observed when one or more points \hat{p}_{nsx} are not on the ICC of the item, where \hat{p}_{nsx} is the probability that individual n chooses the category x to item s , as specified by the Rasch model, with estimated parameters. The CPC provides the probability to choose each of the possible categories according to the difference between ability of the subjects, average difficulty of the item and thresholds among the categories. The thresholds correspond to the measures to which the adjacent categories are equally likely. Compared to the ICC the ordinate represents the expected score for the item, it is obtained by accumulating, for each ability level in abscissa, the product of the estimated probability for each response and the corresponding raw score.

To improve the goodness of fit of a model one can proceed to the elimination of all items (and/or individuals) that do not fit well through an iterative procedure. Often the set of excluded items helps to measure a separate dimension. However, in extreme cases, it can happen it is not possible to identify any set of items consistent with the hypothesis of the Rasch model: this can be caused by a ill calibrated questionnaire or a mixture of individuals apparently belonging to the same population, but in reality related to different populations.

The latter case can be a symptom of a different functioning of the items corresponding to distinct groups of individuals: this phenomenon is called Differential Item Functioning or DIF. More precisely, an item is considered biased if, conditionally to a certain level of ability, the probability of choosing a certain category of response differs systematically between subgroups of individuals (eg., Between males and females). If the presence of DIF is statistically significant, it will be necessary to identify homogeneous groups of individuals that present a good fit.

In literature there are several DIF diagnostics (Glas & Verhelst, 1995), but the most used and implemented in the most commonly used software (Wu, Adams & Wilson, 1998) is based on the residual analysis among the subgroups identified by one or more aggregation variables.

In order to compare the abilities of individuals and the difficulties of the items, one can use the person-item map, a simultaneous graphical representation of both individuals and items. It allows to assess both if an item is more difficult than another one and if an individual is more able than another one.

By convention, the average difficulty of the items in a test is equal to 0 logit: more difficult items than the average difficulty have positive logit values, easier items show negative values. The abilities of individuals are estimated by the model according to the difficulties of the items: a person with an ability equal to 0 logit has a probability equal to 0.5 to successfully pass an item of medium difficulty. More able individuals show positive logit values, less able individuals have negative values. If a person and an item have the same measure on the logit scale, then the person has a probability of 50% to successfully pass the item.

3. Data analysis

3.1. A look at data

The PISA 2006 database includes information on nearly 400,000 students from 57 countries (30 OECD countries and 27 partner countries).

Italy participated to PISA 2006 with a sample of 21,773 students, from 806 schools, stratified by geographical macro-areas (Northwest, Northeast, Central, South, South Island) and fields of study (high schools, technical colleges, vocational schools, secondary schools, vocational training). Moreover, the Italian sample is representative of 11 regions (Basilicata, Campania, Emilia Romagna, Friuli Venezia Giulia, Liguria, Lombardia, Piemonte, Puglia, Sardegna, Sicilia and Veneto) and two autonomous provinces of Bolzano and Trento.

The cognitive test is divided into a variable number of items for each domain. Item formats are multiple-choice, short closed-constructed response, and open-constructed response. Most of the items have only one correct answer (with score 1), then there are some items that allow two correct answers, but with different scores (1 and 2), and some science items that allow three correct answers with scores 1, 2 and 3. In addition, code 9 is used if none of the choices is circled and code 8 if two or more choices are circled. Finally code 7 is reserved for the cases when due to poor printing an item presented to a student is illegible, and therefore the student do not have access to the item.

The mathematics test consist of 48 items (44 have only one correct answer and 4 allow two correct answers). The reading test consist of 28 items (22 admit only one correct answer and the remaining 6 two correct answers). The science test is composed of 192 items.

The descriptive analysis of national and international database shows that each item has about the 69% of 7, so we proceeded to a descriptive analysis for individual and domain. The tables 1, 2 and 3 show the results for student, respectively in mathematics, reading and science at national level.

Table 1. Percentage of illegible items in the mathematics test

Percentage of illegible items for student	Percentage
0 - 50%	45.8%
50 - 75%	30.7%
75 - 100%	22.9%

Table 2. Percentage of illegible items in the reading test

Percentage of illegible items for student	Percentage
0 - 50%	53.7%
50 - 75%	0%
75 - 100%	46.3%

Table 3. Percentage of illegible items in the science test

Percentage of illegible items for student	Percentage
0 - 50%	15.3%
50 - 75%	69.4%
75 - 100%	15.3%

Given the massive presence of missing data, for next analyses we decided to use only the students who have had the opportunity to respond to at least 50% of the items.

3.2 Mathematics performance

In this paragraph we analyze Italian student performance in mathematics. The analysis is conducted on the 9963 students who answered at least 50% of the items and 48 items.

The results of the Rasch analysis show an item reliability equal to 1 and a person reliability equal to 0.82, so the test has excellent proprieties of reproducibility. The INFIT and OUTFIT statistics for each item do not present values outside the range [0.6, 1.4], so there is a good fit between data and model for all the items used (Table 4).

Table 4. Item statistics for mathematics

Person: REAL SEP.: 2.12 REL.: .82 ... Item: REAL SEP.: 35.95 REL.: 1.00														
		Item STATISTICS: MISFIT ORDER												
Entry number	Total score	Count	Measure	Model S.E.	Infit		Outfit		PTMEA		Exact match		Item	G
					MNSQ	ZSTD	MNSQ	ZSTD	CORR.	OBS%	EXP%			
20	505	4908	2.37	.05	1.04	1.2	2.05	9.5	A	.28	90.4	90.2	M421Q02T	0
40	4282	4753	-3.37	.05	1.08	2.2	1.89	7.9	B	.30	90.8	91.1	M800Q01	0
22	3733	4860	-2.04	.04	1.12	5.7	1.70	9.9	C	.34	79.4	80.5	M423Q01	0
21	1330	4905	.88	.04	1.13	7.1	1.69	9.9	D	.35	76.3	78.1	M421Q03	0
39	1360	4889	.83	.04	1.10	5.6	1.49	9.9	E	.38	75.7	77.7	M710Q01	0
33	1934	4635	.01	.03	1.17	9.9	1.37	9.9	F	.38	68.0	72.6	M564Q01	0
34	1949	4624	-.01	.03	1.20	9.9	1.35	9.9	G	.36	66.3	72.5	M564Q02	0
12	2630	4892	-.66	.03	1.22	9.9	1.34	9.9	H	.36	63.5	71.7	M305Q01	0
36	2900	4877	-.99	.03	1.14	8.8	1.23	7.5	I	.43	68.5	73.5	M598Q01	0
1	3599	4998	-1.71	.04	1.07	3.8	1.21	5.3	J	.42	76.0	77.8	M033Q01	0
8	2198	4874	-.18	.03	1.11	7.6	1.19	6.5	K	.43	69.1	72.4	M273Q01T	0
10	3640	4923	-1.90	.04	1.00	-.2	1.19	4.0	L	.47	80.9	79.6	M302Q02	0
48	1297	4664	.88	.04	1.12	6.2	1.14	3.1	M	.39	74.3	77.6	M833Q01T	0
18	2124	4850	-.10	.03	1.09	5.9	1.13	4.6	N	.45	69.4	72.6	M420Q01T	0
5	1048	4955	1.86	.03	.98	-.6	1.10	1.2	O	.48	84.8	82.9	M155Q03T	0
17	1949	4910	.13	.03	1.02	1.2	1.10	3.4	P	.47	72.5	72.7	M411Q02	0
29	3670	4997	-1.80	.04	1.01	.3	1.09	2.3	Q	.46	79.0	78.6	M474Q01	0
15	2088	4865	-.05	.03	1.05	3.7	1.08	2.7	R	.47	70.8	72.7	M408Q01T	0
6	2460	4944	-.42	.03	1.04	2.7	1.05	2.0	S	.47	70.1	71.7	M155Q04T	0
37	2384	4744	-.45	.03	1.02	1.8	1.03	1.4	T	.48	71.0	71.6	M603Q01T	0
47	1265	4766	.94	.04	1.03	1.8	.92	-1.9	U	.45	76.6	78.5	M828Q03	0
4	4299	4988	-.13	.02	1.03	1.3	.98	-.4	V	.64	57.8	58.3	M155Q02T	0
32	3162	4806	-1.33	.04	.99	-.3	1.03	.7	W	.50	75.4	75.4	M559Q01	0
27	1015	4775	1.84	.03	1.01	.4	.98	-.2	X	.46	83.7	82.9	M462Q01T	0
35	1934	4676	.03	.03	.93	-4.8	1.00	.0	x	.53	76.0	72.7	M571Q01	0
31	2919	4875	-.99	.03	.99	-.4	1.00	.0	w	.49	73.3	72.6	M496Q02	0
42	1820	4806	.24	.03	.99	-1.0	.93	-2.4	v	.50	74.1	73.7	M810Q01T	0
7	1429	4853	.73	.04	.97	-1.5	.98	-.4	u	.48	77.5	77.0	M192Q01T	0
26	3151	4885	-1.26	.03	.97	-2.1	.96	-1.2	t	.52	75.8	74.9	M447Q01	0
2	1575	4714	.52	.04	.96	-2.6	.88	-4.0	s	.50	75.5	74.7	M034Q01T	0
16	2338	4943	-.29	.03	.96	-3.2	.94	-2.7	r	.53	73.1	71.7	M411Q01	0
46	2255	4775	-.28	.03	.95	-3.5	.92	-3.1	q	.54	74.1	72.3	M828Q02	0
38	1537	4732	.55	.04	.95	-3.2	.87	-4.2	p	.52	76.3	75.6	M603Q02T	0
25	186	4814	3.61	.08	.93	-1.1	.74	-2.1	o	.29	96.2	96.2	M446Q02	0
45	940	4802	1.45	.04	.93	-3.2	.75	-4.6	n	.48	83.2	82.8	M828Q01	0
30	2029	4880	.01	.03	.92	-5.8	.87	-5.4	m	.55	75.0	72.7	M496Q01T	0
3	3043	4957	-1.06	.03	.91	-6.1	.85	-5.6	l	.55	76.4	73.3	M155Q01	0
9	4607	4937	-4.03	.07	.91	-1.9	.65	-3.5	k	.43	94.8	94.3	M302Q01T	0
24	2937	4838	-1.03	.03	.90	-7.1	.86	-4.9	j	.56	77.8	73.6	M446Q01	0
11	1086	4923	1.24	.04	.89	-5.3	.74	-5.3	i	.50	82.6	81.0	M302Q03	0
43	2724	4741	-.84	.03	.89	-7.4	.84	-6.0	h	.57	76.4	72.9	M810Q02T	0
41	1029	4891	1.31	.04	.89	-5.4	.70	-6.7	g	.50	82.5	81.4	M803Q01T	0
28	1203	4763	1.03	.04	.89	-5.9	.75	-5.8	f	.53	81.7	79.2	M464Q01T	0
23	1849	4825	.23	.03	.87	-9.4	.79	-8.2	e	.57	78.0	73.0	M442Q02	0
44	1281	4733	1.65	.03	.84	-5.5	.54	-7.3	d	.56	82.5	78.8	M810Q03T	0
14	624	4768	2.04	.05	.84	-5.7	.55	-7.5	c	.49	88.9	87.8	M406Q02	0
13	1138	4814	1.12	.04	.83	-8.6	.68	-8.0	b	.55	83.6	80.2	M406Q01	0
19	2542	4914	-.56	.03	.82	-9.9	.74	-9.9	a	.62	79.3	72.4	M421Q01	0
MEAN	2145.8	4838.8	.00	.04	.99	-.3	1.04	.2			77.2	77.0		
S.D.	1042.8	93.6	1.41	.01	.10	5.3	.32	5.8			7.5	6.6		

The measures, the abilities of the students and the difficulties of the items can be displayed graphically through the person-item map (Figure 1). It is noted that in the central part of the graph there is most of the students represented on the left by # (each # represents 48 students), and most of the items represented on the right by label of the item. The test is quite broad, though slightly upon the mean level of students, the items and the students are quite well approximated by a normal distribution. The item M446Q02 is the most difficult, on the contrary the items M800Q01 and M302Q01T are the simplest. At the



bottom of the chart there is a small group of students, at which there are no items able to measure their ability.



