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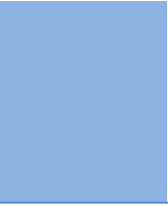
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Sim_Dsc: Simulator for Optimizing the Performance of Disk Scheduling Algorithms

By P.K. Suri, Sumit Mittal

Kurukshetra University

Abstract - Disk scheduling involves a careful examination of pending requests to determine the most efficient way to service these requests. A disk scheduler examines the positional relationship among waiting requests, then reorders the queue so that the requests will be serviced with minimum seek. The purpose of the study is to obtain the best scheduling algorithm based on the seek time, rotation time and transfer time for moveable head disks. Keeping in view an attempt has been made to design a simulator for optimizing the performance of disk scheduling algorithms using Box-Muller transformation. The input for the simulator has been derived by using an algorithm for generating pseudo random numbers which follows box-muller transformations. Simulator takes access time which is generated using seek time, rotation time and transfer time, as the request of cylinder numbers, current position of read/write head as inputs. On the basis of these inputs, total head movement of each disk scheduling algorithm is calculated under various loads.

Keywords : disk scheduling algorithms, seek time, rotational delay, transfer time, access time, head movement, box-muller transformation.

GJCST-A Classification : F.2.1,G.1.6



SIMDSCSIMULATOR FOR OPTIMIZING THE PERFORMANCE OF DISK SCHEDULING ALGORITHMS

Strictly as per the compliance and regulations of:



Sim_Dsc: Simulator for Optimizing the Performance of Disk Scheduling Algorithms

P.K. Suri^α, Sumit Mittal^Ω

Abstract - Disk scheduling involves a careful examination of pending requests to determine the most efficient way to service these requests. A disk scheduler examines the positional relationship among waiting requests, then reorders the queue so that the requests will be serviced with minimum seek. The purpose of the study is to obtain the best scheduling algorithm based on the seek time, rotation time and transfer time for moveable head disks. Keeping in view an attempt has been made to design a simulator for optimizing the performance of disk scheduling algorithms using Box-Muller transformation. The input for the simulator has been derived by using an algorithm for generating pseudo random numbers which follows box-muller transformations. Simulator takes access time which is generated using seek time, rotation time and transfer time, as the request of cylinder numbers, current position of read/write head as inputs. On the basis of these inputs, total head movement of each disk scheduling algorithm is calculated under various loads.

Keywords : disk scheduling algorithms, seek time, rotational delay, transfer time, access time, head movement, box-muller transformation.

1. INTRODUCTION

Among major responsibilities of operating system disk scheduling is one of the important tasks to use disk efficiently. For meeting these objective disk drives should have fast access time and disk bandwidth. Access time is improved by scheduling the service of disk I/O in a good manner. Many processes make request for reading/writing data on disk simultaneously. As these requests sometimes makes requests faster than serviced by the disk. Therefore, a request queue has to hold disk requests. To reduce the time spent seeking records, the request queue is ordered in some manner. This process is called Disk scheduling.

A disk-scheduling algorithm decides that which request of cylinder is to be serviced when there are so many requests. Various disk-scheduling algorithms are used. However, there will be common criteria for evaluating the performance of all these algorithms that is total head movement. Each algorithm aims to minimise the total head movement. The algorithms can be evaluated by running them on a particular string of randomly generated requests and computing the access time of the moveable head disks.

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Access Time has two major components. First one is Seek time and another one is Rotational Latency Time. The Seek Time is the time taken by read/write head to reach at a requested Cylinder/Track number and later one the time taken by the disk to rotate the desired sector under the read/write head. The disk bandwidth is defined as the total number of bytes transferred, divided by the total time between first request and completion of last transfer. Both the access time and disk bandwidth can be improved by scheduling the service of disk I/O in a good manner [7]. The time taken by a disk to move the required data under the read/write head is called rotational latency time. A disk's average rotational latency is simply half the time it takes to complete one revolution.

a) FCFS algorithm

This algorithm treats the requests of cylinders as a FIFO queue. Besides simplicity, this policy is preferred because this ensures that no request can be postponed indefinitely. This policy suffers from global zigzag effect.

b) SSTF algorithm

This algorithm selects the request, which has shortest seek from the current position of R/W head. As this policy can leads to indefinite postponement of the requests, which are not closer to R/W head. This policy gives a substantial improvement in performance, but it leads to problem of starvation.

c) SCAN algorithm

In this algorithm request is chosen for service that requires the shortest seek in preferred direction & do not change the direction until it reaches at the end of the disk. After that head moves in reverse direction and services all the requests in the opposite direction. This policy is also called as elevator algorithm.

d) C-SCAN algorithm

In C-Scan head moves only in one direction to service the requests. When head moves in reverse direction it does not service the incoming requests. When head has completed its inward sweep, it jumps to outermost cylinder without servicing the requests and then it resumes its inward sweep.

e) Look (Up/Down) algorithm

In this, head goes only as far as the final request in each direction. Then, it reverses direction immediately, without going all the way to end of disk. It

is appropriate to call the elevator algorithm as it continuous in one direction until it reaches the last request in that direction, then reverse direction.

f) C-Look algorithm

This algorithm reduces the bias against request located at the extreme ends of platters. When there is no request on a current sweep in either direction (inward or outward) the read/write head moves to the request closest to the outer/inner cylinder and again begins the next sweep.

II. RELATED WORK

David M. Jacobson and John Wilkes [1] have discussed the disk scheduling algorithm based on rotational position in their research paper. Disk scheduling based on rotational position as well as disk arm position is shown to provide improved performance. The access time based algorithms match or outperform all the seek-time ones. The best of them is Aged Shortest Access Time First, or ASATF, which forms a continuum between FCFS and SATF. It is equal or superior to the others in both mean response time and variance over the entire useful range.

Margo Seltzer, Peter Chen and John Ousterhout [2] have jointly written a research paper "Disk Scheduling Revisited". In this paper, the invention of the movable head disk has been discussed. These techniques have been applied to systems with large memories and potentially long disk queues. Disk bandwidth utilisation can be improved by applying some traditional disk scheduling techniques, which attempt to optimise head movement and guarantee fairness in response time.

Daniel T. Joyce [3] in his article "An Investigation of Disk Scheduling Algorithms Laboratory" discussed the behaviour of disk scheduling algorithms by using a simulation program. The program is used to generate data that reflects the performance of the FCFS and SSTF algorithms under a variety of conditions. For each algorithm under each situation the program simulates how the algorithm would handle the situation and calculates the expected service time b/w requests, the expected waiting time for a request and the standard deviation of these waiting times.

Toby J. Teorey and Tad B. Pinkerton [4] has discussed five well-known scheduling policies for movable head disks. These policies are compared using the performance criteria of expected seek time and expected waiting time. The variance of waiting time is introduced as another meaningful measure of performance, showing possible discrimination against individual requests. Then the choice of a utility function to measure total performance including system oriented and individual request oriented measures is described.

Helen D. Karatza [5] has discussed scheduling in a distributed system. A simulation model is used to

address performance issues associated with scheduling. Three policies which combine processor and I/O scheduling are used to schedule parallel jobs for a variety of workloads.

Hu Ming [6] has discussed disk-scheduling algorithms based on both disk arm and rotational positions. Their time-resolving powers are more precise in comparison with those for disk-scheduling algorithms based only on disk arm position. For modern disks, increase of disk rotation rate makes overhead of disk access to data transfer heavier. Therefore, it seems more important to improve both parallel processing capability of disk I/O and disk-scheduling performance at the same time.

III. PROPOSED MODEL

In this research effort, the problem under study is to optimize the performance of various disk scheduling algorithms before these are actually followed in any operating system and to design the simulator to mimic the real behaviour of the system. Because the seek distance between the position of head and position of requesting cylinder at the time of request is the basic need for evaluating the performance of the I/O system. Thus an efficient Disk Scheduling algorithm can enhance the performance of overall system whereas a poorly design scheme can degrade the performance. Thus to study the various algorithms, simulator is designed.

A simulation of any process in which there are inherently random components requires a method of generating random numbers. Thus whenever simulator is used, as a tool for research, there is need for generating random numbers that are conveniently and efficiently generated from a desired probability distribution. The present research work uses box-muller transformation for generation of cylinder numbers.

Suppose R_1 and R_2 are independent random variables that are uniformly distributed in the interval $[0, 1]$.

$$S = (-2 \log_e (R_1))^{1/2} * \cos(2\pi R_2)$$

Here S is independent random variables with a normal distribution of standard deviation 1. In present research work, the foremost criterion for the evaluation of disk scheduling algorithms is the access time calculated by seek time, rotational delay and transfer time that are produced by each policy under same set of conditions and same workload. The workload here is the cylinder numbers whose data is to be accessed to perform I/O operation. This calculated access time is used to find out the total head movement for various disk scheduling algorithms.

$$T_A = T_s + T_R + T_T$$

Where

T_A (access time): sum of seek time, rotational latency time and transfer time.

T_S (seek time): time for the disk arm to move the heads to the cylinder containing the desired sector.

T_R (rotational delay): time waiting for the disk to rotate the desired sector to the disk head.

T_T (transfer time): the time it takes to transfer a block of bits to and from the disk.

Among these three, seek time has large significant effect on the total access time of the disk. As seek time is the time relating to cylinder number. Therefore cylinder number and number of seek movements are central point of consideration.

Simulator for Optimizing the Comparative Performance of Disk Scheduling Algorithms

N : no. of cylinders

$NODE$: current position of moveable read/write head

R_1/R_2 : two independent random variables in the interval $[0, 1]$

$T_S(i)$: seek time of N cylinders

$T_A(i)$: access time of N cylinders

T_R : rotational speed of the disk

T_T : transfer time between adjacent cylinders

$RUNS$: no. of times the simulation process is repeated

$RAND$: random number

L_TIME : latency time to move the head from one to another cylinder

$CL[i]$: left requests with respect to head position.

$CR[j]$: right requests with respect to head position.

Algorithm to compute the access time to read/write a disk

Step 1. Read no. of cylinders for different workload.

Step 2. Generate random numbers using the random number generation method in the interval of $[0, 1]$.

Step 3. Compute the mean and standard deviation of m-pseudo random numbers.

Step 4. Apply Box-Muller transformation to calculate the value of S , using two random variates between $[0, 1]$.

Step 5. Using the values of mean, standard deviation and S , calculate the value of x and store in an array $x[i]$, which can use as the number of requests.

Step 6. Call modules for all seven policies named FCFS (), SSTF (), SCAN (), C-SCAN (), LOOK UP (), LOOK DOWN () and C-LOOK ().

Step 7. Compute access time based on seek time, rotational delay and transfer time produced by each policy is returned to the main module.

Step 8. Each algorithm is run for 20000 times and result of every 1000th run of each algorithm is displayed in a table.

Step 9. Stop

IV. RESULTS

The best way to compare the result of different algorithms is to present them in form of table depicting the result in the form of rows and columns. Different test cases are simulated by varying the number of randomly generated cylinders and accordingly results are shown as in Table 1/Table 2/Table 3.

Test case 1: No. of cylinders (Low Laod) = 200

Test case 2: No. of cylinders (Medium Laod) = 700

Test case 3: No. of cylinders (Heavy Laod) = 1200

Test Case 1: It is shown in the table 1 regarding total head movement of different disk scheduling algorithms in the case of low load on various simulation runs.

Simulation Runs	FCFS	SSTF	SCAN	C-SCAN	LOOK (UP)	LOOK (DN)	C-LOOK
1000	4065	574	289	376	187	107	194
2000	4677	459	2684	325	199	146	229
3000	4629	1077	293	410	199	119	211
4000	3867	479	281	361	182	121	201
5000	4328	415	299	396	226	155	252
6000	4253	536	285	369	184	113	197
7000	4133	586	282	310	187	128	208
8000	4095	530	290	378	194	118	206
9000	4372	612	282	456	180	114	193
10000	4604	448	293	385	208	137	229
11000	4260	426	302	402	218	130	230
12000	4492	558	278	355	184	134	211
13000	4438	450	281	379	183	123	203
14000	3837	403	278	355	171	108	185
15000	4713	517	290	402	203	136	225
16000	4130	539	290	379	198	126	215
17000	4690	444	298	395	204	114	211
18000	4139	481	293	326	200	121	212
19000	4580	548	298	393	222	150	245
20000	4518	482	292	382	199	122	212

Table 1 : Total head movement for low load (No. of cylinders: 200)

Test Case 2: It is shown in the table 2 regarding algorithms in the case of medium load on various total head movement of different disk scheduling simulation runs.

Simulation Runs	FCFS	SSTF	SCAN	C-SCAN	LOOK (UP)	LOOK (DN)	C-LOOK
1000	15730	1057	287	1297	199	137	223
2000	15520	1069	290	1359	215	160	249
3000	14302	947	299	1415	213	129	227
4000	15615	976	285	1325	208	161	245
5000	15438	1210	292	1427	205	134	225
6000	15253	1026	296	1382	215	142	237
7000	15683	1106	294	1350	231	180	273
8000	14991	1117	297	1402	233	175	271
9000	15959	1132	304	1372	238	164	267
10000	15072	1043	289	1415	221	175	263
11000	14662	1098	293	1318	210	141	233
12000	14926	1128	300	1365	233	166	265
13000	14034	1057	288	1380	200	136	223
14000	16468	1026	297	1426	220	149	245
15000	15466	1100	289	1402	206	145	233
16000	15024	1178	290	1379	201	132	221
17000	15252	1065	284	1424	205	158	241
18000	14442	1106	286	1408	198	138	223
19000	15238	1352	291	1392	211	149	239
20000	15617	1094	289	1310	206	145	233

Table 2: Total head movement for medium load (No. of cylinders: 700)

Test Case 3: It is shown in the table 3 regarding algorithms in the case of heavy load on various total head movement of different disk scheduling simulation runs.

Simulation Runs	FCFS	SSTF	SCAN	C-SCAN	LOOK (UP)	LOOK (DN)	C-LOOK
1000	25629	1528	289	2397	209	151	239
2000	27118	1664	301	2382	240	177	277
3000	26256	1728	300	2356	223	146	245
4000	25234	1802	294	2326	228	174	267
5000	25969	1663	292	2415	215	154	245
6000	26546	1652	302	2340	233	160	261
7000	27861	1590	293	2502	224	169	261
8000	26404	1584	298	2448	228	162	259
9000	26019	1608	299	2397	229	161	259
10000	26055	1568	293	2415	215	151	243
11000	26978	1776	309	2345	242	157	265
12000	26299	1595	291	2300	210	147	237
13000	25891	1760	297	2417	222	153	249
14000	25360	1556	300	2397	233	166	265
15000	25035	1636	291	2396	226	179	268
16000	26601	1658	303	2318	248	187	289
17000	25792	1555	294	2368	217	152	245
18000	26916	1530	310	2420	250	170	279
19000	26671	1707	294	2382	213	144	237
20000	27463	1865	290	2392	212	154	243

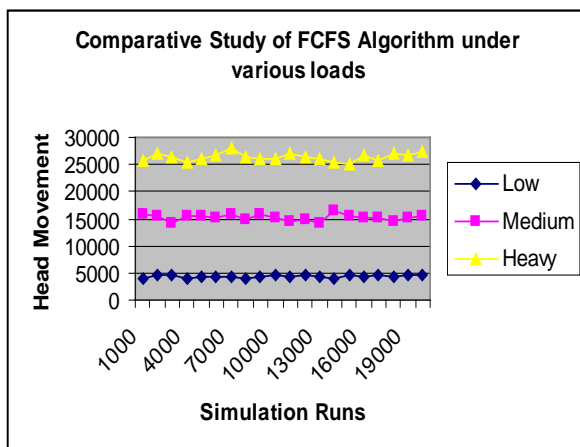
Table 3: Total head movement for heavy load (No. of cylinders: 1200)

V. DISCUSSION AND CONCLUSION

After analysing the results and findings of the simulator, it might be concluded no single policy is best in all situations. The performance do not depend upon only on the number of requests but it also depends on

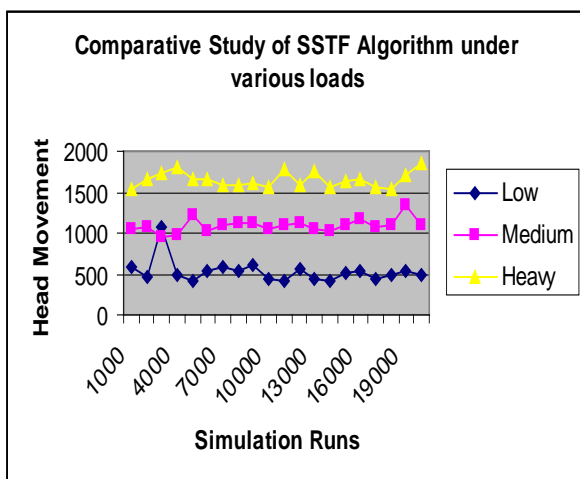
the position of read/write head & direction of the movement of head and it varies with the variation in number of requests even the current head position is same. It has been also observed that if there is only one outstanding request, then all the policies behave the same.

FCFS policy can be considered best for the system, which has fewer loads of Input-output requests, but in heavy load of requests, FCFS tends to saturate the device. SSTF produced least number of head movement in maximum runs as compared to FCFS. Therefore this policy is the optimal policy. But this policy can not be considered optimal as this policy has the starvation problem. LOOK has no starvation problem. But this policy has the overhead of decision variable, which is used to decide the direction (inward or outward) of read/write head. LOOK (Down) algorithm is always better than as compared to LOOK (UP) algorithm. C-Look disk scheduling algorithm performs better for those systems which puts medium and heavy load of requests on the disk. The graph 1 depicts the head movement for different number of simulation runs for FCFS algorithm under various loads.



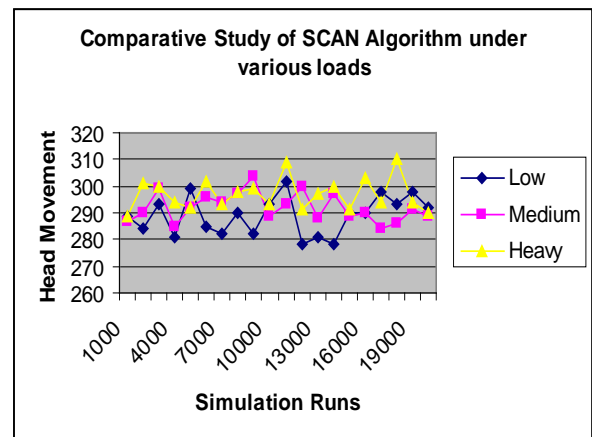
Graph No. 1

The graph 2 depicts the head movement for different number of simulation runs for SSTF algorithm under various loads.



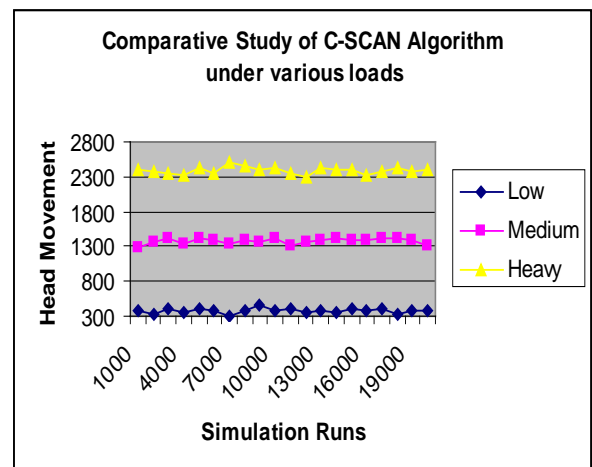
Graph No. 2

The graph 3 depicts the head movement for different number of simulation runs for SCAN algorithm under various loads.



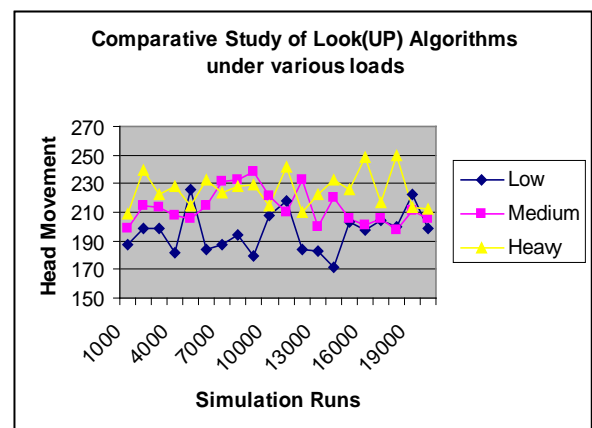
Graph No. 3

The graph 4 depicts the head movement for different number of simulation runs for C-SCAN algorithm under various loads.



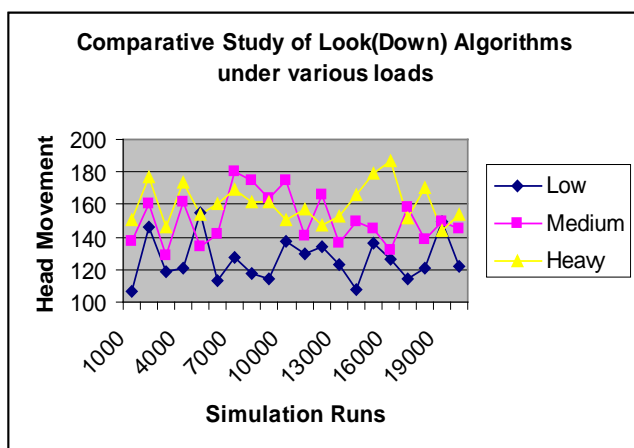
Graph No. 4

The graph 5 depicts the head movement for different number of simulation runs for Look (UP) algorithm under various loads.



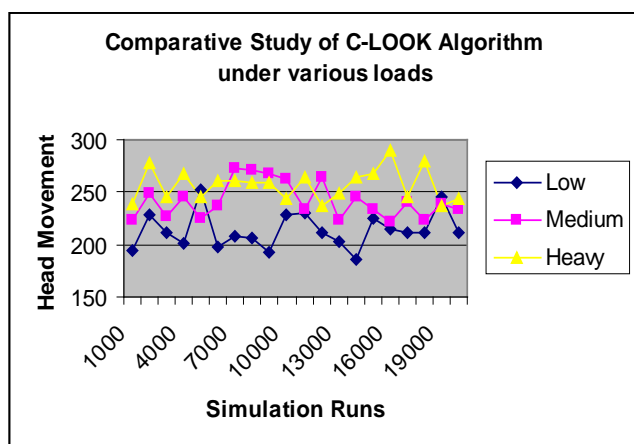
Graph No. 5

The graph 6 depicts the head movement for different number of simulation runs for Look (Down) algorithm under various loads.



Graph No. 6

The graph 7 depicts the head movement for different number of simulation runs for C-Look algorithm under various loads.



Graph No. 7

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Improved Privacy in Wireless Sensor Network Using QoS Routing Protocols

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Keywords : *anonymity; eavesdropping; hop-by-hop trace back; privacy; routing; wireless sensor networks.*

GJCST-F Classification : *C.2.2*



Strictly as per the compliance and regulations of:



Improved Privacy in Wireless Sensor Network Using QoS Routing Protocols

Tenali. Nagamani^α, Damineni.SreeLakshmi^Ω

Abstract - Full network level privacy has often been categorized into four sub-categories: *Identity*, *Route*, *Location* and *Data* privacy. Achieving full network level privacy is a challenging problem due to the conditions imposed by the sensor nodes (e.g., energy, memory and computation power), sensor networks (e.g., mobility and topology) and QoS issues (e.g., packet reach-ability and timeliness). This proposed paper consists of two algorithms IRL algorithm and data privacy mechanism that addresses this problem. The proposed system provides additional trustworthiness, less computation power, less storage space and more reliability. Also, we proved that our proposed solutions provide protection against various privacy disclosure attacks, such as eavesdropping and hop-by-hop trace back attacks.

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I. INTRODUCTION

In order to present the adversary from back-tracing, the route, location and data privacy mechanism must be enforced. With the spreading application of Wireless Sensor Networks (WSNs) in various sensitive areas such as health-care, military, habitat monitoring, etc. Network level privacy often been categorized into 4 categories:

1. Sender node identity privacy: no intermediate node can get any information about who is sending the packets except the source, its immediate neighbors and the destination.
2. Sender node location privacy: no intermediate node can have any information about the location (in terms of physical distance or number of hops) about the sender node except the source, its immediate neighbors and the destination.
3. Route privacy: no node can predict the information about the complete path (from source to destination). Also, a mobile adversary gets no clue to trace back the source node either from the contents and/or directional information of the captured packet(s).
4. Data packet privacy: no node can see the information inside in a payload of the data packet except the source and the destination.

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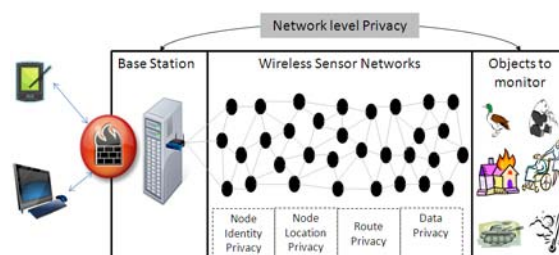
An energy-efficient privacy solution is needed to address these patterns in Wireless Sensor Network. Advanced features in cryptographic system were introduced in this paper are:

- A new Identity, Route and Location (IRL) privacy algorithm is proposed that ensures the source, identity and location. This algorithm allows the packets to destination only through trusted intermediate nodes.
- The extension of our proposed IRL algorithm is a new reliable Identity, Route and Location (r-IRL) privacy algorithm. This algorithm has the ability to forward packets from multiple secure paths to increase the packet reach-ability.
- A data privacy mechanism is used to unique in the sense that it provides secure data and packet authentication.

a) Network and Assumptions Model

A wireless sensor network (WSN) is composed of large number of small sensor nodes that are of limited resource and densely deployed in an environment. This sensor node uses IEEE 802.11 standard link layer protocol, which keeps packets in its cache until the sender receives an acknowledgment (ACK). The sender node will retransmit the packet, if the ACK does not receive within threshold.

Figure 1 : Typical WSN scenario.



II. PROPOSED SCHEME

a) Concepts and Definitions

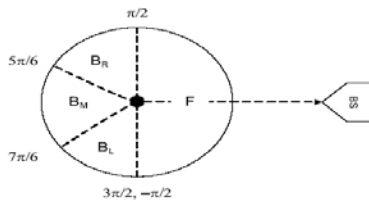
The proposed algorithms use two notions: Direction and Trust. These notions are used to provide reliable secure paths for achieving robust route privacy. Direction helps to forward packet to the destination in a timely manner and trust will help to forward the packets via reliable nodes.

Direction: The first notion used in our algorithms is that of direction. The physical location of the base station is the reference point for each sensor node. Based on this reference point, each node classifies its neighboring nodes into four categories: (1) forward neighboring nodes (F), (2) right side backward neighboring nodes (Br), (3) left side backward neighboring nodes (Bl), and (4) middle backward neighboring nodes (Bm). The objective of this categorization is to provide more path diversity as discussed in Section 4.2. A node x classifies its neighboring node y in following fashion :

$$C_{x,y} = \begin{cases} F & -\frac{\pi}{2} \leq \theta \leq \frac{\pi}{2} \\ B_r & \frac{\pi}{2} < \theta \leq \frac{5\pi}{6} \\ B_m & \frac{5\pi}{6} < \theta \leq \frac{7\pi}{6} \\ B_l & \frac{7\pi}{6} < \theta < \frac{3\pi}{2} \end{cases}$$

Where θ is the angle between the node x and its neighboring node y with respect to the line joining node x and the base station as shown in Figure 2.

Figure 2 : Neighbor node classification



Trust: The second notion used in our algorithms is that of trust. The definition of a trust here is based on our other paper and restated here. A node can be classified into one of the three categories: trustworthy, untrustworthy, and uncertain. A node is considered trustworthy if it interacts successfully most of the time with the other nodes. A node is considered untrustworthy if it tries to do as many unsuccessful interactions as possible with the other nodes. An untrustworthy node could be a faulty or malicious node. A node is considered uncertain if it performs both successful and unsuccessful interactions. Detailed definition of the successful and unsuccessful interactions and trust calculation methodology is available in our paper and provided here in a simplified form.

A sender will consider an interaction successful if the sender receives confirmation that the packet is successfully received by the neighbor node and forwarded towards the destination in an unaltered fashion. The first requirement of successful reception is achieved on the reception of the link layer acknowledgment (ACK). The second requirement of forwarding towards the destination is achieved with the help of enhanced passive acknowledgment (PACK) by overhearing the transmission of a next hop on the route, since they are within the radio range. If the sender node does not overhear the retransmission of the packet within a threshold time from its neighboring node or if

the overheard packet is found to be illegally fabricated (by comparing the payload that is attached to the packet), then the sender node will consider that interaction as unsuccessful.

With this simple approach, several attacks can be prevented, i.e., the black hole attack is straightforwardly detected when malicious node drops the incoming packets and keeps sending self-generated packets. Similarly, sink hole attack, an advanced version of the black hole attack, is also easily detectable by looking at the passive acknowledgment. Likewise, the selective forwarding attack and gray-hole attack [27] can also be eliminated with the aid of above mentioned approach. Based on these successful and unsuccessful interactions node x can calculate the trust value of node y in following fashion:

$$T_{x,y} = \left[100 \left(\frac{S_{x,y}}{S_{x,y} + U_{x,y}} \right) \left(1 - \frac{1}{S_{x,y} + 1} \right) \right]$$

Where $[.]$ is the nearest integer function, $S_{x,y}$ is the total number of successful interactions of node x with y during time δt , and $U_{x,y}$ is the total number of unsuccessful interactions of node x with y during time δt . After calculating trust value, a node will quantize trust into three states as follows:

$$Mp(T_{x,y}) = \begin{cases} \text{trustworthy} & 100 - f \leq T_{x,y} \leq 100 \\ \text{uncertain} & 50 - g \leq T_{x,y} < 100 - f \\ \text{untrustworthy} & 0 \leq T_{x,y} < 50 - g \end{cases}$$

Where, f represents half of the average values of all trustworthy nodes and g represents one-third of the average values of all untrustworthy nodes. Both f and g are calculated as follows:

$$f_{j+1} = \begin{cases} \left[\frac{1}{2} \left(\frac{\sum_{i \in R_x} T_{x,i}}{|R_x|} \right) \right] & 0 < |R_x| \leq n - 1 \\ f_j & |R_x| = 0 \end{cases}$$

$$g_{j+1} = \begin{cases} \left[\frac{1}{3} \left(\frac{\sum_{i \in M_x} T_{x,i}}{|M_x|} \right) \right] & 0 < |M_x| \leq n - 1 \\ g_j & |M_x| = 0 \end{cases}$$

The steady-state operation, these values can change with every passing unit of time which creates dynamic trust boundaries. After each passage of time, Δt , nodes will recalculate the values of f and g . This trust calculation procedure will continue in this fashion.

The time window length (Δt) could be made shorter or longer based on the network analysis scenarios. If Δt is too short, then the calculated trust value may not reflect the reliable behavior. On the other hand, if it is too long, then it will consume too much memory to store the interaction record at the sensor node. Therefore, various parameters can be used to adjust the length of Δt .

Where $[.]$ is the nearest integer function, R_x represents the set of trustworthy nodes for node x , M_x the set of untrustworthy nodes for node x , and n is the total number of nodes that contains trustworthy, untrustworthy and uncertain nodes. The initial trust

values of all nodes are 50. The values of f and g are adaptive.

b) Identity, Route, and Location Privacy (IRL)

The proposed identity, route and location privacy scheme works in two phases. The first is neighbor node state initialization phase, and the second is routing phase.

Route Privacy: In initialization phase, let the node i have m neighboring nodes in which t nodes are trusted. So, $0 \leq t \leq m$ and $M(t) = M(tF) \cup M(tBr) \cup M(tBl) \cup M(tBm)$. Here $M(tF)$, $M(tBr)$, $M(tBl)$, and $M(tBm)$ represent the set of trusted nodes that are in the forward, right backward, left backward, and middle backward directions, respectively. These neighbor sets ($M(tF)$, $M(tBr)$, $M(tBl)$, and $M(tBm)$) are initialized and updated whenever a change occurs in neighborhood. For example, the entrance of a new node, change of a trust value, etc.

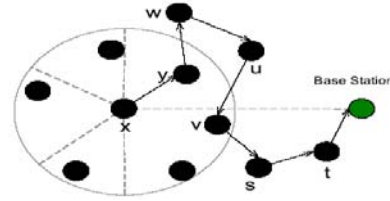
Whenever a node needs to forward a packet, the routing phase for source node and for intermediate node) of IRL algorithm is called.

Whenever a source node wants to forward the packet, it will first check the availability of the trusted neighboring nodes in its forward direction set $M(tF)$. If trusted nodes exist then it will randomly select one node as a next hop from the set $M(tF)$ and forward the packet towards it. If there is no trusted node in its forward direction, then the source node will check the availability of a trusted node in the right ($M(tBr)$) and left ($M(tBl)$) backward sets. If the trusted nodes are available then the source node will randomly select one node as a next hop from these sets and forward the packet towards it. If the trusted node does not exist in these sets either, then the source node will randomly select one trusted node from the backward middle set ($M(tBm)$) and forward the packet towards it. If there are no trusted nodes available in all of the sets then the packet will be dropped.

When an intermediate node receives the packet (either from the source node or from another en-route node), it will first check whether the packet is new or old. If it is new, then the node will first check the availability of the trusted node from the forward direction set (MF) excluding the *prevhop* node if it belongs to forward set. If trusted nodes exist in the forward set then the node will randomly select any one trusted node as a next hop and forward the packet towards it. If there is no trusted node available in the forward direction, then it will check to which set the sender of the packet belongs to. For example, If the packet, forwarded by a node, belongs to the right backward set, then it will first check whether the left or middle backward sets contain any trusted nodes. If so, it will randomly select one node from those sets and forward the packet towards it. If there is no trusted node in those two sets, then the node will randomly select a trusted node from the right backward set ($M(tBr)$) excluding the one from which

the node received the current packet and forward the packet towards it. Similar operations will be performed, if the packet, forwarded by a node, belongs to the left and middle backward or forward sets. An example IRL routing scenario is shown in Figure 3.

Figure 3 : Sample routing scenario of IRL scheme.

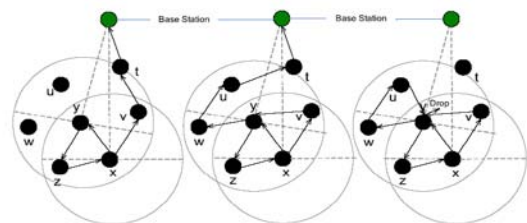


This routing strategy may result in the creation of a cycle (loop). However, due to the randomness in the selection of the next-hop and the presence of the different four direction sets, the probability of creation of any cycle is very low. Nevertheless, in order to fully avoid the occurrence of the cycles, each node (prior to forwarding of a packet) will save the signature of the packet in the buffer for the δt time, that is:

$$\delta t = 2 \left(\frac{D}{d} \times p_t \right)$$

Where D is the distance between the forwarding node and the base station, d is the distance between the forwarding node and the next hop, and p_t is the propagation transfer time between the forwarding node and the next hop. This signature consists of two fields: (1) sequence number of the packet, and (2) the payload. The potential of the signature to compare and identify the same packet is detailed in the later section. Corresponding to this signature, three more fields are also stored in the buffer: (1) Previous hop identity, (2) next hop identity where the packet is forwarded, and (3) Counter, that tells how many times the same packet is received by the node. This information will later be used to get rid of any cycle. The size of the buffer is mainly dependent on the network traffic conditions. However, it is expected to be low due to the sensor nodes send data either in periodic intervals or upon the occurrence of some event.

Figure 4 : Three sample cycle detection and prevention scenarios.



If the node received the packet whose signature exists in the buffer, then including the previous hop node, two other nodes will also be excluded from the

selection of the next hop process: 1) the node from which last time the packet was received the node from which last time the packet was forwarded. If the same packet is received three times by the same node then the packet will be dropped. Three sample scenarios of the loop creation, detection and prevention are shown in Figure 4. Creation of loops and traversing of the packets in the backward direction is not a completely negative effect. Rather, it provides positive effects in terms of strengthening the route and source location privacy, because these effects will help to increase the safety period, which is the time for an adversary to reach at the source node.

Identity Privacy: Whenever a node receives the packet p from the source node or en-route node then the receiving node will replace the previous hop's identity $prevhop$ contained in the packet with its own. After that, the node will get the next forwarding node $nexthop$ and update the header of the packet $p = \{prevhop, nexthop, payload\}$. After modification of the two header fields, the node will forward the packet. In this way, all the intermediate forwarding nodes replace the source and next hop's identity contained in the packet p . This process will go on until the packet reaches the base station.

Location Privacy: The neighboring nodes which are in each other's radio range can easily approximate the location of each other by measuring the received signal strength and the angle of arrival. If the adversary is within the range of the source node, then adversary can easily estimate the location of the source. Once the packet has crossed the radio range of the original source node, then becomes very difficult for an attacker to estimate the location of the node either in terms of the physical distance or in terms of the number of hops of an original source node. The main reason for this is that the path selection is random and packets are forwarded by only trusted nodes which only contain the information of the last and the next hop.

c) *Reliable Identity, Route, and Location Privacy (r-IRL)*

It is also possible that some applications require more reliability in terms of packet reach-ability; and the packet could be dropped due to either network congestion or malicious behavior of an en-route node. Thus, in order to achieve more reliability, the packet should be forwarded from multiple paths simultaneously, which will give trustworthiness in the sense that at least the packet should reach the base station by any one of the paths, although, this may increase some communication overhead. Our reliable IRL (r-IRL) algorithm is the extended version of our proposed IRL algorithm, in which we introduce one more parameter, reliability r . The source node i will multi-cast a packet to all r randomly selected neighboring trusted nodes that are in the forward direction. If there are no adequate trusted nodes present in the forward direction, then it will

select the remaining trusted nodes from the backward direction. The rest of the mechanism of the r-IRL algorithm is the same as the IRL algorithm.

d) *Data Privacy*

The payload contains the identity of the source node (IDx) and the actual data (d). Identity is encrypted with the public key ($k+bs$) of the base station and data is encrypted with the secret key (kx,bs) shared between the sender node and the BS. Both are appended with the payload as shown below:

$$Payload = [E(IDx, k+bs), E(d, kx,bs)]$$

If we assume that the adversary knows the range of identities assigned to the sensor nodes, public key of the base station and information about cipher algorithm used in the network, an adversary can then successfully obtain the identity of the source by performing simple brute-force search attack by comparing the pattern of encrypted identity with a known range of identities. Therefore in order to provide protection against brute-force search attack, we append a random number (Rn) (equivalent to the size of identity) with the identity of a node and then perform encryption. Now the payload is:

$$Payload = [E(IDx//Rn, k+bs), E(d, kx,bs)]$$

Where $//$ is the append operation. Inclusion of random number may introduce additional computational overhead. However, the amount of overhead is mainly dependent on random number generation technique. Recently, very nice random generation techniques have been specially designed for low power sensor networks, such as. These techniques could be used to generate random number for each packet. Also, overall computational overhead is dependent on the number of packets generated by the sensor nodes.

Our proposed data privacy approach provides several benefits. Firstly, data secrecy is achieved in the presence of identity anonymity. This feature is not available in earlier proposed privacy schemes. Secondly, the base station will receive both the identity of the actual source node and message authentication. If the packet has been successfully decrypted with the shared secret key, it means that packet is received from genuine sensor node.

III. ANALYSIS AND EVALUATION

a) *Energy Consumption Analysis*

This section, shows the efficiency of our routing strategies with existing schemes. Energy is computed based on the communication overhead (including transmission and reception cost, path length) introduced by our proposed routing protocols and compared it with other existing schemes.

Table 4 : Simulation parameters

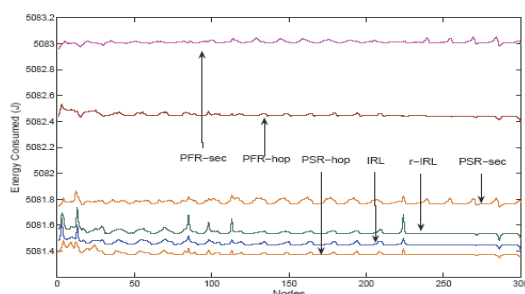
Network specific	Number of nodes	300
	Distance b/w nodes	50 units
	Mobility of nodes	zero
Node specific	Sensor node's Initial battery	$1 \times 106J$
	Power consumption for trans.	1.6W
	Power consumption for recv.	1.2 W
	Idle power consumption	1.15W
	Carrier sense threshold	$3.65e-10W$
	Receive power threshold	$1.55e-11W$
	Frequency	$9.14e8$
	Trans. & Recv. antenna gain	1.0
Protocol & Application specific	Application	CBR
	Reliability param. r for r-IRL	3
	$hwalk$ param. for PFR & PSR	10

The proposed paper has implemented our IRL and r-IRL routing schemes on Sensor Network Simulator and Emulator (SENSE). At the application layer we used constant bit rate component (CBR) that generate constant traffic during simulation between randomly selected source node(s) and the base station. For the simplicity, assume that both sensor nodes and the base station are static. Network consists of 300 sensor nodes that are organized into 15 by 20 grid manner.

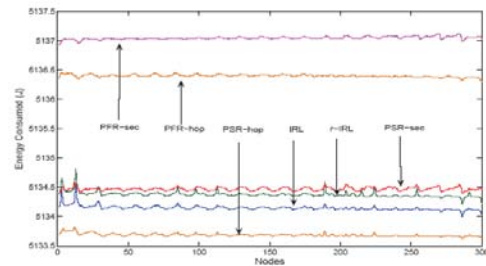
Comparison of proposed IRL and r-IRL algorithms with the four variations of phantom routing schemes that are:

1. Phantom single path routing scheme with hop-based approach (PSR-hop).
2. Phantom single path routing scheme with sector-based approach (PSR-sec).
3. Phantom flood routing scheme with hop-based approach (PFR-hop).
4. Phantom flood routing scheme with sector-based approach (PFR-sec).

Figure 6 : Energy consumption analysis: simulation time: 5,000.



(a) Source node 5

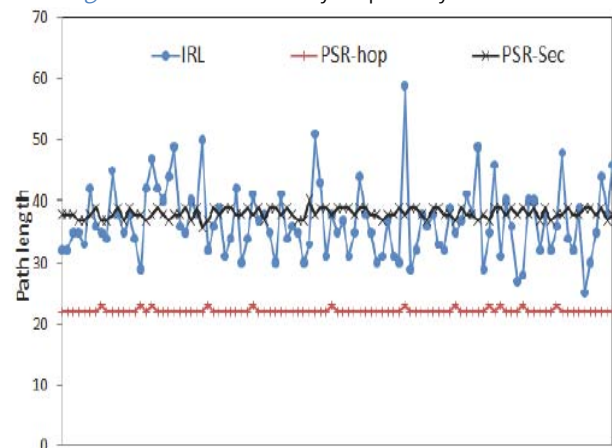


(b) Source node 10

The energy consumption analysis with different scenarios are shown in Figure 6. For the r-IRL scheme we select $r = 3$, which means a single packet will reach the destination via three different routes simultaneously. For phantom routing schemes, we select parameter $hwalk=10$ (as recommended). Figure 6 clearly indicates that, the IRL and r-IRL schemes consume less energy as compared to the PSR-sec, PFR-hop and PFR-sec schemes but slightly consume higher energy as compared to the PSR-hop scheme. This is due to the fact that the IRL and r-IRL algorithms provides more path diversity and packets sometimes took longer paths.