Figure 1. Person-item map for mathematics

In order to verify that the thresholds are ordered and between them there is a suitable distance, we show the CPCs. To not bore the reader, the figure 1 shows only the CPCs of the four items that allow two correct answers with score 1 and 2. It is easy to check that, for each them, the curve of probability of category 0 meets, first, the curve of probability of category 2 and, then, that of category 1. The category 1, therefore, is never the most likely. To improve the interpretation of the measures it could be appropriate a pooling of the categories 0 and 1 of these items.

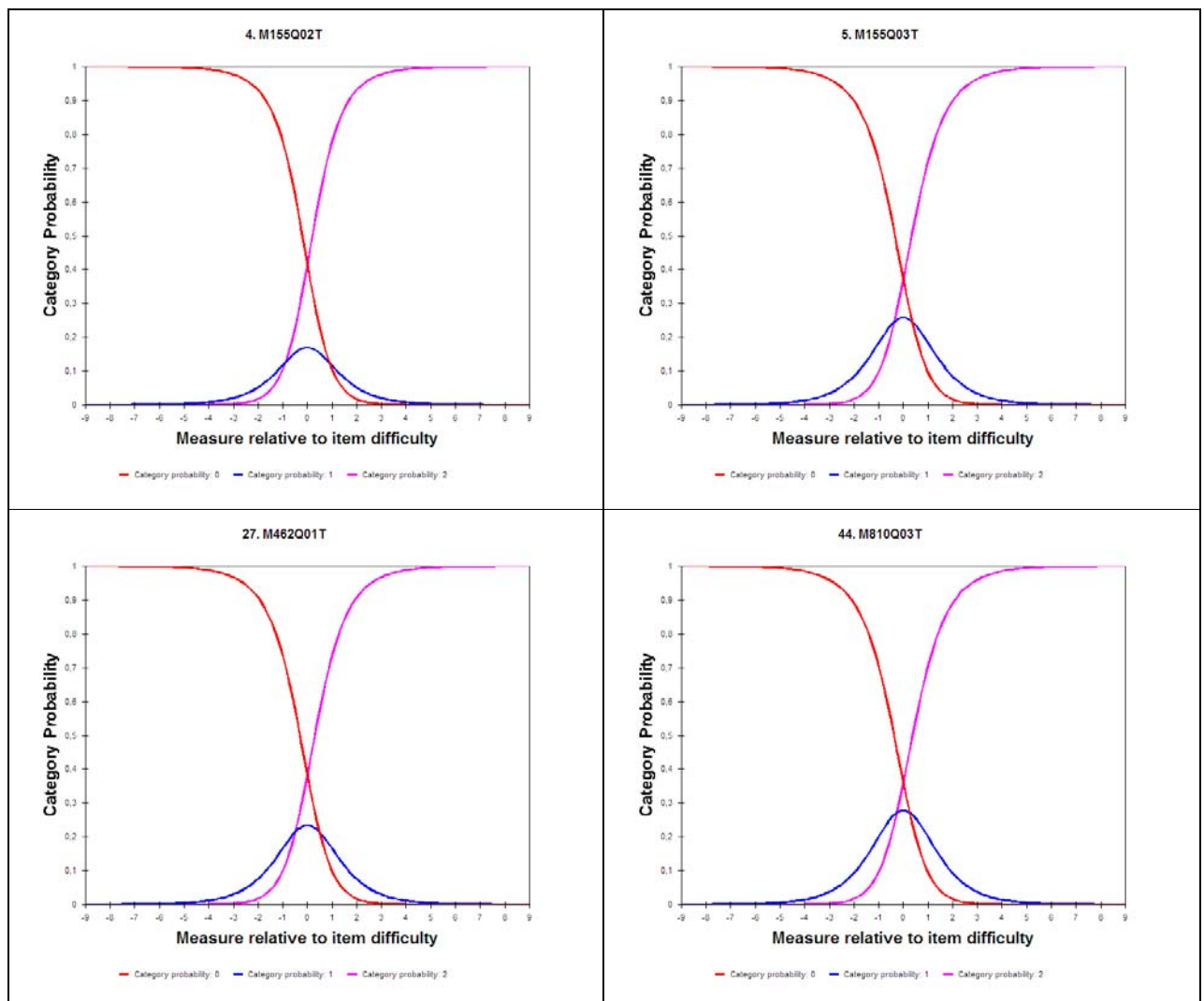


Figure 2. Mathematics performance: CPCs of the four items that allow two correct answers

The final step of the analysis is the comparison of estimates among two or more groups to examine whether the items have a significantly different functioning. This phenomenon is called Differential Item Functioning (DIF). In our case it is interesting to examine the functioning of items among the various Italian regions.

For this purpose, the table 5 allows us to test the hypothesis that the items have the same functioning among the several Italian regions. The table shows that 29 items have a

statistically different functioning among the various Italian regions at a significance level of 5%. These items are in red in Table 5.

Table 5. Mathematics performance: DIF among Italian regions

Person CLASSES	SUMMARY DIF CHI-SQUARE	D.F.	PROB.	Item	
				Number	Name
14	28.1544	13	.0086	1	M033Q01
14	22.7786	13	.0444	2	M034Q01T
14	13.4938	13	.4104	3	M155Q01
14	33.1855	13	.0016	4	M155Q02T
14	17.7838	13	.1659	5	M155Q03T
14	25.8611	13	.0177	6	M155Q04T
14	29.5869	13	.0054	7	M192Q01T
14	30.0954	13	.0046	8	M273Q01T
14	18.9098	13	.1259	9	M302Q01T
14	5.8639	13	.9510	10	M302Q02
14	24.3117	13	.0284	11	M302Q03
14	26.5099	13	.0145	12	M305Q01
14	20.1734	13	.0909	13	M406Q01
14	28.4531	13	.0078	14	M406Q02
14	32.6207	13	.0019	15	M408Q01T
14	7.9870	13	.8444	16	M411Q01
14	27.3574	13	.0111	17	M411Q02
14	31.8679	13	.0025	18	M420Q01T
14	60.1173	13	.0000	19	M421Q01
14	43.5071	13	.0000	20	M421Q02T
14	37.5156	13	.0003	21	M421Q03
14	69.3453	13	.0000	22	M423Q01
14	38.4619	13	.0002	23	M442Q02
14	36.1250	13	.0006	24	M446Q01
14	14.8873	13	.3144	25	M446Q02
14	17.9794	13	.1583	26	M447Q01
14	58.3753	13	.0000	27	M462Q01T
14	14.3670	13	.3485	28	M464Q01T
14	8.9987	13	.7730	29	M474Q01
14	15.2868	13	.2898	30	M496Q01T
14	13.5657	13	.4051	31	M496Q02
14	29.7048	13	.0052	32	M559Q01
14	33.9595	13	.0012	33	M564Q01
14	10.7980	13	.6277	34	M564Q02
14	13.0336	13	.4452	35	M571Q01
14	36.8692	13	.0004	36	M598Q01
14	16.8599	13	.2058	37	M603Q01T
14	21.0081	13	.0727	38	M603Q02T
14	18.6118	13	.1356	39	M710Q01
14	63.1808	13	.0000	40	M800Q01
14	30.2286	13	.0044	41	M803Q01T
14	24.2321	13	.0290	42	M810Q01T
14	35.2862	13	.0008	43	M810Q02T
14	19.3109	13	.1137	44	M810Q03T
14	81.3797	13	.0000	45	M828Q01
14	13.0175	13	.4464	46	M828Q02
14	23.3072	13	.0381	47	M828Q03
14	24.1007	13	.0302	48	M833Q01T

To understand the magnitude of the differences between the regions is interesting to look at figure 3. This graph shows the difficulty of each item for each region. From the figure 3. it would seem that there are no appreciable differences of the items between different regions. However, a small value of DIF could be statistically significant, while a large value of DIF could be not statistically significant, so it is important to look at the chi-square test above illustrated (table 5.).

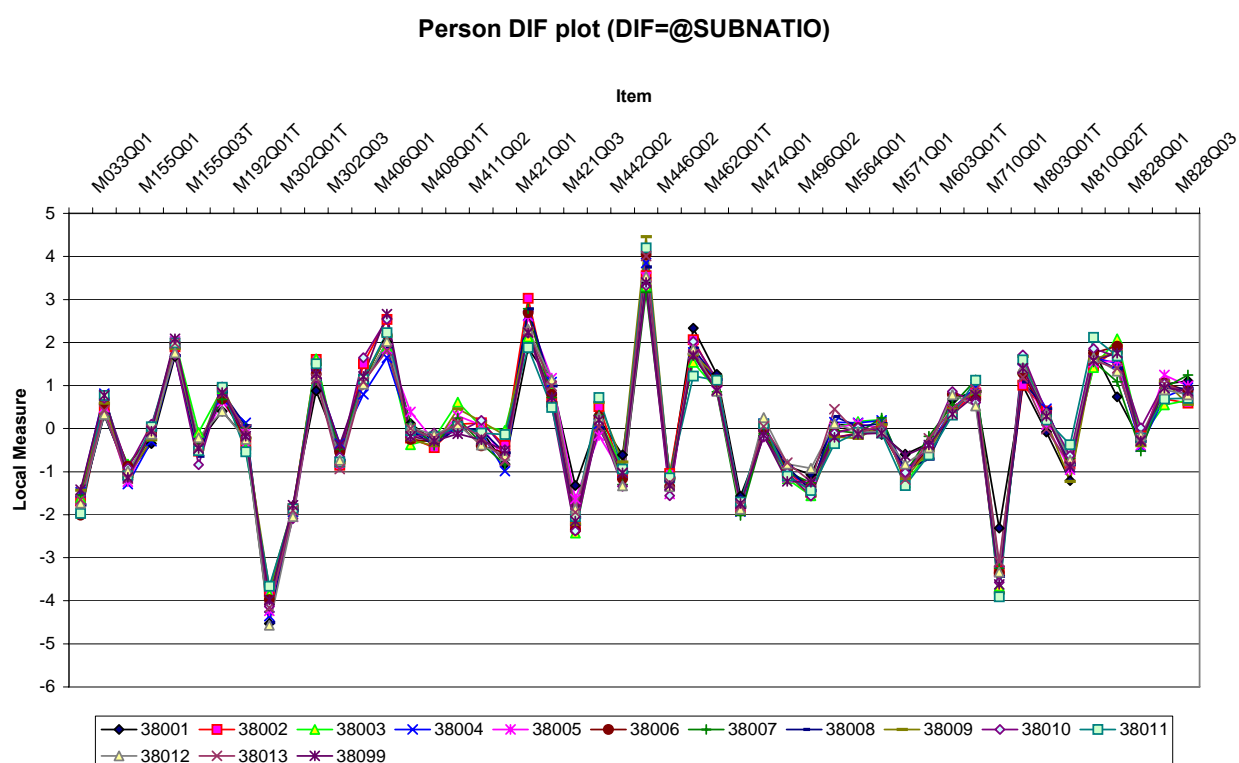


Figure 3. Mathematics performance: difficulty of each item for each region.

3.3 Reading performance

In this paragraph we analyze Italian student performance in reading. The analysis is conducted on the 11686 students who answered at least 50% of the items and 28 items.

The results of the Rasch analysis show an item reliability equal to 1 and a person reliability equal to 0.78, so the test has good proprieties of reproducibility. From the table 6. we can observe that the INFIT and OUTFIT statistics present values outside the range [0.6, 1.4] for the following four items: R111Q06B, R067Q04, R227Q02T R067Q05. This could be due to a different functioning of the items among the various Italian regions. This hypothesis will be verified by analysis of DIF. Moreover, it would be appropriate to remove or replace these items because they could distort the measures obtained. However, we prefer not to make these changes to remain faithful to the test calibrated at international level. The stakeholders can focus on the contents of such items to address the educational proposals towards the disciplinary facets which are more problematic.

For this purpose the PISA compendium has been published. It gathers the PISA tests that have been issued in various editions and administered in Main Studies, i.e. those which have been published and will not be reused in subsequent cycles. The compendium is divided into three parts: Reading, Math and Science. Each test is accompanied by the description of items, by the guide for the correction of responses and by the data on student responses at different levels: average of the OECD countries, the national percentages, the percentages for macro-areas. The original numbering of the items has been left, so it is easy to establish the correspondence between item and content.

Table 6. Item statistics for Reading

Person: REAL SEP.: 1.88 REL.: .78 ... Item: REAL SEP.: 40.43 REL.: 1.00												
Item STATISTICS: MEASURE ORDER												
ENTRY NUMBER	TOTAL SCORE	COUNT	MEASURE	MODEL S.E.	Infit		Outfit		PTMEA CORR.	Exact match		Item
					MNSQ	ZSTD	MNSQ	ZSTD		OBS%	EXP%	
8	1598	6551	2.16	.03	.86	-8.7	.80	-4.6	.44	80.3	76.9	R102Q04A
20	2285	6497	1.54	.03	.81	-9.9	.78	-6.7	.52	78.0	71.8	R220Q01
9	2318	6508	1.51	.03	.84	-9.9	.81	-5.8	.50	76.7	71.7	R102Q05
12	2229	6360	1.38	.03	1.19	9.9	1.66	9.9	.30	65.8	72.9	R104Q02
13	2284	6324	1.32	.03	1.32	9.9	1.17	4.8	.48	71.5	72.6	R104Q05
17	2594	6625	1.31	.03	.84	-9.9	.83	-5.6	.51	75.9	71.1	R219Q01E
25	2792	6631	.95	.03	1.21	9.9	1.67	9.9	.32	63.0	71.8	R227Q01
2	2889	6524	.86	.03	.89	-7.2	.91	-3.3	.52	72.5	71.6	R055Q02
15	3370	6561	.49	.03	1.16	9.3	1.08	3.4	.56	70.6	71.6	R111Q02B
18	3880	6643	.36	.03	.74	-9.9	.72	-9.9	.58	76.7	71.4	R219Q01T
27	3614	6619	.32	.03	.73	-9.9	.69	-9.9	.62	78.2	71.7	R227Q03
21	3855	6398	.32	.03	.76	-9.9	.86	-6.4	.55	76.1	71.5	R220Q02B
14	3776	6590	.20	.03	.75	-9.9	.78	-9.9	.60	78.0	72.0	R111Q01
3	3771	6456	.18	.03	.75	-9.9	.72	-9.9	.59	77.2	71.9	R055Q03
22	4181	6387	.07	.03	.76	-9.9	.82	-8.7	.53	76.8	72.2	R220Q04
24	4363	6369	-.07	.03	.74	-9.9	.79	-9.9	.53	77.2	72.5	R220Q06
19	4513	6618	-.10	.03	.69	-9.9	.71	-9.9	.57	78.9	72.6	R219Q02
28	4233	6590	-.14	.03	.69	-9.9	.69	-9.9	.61	79.2	72.5	R227Q06
4	4276	6448	-.20	.03	.59	-9.9	.57	-9.9	.66	83.1	72.5	R055Q05
23	4971	6377	-.52	.03	.53	-9.9	.55	-9.9	.60	84.3	73.4	R220Q05
11	4722	6412	-.55	.03	.70	-9.9	.72	-9.9	.53	79.1	73.0	R104Q01
16	5262	6541	-.91	.03	2.45	9.9	2.47	9.9	.65	34.7	72.9	R111Q06B
10	5591	6501	-.92	.03	.51	-9.9	.53	-9.9	.51	86.5	73.5	R102Q07
1	5308	6537	-.94	.03	.59	-9.9	.62	-9.9	.53	83.1	72.9	R055Q01
5	5948	6609	-1.12	.03	.47	-9.9	.50	-9.9	.47	87.9	73.2	R067Q01
6	7343	6594	-2.08	.03	2.01	9.9	2.06	9.9	.56	43.8	68.6	R067Q04
26	7558	6627	-2.45	.03	1.57	9.9	1.61	9.9	.58	54.2	67.7	R227Q02T
7	8707	6570	-2.98	.03	2.29	9.9	2.32	9.9	.65	41.5	66.8	R067Q05

In order to compare the student ability and the item difficulty, we present the person-item map (Figure 5). The test is significantly upon the mean level of students, in fact, there is a large group of students for who there are no items calibrated on their ability level, on the contrary, there are some very difficult items (R102Q04A, R102Q05, R220Q01, R104Q02, R104Q05, R219Q01E, R227Q01, R055Q02) at which they are no students or there is a very small number. The students are asymmetrically distributed.

The analysis of the CPCs does not show problematic aspects: the thresholds are ordered and their distance is sufficient (Figure 9). Indeed Linacre (1999) indicates that the thresholds should grow at least 1.4 logit for different categories, but not more than 5 logit to ensure continuity of the variable.

Finally, we examine the functioning of items among the various Italian regions. The table 7 shows that almost all the items (red-ink in the table 7) have a statistically different functioning among the various Italian regions at a significance level of 5%, so it would be desirable to identify a battery of items that can operate in not statistically different way between the Italian regions. The DIF analysis seems to verify the hypothesis that a different functioning of the items among the various Italian regions lead to their poor fit. In fact, the four items that have a bad fit also have a significant DIF.

The magnitude of the differences between regions is represented in Figure 6, where one can see that the variability of item difficulty is greater when the item has a different functioning between the regions. For example, Bolzano (38001) seems the more different region than the others.



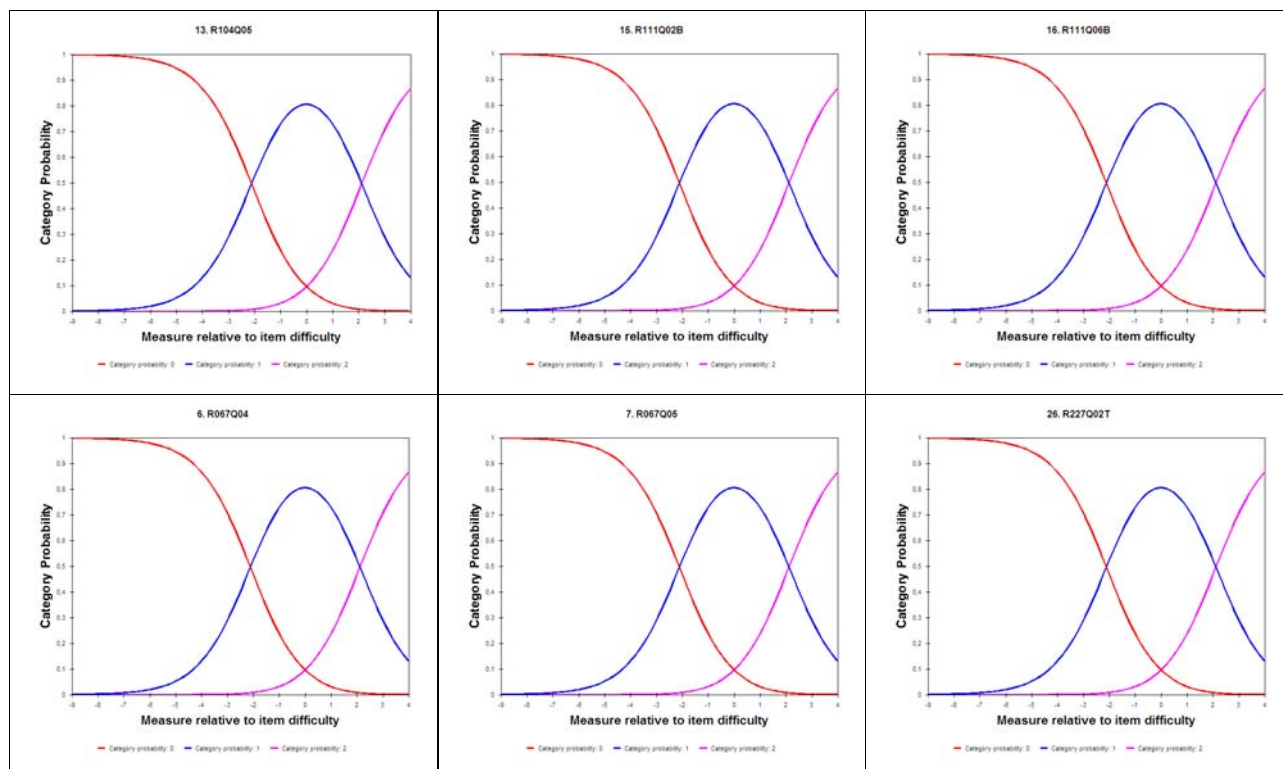


Figure 4. Reading performance: CPCs of the six items that allow two correct answers

Table 7. Reading performance: DIF among Italian regions

Person CLASSES	SUMMARY DIF CHI-SQUARE	D.F.	PROB.	Item	
				Number	Name
14	33.2730	13	.0015	1	R055Q01
14	37.5273	13	.0003	2	R055Q02
14	28.6081	13	.0074	3	R055Q03
14	18.8289	13	.1285	4	R055Q05
14	127.210	13	.0000	5	R067Q01
14	81.1536	13	.0000	6	R067Q04
14	49.2266	13	.0000	7	R067Q05
14	80.4530	13	.0000	8	R102Q04A
14	28.2015	13	.0085	9	R102Q05
14	134.343	13	.0000	10	R102Q07
14	27.0227	13	.0123	11	R104Q01
14	37.8045	13	.0003	12	R104Q02
14	25.2743	13	.0212	13	R104Q05
14	14.0466	13	.3705	14	R111Q01
14	33.8958	13	.0012	15	R111Q02B
14	196.096	13	.0000	16	R111Q06B
14	76.4062	13	.0000	17	R219Q01E
14	63.5525	13	.0000	18	R219Q01T
14	31.4878	13	.0029	19	R219Q02
14	90.7701	13	.0000	20	R220Q01
14	11.4726	13	.5713	21	R220Q02B
14	26.6078	13	.0141	22	R220Q04
14	18.8257	13	.1286	23	R220Q05
14	33.4510	13	.0015	24	R220Q06
14	32.9223	13	.0017	25	R227Q01
14	34.3976	13	.0010	26	R227Q02T
14	12.7273	13	.4691	27	R227Q03
14	13.1172	13	.4388	28	R227Q06