Our proposed routing strategies (IRL and r-IRL) have both features. Because of the concept of *direction* (Section 3.1), proposed schemes provide more length variation and because of the *randomness* (Section 3.2) proposed schemes provide high path variation. Incorporation of both features offer high path diversity.

Figure 7 : Path diversity of privacy schemes.



In order to analyze the path diversity behavior, assume 300 sensor nodes in a 10 by 30 grid manner. In the simulation, a single source node (ID: 224) generates 100 data packets for the base station. Figure 7 shows the path diversity (in terms of path length) of the IRL, PSR-hop and PSR-sec schemes.

The average path taken by the PSR-hop, IRL and PSR-sec is 22.12, 36.81 and 38.17, respectively. It indicates that the IRL scheme incurs more delay as compared with the PSR-hop scheme and less delay as compared with the PSR-sec scheme. This figure also indicates that the IRL scheme has more path variation as

compared with the other schemes, which creates more difficulties for the adversary to trace back the source from the captured packets.

Figure 7 also shows that some packets took longer paths in the IRL scheme as compared with others. This is due to the fact that the source or en-route node did not find any trusted node in its forward direction, so the packet is relayed back in the backward direction.

Figure 8 shows the result of 100 simulation runs in each node has equal probability to be trusted and untrusted. It shows that, as the neighborhood size increases, the probability of the packet to move in the backward direction decreases sharply.

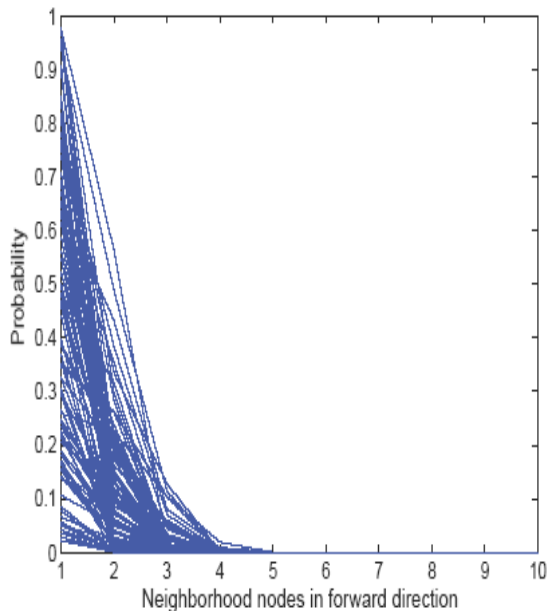


Figure 8 : Probability of a packet to move in the backward direction.

IV. CONCLUSIONS AND FUTURE WORK

Existing privacy schemes of WSNs only provides partial network level privacy. Providing full network level privacy is a critical and challenging issue due to the constraints imposed by the sensor nodes (e.g., energy, memory and computation power), sensor network (e.g., mobility and topology) and QoS issues (e.g., packet reach-ability and timeliness). Therefore, in this paper we proposed the first full network level privacy solution that is composed of two new identity, route and location privacy algorithms and data privacy mechanism. Our solutions provide additional trustworthiness and reliability at modest cost of energy and memory. Future work, will evaluate proposed schemes from the perspective of computation cost that is required to perform encryption and random number generation.

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Software Effort Estimation Using Particle Swarm Optimization with Inertia Weight

By Prasad Reddy.P.V.G.D, Ch.V.M.K.Hari

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Abstract - Software is the most expensive element of virtually all computer based systems. For complex custom systems, a large effort estimation error can make the difference between profit and loss. Cost (Effort) Overruns can be disastrous for the developer. The basic input for the effort estimation is size of project. A number of models have been proposed to construct a relation between software size and Effort; however we still have problems for effort estimation because of uncertainty existing in the input information. Accurate software effort estimation is a challenge in Industry. In this paper we are proposing three software effort estimation models by using soft computing techniques: Particle Swarm Optimization with inertia weight for tuning effort parameters. The performance of the developed models was tested by NASA software project dataset. The developed models were able to provide good estimation capabilities.

Keywords : PM- Person Months, KDLOC-Thousands of Delivered Lines of Code, PSO - Particle Swarm Optimization, Software Cost Estimation

GJCST-C Classification : D.2.9



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Prasad Reddy.P.V.G.D^a, Ch.V.M.K.Hari^Ω

Abstract - Software is the most expensive element of virtually all computer based systems. For complex custom systems, a large effort estimation error can make the difference between profit and loss. Cost (Effort) Overruns can be disastrous for the developer. The basic input for the effort estimation is size of project. A number of models have been proposed to construct a relation between software size and Effort; however we still have problems for effort estimation because of uncertainty existing in the input information. Accurate software effort estimation is a challenge in Industry. In this paper we are proposing three software effort estimation models by using soft computing techniques: Particle Swarm Optimization with inertia weight for tuning effort parameters. The performance of the developed models was tested by NASA software project dataset. The developed models were able to provide good estimation capabilities.

Index Terms : PM- Person Months, KDLOC-Thousands of Delivered Lines of Code, PSO - Particle Swarm Optimization, Software Cost Estimation.

I. INTRODUCTION

The modern day software industry is all about efficiency. With the increase in the expanse and impact of modern day software projects, the need for accurate requirement analysis early in the software development phase has become pivotal. The provident allocation of the available resources and the judicious estimation of the essentials form the basis of any planning and scheduling activity. For a given set of requirements, it is desirable to cognize the amount of time and money required to deliver the project prolifically. The chief aim of software cost estimation is to enable the client and the developer to perform a cost – benefit analysis. The software, the hardware and the human resources involved add up to the cost of a project. The cost / effort estimates are determined in terms of person-months (pm) which can be easily interchanged to actual currency cost.

The basic input parameters for software cost estimation is size, measured in KDLOC (Kilo Delivered Lines Of Code). A number of models have been evolved to establish the relation between Size and Effort [13]. The parameters of the algorithms are tuned using Genetic Algorithms [5], Fuzzy models [6], Soft-Computing Techniques [7][9][10], Computational Intelligence Techniques[8],Heuristic Algorithms, Neural Networks, Radial Basis and Regression [11][12] .

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a) Basic Effort Model

A common approach to the estimation of the software effort is by expressing it as a single variable function of the project size. The equation of effort in terms of size is considered as follows:

$$\text{Effort} = a * (\text{Size})^b \quad (1)$$

Where a, b are constants. The constants are usually determined by regression analysis applied to historical data.

b) Standard PSO with Inertia Weights

In order to meet the needs of modern day problems several optimization techniques have come been introduced. When the search space is too large to search exhaustively, population based searches may be a good alternative, however, population based search techniques cannot guarantee you the optimal (best) solution. We will discuss a population based search technique, Particle Swarm Optimization (PSO) with Inertia Weights [Shi and Eberhart 1998]. Particle Swarm has two primary operators: Velocity update and Position update. During each generation each particle is accelerated toward the particles previous best position and the global best position. At each iteration a new velocity value for each particle is calculated based on its current velocity, the distance from its previous best position, and the distance from the global best position. The new velocity value is then used to calculate the next position of the particle in the search space. The inertia weight is multiplied by the previous velocity in the standard velocity equation and is linearly decreased throughout the run. This process is then iterated a set number of times or until a minimum error is achieved.

The basic concept of PSO lies in accelerating each particle towards its Pbest and Gbest locations with regard to a random weighted acceleration at each time. The modifications of the particle's positions can be mathematically modeled by making use of the following equations:

$$V_i^{k+1} = w * V_i^k + c_1 * \text{rand}()_1 * (Pbest - S_i^k) + c_2 * \text{rand}()_2 * (Gbest - S_i^k) \quad (2)$$

$$S_i^{k+1} = S_i^k + V_i^k \quad (3)$$

Where,

S_i^k is current search point,

S_i^{k+1} is modified search point,

V_i^k is the current velocity,

V^{k+1} is the modified velocity,

V_{pbest} is the velocity based on Pbest ,

V_{gbest} = velocity based on Gbest,

w is the weighting function,

c_j is the weighting factors,

$\text{Rand}()$ are uniformly distributed random numbers between 0 and 1.

II. THE STANDARD PSO WITH INERTIA WEIGHT FOR SOFTWARE EFFORT ESTIMATION

The software effort is expressed as a function of a single variable of effort in terms of the project size as shown in equation-1. The parameters a , b are measured by using regression analysis applied to historical data. In order to tune these parameters we use the standard PSO with inertia weights. A nonzero inertia weight introduces a preference for the particle to continue moving in the same direction it was going on the previous iteration. Decreasing the inertia over time introduces a shift from the exploratory (global search) to the exploitative (local search) mode. The updating of weighting function is done with the following formula.

$$W_{\text{new}} = [(T_{\text{mi}} - T_{\text{ci}}) * (W_{\text{iv}} - W_{\text{fv}})] / T_{\text{mi}} + W_{\text{fv}} \quad (4)$$

Where

W_{new} is new weight factor,

T_{mi} is the maximum number of iteration specified,

T_{ci} is the current iteration number,

W_{iv} is the initial value of the weight,

W_{fv} is the final value of the weight.

Empirical experiments have been performed with an inertia weight set to decrease linearly from 0.9 to 0.4 during the course of simulation. In the first experiment we keep the parameters c_1 and c_2 (weighting factors) fixed, while for the following experiment we change c_1 and c_2 (weighting factors) during subsequent iterations by employing the following equations [Rotnaweera, A. Halgamog S.K. and Watson H.C, 2004].

$$C_1(t) = 2.5 - 2 * (t / \text{max_iter}), \text{ which is the cognitive learning factor.} \quad (5)$$

$$C_2(t) = 0.5 + 2 * (t / \text{max_iter}), \text{ which is the social coefficient.} \quad (6)$$

The particles are initialized with random position and velocity vectors the fitness function is evaluated and the Pbest and Gbest of all particles is found out. The particles adjust their velocity according to their Pbest and Gbest values. This process is repeated until the particles exhaust or some specified number of iterations takes place. The Gbest particle parameters at the end of the process are the resultant parameters.

III. MODEL DESCRIPTION

In this model we have considered "The standard PSO with inertia weights" with /without changing the weighting factors (c_1 , c_2). PSO is a robust stochastic optimization technique based on the movement of swarms. This swarm behavior is used for tuning the parameters of the Cost/Effort estimation. As the PSO is a random weighted probabilistic model the previous benchmark data is required to tune the parameters, based on that data, swarms develop their intelligence and empower themselves to move towards the solution. The following is the methodology employed to tune the parameters in each proposed models following it.

a) METHODOLOGY (ALGORITHM)

Input: Size of Software Projects, Measured Efforts, Methodology (Effort Adjustment factor-EAF).

Output: Optimized Parameters for Estimating Effort.

The following is the methodology used to tune the parameters in the proposed models for Software Effort Estimation.

Step 1: Initialize "n" particles with random positions P_i and velocity vectors V_i of tuning parameters. We also need the range of velocity between $[-V_{\text{max}}, V_{\text{max}}]$. The Initial positions of each particle are Personally Best for each Particle.

Step 2: Initialize the weight function value w with 0.5 and weighting parameters cognitive learning factor c_1 , social coefficient c_2 with 2.0.

Step 3: Repeat the following steps 4 to 9 until number of iterations specified by the user or Particles Exhaust.

Step 4: for $i = 1, 2, \dots, n$ do // For all the Particles

For each particle position with values of tuning parameters, evaluate the fitness function. The fitness function here is Mean Absolute Relative Error (MARE). The objective in this method is to minimize the MARE by selecting appropriate values from the ranges specified in step 1.

Step 5: Here the Pbest is determined for each particle by evaluating and comparing measured effort and estimated effort values of the current and previous parameters values. If fitness (p) better than fitness (Pbest) then: Pbest = p .

Step 6: Set the best of 'Pbests' as global best – Gbest. The particle value for which the variation between the estimated and measured effort is the least is chosen as the Gbest particle.

Step 7: Update the weighting function is done by the following formula

$$W_{\text{new}} = [(T_{\text{mi}} - T_{\text{ci}}) * (W_{\text{iv}} - W_{\text{fv}})] / T_{\text{mi}} + W_{\text{fv}} \quad (7)$$

Step 8: Update the weighting factors is done with the following equations for faster convergence.

$$C_1(t) = 2.5 - 2 * (T_{ci} / T_{mi}) \quad (8)$$

$$C_2(t) = 0.5 + 2 * (T_{ci} / T_{mi}), \quad (9)$$

Step 9: Update the velocity and positions of the tuning parameters with the following equations for $j = 1, 2, \dots, m$ do // For number of Parameters, our case m is 2 or 3 or 4

begin

$$V_{ji}^{k+1} = w * V_{ji}^k + c_1 * \text{rand}()_1 * (P_{best} - S_{ji}^k) + c_2 * \text{rand}()_2 * (G_{best} - S_{ji}^k) \quad (10)$$

$$S_{ji}^{k+1} = S_{ji}^k + V_{ji}^{k+1} \quad (11)$$

end;

Step 10: Give the Gbest values as the optimal solution.

Step 11: Stop

b) PROPOSED MODELS

i. MODEL 1:

A prefatory approach to estimating effort is to make it a function of a single variable, often this variable is project size measure in KDLOC (kilo delivered lines of code) and the equation is given as,

$$\text{Effort} = a (\text{size})^b$$

Now in our model the parameters are tuned using above PSO methodology. The Update of velocity and positions of Parameter "a" is

$$V_{ai}^{k+1} = w * V_{ai}^k + c_1 * \text{rand}()_1 * (P_{best} - S_{ai}^k) + c_2 * \text{rand}()_2 * (G_{best} - S_{ai}^k) \quad (12)$$

$$S_{ai}^{k+1} = S_{ai}^k + V_{ai}^{k+1}$$

The Update of velocity and positions of Parameter "b" is

$$V_{bi}^{k+1} = w * V_{bi}^k + c_1 * \text{rand}()_1 * (P_{best} - S_{bi}^k) + c_2 * \text{rand}()_2 * (G_{best} - S_{bi}^k)$$

$$S_{bi}^{k+1} = S_{bi}^k + V_{bi}^{k+1}$$

COST FACTORS	DESCRIPTION	RATING				
		VERY LOW	LOW	NOMINAL	HIGH	VERY HIGH
Product						
RELY	Required software reliability	0.75	0.88	1	1.15	1.4
DATA	Database size	-	0.94	1	1.08	1.16
CPLX	Product complexity	0.7	0.85	1	1.15	1.3
Computer						
TIME	Execution time constraint	-	-	1	1.11	1.3
STOR	Main storage constraint	-	-	1	1.06	1.21
VIRT	Virtual machine volatility	-	0.87	1	1.15	1.3
TURN	Computer turnaround time	-	0.87	1	1.07	1.15
Personnel						
ACAP	Analyst capability	1.46	1.19	1	0.86	0.71
AEXP	Application experience	1.29	1.13	1	0.91	0.82
PCAP	Programmer capability	1.42	1.17	1	0.86	0.7
VEXP	Virtual machine volatility	1.21	1.1	1	0.9	-
LEXP	Language experience	1.14	1.07	1	0.95	-
Project						
MODP	Modern programming practice	1.24	1.1	1	0.91	0.82
TOOL	Software tools	1.24	1.1	1	0.91	0.83
SCED	Development schedule	1.23	1.08	1	1.04	1.1

Table 1 : Effort Multipliers

ii. MODEL 2:

Instead of having resources estimates as a function of one variable, resources estimates can depend on many different factors, giving rise to multivariable models. Such models are useful as they take into account the subtle aspects of each project such as their complexity or other such factors which usually create a non linearity. The cost factors considered are shown below. The product of all the above cost factors is the Effort Adjustment Factor (EAF). A model of this category starts with an initial estimate determined by using the strategic single variable model equations and adjusting the estimates based on other variable which is methodology.

The equation is,

$$\text{Effort} = a * (\text{size})^b + c * (\text{ME}).$$

Where ME is the methodology used in the project. The parameters a, b, c are tuned by using above PSO methodology. The Update of velocity and positions of Parameter "a", "b" are shown in Model 1 and Parameter "c" is

$$V_{ci}^{k+1} = w * V_{ci}^k + c_1 * \text{rand}()_1 * (P_{best} - S_{ci}^k) + c_2 * \text{rand}()_2 * (G_{best} - S_{ci}^k)$$

$$S_{ci}^{k+1} = S_{ci}^k + V_{ci}^{k+1}$$

iii. MODEL 3

There are a lot of factors causing uncertainty and non linearity in the input parameters. In some projects the size is low while the methodology is high and the complexity is high, for other projects size is huge but the complexity is low. As per the above two models size and effort are directly proportional. But such a condition is not always satisfied giving rise to eccentric inputs. This can be accounted for by introducing a biasing factor (d). So the effort estimation equation is:

$$\text{Effort} = a * (\text{size})^b + c * (\text{ME}) + d$$

a,b,c,d parameters are tuned by using above PSO methodology.

The Update of velocity and positions of Parameter "a", "b", "c" are shown in Model 1,2 and Parameter "d" is

$$V_{di}^{k+1} = w * V_{di}^k + c_1 * \text{rand}()_1 * (P_{best} - S_{di}^k) + c_2 * \text{rand}()_2 * (G_{best} - S_{di}^k)$$

$$S_{di}^{k+1} = S_{di}^k + V_{di}^{k+1}$$

IV. MODEL ANALYSIS

a) Implementation

We have implemented the above methodology for tuning parameters a,b,c and d in "C" language. For the parameter 'a' the velocities and positions of the

particles are updated by applying the following equations:

$$V_{ai}^{k+1} = w * V_{ai}^k + c_1 * \text{rand}_1 * (Pbest_a - S_{ai}^k) + c_2 * \text{rand}_2 * (Gbest - S_{ai}^k)$$

$$S_{ai}^{k+1} = S_{ai}^k + V_{ai}^{k+1}, w=0.5, c_1=c_2=2.0.$$

And similarly for the parameters b,c and d the values are obtained for the first experiment and weight factor w changed during the iteration and C1 and C2 are constant. For the second experiment we changed the C1, C2 weighting factors by using equations 4 and 5.

b) Performance Measures

We consider three performance criterions:

1) Variance accounted – For(VAF)

$$\%VAF = \left[1 - \frac{\text{var}(ME-EE)}{\text{var}(ME)} \right] \times 100$$

2) Mean Absolute Relative Error

$$\%MARE = \text{mean} \left[\frac{\text{abs}(ME-EE)}{(ME)} \right] \times 100$$

3) Variance Absolute Relative Error (VARE)

$$\%VARE = \text{var} \left[\frac{\text{abs}(ME-EE)}{(ME)} \right] \times 100$$

Where ME represents Measured Effort, EE represents Estimated Effort.

V. MODEL EXPERIMENTATION

EXPERIMENT – 1:

For the study of these models we have taken data of 10 NASA [13]

Project No	Size In KDLOC	Methodology (ME)	Measured Effort
13	2.1	28	5
10	3.1	26	7
11	4.2	19	9
17	12.5	27	23.9
3	46.5	19	79
4	54.5	20	90.8
6	67.5	29	98.4
15	78.6	35	98.7
1	90.2	30	115.8
18	100.8	34	138.3

Table 2 : NASA software projects data

By running the “C” implementation of the above methodology we obtain the following parameters for the proposed models.

Model 1: a=2.646251 and b=0.857612 .
The range of a is [1, 10] and b is [-5,5] .

Model 2: a=2.771722, b=0.847952 and c= -0.007171.
The range of a is [1, 10], b is [-5,5] and c is [-1,1].

Model 3: a =3.131606, b=0.820175, c=0.045208 and d= -2.020790. The ranges are a[1,10],b[-5,5], c[-1,1] and d[1,20]. respectively.

EXPERIMENT -2:

The following are the results obtained by running the above PSO algorithm implemented in “C” with changing weighting factors on each iteration.

Model 1: a=2.646251 and b=0.857612.

The range of a is [1,10] and b is [-5,5]

Model 2: a=1.982430, b=0.917533 and c= 0.056668.

The range of a, b, c is [1, 10], [-5, 5] and [-1, 1] respectively.

Model 3: a= 2.529550, b= h0.867292, c= -0.020757 and d=0.767248.

The ranges of a,b,c,d is [1,10] , [-5,5] , [-1,1] and [0,20] respectively.

VI. RESULTS AND DISCUSSIONS

The following table shows estimated effort of our proposed model:

EXPERIMENT -1:

SIZE	MEASURED EFFORT	METHODOLOGY	ESTIMATED EFFORT OF OUR MODELS C1,C2 ARE CONSTANT DURING THE ITERATION (CASE-I)			ESTIMATED EFFORT OF OUR MODELS C1,C2 ARE CHANGED DURING THE ITERATION (CASE-II)		
			MODEL-I	MODEL-II	MODEL-III	MODEL-I	MODEL-II	MODEL-III
2.1	5	28	5.000002	4.998887	5.000007	5.000002	5.502722	5.000001
3.1	7	26	6.982786	7.047925	7.07543	6.982786	7.071439	6.975912
4.2	9	19	9.060186	9.222874	8.999259	9.060186	8.47359	9.154642
12.5	23.9	27	23.08629	23.40447	24.05549	23.08629	21.65101	22.82118
46.5	79	19	71.2293	71.75396	71.84614	71.2293	68.24138	71.03909
54.5	90.8	20	81.61792	82.10557	82.04368	81.61792	78.82941	81.44935
67.5	98.4	29	98.05368	98.39988	98.39998	98.05368	96.18965	97.79541
78.6	98.7	35	111.7296	111.9449	111.8526	111.7296	110.7037	111.4518
90.2	115.8	30	125.7302	125.8721	125.048	125.7302	125.0572	125.6834
100.8	138.3	34	138.3002	138.3003	137.2231	138.3002	138.523	138.2999

Table 3 : Estimated Efforts of Proposed Models

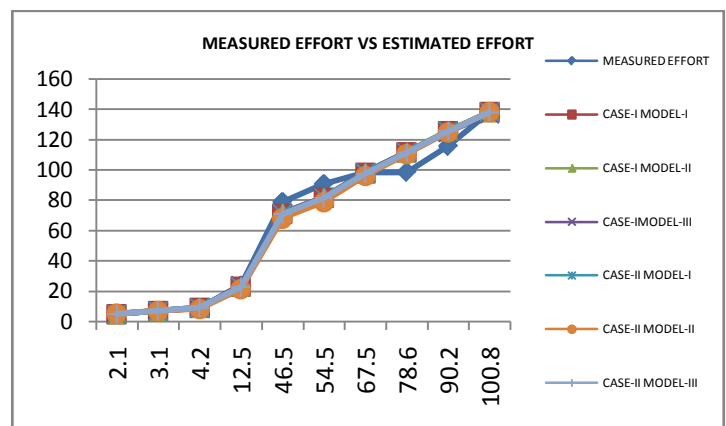


Fig 1 : Measured Effort Vs Estimated Efforts of Proposed Models

COMPARISON WITH OTHER MODELS

Refer Table 4 for the comparison with other models.

VII. PERFORMANCE ANALYSIS

Model	VAF (%)	Mean Absolute Relative Error (%)	Variance Absolute Relative Error (%)
Bailey –Basili Estimate	93.147	17.325	1.21
Alaa F. Sheta G.E. Model I Estimate	98.41	26.488	6.079
Alaa F. Sheta Model II Estimate	98.929	44.745	23.804
Harish model1	98.5	12.17	80.859
Harish model2	99.15	10.803	2.25
CASE-I MODEL -I	98.92	4.6397	0.271
CASE-I MODEL-II	98.92	4.6122	0.255
CASE-I MODEL-III	98.9	4.4373	0.282
CASE-II MODEL -I	98.92	4.6397	0.271
CASE-II MODEL-II	98.89	7.5	0.253
CASE-II MODEL-III	98.95	4.9	0.257

Table 5 : Performance Measures

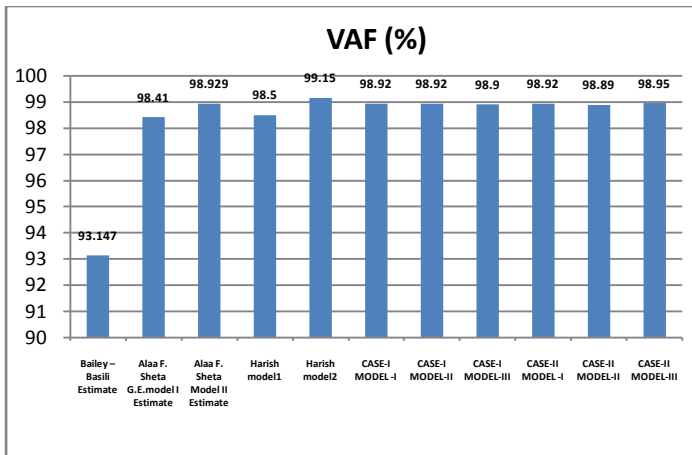


Fig 2 : Variance Accounted For %

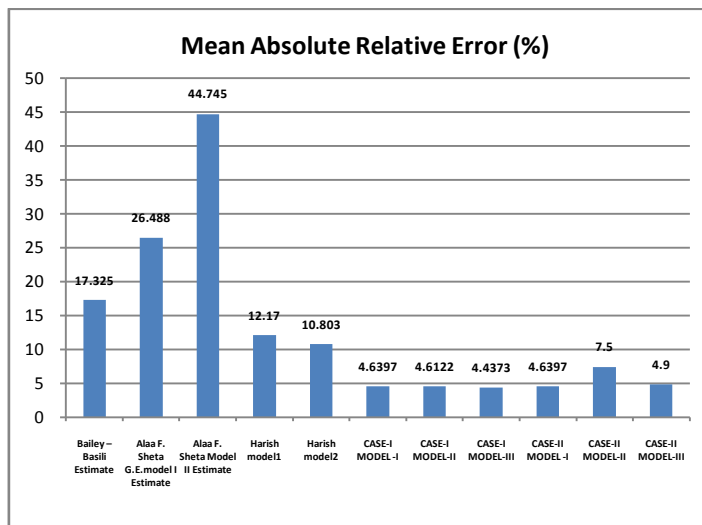


Fig 3 : Mean Absolute Relative Error (MARE)

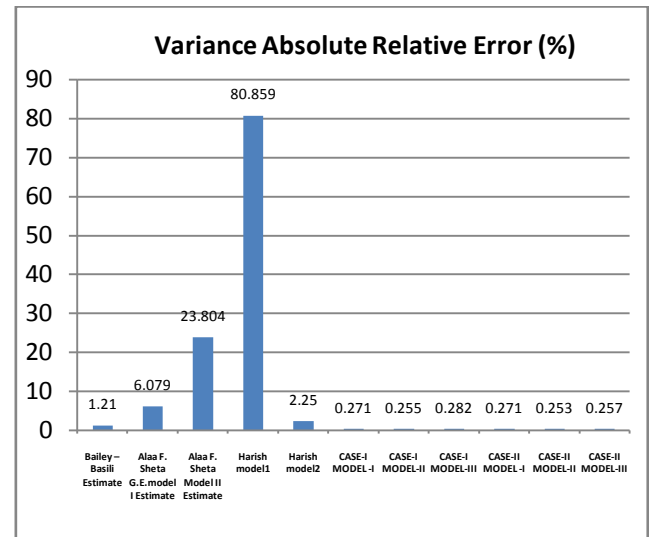


Fig 4 : Variance Absolute Relative Error %

VIII. CONCLUSION

Software cost estimation is based on a probabilistic model and hence it does not generate exact values. However if good historical data is provided and a systematic technique is employed we can generate better results. Accuracy of the model is measured in terms of its error rate and it is desirable to be as close to the actual values as possible. In this study we have proposed new models to estimate the software effort. In order to tune the parameters we use particle swarm optimization methodology algorithm. It is observed that PSO gives more accurate results when juxtaposed with its other counterparts. On testing the performance of the model in terms of the MARE, VARE and VAF the results were found to be futile. These techniques can be applied to other software effort models.

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Table 4 : Measured Efforts of Various Models

Measure d effort	Bailey-Basili Estimate	Alaa F. hetaG. E.Model Estimate	Alaa F. Sheta Model 2 Estimate	Harish model1	Harish model2	CASE-I MODEL-I	CASE-I MODEL-II	CASE-I MODEL-III	CASE-II MODEL-I	CASE-II MODEL-II	CASE-II MODEL-III
5	7.226	8.44	11.271	6.357	4.257	5.000002	4.998887	5.000007	5.000002	5.502722	5.000001
7	8.212	11.22	14.457	8.664	7.664	6.982786	7.047925	7.07543	6.982786	7.071439	6.975912
9	9.357	14.01	19.976	11.03	13.88	9.060186	9.222874	8.999259	9.060186	8.47359	9.154642
23.9	19.16	31.098	31.686	26.252	24.702	23.08629	23.40447	24.05549	23.08629	21.65101	22.82118
79	68.243	81.257	85.007	74.602	77.452	71.2293	71.75396	71.84614	71.2293	68.24138	71.03909
90.8	80.929	91.257	94.977	84.638	86.938	81.61792	82.10557	82.04368	81.61792	78.82941	81.44935
98.4	102.175	106.707	107.254	100.329	97.679	98.05368	98.39988	98.39998	98.05368	96.18965	97.79541
98.7	120.848	119.27	118.03	113.237	107.288	111.7296	111.9449	111.8526	111.7296	110.7037	111.4518
115.8	140.82	131.898	134.011	126.334	123.134	125.7302	125.8721	125.048	125.7302	125.0572	125.6834
138.3	159.434	143.0604	144.448	138.001	132.601	138.3002	138.3003	137.2231	138.3002	138.523	138.2999



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Studies on Colour Image Segmentation Method Based on Finite Left Truncated Bivariate Gaussian Mixture Model with K-Means

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Abstract - Colour Image segmentation is one of the prime requisites for computer vision and analysis. Much work has been reported in literature regarding colour image segmentation under HSI colour space and Gaussian mixture model (GMM). Since the Hue and Saturation values of the pixel in the image are non-negative. And may not be meso-kurtic, it is needed left truncate the Gaussian variate and is used to represent these two features of the colour image. The effect of truncation can not be ignored in developing the model based colour image segmentation. Hence in this paper a left truncated bivariate Gaussian mixture model is utilized to segment the colour image. The correlation between Hue and Saturation plays a predominant role in segmenting the colour images which is observed through experimental results. The expectation maximization (EM) algorithm is used for estimating model parameters. The number of image segments can be initialization of the model parameters are done with K-means algorithm. The performance of the proposed algorithm is studied by calculating the segmentation performance techniques like probabilistic rand index (PRI), global consistency error (GCE) and variation of information (VOI). The utility of the estimated joint probability density function of feature vector of the image is demonstrated through image retrievals. The image quality measures obtained for six images taken from Berkeley image dataset reveals that the proposed algorithm outperforms the existing algorithms in image segmentation and retrievals.

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Studies on Colour Image Segmentation Technique Based on Finite Left Truncated Bivariate Gaussian Mixture Model with K-Means

G.V.S. Rajkumar ^α, K.Srinivasa Rao ^Ω, P.Srinivasa Rao ^β

Abstract - Colour Image segmentation is one of the prime requisites for computer vision and analysis. Much work has been reported in literature regarding colour image segmentation under HSI colour space and Gaussian mixture model (GMM). Since the Hue and Saturation values of the pixel in the image are non-negative. And may not be meso-kurtic, it is needed left truncate the Gaussian variate and is used to represent these two features of the colour image. The effect of truncation can not be ignored in developing the model based colour image segmentation. Hence in this paper a left truncated bivariate Gaussian mixture model is utilized to segment the colour image. The correlation between Hue and Saturation plays a predominant role in segmenting the colour images which is observed through experimental results. The expectation maximization (EM) algorithm is used for estimating model parameters. The number of image segments can be initialization of the model parameters are done with K-means algorithm. The performance of the proposed algorithm is studied by calculating the segmentation performance techniques like probabilistic rand index (PRI), global consistency error (GCE) and variation of information (VOI). The utility of the estimated joint probability density function of feature vector of the image is demonstrated through image retrievals. The image quality measures obtained for six images taken from Berkeley image dataset reveals that the proposed algorithm outperforms the existing algorithms in image segmentation and retrievals.

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I. INTRODUCTION

Image segmentation is a process of extracting useful information from the images through features and dividing the whole image into various homogeneous groups in which, the pixels within the group are more homogeneous and are heterogeneous between the

groups. It is an important technology for image processing and understanding. The structural characteristics of objects and surfaces in an image can be determined by segmenting the image using image domain properties. One of the major advantages of image segmentation is denoising. Denoising is the process of removing unwanted noise from the image. Segmentation specifically attempts to separate structure from noise on a local scale. It is one of the most important steps in computer vision and analysis.

For the last three decades lot of work has been reported in literature regarding image segmentation methods (Lucchese L. et al (2001), Srinivas Y. and Srinivas Rao K. (2007), Majid Fakheri et al (2010), Siddhartha Bhattacharyya (2011)). The image segmentation methods can be divided into two categories depending upon the type of image. The images can be broadly categorized into two types namely, gray level images and colour images. A gray level image is usually characterized by pixel intensity (Farag A.A. et al (2004), Seshashayee M. et al (2011), Srinivas Yerramalle et al (2010)). But in colour images the colour is a perceptual phenomenon related to human response to different wavelengths in the visible electro-magnetic spectrum. In colour images the features that represent the image pixel are highly influenced by three feature descriptions namely, intensity, colour and texture. Among these features colour is the most important one in segmenting the colour images since intensity and texture features also be embedded in colour features. (Fesharaki and Hellestrand (1992), Kato Z. et al (2006), Kang Feng et al (2009), Kaikuo Xu et al (2011)). A better colour space than the RGB space in representing the colours of human perception is the HSI space, in which the colour information is represented by Hue and Saturation values. Thus the human perception of image can be characterized through a bivariate random variable consisting of Hue and Saturation which can be measured using generic structure of a colour appearance model (Sangwine et al (1998)).

Ferri and Vidal (1992), Lee E. et al (2010), Dipti P. and Mridula J. (2011) and others have reviewed colour image segmentation techniques. Among these

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model based image segmentation methods are more efficient than the edge based or threshold or region based methods (Lucchese L. et al (2001)). In model based image segmentation the whole image is divided into different image regions and each image region is characterized by a suitable probability distribution. For ascribing a probability model to the feature vector of the pixels in the image region, it is needed to study the statistical characteristics of the feature vector.

In image segmentation it is customary to consider that the whole image is characterized by a finite Gaussian mixture model. That is, the feature vector of each image region follows a Gaussian distribution (Haralick and Shapiro (1985), Shital Raut et.al (2009), Kato Z. et al (2006), Mantas Paulinas and Audrius Usinskas (2007), Rahman Farnoosh et al (2008), Sujaritha M. and Annadurai S. (2010)). The image segmentation methods based on Gaussian mixture model work well only when the feature vector of the pixels are having infinite range and the distribution of the feature vector is symmetric and meso-kurtic. But in many colour images the feature vector represented by Hue and Saturation are having finite values (say nonnegative) and may not be mesokurtic and symmetric. Hence, to have an accurate image segmentation of these sorts of colour images it is needed to develop and analyze image segmentation methods based on truncated bivariate mixture distributions.

Here, it is assumed that the feature vector in different image regions follows a left truncated bivariate Gaussian distribution and the feature vector of the whole image is characterized by a finite left truncated bivariate Gaussian mixture model. This assumption is made since the Hue and Saturation values of the pixel which represents the bivariate feature vector can take nonnegative values only. Hence, the range of the Hue and Saturation values are to be left truncated at zero. The effect of the truncated nature of Hue and Saturation cannot be ignored, since the leftover probability is significantly higher than zero in the left tail end of the distribution. This left truncated nature of the bivariate feature vector can approximate the pixels of the colour image more close to the reality.

In this method of segmentation, the number of image regions is obtained by *K*-means algorithm for which the initial value of the number of components is identified from the number of peaks in the image histogram. The model parameters are estimated by using Expectation Maximization (EM) algorithm. The EM-algorithm is one of the most preferred method of estimating the model parameters in mixture distributions (McLachlan G. and Krishnan T. (1997)). The EM-algorithm requires the updated equations of the model parameters which are derived for the left truncated bivariate Gaussian mixture model. The initialization of

the model parameters for carrying the EM-algorithm is done through feature vector of the pixel intensities of the image regions obtained through *K*-means clustering and moment method of estimation. An image segmentation algorithm with component likelihood maximization under Bayesian frame work is also developed and analyzed.

The efficiency of the developed image segmentation algorithm is studied by conducting experimentation with six images namely, OSTRICH, POT, TOWER, BEARS, DEER and BIRD which are taken randomly from Berkeley image data set. The segmentation performance measures namely, probabilistic rand index (PRI), global consistency error (GCE) and variation of information (VOI) are computed for the six images and presented. A comparative study of these measures with those obtained from the finite Gaussian mixture model reveals that this algorithm performs better than the Gaussian mixture model with *K*-means and having clear boundaries.

Using the estimated joint probability density functions of the feature vector of pixels of each image, the images are retrieved. The efficiency of the developed algorithm in image retrieval is also studied by computing the image quality metrics like maximum distance, image fidelity, mean square error, signal to noise ratio and image quality index and the results are presented. A comparative study of these quality measures with those obtained from the Gaussian mixture model with *K*-means revealed that this algorithm performs better.

II. FINITE LEFT TRUNCATED BIVARIATE GAUSSIAN MIXTURE MODEL

The effect of truncation in bivariate Gaussian distribution has been discussed by several researchers (Norman L.Johnson, Samuel Kotz and Balakrishnan (1994)). The probability density function of the left truncated Gaussian distribution (truncated at zero) is,

$$g(x, y; \theta) = \frac{f(x, y)}{\int_0^\infty \int_0^\infty f(x, y) dx dy}, 0 < x < \infty; 0 < y < \infty \quad (1)$$

Where, zero is the truncation point for both the Hue and saturation, $f(x, y)$ is the probability density function of the bivariate Normal distribution is

$$f(x, y) = \frac{1}{2\pi\sqrt{1-\rho^2}\sigma_1\sigma_2} \exp\left\{-\frac{1}{2(1-\rho^2)}\left[\left(\frac{x-\mu_1}{\sigma_1}\right)^2 - 2\rho\left(\frac{x-\mu_1}{\sigma_1}\right)\left(\frac{y-\mu_2}{\sigma_2}\right) + \left(\frac{y-\mu_2}{\sigma_2}\right)^2\right]\right\}$$

$$\begin{aligned} &-\infty < x < +\infty; -\infty < y < +\infty, \\ &\sigma_1 > 0; \sigma_2 > 0; -1 < \rho < 1, \\ &-\infty < \mu_1 < +\infty; -\infty < \mu_2 < +\infty \end{aligned} \quad (2)$$

The value of $\left[1 - \int_0^{\infty} \int_0^{\infty} f(x, y) dx dy\right]$ is significant

based on the values of the parameters. This distribution includes the skewed, asymmetric bivariate distributions as particular cases for limiting and specific values of the parameters. The various shapes of the frequency curves of the left truncated bivariate Gaussian distribution are shown in Figure1.

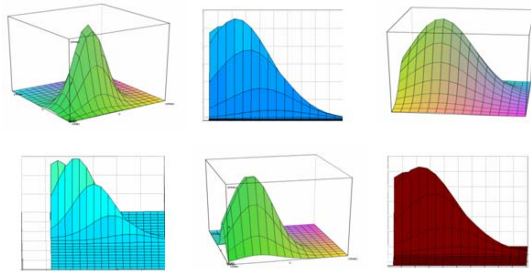


Fig1 : Shapes of left truncated bivariate Gaussian frequency surfaces

Following the heuristic arguments given by Bengt Muthen (1990), the mean value of 'X'(hue) is obtained as

$$E(X) = \mu_1 + \sigma_1 A \quad (3)$$

Where,

$$A = \phi\left(\frac{-\mu_1}{\sigma_1}\right) \left[1 - \Phi\left[\left(\frac{-\mu_1}{\sigma_1}\right) - \rho\left(\frac{-\mu_1}{\sigma_1}\right)c\right]\right] + \rho\phi\left(\frac{-\mu_2}{\sigma_2}\right) \left[1 - \Phi\left[\left(\frac{-\mu_1}{\sigma_1}\right) - \rho\left(\frac{-\mu_2}{\sigma_2}\right)c\right]\right]$$

and $c = (1 - \rho^2)^{-1/2}$, ϕ , Φ are the ordinate and area of standard Normal distribution. Similarly the mean value 'Y'(saturation) is

$$E(Y) = \mu_2 + \sigma_2 B \quad (4)$$

Where,

$$B = \phi\left(\frac{-\mu_2}{\sigma_2}\right) \left[1 - \Phi\left[\left(\frac{-\mu_2}{\sigma_2}\right) - \rho\left(\frac{-\mu_2}{\sigma_2}\right)c\right]\right] + \rho\phi\left(\frac{-\mu_1}{\sigma_1}\right) \left[1 - \Phi\left[\left(\frac{-\mu_2}{\sigma_2}\right) - \rho\left(\frac{-\mu_1}{\sigma_1}\right)c\right]\right]$$

and c is as given in equation (3)

The Variance of X is

$$\begin{aligned} V(X) &= \sigma_1^2 R - 2A \sigma_1 A + A^2 \\ &= \sigma_1^2 R - A^2 (2\sigma_1 - 1) \end{aligned} \quad (5)$$

Where,

$$\begin{aligned} R &= \left[\pi + \rho\left(\frac{-\mu_1}{\sigma_1}\right) \phi\left(\frac{-\mu_1}{\sigma_1}\right) \left[1 - \Phi\left[\left(\frac{-\mu_1}{\sigma_1}\right) - \rho\left(\frac{-\mu_1}{\sigma_1}\right)c\right]\right] + \rho^2\left(\frac{-\mu_2}{\sigma_2}\right) \phi\left(\frac{-\mu_2}{\sigma_2}\right) \left[1 - \Phi\left[\left(\frac{-\mu_1}{\sigma_1}\right) - \rho\left(\frac{-\mu_2}{\sigma_2}\right)c\right]\right] \right. \\ &\quad \left. - c^{-1} \rho \phi\left(\frac{-\mu_1}{\sigma_1}\right) \left[-\phi\left[\left(\frac{-\mu_1}{\sigma_1}\right) - \rho\left(\frac{-\mu_2}{\sigma_2}\right)c\right] \right] \right] \end{aligned}$$

and c and A is given in equation (3). The Variance of Y is

$$V(Y) = \sigma_2^2 T - 2B \sigma_2 B + B^2$$

$$= \sigma_2^2 T - B^2 (2\sigma_2 - 1) \quad (6)$$

Where,

$$\begin{aligned} T &= \left[\pi + \rho\left(\frac{-\mu_2}{\sigma_2}\right) \phi\left(\frac{-\mu_2}{\sigma_2}\right) \left[1 - \Phi\left[\left(\frac{-\mu_2}{\sigma_2}\right) - \rho\left(\frac{-\mu_2}{\sigma_2}\right)c\right]\right] + \rho^2\left(\frac{-\mu_1}{\sigma_1}\right) \phi\left(\frac{-\mu_1}{\sigma_1}\right) \left[1 - \Phi\left[\left(\frac{-\mu_2}{\sigma_2}\right) - \rho\left(\frac{-\mu_1}{\sigma_1}\right)c\right]\right] \right. \\ &\quad \left. - c^{-1} \rho \phi\left(\frac{-\mu_2}{\sigma_2}\right) \left[-\phi\left[\left(\frac{-\mu_1}{\sigma_1}\right) - \rho\left(\frac{-\mu_2}{\sigma_2}\right)c\right] \right] \right] \end{aligned}$$

c and B are as given in equations (3) and (4) respectively. The Covariance of (X, Y) is

$$\text{COV}(X, Y) = \sigma_1 \sigma_2 U - AB (\sigma_1 + \sigma_2 - 1) \quad (7)$$

where,

$$\begin{aligned} U &= \left[\rho\pi + \rho\left(\frac{-\mu_1}{\sigma_1}\right) \phi\left(\frac{-\mu_1}{\sigma_1}\right) \left[1 - \Phi\left[\left(\frac{-\mu_1}{\sigma_1}\right) - \rho\left(\frac{-\mu_1}{\sigma_1}\right)c\right]\right] - c^{-1} \phi\left(\frac{-\mu_1}{\sigma_1}\right) \left[-\phi\left[\left(\frac{-\mu_1}{\sigma_1}\right) - \rho\left(\frac{-\mu_1}{\sigma_1}\right)c\right] \right] \right. \\ &\quad \left. + \rho\left(\frac{-\mu_2}{\sigma_2}\right) \phi\left(\frac{-\mu_2}{\sigma_2}\right) \left[1 - \Phi\left[\left(\frac{-\mu_1}{\sigma_1}\right) - \rho\left(\frac{-\mu_2}{\sigma_2}\right)c\right]\right] \right] \end{aligned}$$

c , A and B are as given in equations (3) and (4) respectively.