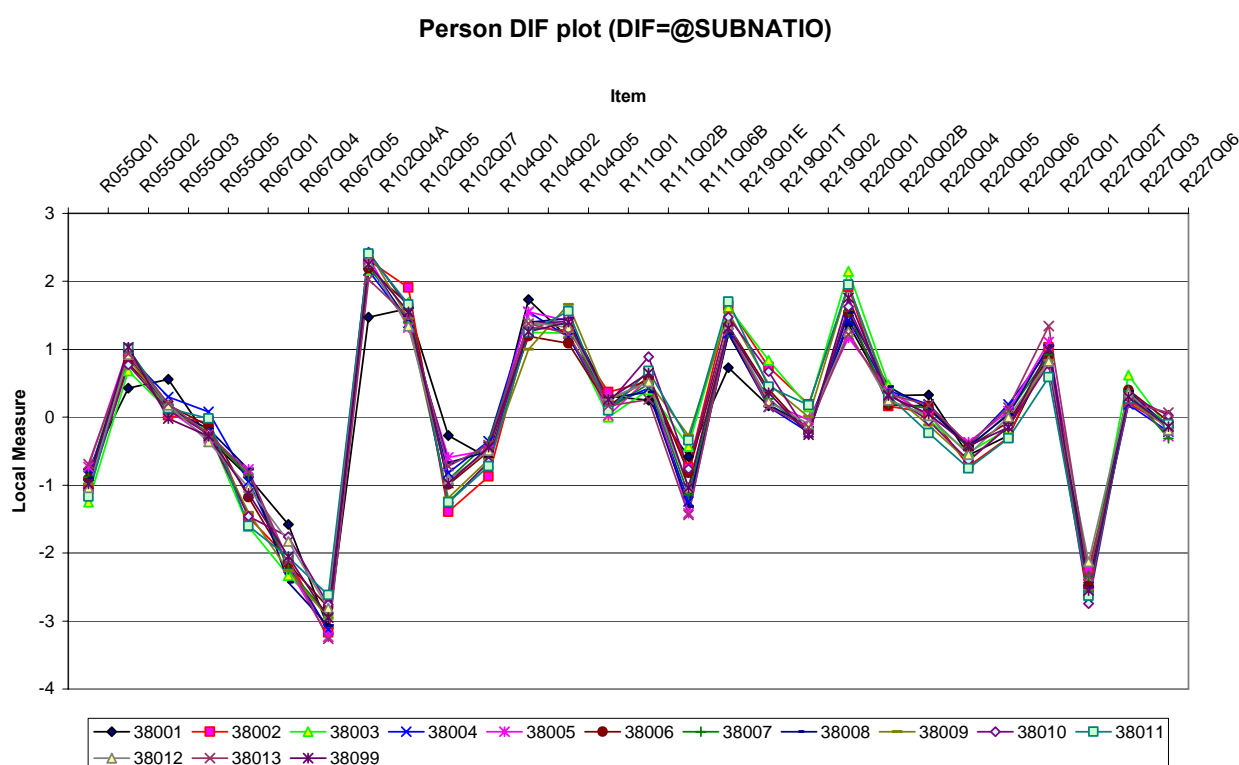


Figure 6. Reading performance: difficulty of each item for each region

3.4. Science performance

The analysis Italian student performance in science is conducted on the 3311 (15,3%) students who answered at least 50% of the items . Though the results of the Rasch analysis show an item reliability equal to 1 and a person reliability equal to 0.92, 22 items present the INFIT and OUTFIT statistics outside the range [0.6, 1.4]. These items are in table 8.

Table 8. The INFIT and OUTFIT statistics of 22 science items outside the range [0.6, 1.4]

ENTRY NUMBER	TOTAL SCORE	COUNT	Measure	Model S.E.	Infit		Outfit		PTMEA CORR.	Exact match		Item
					MNSQ	ZSTD	MNSQ	ZSTD		OBS%	EXP%	
94	745	1551	1.83	.05	1.92	9.9	1.83	9.9	.35	54.9	60.0	S519Q01
56	868	1642	1.66	.05	2.22	9.9	2.09	9.9	.42	41.4	58.8	S447Q05
2	1085	1669	1.26	.04	1.80	9.9	1.78	9.9	.38	37.8	58.6	S114Q04T
76	1297	1669	.87	.04	1.50	9.9	1.51	9.9	.32	43.8	61.3	S485Q05
86	1291	1466	.54	.04	2.29	9.9	2.27	9.9	.45	10.0	63.0	S498Q04
59	1637	1642	.23	.04	1.79	9.9	1.80	9.9	.38	27.8	63.4	S465Q01
149	1908	1653	-.19	.04	1.75	9.9	1.76	9.9	.25	45.4	61.8	S521QNB
148	2014	1651	-.36	.04	1.63	9.9	1.63	9.9	.32	47.4	60.1	S521QNA
119	1975	1606	-.38	.04	1.72	9.9	1.73	9.9	.31	45.1	59.5	S438QNB
109	2006	1605	-.43	.04	1.85	9.9	1.85	9.9	.37	44.2	58.9	S413QNC
118	2119	1609	-.59	.04	1.73	9.9	1.76	9.9	.27	42.0	57.0	S438QNA
120	2333	1607	-.91	.04	1.54	9.9	1.56	9.9	.36	43.6	53.4	S438QNC
116	2473	1626	-1.07	.04	1.46	9.9	1.48	9.9	.34	41.0	51.7	S437QNB
135	2547	1649	-1.12	.04	1.50	9.9	1.50	9.9	.35	40.0	51.4	S485QNC
104	2615	1642	-1.23	.04	1.65	9.9	1.66	9.9	.34	37.7	50.9	S408QNA
136	2356	1444	-1.33	.04	1.51	9.9	1.52	9.9	.29	41.4	50.9	S498QNA

ENTRY NUMBER	TOTAL SCORE	COUNT	Measure	Model S.E.	Infit		Outfit		PTMEA CORR.	Exact match		Item
					MNSQ	ZSTD	MNSQ	ZSTD		OBS%	EXP%	
114	2657	1618	-1.33	.04	1.41	9.9	1.43	9.9	.44	43.3	50.8	S428QNC
129	2738	1654	-1.37	.04	1.54	9.9	1.57	9.9	.28	39.2	50.5	S476QNC
153	2760	1657	-1.39	.04	1.44	9.9	1.47	9.9	.30	40.3	50.4	S527QNA
150	2343	1396	-1.43	.04	1.49	9.9	1.51	9.9	.35	42.1	50.6	S524QNA
115	2830	1626	-1.54	.04	1.46	9.9	1.47	9.9	.28	42.7	50.2	S437QNA
117	3010	1628	-1.77	.04	1.41	9.9	1.42	9.9	.40	43.6	50.2	S437QNC

The person-item map (Figure 7.) shows that items are distributed into two main blocks: half (maybe more than half) of the items are more difficult than the average, about half of the items are easier than the average, while items of average difficulty (by convention, the average difficulty of items in a test is set equal to 0) are missing. Looking at the distribution of the items one feels that the test measure two different dimensions. Looking at the distribution of individuals, it is easy to see that it is normal and symmetric with respect to -1 rather than 0. Consequently, it would be appropriate to introduce items of average difficulty and to reduce the number of easy and difficult items.

Persons - MAP - Items

	<more>		<rare>									
4			+ T									
				S527Q01T								
				S114Q05T								
				S458Q01								
3			+ S	S425Q04	S524Q07							
				S326Q04T	S425Q03	S519Q03						
				S131Q04T	S268Q02T	S269Q04T	S304Q03A	S408Q03				
				S413Q04T	S447Q02	S498Q03	S510Q04T					
				S213Q01T	S269Q03T	S408Q04T	S413Q06	S425Q02				
				S438Q03T	S447Q04	S465Q04	S495Q01T					
2			+ S	S304Q01	S408Q05	S416Q01	S421Q01	S428Q05				
				S437Q03	S493Q01T	S495Q03	S498Q02T	S510Q01T				
				S514Q03								
				S114Q03T	S131Q02T	S269Q01	S304Q03B	S447Q05				
				S477Q04	S478Q01	S493Q05T	S495Q02T	S514Q04				
				S519Q01	S519Q02T	S524Q06T						
				S268Q06	S326Q01	S326Q02	S326Q03	S415Q08T				
				S421Q03	S426Q03	S426Q07T	S437Q04	S438Q02				
				S447Q03	S465Q02	S466Q05	S478Q02T	S485Q02				
				S495Q04T	S521Q02	S527Q03T						
				S114Q04T	S268Q01	S304Q02	S408Q01	S425Q05				
				S428Q03	S437Q01	S437Q06	S458Q02T	S476Q03				
				S478Q03T	S485Q03	S508Q02T	S508Q03	S527Q04T				
1			+ S	S213Q02	S413Q05	S415Q02	S415Q07T	S428Q01				
				S466Q01T	S466Q07T	S476Q01	S476Q02	S477Q02				
				S514Q02								
				S256Q01	S426Q05	S438Q01T	S477Q03	S485Q05				
				S493Q03T								
			.T	S498Q04	S521Q06							
			.#	S465Q01								

0	.####	+ M					
	.##### S		S521QNA	S521QNB			
	.#####		S413QNC	S438QNA	S438QNB		
	##### M						
-1	.#####	+	S408QNB	S408QNC	S437QNB	S438QNC	S485QNC
	.#####		S408QNA	S413QNA	S413QNB	S428QNB	S428QNC
			S476QNC	S498QNA	S508QNB	S514QNB	
	.##### S		S408QSC	S428QNA	S437QNA	S456QNA	S466QNB
			S508QNA	S508QNC	S514QNA	S524QNA	S524QNB
			S524QNC	S527QNA			
	###		S408QSB	S416QNA	S426QSB	S437QNC	S438QSB
			S456QNB	S485QNA	S485QNB	S498QNB	S498QNC
			S527QNC				
-2	.# T	+ S	S408QSA	S425QSC	S456QNC	S456QSB	S456QSC
			S466QNA	S476QNA	S476QNB	S478QNA	S478QNB
			S478QNC	S514QNC			
	.		S416QSA	S416QSB	S438QSC	S466QNC	S477QSB
			S485QSB	S485QSC	S519QNB	S519QNC	S527QNB
	.		S416QNB	S421QSC	S426QSA	S438QSA	S465QSA
			S465QSB	S477QSC	S498QSA	S498QSB	
	.		S425QSB	S426QSC	S456QSA	S519QNA	S519QSB
			S527QSB				
-3	.	+	S416QSC	S421QSA	S476QSA	S527QSC	
	.		S425QSA	S477QSA	S519QSA	S519QSC	
	.						
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-4	.	+ T	S476QSC				
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EACH '#' IS 47.

Figure 7. Person-item map for science

As we wrote above about reading performance, it would be appropriate to remove or replace these items which present the INFIT and OUTFIT statistics outside the range [0.6, 1.4], because they could distort the measures obtained. However, we prefer not to make these changes to remain faithful to the test calibrated at international level. The stakeholders can focus on the contents of such items to address the educational proposals towards the disciplinary facets which are more problematic.

In the light of these results we believe it is not necessary to show the CPC curves and to compare the different Italian regions.

4. Final remarks

In this paper we analyzed the cognitive test for evaluating the Italian student performance in reading, mathematics and science, comparing the results of different local governments.

The descriptive analysis of national and international database showed that each item has about the 69% of 7, so we proceeded to a descriptive analysis for individual and domain in order to identify students who have had the opportunity to respond to at least 50% of the items.