Since the entire image is a collection of regions, which are characterized by left truncated bivariate normal distribution, it can be characterized through a K-Component finite left truncated bivariate Gaussian distribution and its probability density function is of the form

$$h(x, y) = \sum_{i=1}^K \alpha_i g_i(x_i, y_i; \theta) \quad (8)$$

Where, K is the number of regions, $\alpha_i > 0$ are weights such that $\sum_{i=1}^K \alpha_i = 1$ and $\theta = \{\mu_{1i}, \mu_{2i}, \sigma_{1i}^2, \sigma_{2i}^2, \rho_i\}$ is the set of parameters. $g_i(x_i, y_i / \theta_i)$ given in equation (1) represent the probability density function of the i^{th} image region. α_i is the probability of occurrence of the i^{th} component of the finite left truncated bivariate Gaussian mixture model (FLTBGMM) i.e., the probability that the feature belongs to the i^{th} image region. The mean vector representing the entire image is

$$E(W^T) = \begin{bmatrix} \sum_{i=1}^K \alpha_i E_i(X) \\ \sum_{i=1}^K \alpha_i E_i(Y) \end{bmatrix} \quad (9)$$

Where, $E(X_i)$ and $E(Y_i)$ are given in equations (3) and (4) for the i^{th} image region.

III. ESTIMATION OF THE MODEL PARAMETERS BY EM-ALGORITHM

To obtain the estimation of the model parameters, we utilized the EM-algorithm by maximizing the expected likelihood function for carrying out the EM-algorithm. The likelihood function of bivariate observations $(x_1, y_1), (x_2, y_2), (x_3, y_3), \dots, (x_N, y_N)$ drawn from an image with probability density function

$$L(\theta) = \prod_{s=1}^N h(x_s, y_s; \theta)$$

$$= \prod_{s=1}^N \left(\sum_{i=1}^K \alpha_i g_i(x_s, y_s; \theta) \right) \quad (10)$$

$$= \prod_{s=1}^N \left(\sum_{i=1}^K \alpha_i \frac{\exp \left\{ \frac{-1}{2(1-\rho_i^2)} \left[\left(\frac{x_s - \mu_{1i}}{\sigma_{1i}} \right)^2 - 2\rho_i \left(\frac{x_s - \mu_{1i}}{\sigma_{1i}} \right) \left(\frac{y_s - \mu_{2i}}{\sigma_{2i}} \right) + \left(\frac{y_s - \mu_{2i}}{\sigma_{2i}} \right)^2 \right] \right\}}{2\pi\sqrt{1-\rho_i^2}\sigma_{1i}\sigma_{2i} \int_0^{\frac{\sigma_{1i}}{\sigma_{2i}}} \int_0^{\frac{\sigma_{2i}}{\sigma_{1i}}} f_i(x, y; \theta) dx dy} \right)$$

This implies

$$\log L(\theta) = \sum_{s=1}^N \log \left(\sum_{i=1}^K \alpha_i g_i(x_s, y_s; \theta) \right) \quad (11)$$

The updated equations of EM-algorithm for estimating the model parameters are

$$\alpha_k^{(l+1)} = \frac{1}{N} \sum_{s=1}^N [t_k(x_s, y_s; \theta^{(l)})]$$

$$= \frac{1}{N} \sum_{s=1}^N \left(\frac{\alpha_k^{(l)} g_k(x_s, y_s; \theta^{(l)})}{\sum_{i=1}^K \alpha_i^{(l)} g_i(x_s, y_s; \theta^{(l)})} \right) \quad (12)$$

Where, $g_k(x_s, y_s; \theta^{(l)})$ is as given in equation (1).

For updating μ_{1k} we have,

$$\mu_{1k}^{(l+1)} \sum_{s=1}^N t_k(x_s, y_s; \theta^{(l)}) - \sum_{s=1}^N t_k(x_s, y_s; \theta^{(l)}) x_s + \sum_{s=1}^N t_k(x_s, y_s; \theta^{(l)}) \sigma_{1k}^{(l)} \left[\frac{\rho_k^{(l)} (y_s - \mu_{2k}^{(l)})}{\sigma_{2k}^{(l)}} + [A - \rho_k^{(l)} B] \right] = 0 \quad (13)$$

Similarly for updating μ_{2k} , we have,

$$\mu_{2k}^{(l+1)} \sum_{s=1}^N t_k(x_s, y_s; \theta^{(l)}) - \sum_{s=1}^N t_k(x_s, y_s; \theta^{(l)}) y_s + \sum_{s=1}^N t_k(x_s, y_s; \theta^{(l)}) \sigma_{2k}^{(l)} \left[\frac{\rho_k^{(l)} (x_s - \mu_{1k}^{(l)})}{\sigma_{1k}^{(l)}} + [B - \rho_k^{(l)} A] \right] = 0 \quad (14)$$

Where,

$$t_k(x_s, y_s; \theta^{(l)}) = \frac{\alpha_k^{(l)} g_k(x_s, y_s; \theta^{(l)})}{h(x_s, y_s; \theta^{(l)})} = \frac{\alpha_k^{(l)} g_k(x_s, y_s; \theta^{(l)})}{\sum_{i=1}^K \alpha_i^{(l)} g_i(x_s, y_s; \theta^{(l)})}$$

$$g_k(x_s, y_s; \theta^{(l)}) = \frac{\exp \left\{ \frac{-1}{2(1-\rho_k^2)} \left[\left(\frac{x_s - \mu_{1k}}{\sigma_{1k}} \right)^2 - 2\rho_k \left(\frac{x_s - \mu_{1k}}{\sigma_{1k}} \right) \left(\frac{y_s - \mu_{2k}}{\sigma_{2k}} \right) + \left(\frac{y_s - \mu_{2k}}{\sigma_{2k}} \right)^2 \right] \right\}}{2\pi\sigma_{1k}\sigma_{2k}\sqrt{1-\rho_k^2} \int_0^{\frac{\sigma_{1k}}{\sigma_{2k}}} \int_0^{\frac{\sigma_{2k}}{\sigma_{1k}}} f_k(x, y; \theta) dx dy}$$

A and B are as given in equations (3) and (4) respectively. The updated equations for σ_{1k}^2 at $(l+1)^{th}$ iteration is,

$$\sum_{s=1}^N t_k(x_s, y_s; \theta^{(l)}) \left[\left[\left(\frac{x_s - \mu_{1k}^{(l)}}{\sigma_{1k}^{(l+1)}} \right)^2 - \frac{\rho_k (x_s - \mu_{1k}^{(l)}) (y_s - \mu_{2k}^{(l)})}{\sigma_{1k}^{(l+1)} \sigma_{2k}^{(l)}} \right] - D + \rho_k^{(l)} E \right] = 0 \quad (15)$$

Where, $t_k(x_s, y_s; \theta^{(l)})$ is given in equation (14),

$$D = \pi \sigma_{1k} \sigma_{2k} + \sigma_{1k} \sigma_{2k} c^{-1} \rho_k \left[\phi \left(\frac{-\mu_{1k}}{\sigma_{1k}} \right) \left[\phi \left(\left(\frac{-\mu_{1k}}{\sigma_{1k}} \right) - \rho_k \left(\frac{-\mu_{1k}}{\sigma_{1k}} \right) c \right) \right] \right] \\ + \sigma_{2k} (\rho_k^2 + 1) (-\mu_{1k}) \phi \left(\frac{-\mu_{1k}}{\sigma_{1k}} \right) \left[1 - \Phi \left(\left(\frac{-\mu_{1k}}{\sigma_{1k}} \right) - \rho_k \left(\frac{-\mu_{1k}}{\sigma_{1k}} \right) c \right) \right],$$

And

$$E = \rho_k \pi \sigma_{1k} \sigma_{2k} + \sigma_{1k} \sigma_{2k} c^{-1} \left[-\phi \left(\frac{-\mu_{1k}}{\sigma_{1k}} \right) \left[-\phi \left(\left(\frac{-\mu_{2k}}{\sigma_{2k}} \right) - \rho_k \left(\frac{-\mu_{1k}}{\sigma_{1k}} \right) c \right) \right] \right] \\ + \rho_k \sigma_{1k} \left[\left(\frac{-\mu_{2k}}{\sigma_{2k}} \right) \phi \left(\frac{-\mu_{2k}}{\sigma_{2k}} \right) \left[1 - \Phi \left(\left(\frac{-\mu_{1k}}{\sigma_{1k}} \right) - \rho_k \left(\frac{-\mu_{2k}}{\sigma_{2k}} \right) c \right) \right] \right] \\ + \rho_k \sigma_{2k} \left[(-\mu_{1k}) \phi \left(\frac{-\mu_{1k}}{\sigma_{1k}} \right) \left[1 - \Phi \left(\left(\frac{-\mu_{2k}}{\sigma_{2k}} \right) - \rho_k \left(\frac{-\mu_{1k}}{\sigma_{1k}} \right) c \right) \right] \right]$$

The updated equations for σ_{2k}^2 at $(l+1)^{\text{th}}$ iteration is

$$\sum_{s=1}^N t_k(x_s, y_s; \theta^{(l)}) \left[\left[\left(\frac{y_s - \mu_{2k}^{(l)}}{\sigma_{2k}^{(l+1)}} \right)^2 - \frac{\rho_k (x_s - \mu_{1k}^{(l)}) (y_s - \mu_{2k}^{(l)})}{\sigma_{1k}^{(l)} \sigma_{2k}^{(l+1)}} \right] - G + \rho_k^{(l)} E \right] = 0 \quad (16)$$

where,

$$G = \pi \sigma_{1k} \sigma_{2k} + \sigma_{1k} \sigma_{2k} c^{-1} \rho_k \left[-\phi \left(\frac{-\mu_{2k}}{\sigma_{2k}} \right) \left[-\phi \left(\left(\frac{-\mu_{2k}}{\sigma_{2k}} \right) - \rho_k \left(\frac{-\mu_{2k}}{\sigma_{2k}} \right) c \right) \right] \right] \\ + \sigma_{1k} (\rho_k^2 + 1) (-\mu_{2k}) \phi \left(\frac{-\mu_{2k}}{\sigma_{2k}} \right) \left[1 - \Phi \left(\left(\frac{-\mu_{2k}}{\sigma_{2k}} \right) - \rho_k \left(\frac{-\mu_{2k}}{\sigma_{2k}} \right) c \right) \right],$$

$t_k(x_s, y_s; \theta^{(l)})$ and E are as given in equations (14) and (15) respectively and

Therefore the updated equation for estimating ρ_k is

$$\sum_{s=1}^N t_k(x_s, y_s; \theta^{(l)}) \left[-\frac{\rho_k}{(1-\rho_k^2)^2} \left[\left(\frac{x_s - \mu_{1k}}{\sigma_{1k}} \right)^2 + \left(\frac{y_s - \mu_{2k}}{\sigma_{2k}} \right)^2 \right] \right. \\ \left. - \frac{1+\rho_k^2}{(1-\rho_k^2)^2} \left[\left(\frac{x_s - \mu_{1k}}{\sigma_{1k}} \right) \left(\frac{y_s - \mu_{2k}}{\sigma_{2k}} \right) \right] + \frac{\rho_k (D+F)}{(1-\rho_k^2)^2} + \frac{(1+\rho_k^2)E}{(1-\rho_k^2)^2} \right] = 0 \quad (17)$$

Where, $t_k(x_s, y_s; \theta^{(l)})$, D , E and G are as given in equations (14), (15) and (16) respectively and

$$\begin{aligned}
F = & \pi \sigma_{1k} \sigma_{2k} - \sigma_{1k} \left[\left(\mu_{2k} \right) \phi \left(\frac{-\mu_{2k}}{\sigma_{2k}} \right) \left[1 - \Phi \left[\left(\frac{-\mu_{2k}}{\sigma_{2k}} \right) - \rho_k \left(\frac{-\mu_{2k}}{\sigma_{2k}} \right) \right] \right] \right] \\
& + \sigma_{1k} \rho_k^2 \left[\left(-\mu_{2k} \right) \phi \left(\frac{-\mu_{2k}}{\sigma_{2k}} \right) \left[1 - \Phi \left[\left(\frac{-\mu_{2k}}{\sigma_{2k}} \right) - \rho_k \left(\frac{-\mu_{2k}}{\sigma_{2k}} \right) \right] \right] \right] \\
& + \sigma_{1k} \sigma_{2k} \rho_k c^{-1} \left[-\phi \left(\frac{-\mu_{2k}}{\sigma_{2k}} \right) \left[-\phi \left[\left(\frac{-\mu_{2k}}{\sigma_{2k}} \right) - \rho_k \left(\frac{-\mu_{2k}}{\sigma_{2k}} \right) \right] c \right] \right]
\end{aligned}$$

Solving equations (12), (13), (14), (15), (16) and (17) iteratively we get the revised estimates $\alpha_k, \mu_{1k}, \mu_{2k}, \sigma_{1k}^2, \sigma_{2k}^2$ and ρ_k .

IV. INITIALIZATION OF PARAMETERS BY K-MEANS

The efficiency of the EM-algorithm in estimating the parameters is heavily dependent on the number of regions in the image. The number of mixture components taken for K -means algorithm is obtained, by plotting the histogram of the pixel intensities of the whole image. The mixing parameter α_k and the region parameters $\mu_{1k}, \mu_{2k}, \sigma_{1k}^2, \sigma_{2k}^2, \rho_k$ are unknown as prior. A commonly used method in initializing parameters is by drawing a random sample in the entire image data (McLachlan G. and Peel D. (2000)). This method performs well, if the sample size is large, and the computation time is heavily increased. When the sample size is small, some small regions may not be sampled. To overcome this problem, we use K -means algorithm to divide the whole image into various homogeneous regions. In K -means algorithm the centroids of the clusters are recomputed as soon as pixel joins the cluster.

The initial values of α_i can be taken as $\alpha_i = \frac{1}{K}$, where, K is the number of image regions obtained from the K -means algorithm (Rose H. Turi (2001)). K -means algorithm uses an iterative procedure that minimizes the sum of distances from each object to its cluster centroid, over all clusters. This procedure consists of the following steps.

- 1) Randomly choose K data points from the whole dataset as initial clusters. These data points represent initial cluster centroids.
- 2) Calculate Euclidean distance of each data point from each cluster centre and assign the data points to its nearest cluster centre.
- 3) Calculate new cluster centre so that squared error distance of each cluster should be minimum.
- 4) Repeat step 2 and 3 until clustering centers do not change.
- 5) Stop the process.

In the above algorithm, the cluster centers are only updated once all points have been allocated to their closed cluster centre. The advantage of K -means are that it is a very simple method, and it is based on intuition about the nature of a cluster, which is that the

within cluster error should be as small as possible. The disadvantage of this method is that the number of clusters must be supplied as a parameter, leading to the user having to decide what the best number of clusters for the image is (Rose H. Turi, (2001)). Success of K -means algorithm depends on the parameter K , number of clusters in image. After determining the final values of K (number of regions), we obtain the initial estimates of the parameters $\mu_{1k}, \mu_{2k}, \sigma_{1k}^2, \sigma_{2k}^2, \rho_k$ and α_k for each image region and with the method of moments given by Bengt Muthen (1990) for Truncated Bivariate Normal Distribution with initial parameters as $\alpha_i = 1/K$ for $i=1,2,\dots,K$

$\mu_{1k} = \bar{x}_{1k}$ is the k^{th} region sample mean of the Hue angle.

$\mu_{2k} = \bar{y}_{1k}$ is the k^{th} region sample mean of the Saturation.

$\sigma_{1k} = s_{1k}$ (Sample Standard Deviation of the k^{th} segment of Hue - angle)

$\sigma_{2k} = s_{2k}$ (Sample Standard Deviation of the k^{th} segment of - Saturation)

ρ_k is the correlation coefficient between Hue and Saturation of the k^{th} image region.

Substituting these values as the initial estimates, we obtain the refined estimates of the parameters by using the EM-algorithm.

V. SEGMENTATION ALGORITHM

After refining the parameters the prime step is image segmentation, by allocating the pixels to the segments. This operation is performed by segmentation algorithm. The image segmentation algorithm consists of four steps

Step 1) Plot the histogram of the whole image.

Step2) Obtain the initial estimates of the model parameters using K -Means algorithm and moment estimators as discussed in section IV.

Step3) Obtain the refined estimates of the model parameters $\mu_{1k}, \mu_{2k}, \sigma_{1k}^2, \sigma_{2k}^2, \rho_k$ and α_k for $i=1,2,\dots,K$ using the EM-algorithm with the updated equations given in section III.

Step4) Assign each pixel into the corresponding j^{th} region (segment) according to the maximum likelihood of the j^{th} component L_j .

That is

$$L_j = \max_{j \in k} \left\{ \frac{\exp \left\{ \frac{-1}{2(1-\rho_k^2)} \left[\left(\frac{x_s - \mu_{1k}}{\sigma_{1k}} \right)^2 - 2\rho_k \left(\frac{x_s - \mu_{1k}}{\sigma_{1k}} \right) \left(\frac{y_s - \mu_{2k}}{\sigma_{2k}} \right) + \left(\frac{y_s - \mu_{2k}}{\sigma_{2k}} \right)^2 \right] \right\}}{2\pi\sigma_{1k}\sigma_{2k}\sqrt{1-\rho_k^2} \int_0^\infty \int_0^\infty f_k(x, y, \theta) dx dy} \right\}$$

VI. EXPERIMENTAL RESULTS

To demonstrate the utility of the image segmentation algorithm developed in this paper, an experiment is conducted with six images taken from Berkeley images dataset (<http://www.eecs.berkeley.edu/Research/Projects/CS/vision/bsds/BSDS300/html/dataset/images.html>). The images namely, OSTRICH, POT, TOWER, BEARS, DEER and BIRD are considered for image segmentation. The feature vectors of the whole image is taken as input for image segmentation. The feature vector of the image are assumed to follow a mixture of left truncated bivariate Gaussian distribution. That is, the image contains K regions and the feature vector of the each image region follow a left truncated bivariate Gaussian distribution with different parameters. The number of segments in each of the six images considered for experimentation is determined by the histogram of pixel intensities. The histograms of the six images are shown in Figure 2.

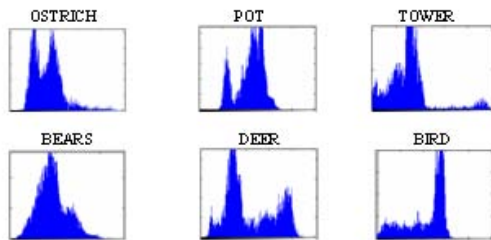


Figure 2 : Histograms Of The Images

The initial estimates of the number of regions K in each image are obtained and given in Table 1.

Table1 : Initial Estimates of K

IMAGE	OSTRICH	POT	TOWER	BEARS	DEER	BIRD
Estimate of K	2	3	4	2	4	3

After assigning these initial values of K to each image data set, the K -means algorithm is performed. The initial values of the model parameters μ_{1i} , μ_{2i} , σ_{1i}^2 , σ_{2i}^2 , ρ_i and α_i for $i=1,2,\dots,K$ for each image region of the images are computed by using the method given in section IV. Using these initial estimates, the refined estimates of the model parameters for each image region are obtained by using EM-algorithm given in section III. The computed

values of the initial estimates and the final estimates of the model parameters K , μ_{1i} , μ_{2i} , σ_{1i}^2 , σ_{2i}^2 , ρ_i and α_i for $i=1,2,\dots,K$ for each image are shown in Tables -2a, 2b, 2c, 2d, 2e and 2f for different images.

Table-2a Estimated Values of The Parameters For OSTRICH Image Number of image regions ($K=2$)				
Parameters	Estimation of Initial Parameters by K-means		Estimation of Final Parameters by EM-Algorithm	
	Regions(j)		Regions(j)	
	1	2	1	2
α_i	1/2	1/2	0.2627	0.7373
μ_{1i}	0.1781	0.1940	0.2054	0.2798
μ_{2i}	0.3321	0.7613	0.2505	0.7775
σ_{1i}^2	0.0016	0.0004	0.0287	0.0772
σ_{2i}^2	0.0126	0.0207	0.0747	0.0768
ρ_i	-0.4310	0.6996	-0.6163	0.2840

Table-2b Estimated Values of The Parameters For POT Image Number of image regions ($K=3$)						
Parameters	Estimation of Initial Parameters by K-means			Estimation of Final Parameters by EM-Algorithm		
	Regions(j)			Regions(j)		
	1	2	3	1	2	3
α_i	1/3	1/3	1/3	0.4888	0.2019	0.3093
μ_{1i}	0.5532	0.4946	0.1517	0.5505	0.5223	0.3089
μ_{2i}	0.2168	0.1125	0.1219	0.1958	0.0810	0.1106
σ_{1i}^2	0.0004	0.0027	0.0029	0.0248	0.0269	0.6202
σ_{2i}^2	0.0008	0.0018	0.0035	0.0358	0.0328	0.0469
ρ_i	0.1666	0.3570	-0.7230	-0.4604	0.9867	0.1373

Table-2c Estimated Values of The Parameters For TOWER Image Number of Image Regions ($K=4$)								
Parameters	Estimation of Initial Parameters by K-means				Estimation of Final Parameters by EM-Algorithm			
	Regions(j)				Regions(j)			
	1	2	3	4	1	2	3	4
α_i	1/4	1/4	1/4	1/4	0.1999	0.1523	0.1872	0.4606
μ_{1i}	0.1519	0.5699	0.1505	0.5738	0.4011	0.6276	0.1616	0.5743
μ_{2i}	0.1937	0.2789	0.6176	0.7724	0.2408	0.3519	0.5047	0.7721
σ_{1i}^2	0.0033	0.0073	0.0011	0.0006	0.1268	0.1045	0.0043	0.0213
σ_{2i}^2	0.0148	0.0148	0.0291	0.0059	0.1055	0.1696	0.6936	0.0539
ρ_i	0.1561	-0.0259	0.0386	-0.1086	0.9061	0.7744	-0.1902	0.1944

Table-2d Estimated Values of The Parameters For BEARS Image Number of Image Regions ($K=2$)				
Parameters	Estimation of Initial Parameters by K-means		Estimation of Final Parameters by EM-Algorithm	
	Regions(j)		Regions(j)	
	1	2	1	2
α_i	1/2	1/2	0.4531	0.5469
μ_{1i}	0.4787	0.2364	0.4867	0.4067
μ_{2i}	0.4532	0.2600	0.4171	0.3375
σ_{1i}^2	0.0027	0.0154	0.0560	0.0667
σ_{2i}^2	0.0129	0.0170	0.0786	0.1439
ρ_i	0.2044	-0.6378	0.1263	0.6274

Table-2e Estimated Values of The Parameters For DEER Image Number of Image Regions ($K=4$)								
Parameters	Estimation of Initial Parameters by K-means				Estimation of Final Parameters by EM-Algorithm			
	Regions(j)				Regions(j)			
	1	2	3	4	1	2	3	4
α_i	1/4	1/4	1/4	1/4	0.0703	0.4769	0.2775	0.1753
μ_{1i}	0.1299	0.1144	0.2324	0.3016	0.1388	0.1969	0.2349	0.3185
μ_{2i}	0.6989	0.4560	0.2354	0.1262	0.6847	0.4538	0.2045	0.1039
σ_{1i}^2	0.0004	0.0001	0.0015	0.0026	0.1153	0.1372	0.0366	0.0650
σ_{2i}^2	0.0104	0.0018	0.0019	0.0011	0.1435	0.0125	0.0525	0.1987
ρ_i	-0.1355	-0.0338	-0.0833	-0.0591	-0.0868	-0.0264	-0.2712	-0.2248

Table-2f						
Estimated Values of The Parameters For BIRD Image						
Number of Image Regions ($K=3$)						
Parameters	Estimation of Initial Parameters by K-means			Estimation of Final Parameters by EM-Algorithm		
	Regions(i)			Regions(i)		
	1	2	3	1	2	3
α_i	1/3	1/3	1/3	0.1029	0.6941	0.2031
μ_{1i}	0.1290	0.5948	0.1425	0.1677	0.6833	0.1946
μ_{2i}	0.6899	0.1143	0.2136	0.6031	0.0965	0.0722
σ_{1i}^2	0.0047	0.0029	0.0048	0.0242	0.2669	0.0237
σ_{2i}^2	0.0330	0.0015	0.0150	0.1369	0.0135	0.5790
ρ_i	0.0834	-0.0504	-0.1409	-0.4398	0.1672	-0.0101

Substituting the final estimates of the model parameters, the probability density function of the feature vector of each image are estimated. Using the estimated probability density functions and the image segmentation algorithm given in section V, the image segmentation is done for each of the six images under consideration. The original and segmented images are shown in Figure 3.

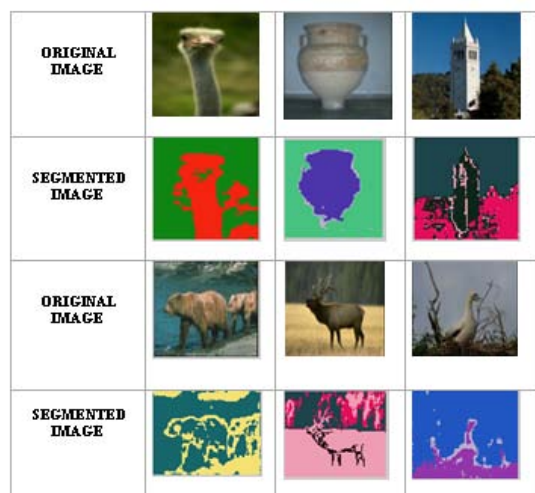


Figure 3 : Original and Segmented Images

VII. PERFORMANCE EVALUATION

After conducting the experiment with the image segmentation algorithm developed in this paper, its performance is studied. The comparison is based on three performance measures namely, Probabilistic Rand Index (PRI) given by Unnikrishnan R. and et.al (2007), the Variation of Information (VOI) given by Meila M. (2005), and Global Consistency error (GCE) given by Martin D. and et al (2001). The objective of the segmentation methods are based on regional similarity measures in relations to their local neighborhood.

The performance of developed algorithm using finite left truncated bivariate Gaussian mixture model (FLTBGMM) is studied by computing the segmentation performance measures namely, PRI, GCE and VOI for the six images under study. The computed values of the performance measures for the developed algorithm and

the earlier existing finite Gaussian mixture model (GMM) with K-means algorithm are presented in Table 3 for a comparative study.

Table 3 : Segmentation performance measures

IMAGE	METHOD	PERFORMANCE MEASURES		
		PRI	GCE	VOI
OSTRICH	GMM	0.9234	0.4317	2.2761
	FLTBGMM-K	0.9782	0.4037	1.7611
POT	GMM	0.9456	0.4281	2.5973
	FLTBGMM-K	0.9796	0.4131	1.9263
TOWER	GMM	0.9615	0.4469	3.7121
	FLTBGMM-K	0.9816	0.4302	2.8194
BEARS	GMM	0.9121	0.4418	3.2693
	FLTBGMM-K	0.9831	0.4337	2.6421
DEER	GMM	0.9774	0.4829	2.2863
	FLTBGMM-K	0.9847	0.4030	1.3947
BIRD	GMM	0.9673	0.4671	2.7197
	FLTBGMM-K	0.9705	0.4226	2.3244

From the above Table 3, It is observed that the PRI values of the proposed algorithm for the six images considered for experimentation are more than that of the values from the segmentation algorithm based on finite Gaussian mixture model with K-means. Similarly GCE and VOI values of the proposed algorithm are less than that of finite Gaussian mixture model. This reveals that the proposed algorithm outperforms the existing algorithm based on the finite Gaussian mixture model.

After developing the image segmentation method it is needed to verify the utility of segmentation in model building of the image for image retrieval. Using the estimated probability density function of the images under consideration the retrieved images are obtained and are shown in Figure 4.

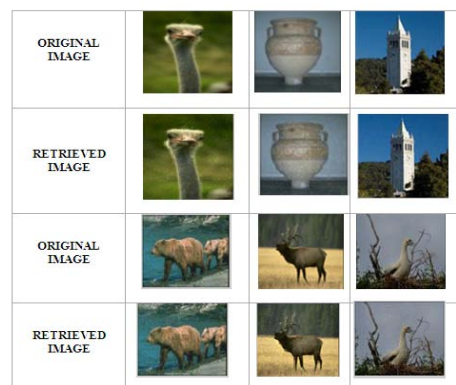


Figure 4 : Original and Retrieved Images

The Performance Evaluation of the retrieved image is done by Subjective Image Quality testing or by Objective Image Quality testing. The Objective Image Quality testing methods are often used since the numerical results of an objective measure are readily computed and allow a consistence comparison of different algorithms. There are several Image Quality

measures available for Performance Evaluation of the Image Segmentation method. An extensive survey of Quality Measures is given by Eskicioglu A.M. and Fisher P.S. (1995). For the Performance Evaluation of the developed Segmentation algorithm, we consider the Image Quality Measures namely (a) Maximum Distance, (b) Image Fidelity, (c) Mean Square Error, (d) Signal to Noise Ratio and (e) Image Quality Index are computed for all the Six images with respect to the developed method and earlier methods and presented in Table- 4.

Table 4 : Comparative study of Image Quality Metrics

IMAGE	Quality Metrics	GMM	FLTBGMM-K	Optimal Criteria
OSTRICH	Maximum Distance	0.5013	0.5067	Close to 1
	Image Fidelity	0.7910	0.8076	Close to 1
	Mean Square Error	0.0932	0.0793	Close to 0
	Signal to Noise Ratio	13.3781	13.9959	As big as possible
	Image Quality Index	0.8102	0.8492	Close to 1
POT	Maximum Distance	0.3290	0.3957	Close to 1
	Image Fidelity	0.6729	0.6786	Close to 1
	Mean Square Error	0.0738	0.0467	Close to 0
	Signal to Noise Ratio	11.7401	13.0240	As big as possible
	Image Quality Index	0.6075	0.6174	Close to 1
TOWER	Maximum Distance	0.8481	0.8757	Close to 1
	Image Fidelity	0.5217	0.3884	Close to 1
	Mean Square Error	0.2101	0.1792	Close to 0
	Signal to Noise Ratio	8.8724	8.8488	As big as possible
	Image Quality Index	0.6271	0.5173	Close to 1
BEARS	Maximum Distance	0.5387	0.8765	Close to 1
	Image Fidelity	0.4277	0.6586	Close to 1
	Mean Square Error	0.0872	0.0484	Close to 0
	Signal to Noise Ratio	9.1217	10.7550	As big as possible
	Image Quality Index	0.5951	0.5906	Close to 1
DEER	Maximum Distance	0.6217	0.6474	Close to 1
	Image Fidelity	0.3982	0.4470	Close to 1
	Mean Square Error	0.0828	0.0547	Close to 0
	Signal to Noise Ratio	10.0629	11.8918	As big as possible
	Image Quality Index	0.3763	0.3840	Close to 1
BIRD	Maximum Distance	0.8429	0.9129	Close to 1
	Image Fidelity	0.1920	0.2349	Close to 1
	Mean Square Error	0.2013	0.0900	Close to 0
	Signal to Noise Ratio	8.9231	9.3864	As big as possible
	Image Quality Index	0.3481	0.4160	Close to 1

From the Table 4, it is observed that all the image quality metrics for the six images are meeting the standard criteria. This implies that using the proposed algorithm the images are retrieved accurately. A comparative study of proposed algorithm with that of algorithm based on finite Gaussian mixture model (GMM) and Finite left truncated bivariate Gaussian mixture model with K-means reveals that the mean square error of the proposed model is less than that of the finite GMM and FLTBGMM. Based on all other quality metrics also it is observed that the performance of the proposed model in retrieving the images is better than the finite Gaussian mixture model.

VIII. CONCLUSION

In this paper we introduce a novel and new colour image segmentation method based on left truncated bivariate Gaussian mixture model. Here it is assumed that the colour image is characterized by HSI colour space, in which the Hue and Saturation values are non negative. they are characterized by left truncated Bivariate Gaussian mixture model. The left truncated bivariate Gaussian distribution includes the Bivariate Gaussian distribution is a limiting case when

the truncation points tends to infinite. It also includes several platy, meso, lefty and skewed distributions as particular cases for different values of the parameters. The model parameters are estimated by using EM-algorithm. The initialization and the number of image segments are determined through K-means algorithm and moment method of estimation. The segmentation algorithm is developed with component maximum likelihood. The experimentation with six colour images reveals that this algorithm outperforms the existing algorithms in both image segmentation and image retrievals. The image quality metrics also supported the utility of the proposed algorithm. It is possible to develop image segmentation algorithm with finite mixture of doubly truncated multivariate Gaussian distribution with more image features which require further investigations.

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Assessing the Quality of a Software Service at the Time of Project Development by Identifying its Reputation

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Abstract -At the time of integration of the software while developing a project the reputation and the quality of execution is tough to identify and which is very risky. As the software industry is introduced with a new type of service delivery model known as SaaS(Software as a service),the problem has increased a lot . Existing system be inclined to rely on rating from customer to experiences of past service which may create major issues in terms of subjectivity and rating unfairness. Few previous works have been considered quality and reputation for selection of services bur none have done service rating process through automation. We proposed an automated quality and reputation framework for rating and selecting a service. In this paper the management of risk has been formulated in context of development of the project using third party software service components and credibility is calculated by a measured reputation system.

Keywords : *Reputation, Service Vendor, Automation, SaaS, Service rating.*

GJCST-D Classification : *H.4.1*



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I. INTRODUCTION

The application of a quantifiable, systematic, disciplined approach to the development, process, and maintenance of software can be stated as software engineering. As software industry has huge competition it has shaped a strong motivation for developing solutions to support more responsive and more competitive businesses. Even with long-standing success of COTS (commercial off-the-shelf) software as a time-effective alternative to custom "in-house" developed solutions is still being compromised by the implicated cost of ownership, installation and maintenance time, and effort.

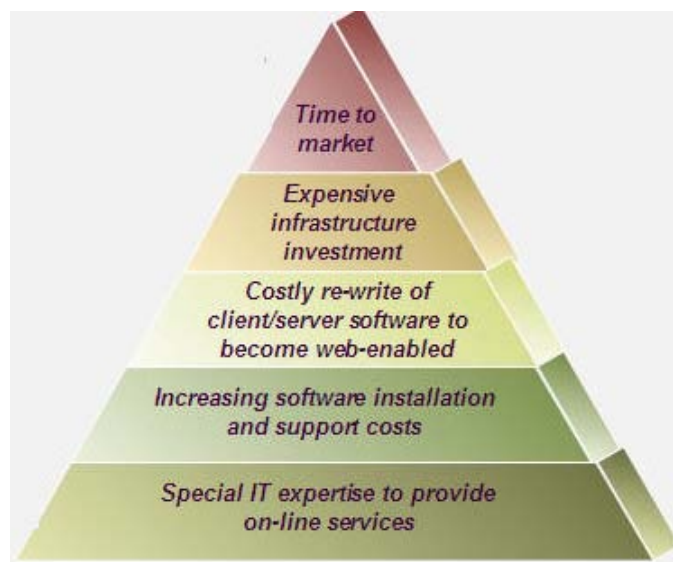


Fig 1 : Challenges for software vendors

That's the reason why software industry has started moving toward a new kind of software delivery model called SaaS(Software as a Service) and which made the things easy to install, maintenance-free, and money-spinning. In Software as a Service (SaaS) software delivery model the software is delivered on-demand and priced on-use, which made it to be widespread implementation of fast Internet access, combined with the widespread acceptance of SOA based solutions. SaaS has gained popularity by reducing the cost of tenure and alleviating the burden of software installation and maintenance. SaaS contributions has expanded dramatically as some of the enterprises have started to outsource their software infrastructure and development projects to SaaS vendors, and the competition has been increased even among vendors of traditional on premises software as in fig 1.

In the world of Software development using service delivery by SaaS model the quality of the software and software provider's credibility is tough and risky. So, the integration of external software in project development is challenging. In this paper risk management has been addressed in context of project development using external software service components. Reputation must be computed on the

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basis of fair and objective feedbacks. Most of the works that addressed until now are on evaluating the fairness of existing Feedbacks. Work in this paper focuses instead on the process of generating objective and fair feedbacks. Feedback can be individual since it is based on consumers' "personal" expectations and opinions. Consumers may have an obstructed view of a service reputation systems are prone to attacks by malicious consumers who may give false ratings and subvert service reputation. Consumers may have little incentive to leave a feedback. In this perspective a framework an automated quality and Reputation based framework for service rating and selection has been proposed.

The main objectives of this paper are:

- In order for a reputation mechanism to be fair and objective, it is essential to compute reputation on the basis of fair and objective feedbacks.
- The simulation results have demonstrated that the devised system has successfully met our primary objectives and can be an important component in a risk management strategy for software development with SaaS.
- A computational model is provided to objectively evaluate the delivered service based on the actual measurement of the conformance of the execution quality to the contracted SLA. A novel algorithm is also devised to automate the rating process based on the expectancy-disconfirmation theory from market science.

II. RELATED WORK

What is the main correlation stuck between "reputation" and "trust"? The major difference between reputation and trust can be illustrated by the following statements: (a) "Because of your good reputation I trust you" (b) "I trust you despite your bad reputation." Here the reputation is a collective measure of trustworthiness and is measured based on the referrals or ratings from other members in a community. According to A.josang and R.ismail, reputation is believed about a person's or thing's character or standing. Hence, trust for an individual is measured from the personal reputation and combination of received referrals, as in the Fig 2.

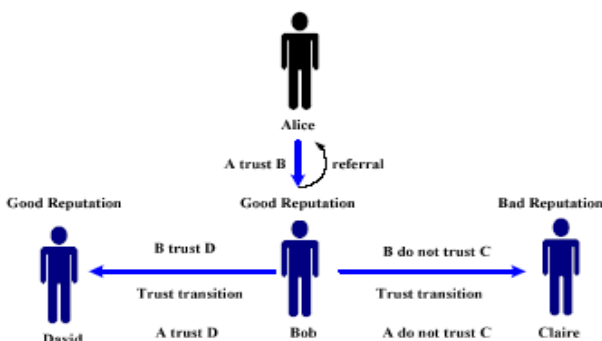


Fig 2 : A Transitive model for consumer reputation

In a centralized reputation management system, the synthetic rating of QoS of web services is aggregated by each rating in the community. To avoid the inapt evaluation by dishonest consumers, it need identify the reputable and disreputable members with their historical comments. Our idea is that consumer reputation is decided by the historical quality of comment, that is, more positive comments gain higher reputation, versa. In other words, lower reputations will worse his/her performance rating on QoS evaluation of web services. When consumers jointing the voting activity can raise their reputation by positive comments and avoid the negative comments. In this work, we proposed a centralized reputation measure for quantifying consumer reputation to properly select the service alternatives, as illustrated in Fig 3.

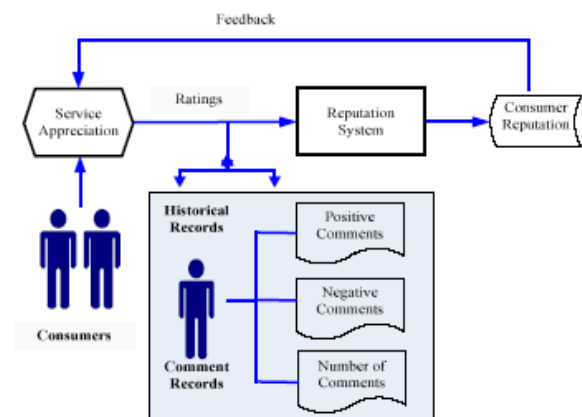


Fig 3 : Consumer Reputation Measure

III. SYSTEM ANALYSIS & DESCRIPTON

For selection of the service many previous works have measured the reputation and quality of the software, but the measurement has been done using some manual tools but none have considered the service rating process in the form of automation. WE introduce a framework for selecting and rating software to provide software service. The important point of the framework which is proposed is to automate both the rating and selection software services which is potentially increasing the objectivity of the service quality reports and concentrating on time-consumption and which finally reduces the risk associated utilization of external software services in development projects.

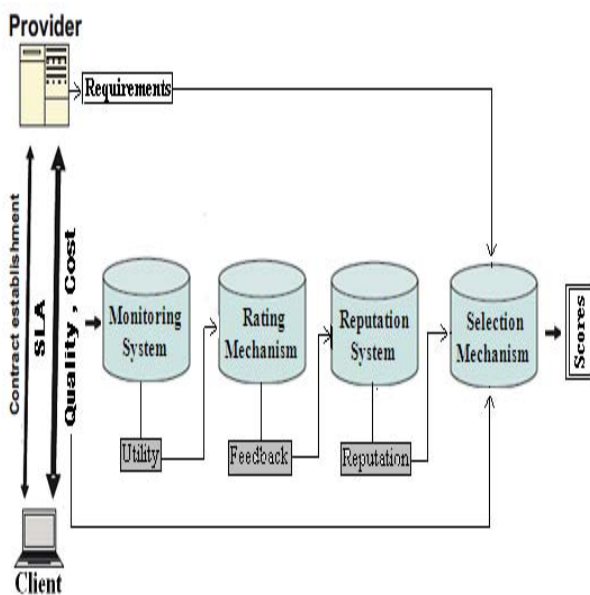


Fig 4 : System Architecture of the framework

While determining a service's suitability to a particular user's preferences in terms of quality and cost the service selection algorithm acts as a user-centric and reputation-aware service recommender. In order for a reputation mechanism to be fair and objective, it is essential to compute reputation on the basis of fair and objective feedbacks. Our work focuses instead on the process of generating objective and fair feedbacks, while most of the works that addressed this latter issue are on evaluating the fairness of existing feedbacks. Here concentrated the calculation of the reputation on works in the area of Service Level Agreement (SLA) monitoring where a computational model is provided to neutrally assess the delivered service based on the actual measurement of the execution quality to the contracted SLA.

In this paper we proposed a framework which has four major modules like Consumer, SLA (Service Level Agreement), Service Providers and Reputational System. Consumer can start the selection based on the trustworthiness features. Consumer selection information will be stored inside database like reputation table. SLA maintains some of the requirements about that particular service. These requirements can be co-incident with SLA requirements and for those services only the service certificate will be approved and that Certificate can be used as Trustworthiness certificate. The services which are provided by SLA can also be present in the service providers itself. User can be satisfied with certified services or trustworthy services. All the user behaviors features can be located inside the trustworthy services. To start the selection at the consumer side we should place the all the features inside that particular service. Reputation can be defined based on the frequent item selection procedure to define the utility measurement identification. Based on

utility measure the feedback about that particular service will be defined. The proposed reputational framework is as shown in fig 4.