Given the considerable presence of missing data, we could opt for different strategies, such as the use of algorithms for estimating the missing data or the analysis of the only available data. Unlike the strategy applied by the OECD that decided to estimate the missing data, we chose to include in the analysis only the students who have had the opportunity to respond to at least 50% of the items.

The results for mathematics performance can be summarized as follows:

- the test has excellent proprieties of reproducibility;
- All the items show a good fit (the INFIT and OUTFIT statistics for each item do not present values outside the range);
- The item map shows that the test is quite broad, though slightly upon the mean level of students;
- There are some items that have a significantly different functioning between the Italian regions at the 5% level.

For reading performance, 4 items have a bad fit, the test is significantly upon the mean level of students and there are some items that have a significantly different functioning between the Italian regions at the 5% level.

For science, the situation is very controversial. The analysis of Italian student performance was conducted on a small sample because of the considerable presence of missing data. We found many items have a bad fit. The person-item map shows it would be appropriate to introduce items of average difficulty.

In the light of the results obtained in the three domain, the stakeholders should address the educational proposals towards the disciplinary facets of mathematics, reading and science which are more problematic in order to reduce the differences between the regions and to improve the Italian student performance. It would also be interesting to compare the results obtained in different domain by using algorithms for estimating the missing data.

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AN ERROR RESILIENT AND MEMORY EFFICIENT SCHEME FOR WAVELET IMAGE CODING¹

Ekram KHAN

Department of Electronics Engineering,
Aligarh Muslim University, Aligarh, India

E-mail: ekhan@lycos.com

Irshad A. ARSHAD

Department of Mathematics and Statistics,
Allama Iqbal Open University, Islamabad, Pakistan

E-mail: irshad_arshad@hotmail.com

Taru VARSHNEY

Department of Electronics Engineering,
Aligarh Muslim University, Aligarh, 202002, India

E-mail: taru_varsh@yahoo.com

Abstract: Set- Partitioning in Hierarchical Trees (SPIHT) is a state-of-art zero-tree based image coder with excellent rate-distortion performance in the noise free environments. However in presence of noise they are extremely sensitive to the bit errors. Even a single bit error may lead a substantial degradation in the quality. Our investigations reveal that the use of three linked lists in SPIHT increases the inter-bitplane dependencies of bits and therefore causes the errors to propagate down the bitstream. In this paper we propose a modified coder which uses only a single list. The list is initialized, processed and exhausted within each bit-plane, thereby removing dependencies of inter-plane bits. The proposed coder is known as Single List SPIHT (SL-SPIHT) image coder. Additionally, this coder is memory efficient as compared to the SPIHT. Simulation results show that under the same channel conditions, SL-SPIHT can improve the quality of reconstructed image by 5-6 dB compared to that of SPIHT.

Key words: Wavelet transform; Image coding; Error resilience; memory efficient

1. Introduction

Real time transmission of images through handheld mobile/portable devices (with limited memory, processing power and battery life) requires an image coding algorithm that can compress images efficiently with reduced memory requirements. The contemporary modern image coding methods (e.g. SPIHT [3]0) support a wide range of functionality but requires significant amount of memory during their execution. Also, wireless channels are highly prone to bit errors; therefore the devolvment of error resilient techniques is an additional challenge for memory constrained environments.

Among the many methods of compression, Discrete Wavelet Transform (DWT) approach has become a popular technique [2-5]. Due to energy compaction nature of wavelet transform coding, it is common that a large number of wavelet coefficients are quantized to zero value. The concept of zero-tree structure was first introduced by Lewis and

Knowles [1] for efficient representation of zero wavelet coefficients after perceptual thresholding. This data structure was later elegantly combined with the bit-plane coding scheme by Shapiro for embedded wavelet image coding, popularly known as embedded zero-tree wavelet (EZW) [2]. Intensive efforts have been drawn into this research field ever since and many zero-tree image coders are being developed. The SPIHT [3] and virtual SPIHT (VSPiHT) [5] algorithms are the simplest and most efficient variations of EZW. These techniques have excellent rate distortion characteristics in noise free environments. However in the presence of noise they are extremely sensitive to bit errors. Even a single bit error may lead to loss of synchronization between the encoder and the decoder beyond erroneous bit. As a result the quality of the reconstructed image degrades substantially unless proper error correcting methods are used. Thus the transmission of zerotree coded images over noisy channels is still a challenging problem.

One of the problems with SPIHT and VSPiHT coders are the large memory requirement due to the uses three continuously growing lists. Also, these lists are responsible for propagation of errors down the bitstream. Since information accumulated in list during any pass is used for coding of significant/insignificant pixels/sets in the subsequent passes. In case any error occurred, it will lead to wrong entries in the lists, thereby severely affecting the decoding process.

Recently, Arora et. al. proposed a memory efficient wavelet image coding scheme [2] which uses a single list for encoding and decoding. Here onward, this algorithm is referred as Single List SPIHT (SL-SPIHT). The emphasis of the paper was more on memory reduction, similar to work done in [7] [8]. In this paper we will explore the error resiliency feature of SL-SPIHT.

As compared to SPIHT, the SL-SPIHT uses only one list to store the significance information of the sets. The use of single and re-usable list in SL-SPIHT not only reduces the memory requirement but also improves the error resiliency. Since the list is re-initialized at the beginning of each pass, in SL-SPIHT algorithm hence the entries are not used in the subsequent passes. This is beneficial as it makes the algorithm pass by pass independent, puts a check on the consecutively growing size of the lists and makes a foundation to study the error resiliency feature of the SL-SPIHT algorithm.

Rest of the paper is organized as follows. Section II briefly reviews the SPIHT algorithm and its error resiliency feature. The SL-SPIHT algorithm is described in Section III. Simulation Results are presented in Section IV, where memory analysis and error resiliency performance of SL-SPIHT algorithm is compared with that of SPIHT algorithm. Finally, paper is concluded in Section V.

3. Review of SPIHT algorithm

SPIHT Algorithm

The SPIHT algorithm is based on hierarchical set partitioning, which can be thought of as a divide-and-conquer strategy. The SPIHT algorithm views wavelet coefficients as a collection of spatial orientation trees, with each tree consisting of coefficients from all sub-bands that correspond to the same spatial location in an image. A partitioning rule is used to divide a given set into smaller subsets so that significant coefficient can be efficiently isolated.

The SPIHT consists of two main stages namely sorting and refinement. For practical implementation, SPIHT maintains three linked lists viz. the list of insignificant pixels (LIP), the list of significant pixels (LSP) and the list of insignificant sets (LIS). At the initialization stage, SPIHT initializes the LIP with all the pixels in the highest level of the pyramid (i.e. LL subband), the LIS with all the coefficients at the highest level of the pyramid except those, which don't have descendents, and LSP as an empty set. During the sorting pass, the algorithm first traverses through the LIP, testing the magnitude of its elements against the current threshold and representing their significance by 0 or 1. Whenever a coefficient is found significant, its sign is coded and it is moved to LSP. The algorithm then examines the LIS and performs a magnitude check on all coefficient of set. If a particular tree/set is found to be significant, it is partitioned into its subsets (children and grandchildren) and tested for

significance. Otherwise a single bit is appended to the bit stream to indicate an insignificant set (or zero-tree). After each sorting pass SPIHT outputs refinement bits at the current level of bit significance of those pixels which had been moved to LSP at higher threshold, resulting in the refinement of significant pixels with bits that reduce maximum error. This process continues by decreasing current threshold by factor of two until desired bit rate is achieved.

Error Analysis

The SPIHT coder generates different type of bits that have different degree of vulnerability to the errors. The effect of error in some bits is more severe, damaging the image globally by disturbing the synchronization between the encoder and decoder. On the other hand, some bits are less sensitive to errors, and damage the image locally without disturbing the synchronization.

- **Significance bits:** During the LIP and LIS testing, pixels or trees are checked for significance. Error in significant bits will result in the propagation of error down to the bit-stream during the decoding process. A significant coefficient or set may be deemed insignificant and vice-versa. This will cause the decoder to update the list with the wrong nodes and can cause a large degradation in the decoded image quality. Thus a single bit error in the significant bits has global effect and may damage the entire image.

- **Sign bits:** The sign bit of a wavelet coefficient is generated, immediately after its significance test during sorting pass, when it is found significant. An error in this bit will change only the phase of the coefficient and does not disturb the synchronization between the encoder and the decoder. The effect of an error in the sign bit corrupts the image locally only during the inverse wavelet transform.

- **Refinement bits:** The refinement bits are generated during refinement pass. An error in these bits has less severe effect and distorts the image locally only. The effect of error in a refinement bit is limited to a single coefficient and does not disturb the progression of the decoding process after it occurs.

- Based on the degree of their sensitivity, the bits generated by SPIHT algorithm can be classified into two classes:

- Critical Bits (CB) and
- Non-Critical Bits (NCB).

The critical bits are those, which causes the loss of synchronization between the encoder and decoder. A single bit error in critical bit causes the failure of the reconstruction process after that point. It consists of significant bits generated during LIP and LIS tests. The non-critical bits on the other hand cause less severe errors. The effect of error in a non-critical bit is limited to a single coefficient and does not disturb the progression of the decoding process after it occurs. The non-critical bits consist of the sign and refinement bits.

3. SL-SPIHT algorithm

The SL-SPIHT algorithm was first proposed by Arora et al. [10] to reduce the working memory requirement of SPIHT algorithm. In this algorithm, the functions of all three lists are performed by a single list. The list is re-initialized at the beginning of each bit-plane to avoid the inter-dependency on the previous bit-planes.

Working of SL-SPIHT Algorithm

The SL-SPIHT algorithm works as follows.

After the ' N_d ' level of wavelet decomposition of input image, depending upon the value of the largest wavelet coefficient, the initial threshold is calculated as 2^n where $n = \left\lceil \log_2 \left(\max_{i,j} |c_{i,j}| \right) \right\rceil$. The LL-sub-band is coded separately. The coefficients of LL-sub-band are represented in $(n+1)$ bit sign-magnitude form. In the first (most significant) bit-plane, the

sign bit and the most significant bit from magnitude of the LL sub-band coefficients are embedded in the bitstream. In the subsequent bit-planes, only the n^{th} most significant bits of LL-sub-band coefficient are transmitted. In order to encode the remaining sub-band coefficients, they are linked through spatial orientation trees with their roots in the LL-sub-band. To define a uniform child-parent relationship, it is assumed that first quarter of LL-sub-band coefficients have no descendents and remaining three-quarter of the coefficients have their children in sub-bands of correspondingly same orientation.

A single list referred as **List of Pixel Sets (LPS)** is used in this algorithm. At the beginning of each bit-plane, LPS is initialized with address of those coefficients of LL-subband that have decedents. For each entry in LPS, test the significance of the set. For insignificant sets, a single bit is appended to the bitstream to indicate a zerotree. However, if a particular set is found to be significant, it is then partitioned into two its subsets (children and grandchildren). First the significance of each of the children is tested and if a particular child is found significant, its sign bit is transmitted and current threshold is subtracted from its magnitude. Then the significance of grandchildren is tested. If the set of grandchildren (provided they exist) is found to be significant, the addresses of children are added at the end of LPS. After testing every entry of LPS (including those which are added in the current pass), it is re-initialized and threshold is reduced by a factor of two. The entire process is repeated for the next pass (bit-plane). It should be noted that same coefficients might become significant in more than one pass, depending upon its absolute value. In order to avoid the multiple transmission of its sign bit, sign bits are masked (or flagged) once they are transmitted at the first significant test of the corresponding coefficient. If in the subsequent passes, the same coefficient again tests significant at lower thresholds, the transmission of sign bit is avoided as it has already been masked.