And the functional requirements of the proposed framework will be as Enter Consumer Details, Update Consumer Required Services, and Enter Service Provider Details, Service updated to SLA, Retrieve Services, Select Service, Utility Measure of Service, Rating Function, Retrieve Feedback, Consumer Preference Updated, Select service and Calculate Score. An empirical study of the risk factors related to the development using external software (COTS-like) components along with associated risk reduction activities has been reported in. It showed that risk reduction at software selection time is negatively correlated with occurrences of most project development-related risks. In fact, selection must be driven by quality constraints, with selection time evaluation of component quality and choice of appropriate service providers all essential to successful integration. However, in practice, the evaluation of service quality cannot be performed until the service is acquired. Consequently, quality evaluation is typically limited to the evaluation of quality offers by comparing the quality level that providers promise to the quality requirements. Compliance cannot be guaranteed at selection time, so it is essential to choose a provider that is trusted to respect its commitments.

IV. SYSTEM DESIGN & IMPLEMENTATION

a) Designing of the Framework

In the system design of a system, a number of classes are identified and grouped together in a class diagram which helps to determine the static relations between those objects. With detailed modeling, the classes of the conceptual design are often split in a number of subclasses.

In order to further describe the behavior of systems, these class diagrams can be complemented by state diagram or UML state machine. Where in our framework we have four classes Service Provider, SLA, Consumer and reputational System as in Fig 5. Here Service provider will check for the service name, cost of service, utility of service and value of time. In SLA class it will monitor the service and measures the utility and produces the rating function and identifies the feedback. In Consumer class consumer will select the category, finds utility, cost and selects the service. In Reputation System it identifies the user preferences then select the service and maintains the time.

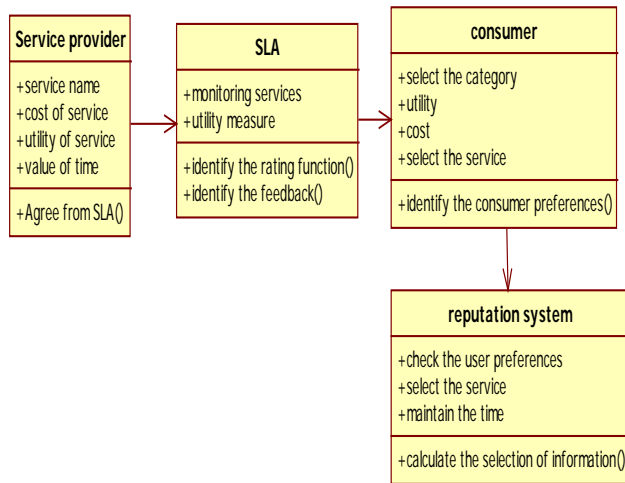


Fig 5 : Inter-operational Class diagram for framework

A use case diagram in the Unified Modeling Language (UML) is a type of behavioral diagram defined by and created from a Use-case analysis. Its purpose is to present a graphical overview of the functionality provided by a system in terms of actors, their goals (represented as use cases), and any dependencies between those use cases. The main purpose of a use case diagram is to show what system functions are performed for which actor. Roles of the actors in the system can be depicted. Use Case diagrams are formally included in two modeling languages defined by the OMG: the Unified Modeling Language (UML) and the Systems Modeling Language (SysML). Major two components for a use case diagram are as follows:

• Use cases

A use case describes a sequence of actions that provide something of measurable value to an actor and is drawn as a horizontal ellipse. Where in our Framework we have set of use-cases like Service Name, Category, Cost of Service, Utility, Value of Time, Monitoring the services, Service Rating, User Preferences, Reputation, Feedback as in Fig 6.

• Actors

An actor is a person, organization, or external system that plays a role in one or more interactions with the system. Where in our framework we have Service Provider and Consumer as actors.

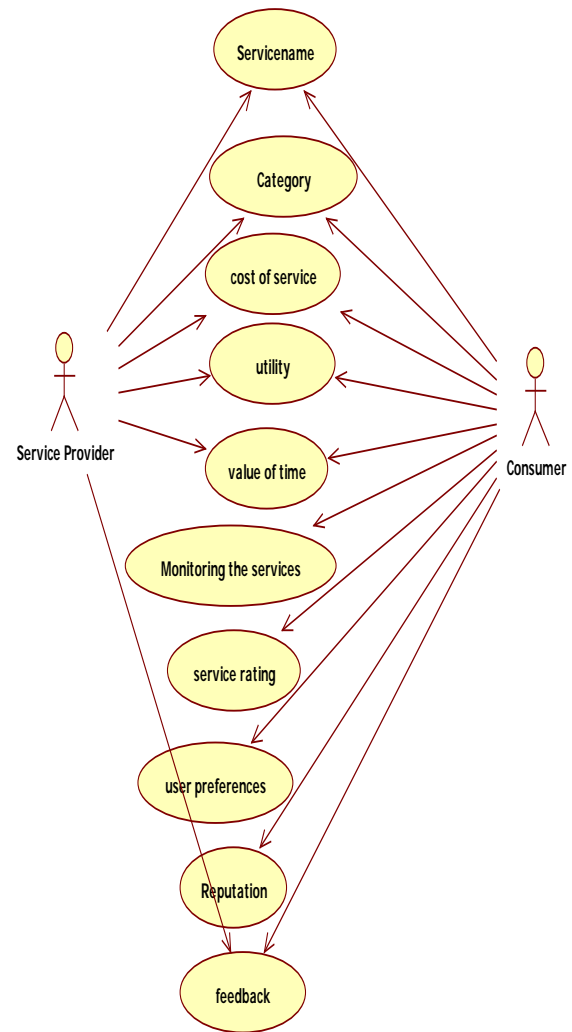


Fig 6 : Inter-operation Use Case diagram for the framework.

b) Implementation of the Framework

The Algorithm representation of the framework is detailed below:

Step 1: Start

Step 2: Enter the Customer Details.

Step 3: Update the Consumer Required Services.

Step 4: Service will be updated into SLA.

Step 5: Consumer will retrieve the services.

Step 6: Consumer choose to select the service.

Step 7: Measurement of Utility for the service.

Step 8: Calculating the Rating Function.

Step9: Retrieving the Feedback for the Service selected.

Step 10: Consumer Preference will be updated.

Step 11: Calculate the Score depending the selected service.

Step 12: Select the service depending on the score achieved.

Step 13: Stop.

V. RESULTS

The following are the screen shots of the system.

Consumer

Required Services

Select Category:

Utility:

Cost:

Consumer Preferences

Select Service:

Preferences...

Fig 7 : Selection of required service by consumer

Service Provider

Service Provider:

Service Name:

Category:

Cost of Service:

Utility:

Value of Time(Year):

Fig 8 : Updating the type of service provided by service provider

Service Level Agreement

Monitoring System

Service Name	Category	Cost	Utility	Value
Mukesh	Programming	3000	12	2
Google Web Service	Internet	5000	21	1

Select Service:

Feedback :

Service_Utility=5000
Service_Cost=21
Consumer Required
Service_Utility=6000
Service_Cost=23

Measured_Service_Utility_Cost=70%
Feedback Result=70%

Fig 9 : The SLA between Consumer and Service Provider

Reputation System

Select Service:

Service Name	Category	Cost	Utility	Value
Google Web Se...	Internet	5000	21	1

Score

Rating Score for [Google_Web_Service]

Score: 80.0

Fig 10 : Calculating the Score of the service using Reputation System

VI. CONCLUSION

In this paper we addressed the risk of incorporating a third party software for development of a project. To overcome the risk factor, proposed an outstanding framework Identifying the Reputation and Assessing the Quality of a Software Service at the Time of Project Development. We highlighted the framework by adding enhanced features like consumer, SLA, Service Provider and Reputation System which made as added additional advantage in rating and selecting the software to be used for integration. The proposed framework have accomplished in confining the service behaviors and translating them into probable customers choice.

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A Robust Approach to Find the Control Points for Wide Variety of 3rd Order Bézier Curves

By Alok Kumar Chowdhury, Prithwi Raj Chakraborty, Md. Ibrahim Khan, Sujan Chowdhury

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Keywords : Bézier curve, curve fitting, segmentation of curve, learning algorithms.

GJCST-B Classification : I.3.5



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Keywords : Bézier curve, curve fitting, segmentation of curve, learning algorithms.

1. INTRODUCTION

The development of Computer Graphics has made computers easier to interact with, and better for understanding and interpreting many types of data which has put a profound impact on many types of media and have revolutionized animation, movies and the video game industry. Generally, Computer Graphics are the representation and manipulation of image data by a computer with help from specialized software and hardware [1].

In Computer Graphics and related fields, a frequently used parametric curve is Bézier curve. Bézier curves are used to model smooth curves that can be scaled indefinitely. As the curve is completely contained in the convex hull of its control points, the points can be graphically displayed and used to manipulate the curve intuitively. When more complex shapes are needed, Bézier curves are patched together. In Animation applications, such as Adobe Flash and Synfig, Bézier curves are used to outline, for example, movement. Besides, True-type Fonts use Bézier curves. Bézier curve is also a very powerful tool for shape approximation and

shape comparison. In identifying of actors drawn in ukiyoe pictures, Bézier curves are widely used [2]. They can be used in face recognition and facial emotion detection [3].

A Bézier curve is defined by its order (Linear, Quadratic, Cubic, etc.) and a set of control points. An nth order Bézier curve has n+1 control points (P_1 to P_{n+1}). The first and last control points are always the end points of the curve. These two end points are can be called initial and terminating point of the curve respectively. The intermediate control points generally do not lie on the curve; they define the shape and direction of the curve. A 1st order (Linear) Bézier curve is simply a straight line between those two given points P_1 and P_2 . A 2nd order (Quadratic) Bézier curve is the path traced by the function $B(t)$, given points P_1 , P_2 , and P_3 . Four points P_1 , P_2 , P_3 and P_4 in the plane or in three-dimensional space define a 3rd order (Cubic) Bézier curve. The 3rd order curve starts at P_1 going toward P_2 and arrives at P_4 coming from the direction of P_3 . P_1 and P_4 are end points [4-5]. Here P_1 and P_4 are end points as well as P_2 and P_3 are intermediate control points. A 3rd order Bézier curve given by the following equation is shown in fig. 1.

$$B(t) = (1-t)^3 P_1 + 3t(1-t)^2 P_2 + 3t^2(1-t) P_3 + t^3 P_4 \quad (1)$$

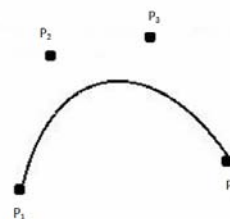


Fig.1 : A sample Bézier Curve

Our discussions will be limited to 3rd order Bézier curve. Though there are several algorithms and approaches to find out control point of 3rd order Bézier curve, these approaches have limitations like incorrect result, requirement of compatible boundary for control points etc. In our previous work [6], we tried to overcome these limitations and it finds control points of 3rd order Bézier curve in an efficient way. But it is still unable to find out control point of 3rd order Bézier curves of some certain shapes. In this paper we are going to propose a modified approach of our previous work to erase the existing limitations. Our new approach is capable of finding out the control points of a large variety of 3rd order Bézier curve shapes efficiently and accurately with minimum error and less iterations.

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II. BACKGROUND

There has been several works in the field of recovering control points of Bézier curves. All the familiar algorithms and characteristics are associated with Bézier curves are generation [7], approximation [8], interpolation [9], subdivision [10-12], degree elevation [13-16], blossoming [17], implication [18] etc. X. Ye represented an approach for directly generating Bézier points of curves and surfaces explicitly from the given compatible arbitrary order boundary information of Hermite curves, Coons-Hermite Cartesian sum patches and Coons-Boolean sum patches [19]. But for Hermite curves, this approach requires the end positions, tangents and higher order derivatives at the end points. Moreover it requires the corner points and the compatible arbitrary order partial derivatives at these points for Coons-Hermite Cartesian sum patches.

In determining inner control points of 3rd order rational Bézier curve, the interpolation method developed by J. Chou and L. A. Piegl works good for special type of 3rd order Bézier curves, not all kind of Bézier curves [20]. Besides, this approach relies on the convex hull and on the variation diminishing properties of Bézier curves. And the most important drawback is that if the curve segment is highly curved in one region and relatively flat in another, the approximation of this method is not good.

The previous approach, proposed by us, contains none of these problems. It can successfully find out control points of 3rd order Bézier curve with minimum iteration and error. Fig. 2(b) and 2(c) are showing the way how the previous approach successfully finds out the control points for the curve of a particular shape indicated in fig. 2(a). The details are described at the previous work [6].

However, for some curves of shapes like fig. 3(a) and 4(a), the previous approach is unable to find out the accurate control points. It is because the segmentation process of the previous approach, which uses vertical segmentation, is different from the current one (described at Methodology section).

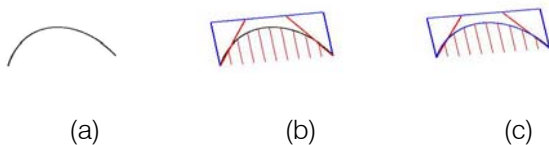


Fig.2 : (a) A typical Bézier curve, (b) Segmentation, (c) Successful-recovery of the curve (previous approach)

Fig. 3 and 4 are showing limitation of the previous approach in finding out the control points for some special shaped curves. The previous approach cannot rescue the whole area for these special shaped

curves, indicated in fig. 3(c) and 4(c). Thus the control points found are also not correct for these curves. The curves drawn in blue color are the recovered curves.

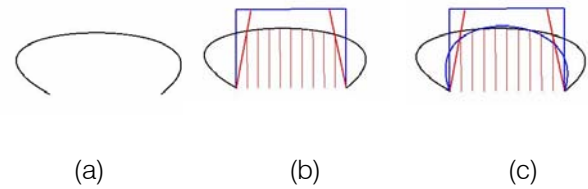


Fig.3 : (a) A typical Bézier curve, (b) Segmentation, (c) Failed-recovery of the curve (previous approach)

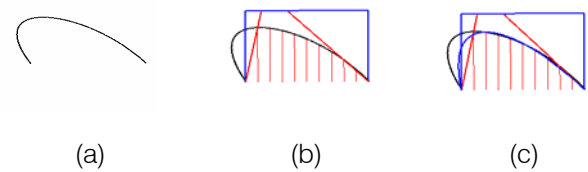


Fig.4 : (a) A typical Bézier curve, (b) Segmentation, (c) Failed-recovery of the curve (previous approach)

But the new approach, with modified segmentation method, is able to find out the actual control points very successfully with minimum iteration and error for all the 3rd order Bézier curves of shapes indicated in fig. 2(a), 3(a) and 4(a). It can rescue the whole area of special shaped curves which is described later in Simulation Results section.

III. METHODOLOGY

As it is mentioned before, a 3rd order Bézier curve contains four control points. The objective is to find out them. In order to do this, the new proposing approach is divided into two stages: modified first stage and second stage.

a) Modified first stage

The new proposing approach is actually different from our previous one because of this stage. This stage is described as following:

1) Finding out of the end points: The end points will be lying on the Bézier curve. End points must be those two which have minimum y values. Now it is to be decided which one of these end points is initial point and which one is terminating point. Among these two end points the point which is nearest from Y axis (i.e. x value is minimum) is our initial point, $P_1(x_1, y_1)$. So, the remaining one is for sure the terminating point, $P_4(x_4, y_4)$. These two end points are shown in fig. 5.

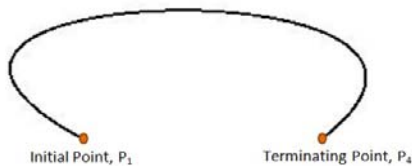


Fig.5 : End points of a Bézier curve

2) Finding out of the height: After finding out the initial and terminating point, the height of the Bézier curve, h is to be found by the following steps:

- A straight line between initial point $P_1(x_1, y_1)$ and terminating point $P_4(x_4, y_4)$ is drawn using the following equation:

$$(y - y_1)(x_4 - x_1) = (x - x_1)(y_4 - y_1) \quad (2)$$

This line is called base line, $P_1 P_4$ shown in fig. 6(a).

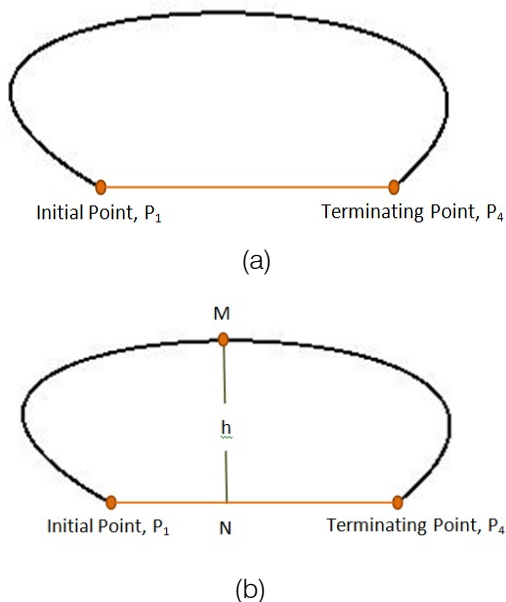


Fig.6 : (a) Base line, (b) Height and peak intersection point of a Bézier curve

- Now the height, h and the peak intersection point, M of the Bézier curve and the parallel line drawn at distance h from the base line are found by a recursive algorithm which makes the new approach faster than the previous one. This algorithm draws some parallel lines at some assumed distances, d from the base line. If a parallel line intersects the curve at two points, this line is stored and the next parallel line is drawn above the currently stored parallel line at a distance d . If the drawn parallel line does not intersect the curve then that line is not stored and the next parallel line is drawn above at a distance $d/2$ from the previously stored parallel line. This algorithm contains three methods – `draw_parallel(_line, _d)`, `intersections(_curve, _line)`, and `distance(_line1, _line2)`. The `draw_parallel(_line, _d)` method takes a line equation and a distance as inputs and returns the parallel line drawn at a distance $_d$ from the line $_line$. If the equation $ax + by + k = 0$ be the

value of input variable $_line$ and d is the value of input variable $_d$, then `draw_parallel(_line, _d)` returns a parallel line using the following equation:

$$ax + by + [d(\sqrt{a^2 + b^2}) + k] = 0 \quad (3)$$

The next method `intersections(_curve, _line)` takes a curve equation and a line equation and returns the number that represents at how many points the line intersects the curve. The peak most segment of the Bézier curve may be appeared as a tiny line rather than a curve. Therefore, the parallel line may intersect the curve at several consecutive intersection points. In that case, the middle one from the consecutive points is taken as the desired single intersection point and the number of intersections is considered as one. The last method `distance(_line1, _line2)` returns the distance between two lines. Now we should look at the recursive algorithm.

- Set `temp_base_line` = `base_line`.
- Repeat step c to g.
- Set `parallel_line` = `draw_parallel(temp_base_line, d)`.
- Set `no_of_intersections` = `intersections (curve, parallel_line)`.
- If `no_of_intersections` = 1, then: Set M = intersection point of curve and `parallel_line`. Set h = `distance(base_line, parallel_line)` and Return h and M .
- Else If `no_of_intersections` = 0, then: Set $d = d/2$.
- Else If `no_of_intersections` = 2, then: Set `temp_base_line` = `parallel_line`.
- Return h and M

Thus fig. 6(b) shows the height, h and the peak intersection point, M found by the above algorithm.

3) Segmentation of the Bézier curve: The segmentation process of the new proposing approach follows horizontal approach rather than vertical approach followed by the previous approach. In segmentation, the following steps are followed:

- A normal is drawn on the base line from the point M . This normal obviously intersects the base line or extended base line and the intersection point between normal and the base line can be easily found. Let this point is N . Now the normal (MN) is divided into n equidistant points, (x_1'', y_1'') , (x_2'', y_2'') ... (x_n'', y_n'') .

The normal and these n equidistant points are shown by fig. 7(a). If $M(x_1''', y_1''')$ and $N(x_2''', y_2''')$ are the two terminal points of the normal then the normal (MN) is divided using the following equation:

$$x_i'' = (m_1 x_1''' + m_2 x_2''') / (m_1 + m_2) \quad (4)$$

$$y_i'' = (m_1 y_1''' + m_2 y_2''') / (m_1 + m_2) \quad (5)$$

The greater value of n is taken, the more accurate segmentation and thus more accurate result is found but in that case computational cost gets higher.

- Now n parallel lines PL_1, PL_2, \dots, PL_n of the base line are drawn which are going through n equidistant points, $(x_1'', y_1''), (x_2'', y_2''), \dots, (x_n'', y_n'')$ on the normal (MN). Each of these n parallel lines of the base line intersects the Bézier curve into two points on both sides of the normal (MN). So, there will be $2n$ intersection points (except initial point, terminating point and M) lying on the Bézier curve for n parallel lines. Fig. 7(b) shows these $2n$ points, which are $(x_1', y_1'), (x_2', y_2') \dots (x_{2n}', y_{2n}')$. These $2n$ points and M are the desired segmentation points.

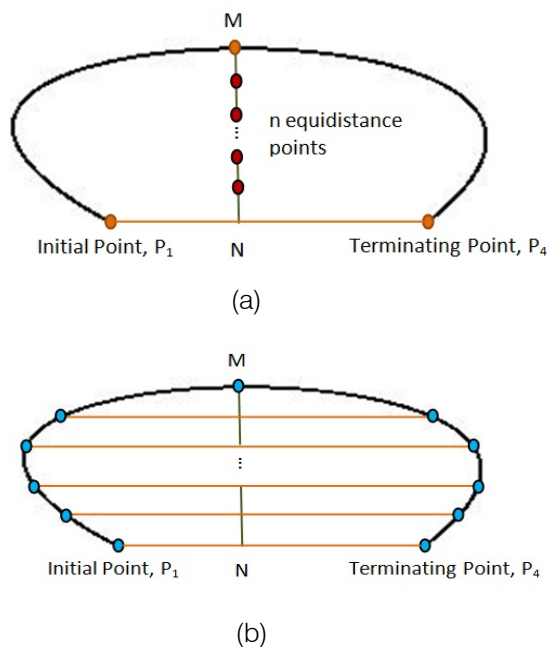


Fig. 7: (a) Normal, equidistant points, (b) Segmentation points of a Bézier curve

4) Assumption of the intermediate control points: This portion of this proposal is same as our previous work. A 3rd order Bézier curve contains two intermediate control points (P_2 and P_3). In this portion these two intermediate control points are assumed. The steps are:

- From the base line, several parallel lines at distance H are drawn. H indicates assumed height with different values - $(3/4)h$, h , $(4/3)h$ and $1.5h$. Here h is the height of the Bézier curve. Fig. 8 is showing such a parallel line drawn at $H = 1.5h$ distance.
- After each parallel line is drawn, two tangents from initial and terminating points are drawn shown in fig. 8. The tangents must intersect the parallel line at two points. These two points are the assumed intermediate control points. Let one point is P_2' , the assumed 2nd

control point and other one is P_3' , the assumed 3rd control point.