At the decoder, the reverse process is performed. At the arrival of first significant bit of a coefficient, it is reconstructed as $\pm 1.5 \times 2^n$. The decoder adds or subtract 2^{n-1} to its current reconstructed value depending upon whether it inputs significant bit in the current pass or not.

It is observed that the working memory requirement in any pass is proportional to the maximum number of entries in LPS during that coding pass. Further, since in SL-SPIHT, LPS is re-initialized, the same memory can be used in the next and subsequent passes, whereas in SPIHT, the lists keep growing progressively thereby increasing memory requirements. Additionally, the entire encoding and decoding of a bit-plane is performed in a single pass, whereas SPIHT requires two passes, sorting and refinement.

Error Analysis of SL-SPIHT

Another attractive feature of the SL-SPIHT algorithm is its improved error resilience. This can be explained as follows. Since SL-SPIHT uses a single re-useable list. The LPS list is initialized, processed and exhausted within each pass, making encoding/decoding of each pass independent to its preceding passes. This is in contrast to the SPIHT algorithm, where the continuously growing lists (LIP, LIS and LSP) make encoding/decoding of each pass dependent over the previous passes. This is because, list entries are processed, updated in a pass (bit-plane), which will be used to code significant/insignificant information in the next pass. This inter-pass dependency of bitstream makes SPIHT coder highly sensitive to the channel errors.

On the other hands, in SL-SPIHT the LPS is initialized at the beginning of each pass, so that even if the list entries during earlier passes are corrupted due to channel errors, it will not affect the decoding of subsequent passes. Therefore, if the decoder can be synchronized again (by using pass-by-pass markers), then there will be decoding error in that pass. This will lead to the reconstruction of good quality image even in presence of errors. Therefore, the use of single re-usable list to achieve pass by pass independency and use of markers to synchronize will lead to better error resiliency in SL-SPIHT.

The use of markers at the end of each pass (bit-plane) will have no impact on the decoding of SPIHT in presence of errors, but it has significant impact on the decoding of SL-SPIHT bitstream. For example, consider a case when error occurs in a bit and all its following

bits until the end of the pass are zero. This situation is more likely to happen in early passes when threshold is high & most of the trees are insignificant (zerotrees) and all the refinement bits are also zero. In this case since all the bits are of same magnitude (i.e.0) and correspond to same type of information (generally zerotree significance) the use of marker will be very much advantageous. As in this case the only error observed is due to loss of synchronization which is again regained in the next pass by the use of marker and since the single list is reinitialized at the beginning of each pass hence the problem of wrong entries in the list is also avoided. This will lead to a large improvement in the quality of the decoded image.

4. Simulation results

In order to highlight the features of SL-SPIHT, we have considered two gray level test images, BARBARA and BOAT, each of depth 8 bits/pixel (bpp) and of size 512×512 . Each image is wavelet transformed with 5 levels of decomposition using 9/7 bi-orthogonal filter. First we compare the memory requirements of SL-SPIHT with that of SPIHT, followed by the error resiliency.

Memory Requirements

In this section we will compare the memory requirements of the auxiliary list(s) in SL-SPIHT with that of SPIHT and NLS 0 algorithms. There are two types of memory requirement in zerotree coders: fixed memory to store wavelet coefficients and variable working memory to store list entries. For the comparison purpose, we consider only variable working memory requirements. Though NLS does not use any lists, but needs fixed size arrays and state tables. For an image of size $M \times N$, the working memory requirement in NLS

0 is $M_{NLS} = \left(\frac{MN}{4} + \frac{MN}{16} \right) W + \frac{MN}{2}$ bytes, where W is the number of bytes used to store each

wavelet coefficient (say two bytes). This working memory is always fixed and is independent of number of bit-planes to be executed. However, in SPIHT and the proposed coder, the size of required working memory at any instant depends upon the current number of entries in the list(s). For the purpose of comparisons, memory requirements at the end of each pass (or bit-plane) and the maximum (or worst case) memory requirement are considered.

SPIHT

In SPIHT three linked lists are used namely LIP, LSP and LIS. Each entry in LIP and LSP is a single coordinate of a wavelet coefficient whereas LIS also requires type (A or B) information to distinguish nodes.

Total memory required in SPIHT, M_{SPIHT} (in bits).

$$M_{SPIHT} = c N_{LIP} + (c-1) N_{LIS} + c N_{LSP} \quad (1)$$

Where

N_{LIP} = number of entries in LIP.

N_{LSP} = number of entries in LSP.

N_{LIS} = number of entries in LIS.

c = number of bits to store addressing information of a coefficient $[2 \cdot \log_2(\max(M,N))]$

Each element in LIS requires $(c-2)$ bits for addressing, since it contains coefficients with descendents and an extra bit is required for defining the 'type' of entries.

In the worst case,

$$N_{LIP} + N_{LSP} = MN$$

$N_{LIS} = MN/4$ (as the coefficients having no descendents or the highest frequency sub-bands will never enter into LIS.) Thus the maximum working memory requirement in SPIHT is

$$M_{SPIHT}^{\max} = (5c-1) \frac{MN}{4} \quad (2)$$

SL-SPIHT

Since SL-SPIHT uses only one list namely LPS. In the list, entries correspond to address of a wavelet coefficient (except the coefficients of LL-sub-band and those which have no offsprings), so each element can be stored by using only $(c-2)$ bits. If N_{LPS} be the numbers of entries in LPS and $M_{SL-SPIHT}$ (in bits) denotes the total working memory required in the SL-SPIHT algorithm, then

$$M_{SL-SPIHT} = (c-2) N_{LPS} \quad (3)$$

In the worst case,

$$N_{LPS} = MN/4$$

Thus, the maximum working memory requirement in SL-SPIHT is

$$M_{MESH}^{max} = (c-2) \frac{MN}{4} \quad (4)$$

For example, for any 512×512 image, $c=2 \cdot \log_2(512)=18$ bits, $W=2$ bytes (say),

$$M_{SPIHT}^{max} = 729088 \text{ bytes}$$

$$M_{NLS}^{max} = 294912 \text{ bytes}$$

$$M_{MESH}^{max} = 131072 \text{ bytes}$$

Thus, $M_{SPIHT}^{max} : M_{NLS}^{max} : M_{MESH}^{max} = 5.56 : 2.25 : 1$. It should be noted that NLS uses fixed working memory equal to M_{NLS}^{max} . Thus it can be seen that the SL-SPIHT algorithm has reduced memory requirement by factors of 5.56 and 2.25 in comparisons to SPIHT and NLS respectively.

At the end of every pass, the number of elements in the list is counted. Based on this, the memory requirement for SPIHT and SL-SPIHT are calculated according to Eqs. (1) and (3) respectively. Table 1 compares the memory requirements for two test images at the end of first 10 passes. It can be observed that SL-SPIHT has about 65-75% memory saving as compared to SPIHT, which is reasonably significant.

Table 1. Memory Requirements for SPIHT and SL-SPIHT for BARBARA and BOAT Images (All values are in K Bytes)

Pass count	BARBARA		BOAT	
	SPIHT	SL-SPIHT	SPIHT	SL-SPIHT
1	1.032	0.384	1.076	0.384
2	1.176	0.384	1.598	0.480
3	1.864	0.480	3.842	1.072
4	4.473	1.200	11.919	3.576
5	15.135	4.408	29.118	8.192
6	51.891	14.440	59.114	17.248
7	101.157	27.744	101.125	28.424
8	168.327	45.816	169.339	57.088
9	260.939	72.040	337.228	94.408
10	413.191	114.632	539.563	129.528

Error Resiliency Performance

In order to compare the error resiliency of SPIHT and SL-SPIHT, we have considered the same set of gray level images. A marker (corresponding to number of bits in each pass) is placed at the end of each pass in the bitstreams of both coders.

The encoded bits are transmitted over Binary Sequential Channel (BSC) with Bit Error Rate (BER) ranging from 10^{-4} to 10^{-1} . The quality is measured in terms of Peak Signal to Noise Ratio (PSNR). All the results are averaged for 20 independent channel conditions. Fig. 1(a) and (b) show the average image quality for different bit rates for BARBARA and BOAT images respectively. Both the images are coded at 0.25 bpp.

It can be observed from these Figures that SL-SPIHT algorithm consistently outperforms SPIHT. The relative improvement is more at lower BERs than at higher BERs. More specifically, SL-SPIHT has up to 5-6 dB improvement over SPIHT for BARBARA image and about 2-3 dB for BOAT Image. This is due to the fact that use of marker in SL-SPIHT results in significant improvement as compared to the SPIHT as discussed earlier. Hence it is capable of totally nullifying the effect of error in special case when all the bits at the end of the pass are zero. This effect leads to improvement in quality in SL-SPIHT as compared to SPIHT both with markers.

5. Conclusion

In this paper, we have shown that SL-SPIHT algorithm is not only memory efficient as reported in [1], but also error resilience. Both these features are obtained due to the use of single re-usable list in SL-SPIHT as opposed to three continuously growing lists in SPIHT. It was observed that SL-SPIHT may result up to 75% of memory saving and up to 5-6 dB quality improvement under the same channel conditions. This type of coders is very useful for real time image transmission over wireless channel through hand held mobile devices such as digital cameras of mobile sets (which has scarcity of memory).

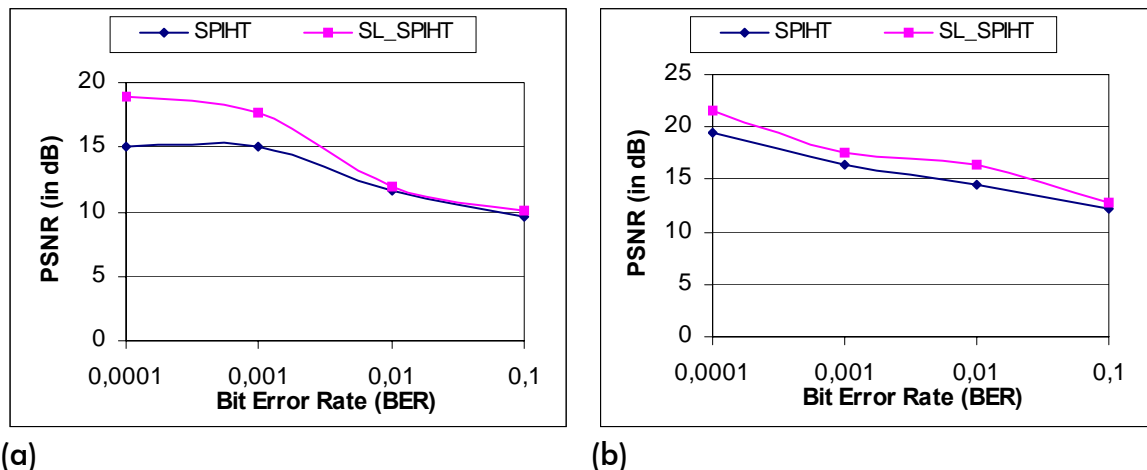
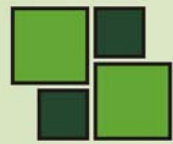


Figure 1. PSNR versus BER comparisons of SPIHT and SL-SPIHT for (a) BARBARA and (b) BOAT Images

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A COST MODEL FOR THE IT DEPARTMENT

Marius MIHUT¹

Economic Informatics Department,
Babeş-Bolyai University of Cluj-Napoca, Cluj-Napoca, Romania.

E-mail: marius.mihut@econ.ubbcluj.ro



Nicolae TOMAI²

Economic Informatics Department,
Babeş-Bolyai University of Cluj-Napoca, Cluj-Napoca, Romania.

E-mail: nicolae.tomai@econ.ubbcluj.ro



Abstract: Traditionally, it is believed that a busy IT Department (Information Technology Department) of an organization, with many services offered for other departments is very profitable, or at least “cost-effective”. A real and precise calculation of costs per product or service have unveiled that the above is not necessarily true. Activity-Based Costing is a costing model that could assign precise costs to products and services. The combination of this method with the model for calculating the total cost of an IT Department, proposed by HP Laboratories in [1] is a powerful tool for improving IT services.

Keywords: Cost model; IT Department cost model; Activity-Based Costing; Cost Object; Cost drivers

1. Introduction

The need for a cost model is given by the fact that it is necessary to identify the cost for each product or service in order to identify the profitability of IT activities and processes.

The objective of this paper is to introduce a precise method for calculating the costs of operating an IT Department. The method described in this paper extends the cost model proposed by HP Laboratories [1] with the Activity-Based Costing method to determine the real cost for each product or service. Practically, the costs calculated using HP method are tracked in activities and work processes.

2. The HP cost model for planning, development and operation of a Data Center

The cost model proposed by HP Laboratories take into account all costs which are involved in operating the data center of an organization: cost of power delivery, cost of cooling, cost of space and cost of operation. For many organizations, data centers are their own IT department. The total cost of ownership of an IT department is summarized in [1] as follows:

$$\text{Cost}_{\text{total}} = \text{Cost}_{\text{space}} + \text{Cost}_{\text{power the hardware}} + \text{Cost}_{\text{cooling}} + \text{Cost}_{\text{operation}} \quad (1)$$

Each of these costs express the "amount" of correspondent resource, consumed by the data center in a specific period of time. The methods for calculating them are detailed in [1]:

- Typically, the cost of space includes the cost of real estate necessary for the data center, for power generation systems and other auxiliary subsystems. It is considered that operating income is realized only in that portion of the data center space which is populated with computing equipment;
- Cost of power delivery includes conditioning, battery back-up, on-site power generation, and redundancy in both delivery as well as generation;
- Cost of cooling is the cost of power consumed by the cooling resources;
- Cost of operation includes personnel costs, depreciation of IT Equipment, software and licensing costs.

This model offers a method for calculating the total costs of a data center. It could be also used to calculating the total costs of an IT department. However, there is a strong need for calculating the cost of each service or product, in order to measure the performance of the IT department. To do that, it is necessary to calculate the exact "consumption" of each resource presented in relation (1) for obtaining an output (product or service). In other words, the indirect costs from (1) must be transformed in direct costs.

3. Activity-Based Costing method

Activity-Based Costing (ABC) is a cost accounting method used for understanding product, service and customer cost and profitability. Activity-Based Costing was first described in [2]. In IT domain, the directions of use for this method are:

- Supporting management decision regarding price, outsourcing or cost and revenue improvement;
- Improving IT processes;
- Calculating the cost of IT services.

Key elements of this method are *activity* and *cost driver*. Activity is defined as a set of complementary tasks, which are done with a specific purpose. Cost driver express the relation between an output (product or service) and the activities consumed by it.

Cost objects represents batches, activities, processes, products, services, customers and suppliers.

This method was developed because the traditional accounting system has a weakness with the assignment of indirect costs. Indirect costs are the expenses that don't directly generate profit, but are necessary for an organization (or department) to continue activity. Traditional accounting method arbitrarily allocates indirect costs to cost objects. Therefore, as the indirect costs increase, the traditional method will yield a less accurate result for the true cost of a cost object.

The ABC method solves this problem by transforming indirect costs to direct ones. It traces, rather than allocates, each expense category to a cost object. This method is applicable when indirect costs are greater than direct costs with 20 percents or more.

Essentially, this method groups the costs in "activity pools" instead of collecting them as indirect costs of a department. Similar processes or activities, which are driven by a common factor, are grouped in the same pool. Next step is to distribute collected costs to each product or service, by using a cost driver, as it is shown in relation 2:

$$\text{Unit Cost}_{\text{driver cost}} = \text{Cost}_{\text{activity pool}} / \text{Quantity}_{\text{driver cost}} \quad (2)$$

A cost driver is the common factor that groups activities or processes in a specific pool. Examples of cost drivers include: number of setups, number of tests, number of inspections, number of uptime hours for servers or computers, number of failures, number of cycle times, cost of providing resources, etc.. The total cost for each pool is distributed to products or services using the volume of cost driver assigned to the pool. For example, if

testing a server requires 30 percent of the testing activity and the cost driver is the number of tests, then the cost of the “testing a server” activity is 30 percent of the testing costs. The differences between the ABC and the traditional costing methods are discussed in reference [3]. These are also summarized below in table 1.

Table 1. Difference between ABC and Traditional costing [3]

	ABC method	Traditional costing method
Cost pools	Costs are grouped into <i>activity cost pools</i> , which correspond to the major activities or business processes.	Costs are grouped in pools using departmental costs. The cost pools are very heterogeneous and are not caused by a single factor.
Allocation bases	Costs are allocated to products or services by using cost drivers	Costs are allocated to products or services by using arbitrary percentage (based on products or services volumes).
Cost Objects	Intend to calculate the “real cost” of cost objects (activities, processes, products, services).	Calculate the “arbitrarily cost” of a cost object (either product or service).
Cost Drivers	Used to group costs into activity cost pool	Used to group costs into indirect/fixed costs.
Decision support	Can support strategic decisions with accurate information, because of the ability to align allocation bases with cost drivers.	Can cause problems of <i>overcosting</i> and <i>undercosting</i> because of the inability to align allocation bases with cost drivers.
Cost Control	By calculating the precise cost of organizational activities, ABC helps management to reduce costs and to prioritize efforts.	Cost control is based on a departmental approach of grouping costs.
Cost of implementation	Implementation is difficult and maintaining is expensive.	Implementation and maintaining are inexpensive because it is a traditional accounting method.

According to [4], ABC’s basic premises are:

- Cost objects consume activities.
- Activities consume resources.
- This consumption of resources is what drives costs.
- Understanding this relationship is critical to successful budget management.

These premises are illustrated in figure 1:

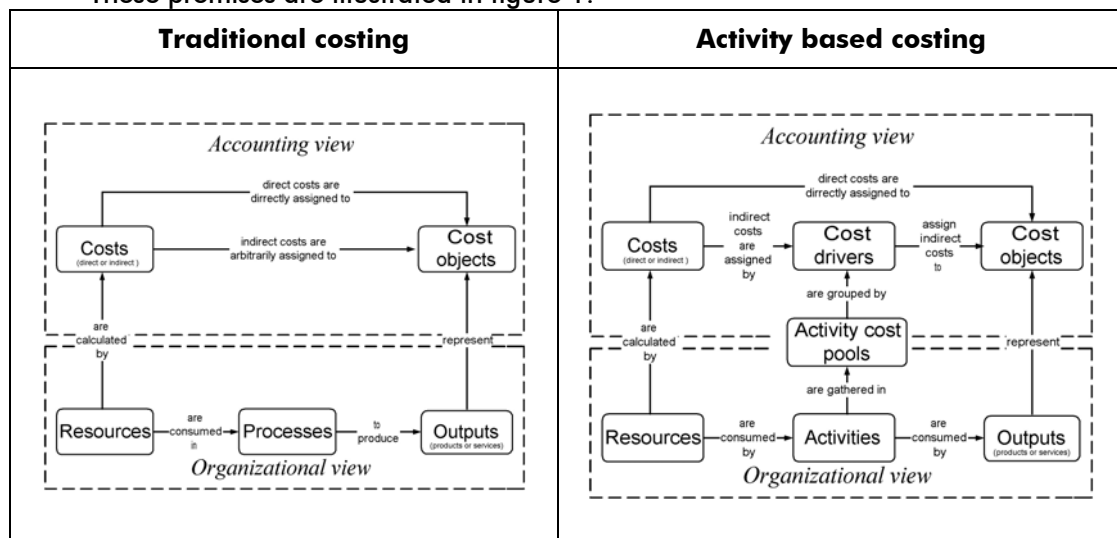


Figure 1. Basic premises of costing methods

The main benefits of Activity-Based Costing method are:

- ABC method offers a way for understanding various costs involved and offer information for: analyzing the costs, identify the activities which add value to a product or service and obtaining the benefits.
- ABC could improve organization performance, by increasing cost visibility.
- ABC could be used as a management tool that provides better allocation of resources.
- ABC offers a competitive cost advantage.

The main disadvantage of **Activity Based Costing** is given by the fact that the implementation is expensive and difficult. However, there are special activity based costing programs, which could be helpful.

4. Results and Discussion

The cost model proposed by the authors of this paper use the ABC method to identify the cost associated with an output service, for example the "Administration of IT server".

As a starting point we take into consideration the terms from relation (1), calculated for a precise period of time (i.e. a month or a year), as exemplified in table 2:

Table 2. IT Department budget

Total Costs of IT Department	
Cost of space	24,000 RON
Cost of power	34,500 RON
Cost of cooling	9,000 RON
Cost of operation, which include:	
- Salary	36,000 RON
- Purchase of new products (hardware and software)	30,000 RON
- Courses for personnel	5,000 RON
- Purchase of spare parts for equipments	9,000 RON
- Programming	7,000 RON
TOTAL COSTS PER PERIOD	138,000 RON

This table shows also the organizational resource consumption for IT department, in order to produce outputs (products and services).

Evidently, the data and information offered by the Table are not enough for calculating the exact cost of the service called "Administration of IT server".

The first three costs from Table will be allocated to service called "Administration of IT server", using parameters like: total area of IT department, area occupied by server, total power consumed by IT department, total power consumed by server, total weight of IT equipment, weight of server .

The cost of space for server is calculated in table 3:

Table 6. Cost of space for server

1. Total cost of space	24,000 RON
2. Area of space occupied with equipments	20 square meters
3. Area of space occupied with server	0.5 square meters
4. Cost of space occupied with the server (= 1 * 3 / 2)	600 RON

The cost of power for server is calculated in Table 7:

Table 7. Cost of power for server

1. Total cost of power	34,500 RON
2. Total power of IT equipments	10 KWatts
3. Total power consumed by server	500 Watts
4. Period of time	8,760 hours
5. Total power consumed by IT equipments (= 2 * 4)	87,600 KWh
6. Total power consumed by server (= 3 * 4)	4,380 KWh
7. Cost of power consumed by server (= 1 * 6 / 5) ³	1,725 RON

The cost of cooling for server is calculated in Table 8:

Table 8. Cost of cooling the server

1. Total cost of cooling	9,000 RON
2. Total weight of IT equipments	1000 Kg
3. Total weight of server	30 Kg
4. Cost of cooling for server (= 1 * 3 / 2)	270 RON

For a precise allocation of the fourth term (cost of operation) to the cost object "Administration of IT server", the ABC method is necessary.

The steps necessary for applying ABC method are:

1. Identify activities.
2. Identify cost drivers.
3. Group activities into activity cost pools.
4. Determine the cost for each activity pool.
5. Identify outputs.
6. Assign activity costs to outputs.

We illustrate this methodology in the following sections.

1. Identify activities

In the first step, the main activities of the IT department must be identified and analyzed. This is the most time-consuming and costly step because it requires a great level of detail. The most common method for identifying IT processes and activities is based on IT employee surveys. The activities are, for example: setup and configuring equipments and programs, service of equipments and programs (maintenance and repairing), developing new programs, supporting the users of the IT systems. The method accuracy depends on the number of activities that are taking into consideration. In table 3 is presented the list of activities used for obtaining the output (service) called "Administration of IT server".

2. Identify cost drivers

In this step, cost drivers are associated with each activity as exemplified in Table 9:

Table 9. Identifying cost drivers

Activity	Cost driver
Setup and configuring equipments	Number of setups
Setup and configuring programs	Number of setups
Service of equipments (maintenance and repairing)	Number of incidents
Service of programs (maintenance and repairing)	Number of incidents
Develop new programs	Programming hours
Support for users	Number of calls

3. Group activities into activity cost pools

In this step, similar activities with the same cost driver are gathered together, as is shown in Table 10:

Table 10. Activity cost pools

Activity pool	Cost driver
Setup and configuring	Number of setups
Service	Number of incidents
Develop new programs	Programming hours
Support for users	Number of calls

4. Determine the cost for each activity pool

The cost of each activity cost pool is obtained from the accounting department. These data is collected in a table format such as the one presented in Table 11 below.

Table 11. Identifying cost objects

Activity	Value (cost of activity)
Setup and configuring	24,500 RON
Service	19,000 RON
Develop new programs	22,500 RON
Support for users	21,000 RON
TOTAL	87,000 RON

5. Identify outputs

The output is a product or a service. Every output could be a cost object. In this example, the output is the service called "Administration of IT server".

6. Assign activity costs to outputs

In this step, the cost drivers determined in the previous step are used to assign cost objects (activity costs) to the output service "Administration of IT server" as presented in Table 12.

Table 12. Calculating the cost of Administration of IT server

Cost object	Value (cost of activity)	Number of activities per department	Number of activities allocated to Administration of IT server	Cost allocated to Administration of IT server
Setup and configuring	24,500 RON	15 setups	4 setups	6533,33 RON
Service	19,000 RON	18 incidents	5 incidents	5277,78 RON
Develop new programs	22,500 RON	160 hours	120 hours	16875 RON
Support for users	21,000 RON	800 calls	98 calls	2572,5 RON
Cost of operation for Administration of IT server				31258,61 RON

Finally, all costs presented in relation (1) are allocated to cost object "Administration of IT server". So, the precise cost of this service can be calculated as is shown in Table 13:

Table 13. Calculating the total cost for output Administration of IT server

Cost of space occupied with the server	600 RON
Cost of power consumed by server	1,725 RON
Cost of cooling for server	270 RON
Cost of operation for Administration of IT server	31,258.61 RON
TOTAL COST for Administration of IT server	33,853.61 RON

Using this methodology a precise cost for each "output" (product or service) can be calculated based on "the consumption" of each activity and resource.

5. Summary and conclusions

The novelty of the approach presented by the authors consist in combining the two costing methods described above that results in an improved approach for calculating and budgeting the IT activities and ultimately in supporting management decisions.

HP Laboratories offer an exhaustive method for calculating the total cost involved in operating a data center that could be used also for IT department of an organization. However it doesn't contain any indication about how to assign the indirect costs to products or services. In order to exactly calculate the exact cost of resources consumed for a product or service (to transform indirect costs into direct costs), the method presented above can be applied

The precision of ABC method is increasing proportionally with the number of activities identified. For good results, a number of, at least, 100-150 activities must be considered.

Alternatively, traditional accounting methods could be applied in place of the ABC method; however, due to the way that they allocate the indirect costs, they are not precise enough. Therefore we consider the Activity-Based Costing model as a better alternative to the traditional methods because it uses an allocation model of indirect costs based on activities.

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¹ **Marius MIHUȚ** is PhD candidate at Faculty of Economics and Business Administration, Babeș-Bolyai University of Cluj-Napoca. His main research areas include network and information security, IT law and costs. He has published 9 papers focused on multidisciplinary approach of IT security issues.

² **Nicolae TOMAI** is full Professor at Faculty of Economics and Business Administration, Babeș-Bolyai University of Cluj-Napoca. His main research areas are: Fundamentals of Computer Science, C#, Computer Networks and Distributed Systems, E-business, Mobile systems His work includes 19 books, 71 scientific papers published, 1 patent, innovation patent 5, active member in 18 research contracts.

³ Although the uptime for an average server is around 99.99 % of total time, during the maintenance and service periods the server actually is powered-on.

Sabry M. ABD-EL-FATTAH

Department of Educational Psychology,
Minia University, Minia, Egypt

E-mail: sabryrahma@hotmail.com

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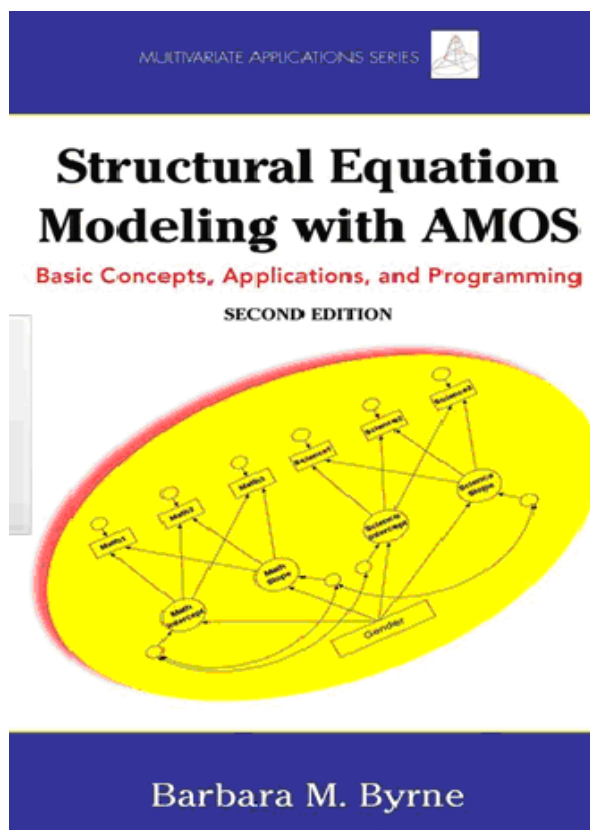
Book Review

**STRUCTURAL EQUATION MODELING WITH AMOS: BASIC
CONCEPTS, APPLICATIONS AND PROGRAMMING**

(2nd ed.)

**by Barbara M. BYRNE,
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One of the most imperative statistical and methodological innovations within area of psychology and education over the last few decades concerns the introduction of structural equation modeling (SEM). SEM links regression analysis to factor analysis. A factor model can be specified for the relationships among sets of items tapping the latent variables underlying these items; at the same time, a regression model can be specified for the relationships among these latent variables. One main advantage of SEM over other multivariate approaches is that it reduces the influence of measurement error on these regression estimates, as such errors are already "partialled out" in the factor model. A further advantage is that models can be specified and tested in details; for example, whereas traditional exploratory factor analysis assumes that all items load on all underlying dimensions, in SEM specific factor models can be specified and tested. Yet another advantage is that the fit of a model to the data can be tested and compared to the fit of competing models, thus allowing researchers to see which theoretical notions are supported empirically (Teo & Khine, 2009).



One of the most popular SEM software is that of James Arbuckle; the AMOS program. It features a sophisticated graphical interface that makes the model specification simple and the analysis output much easier to read. Furthermore, because model specification is so much easier, in my experience, users tend to make much fewer errors in specifying models than using other SEM software with no graphical interface.

Although the user's manual (Arbuckle, 2009) does a fantastic job in discussing the basic features and operation of AMOS, it falls short in two respects. First, although model specification is easy, users still have to translate their research questions into the right model (or series of models). Second, on a more theoretical level, it is important to realize that model specification is part of the job; SEM comes with a highly structured set of methodological strategies and notions that must be understood before it can be applied effectively.

One of the true virtues of the second edition of Barbara Byrne's book, *Structural Equation Modeling with AMOS*, is that it deals quite well with these two issues. According to the Preface "...overall goal is to provide readers with a nonmathematical introduction to basic concepts associated with structural equation modeling (SEM), and to illustrate basic applications of SEM using the AMOS program." All applications in the book were based on the AMOS 17.0 program.

The second edition of the book is different from the first edition in several aspects. First, the number of applications has been expanded to include the testing of: a multitrait-multimethod model, a latent growth curve model, and a second-order model based on categorical data using a Bayesian statistical approach. Second, an update was implemented concerning the automated multi-group approach to test for equivalence. Third, data derived from the use of Likert scaled measures was analyzed using both continuous and categorical approaches. Fourth, there was an update concerning the AMOS text output files which are now imbedded within cell format. Fifth, all applications are based on the graphical mode only of the program. Thus, in contrast to the first edition of this book, example input files for AMOS based on a programming approach (formerly called AMOS Basic) are not included. Finally, all data files used for the applications in this book have been uploaded on a new website: <http://www.psypress.com/sem-with-amos/>

The book is divided into five major sections. The first section discusses the basics of SEM, and presents a general introduction to model specification in AMOS. The remaining four sections deal with applications of AMOS to exemplary real life data sets drawn from Byrne's own archives. The second section is devoted to applications involving single-group analyses (i.e., analyses in which data are available for one group only; no across-group comparisons are made). This section addresses simple applications such as testing first- and second-order factor models and "causal" models (i.e., models in which some variables are presumed to affect other variables in the model causally). The third section discusses more complicated multi-group analyses (analyses in which models are compared across different groups; e.g., does the same model apply to boys and girls? can we cross-validate a particular model using data from another group?). The fourth section tackles construct validity via the multitrait-multimethod model and change over time via the latent growth model. The fifth section addresses the analysis of non-normal data using the bootstrap and missing values or incomplete data.

All chapters in these five sections follow the same format. First, a substantive research question is introduced; then the model corresponding to this research question is

presented and applied to a data set; next the output for this analysis is presented and discussed. Thus, the book can be used as a sort of cookbook; if this is the question you want to address, then this is your model and this is how you should interpret your results. The success of such an approach depends on the range of models considered. That is, by this approach readers are getting familiar with a limited number of models only. Although the readers will be able to modify these models to some degree, they may be more or less helpless when confronted with research questions or data types that require a wholly different approach.

Fortunately, Byrne covers most of the grounds relevant to education and psychology researchers. There is one exception to this general rule. Many psychological theories distinguish between main and moderator (or interaction) effects of the variables included in these models. For example, educational and psychological researchers may examine the effect of pay increase on teachers' job satisfaction, while acknowledging that these effects may vary as a function of teachers' educational qualifications. Given the importance of this issue and the fact that specification of interaction effects is not exactly easy in SEM, I would have expected some attention for this issue in this book.

Overall, Byrne's book is a valuable source to those who seek to analyze their data with SEM techniques because it addresses most of the problems readers will encounter in their own research. The remarkable features including clear style, rich illustrations of a comprehensive range of topics based on real empirical data, and simple-to-follow examples have made this book the first choice by researchers, teachers, students, and other for clearly understandable input and examples on applying SEM with AMOS.

References

1. Teo, T., & Khine, M. S. (Eds.). **Structural equation modeling in educational research: Concepts and applications**, Rotterdam, Netherlands: Sense, 2009
2. Arbuckle, J. **AMOS 17.0 user's guide**, Crawfordville, FL. Amos Development Corporation, 2009