- So, for four values of H (four different distances) - $(3/4)h$, h , $(4/3)h$ and $1.5h$, four pair of assumed intermediate control points (P_2', P_3') are found. Such a pair of points found at distance $1.5h$ is shown by fig. 8.

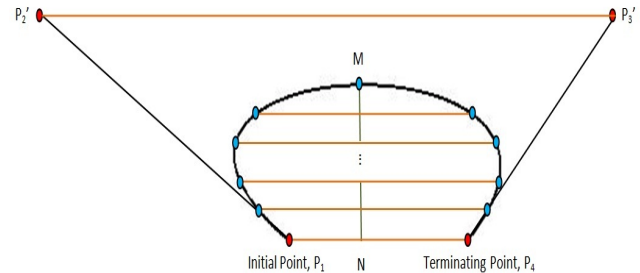


Fig. 8: Parallel line, tangent and assumed intermediate control points of a Bézier curve

From observation it has been found that the original intermediate control points (P_2, P_3) generally lie somewhere around the distances - $(3/4)h$, h , $(4/3)h$ and $1.5h$. The accurate assumption reduces the number of iterations [6]. It is the reason for drawing parallel lines from the base line at these different values of H . Each distance give us a pair of assumed intermediate control points (P_2', P_3').

b) Second Stage

This stage is same as our previous work [6]. As proposing method focus on minimization of error, so the 2nd and 3rd control point must be in the desired location. In order to find the exact location the approach always calculate the error between the two curves iteratively. If the error becomes minimum, according to algorithm it finds the exact location of 2nd and 3rd control points. The value of the error is the summation of the difference between the points obtained from given curve $Q(t)$ and their corresponding points generating from newly found control points in the approximated Bézier curve $Q'(t)$ which is shown in fig. 9. Since Bézier curves can be obtained using a parameter t , where $0 \leq t \leq 1$, the error between two curves $Q(t)$ and $Q'(t)$ can be computed by considering the curves in parametric expressions and finding the corresponding points. For $N+1$ corresponding points, we are getting locations of corresponding points $Q(t) = (Q_x(t), Q_y(t))$ and $Q'(t) = (Q'_x(t), Q'_y(t))$, ($t = 0, 1/N, 2/N, \dots, 1$). The equation that is used for calculating error as follows:

$$\text{Error}, E_1 = \sum_{k=0}^N |Q(t = k/N) - Q'(t = k/N)|$$

$$= \sum_{k=0}^N \sqrt{\{Q_x(k/N) - Q'_x(k/N)\}^2 + \{Q_y(k/N) - Q'_y(k/N)\}^2}$$

The reason why summation is used instead of integration is for computational simplicity. It is applicable when the original Bézier curve is known.

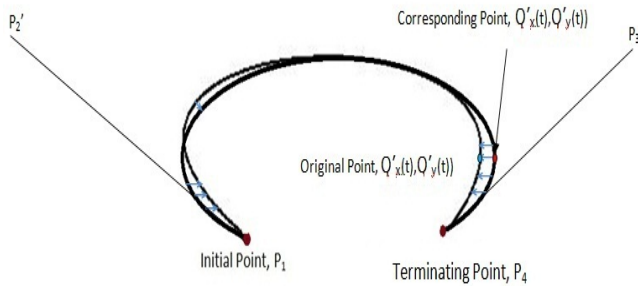


Fig.9 : Computation of the sum of the differences between the corresponding points.

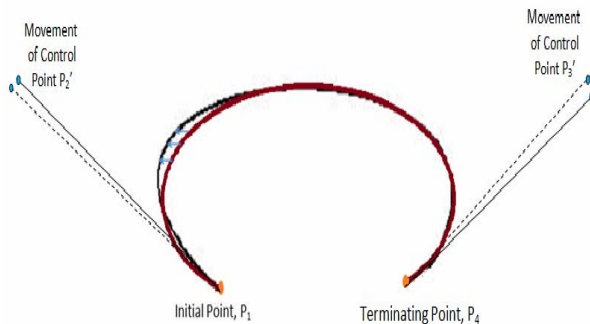


Fig. 10 : Movement of the control point to recover error

In order to recover the exact control points the proposing approach try to move those points at a step ∂ towards both X and Y coordinates which shown in fig. 10. The value of the control point will be updated if the calculated error is minimized, otherwise it will not update. If the step fails to minimize error according to algorithm it reduces the step by half and recursively performing the same operation as stated above.

Increasing the efficiency of the overall program was the main contribution as well as overcoming the limitation of the step-size. Although the approach able to get the accurate result for some Bézier curve by considering step size $\partial x = \partial y = 1$ [21], but it takes more iterations compared to our propose step size, which reduce the efficiency of the overall performance. Here proposing modification works well for most of the Bézier curve and get the error level less than 0.000001 which increase the dynamicity of overall program because it need less iteration. With variable step $\partial x = \partial y = 5, 50, 75$ and 100 in our simulation result we get near to 100% accuracy. The algorithm followed in the second stage is given below.

[Step 1]

Initialization: allsteps [] = {1, 5, 50, 75, 100}
 Count=0(Learning time)
 MAX=100 (Maximum learning time)
 Stepdeterminer = 0 (Variable to determine step)
 $\partial x = \partial y = \text{allsteps} [\text{stepdeterminer}]$ (Variable displacement for the movement of the control points)
 $x = \text{assumed_x}; y = \text{assumed_y}$ (Initial assumption of control points at 1st stage)
 $e = 0.000001$ (Minimum permissible error)

[Step 2]

Count++; if (Count>MAX), then goto step 3
 if $E(x + \partial x, y)$ or $E(x - \partial x, y)$ is minimum then goto step 3
 if $E(x, y + \partial y)$ or $E(x, y - \partial y)$ is minimum then goto step 4
 if $E(x, y)$ is minimum then $\partial x = \partial x / 2, \partial y = \partial y / 2$, goto step 2

[Step 3]

Searching minimum error for x-direction
 While ($E(x + \partial x, y) < E(x, y)$) { Count++; $x = x + \partial x$ }
 While ($E(x - \partial x, y) < E(x, y)$) { Count++; $x = x - \partial x$ }
 if (Count>MAX|| $E(x, y) < e$), then goto Step 5 else goto Step 2

[Step 4]

Searching minimum error for y-direction
 While ($E(x, y + \partial y) < E(x, y)$) { Count++; $y = y + \partial y$ }
 While ($E(x, y - \partial y) < E(x, y)$) { Count++; $y = y - \partial y$ }
 if (Count>MAX|| $E(x, y) < e$), then goto Step 5 else goto Step 2

[Step 5]

If ($E(x, y) > e$ AND stepdeterminer < allsteps.Length) {
 Count = 0; stepdeterminer ++; $\partial x = \partial y = \text{allsteps} [\text{stepdeterminer}]$;
 $x = \text{assumed_x}; y = \text{assumed_y}$; goto Step 2}
 Else { Exit }

IV. SIMULATION RESULTS

The main contribution of the proposed scheme is to find control points of large number of Bézier curve shapes while keeping the efficiency and accurateness of our existing method. We checked several Bézier curves of different shapes with our computer implemented simulation and got desired results as we expected.

Consider the simulation result shown in fig. 11 and fig. 12. Figures of simulation results are picked directly from our computer implemented simulation. At first stage we segment given curve to know its internal points and assume some control points at some probable height using our new proposed schema. Then using the algorithm of 2nd stage we successfully find the control points. Previous Method [6] was unable to extract control points of those shaped curves because of its limited segmentation technique.

Those shaped curves, whose control points can be found by existing method, can also be found by the newly proposed technique while the accuracy and efficiency of both methods remain same. Example of fig. 13 shows it clearly.

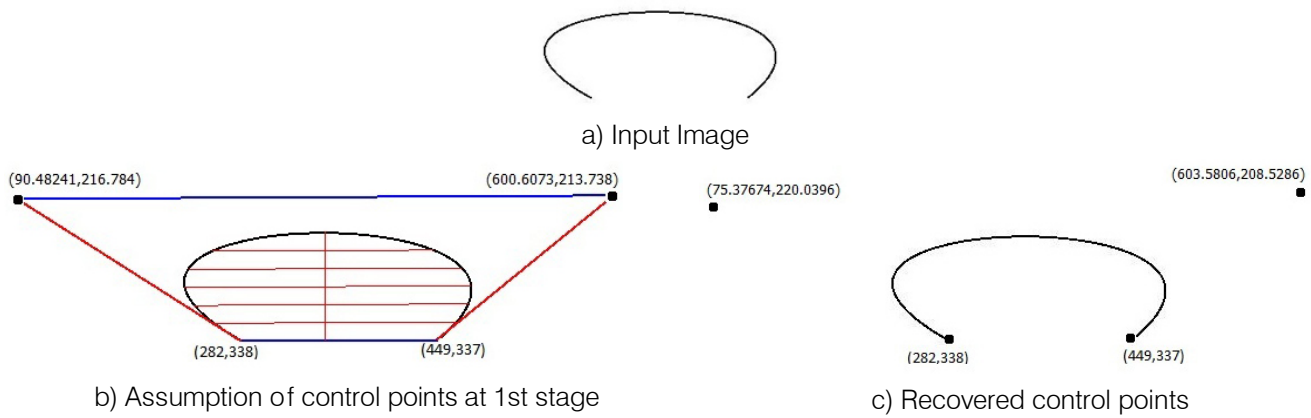


Fig.11 : Simulation Results

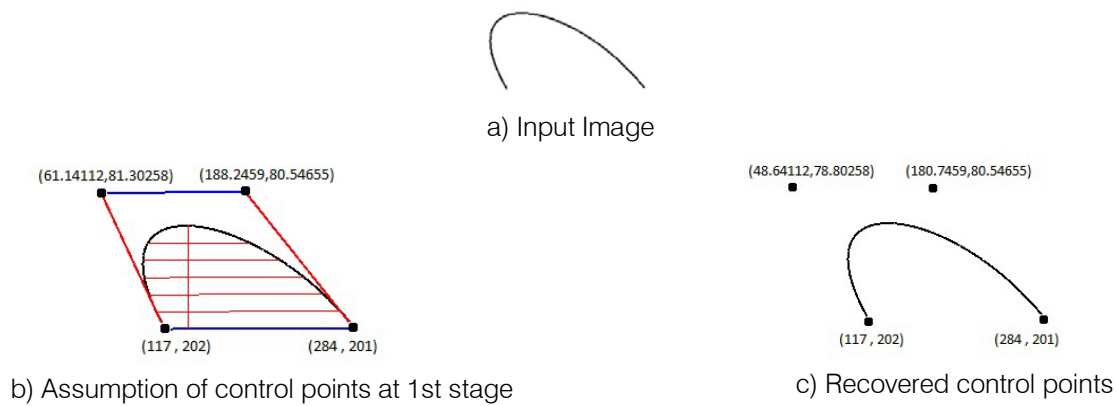


Fig.12 : Simulation Results

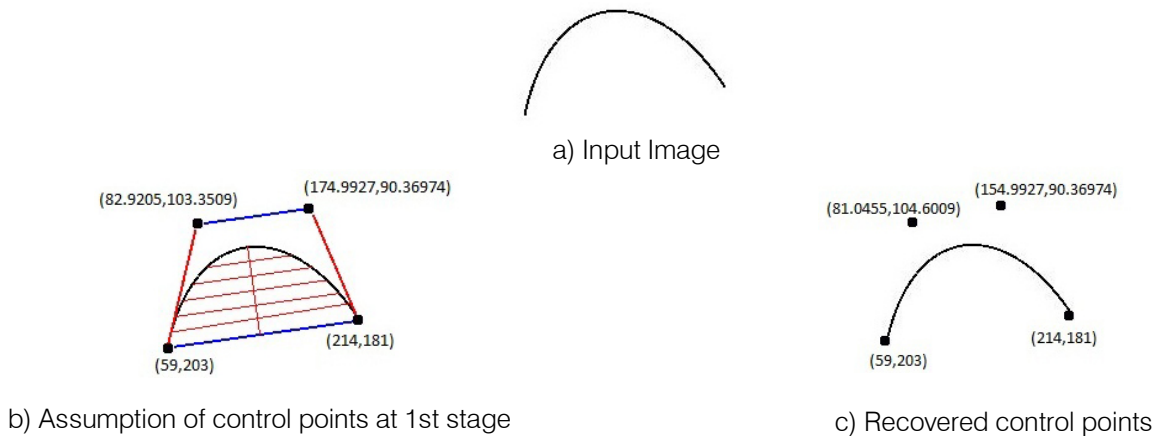


Fig.13 : Simulation Results

To obtain our desired minimum error we checked our input image of Bézier curve with different combination of H (height) and ∂ (step size). Simulation automatically checks Bézier curve in different combination of H and ∂ until the error < 0.001 .

Recovered control points for a combination of H and ∂ is chosen as best result if both error and number of iteration are minimum. Table 1 shows recovered control points on each combination with their corresponding error for the example shown in fig. 11.

Table.1 : Calculated error in different combination of height and step-size

Height	Step Size	1 st control point(P ₁), 2 nd control point(P ₂), 3 rd control point(P ₃), 4 th control point(P ₄)	Iteration	Error
0.75h	5	(282,338),(79.239.6442), (579.5,237.4855), (449,337)	41	0.134
0.75h	50	(282,338),(77.125,216.5192), (606.375,211.2355), (449,337)	26	4E-3
0.75h	75	(282,338),(75.5625,216.9099), (609.5,213.5793), (449,337)	23	4E-3
0.75h	100	(282,338),(77.125,216.5192), (606.375,211.2355), (449,337)	22	4E-3
h	5	(282,338),(78.00368,219.3613), (602.9881,209.3237), (449,337)	40	4E-3
h	50	(282,338),(75.50368,221.8613), (600.4881,206.8237), (449,337)	10	6E-3
h	75	(282,338),(74.72243,225.1816), (598.1443,202.1362), (449,337)	23	14E-3
h	100	(282,338),(75.50368,221.8613), (600.4881,206.8237), (449,337)	11	6E-3
1.33h	5	(282,338),(76.73241,216.784), (609.3573,213.738), (449,337)	10	4E-3
1.33h	50	(282,338), (74.85741,216.784), (606.8573,213.738), (449,337)	8	8E-3
1.33h	75	(282,338),(76.41991,216.784), (607.6385,212.5662), (449,337)	13	4E-3
1.33h	100	(282,338),(74.85741,216.784), (606.8573,213.738), (449,337)	9	8E-3
1.5h	5	(282,338),(73.50174,221.2896), (604.9869,209.2318), (449,337)	16	2E-3
1.5h	50	(282,338),(78.50174,226.2896), (601.2369,204.2318), (449,337)	11	16E-3
1.5h	75	(282,338), (75.37674,220.0396), (603.5806,208.5286), (449,337)	15	2E-3
1.5h	100	(282,338),(78.50174,226.2896), (601.2369,204.2318), (449,337)	12	16E-3

Data from Table 1 shows that minimum error is found at different combination of step-size and height. In this case when height= $1.5h$ and step-size= 5, or height= $1.5h$ and step-size=75 the error is minimum that is $2E-3$. But when height= $1.5h$ and step-size=75, our simulation takes less iteration. That's why we consider (282, 338), (75.37674, 220.0396), (603.5806, 208.5286), (449, 337) as 1st, 2nd, 3rd and 4th control points respectively.

V. CONCLUSION

The new modified approach can recover the control points of 3rd order Bézier curves more accurately and efficiently than those methods which was proposed earlier. In addition, the emphasis of this paper is on the implementation of a new method that can recover the more different shaped Bézier curve which is not possible according to earlier methods. While conducting the experiment different shapes of 3rd order Bézier curves are taken for simulation and method is proved by the simulation result.

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A Comprehensive Analysis of Congestion Control Using Random Early Discard (RED) Queue

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Abstract - Normally all the congestion control method discard the received packet when the queue is full but it is a great problem for speed of data transfer at present. There are many ways to solve this problem. Random Early Detection (RED) algorithm is one of the most famous and powerful method to improve the performance for TCP Connection. In terms of queue management RED drops packet in considered router buffer to adjust the network traffic behavior according to the queue size. We want to investigate how high priority user datagram protocol (UDP) traffic affects the performance of lower priority Transmission Control Protocol (TCP) and proof that RED is the better for controlling the Traffic when they share the same bottleneck link with one or two classes of service.

General Terms : Network Congestion Control, TCP Variants, Network Parameters, Queuing, Drop Tail and User Data Gram Protocol.

Keywords : IETF, RED, AQM, BW, TCP Variants, NS-2, TCL and OTCL.

GJCST-F Classification : C.2.2



Strictly as per the compliance and regulations of:



A Comprehensive Analysis of Congestion Control Using Random Early Discard (RED) Queue

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Abstract - Normally all the congestion control method discard the received packet when the queue is full but it is a great problem for speed of data transfer at present. There are many ways to solve this problem. Random Early Detection (RED) algorithm is one of the most famous and powerful method to improve the performance for TCP Connection. In terms of queue management RED drops packet in considered router buffer to adjust the network traffic behavior according to the queue size. We want to investigate how high priority user datagram protocol (UDP) traffic affects the performance of lower priority Transmission Control Protocol (TCP) and proof that RED is the better for controlling the Traffic when they share the same bottleneck link with one or two classes of service.

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I. INTRODUCTION

A Random Early Detection (RED) is the first active queue management algorithm proposed for deployment in TCP/IP networks. The basic idea behind an active queue management algorithm is to convey congestion notification early to the TCP end points so that they can reduce their transmission rates before queue overflow and sustained packet loss occur. "It is now widely accepted that the RED controlled queue performs better than a drop-tail queue. It is an active queue management algorithm" [1]. "The tail drop algorithm, a router buffer as many packets as it can, and drops the packet when it cannot buffer. If buffers are constantly full, the network is congested" [2]. RED

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addresses these issues. It monitor the average queue size and drops packets based on statistical probabilities. If the buffer is almost empty, all incoming packets reaccepted. As the queue grows, the probabilities for dropping incoming packet are dropped too. RED is more fair than trail drop in the sense of it does not possess a bias against burst traffic that use only a small portion of the bandwidth. The more the more a host transmits, likely it is that packets are dropped. The most common technique of queue management is a trail drop. In this method packets are accepted as long as there is space in the buffer when it becomes full, incoming packets are dropped. This approach results in dropping large number of packets in the time congestion. This can result in lower throughput and TCP synchronization [3]. However TCP includes eleven variants (Tahoe, FullTcp, TCP/Asym, Reno, Reno/Asym, Newreno/Asym, Sack1, DelAck and Sack1/DelAck) as source and five (TCPSink, TCPSink/Asym, Sack1, DelAck and Sack1/DelAck) as destination, implementation in NS-2 [4, 5]. The base TCP has become known as TCP Tahoe. TCP Reno attaches one novel mechanism called Fast Recovery to TCP Tahoe [4]. In addition, TCP Newreno employs the most recent retransmission mechanism of TCP Reno. [6]. The use of Sacks allows the receiver to stipulate several additional data packets that have been received out-of-order within one dupack, instead of only the last in order packet received [5]. TCP Vegas offers its own distinctive retransmission and congestion control strategies. TCP Fack is Reno TCP with forward acknowledgment [7]. Transmission Control Protocol (TCP) Variants Reno, NewReno, Vegas, Fack and Sack1 are implemented in NS-2. RED supervises the average queue size and drops packets based on statistical likelihoods [3].

II. RANDOM EARLY DETECTION

a) RED Parameter Setting

Average queue size avg is formulated [1] as:

$$avg \leftarrow (1 - wq) \times avg + wq \times q \quad (I)$$

Where, wq is the queue weight, q is current queue size. wq should have lower value for bustier traffic; more weight is given in this case for the historic

size of the queue. As avg varies from min_{th} to max_{th} , the packet-marking probability p_b varies linearly from 0 to max_p .

$$p_b \leftarrow \frac{max_p \times (avg - min_{th})}{max_{th} - min_{th}} \quad (II)$$

The final packet-marking probability p_a increases slowly as the count increases since the last marked packet [1]:

$$p_a \leftarrow \frac{p_b}{1 - count \times p_b} \quad (III)$$

III. PERFORMANCE ANALYSIS OF RED MODEL

a) Variation in Threshold Value

Table 1 : Number received packet for various TCP variants with respect to threshold for simulation time 70s

TCP variants	15	20	25	30	35
Reno	854	1185	845	711	733
Newreno	721	763	752	774	741
Vegas	821	777	685	686	625
Fack	800	721	713	644	761
Sack1	864	870	749	813	786

Table 2 : Number received packet for various TCP variant with respect to threshold for simulation time 140s

TCP variants	15	20	25	30	35
Reno	1452	1532	1333	1778	1398
Newreno	1458	1465	1501	1631	1538
Vegas	1345	1578	1350	1498	1538
Fack	1412	1754	1252	2379	1422
Sack1	1501	1339	1595	1358	1179

Table 3 : Number received packet for various TCP variants with respect to threshold for simulation time 210s

TCP variants	15	20	25	30	35
Reno	2659	2635	2376	1946	2300
Newreno	2701	2546	2032	2169	2303
Vegas	2254	2255	2301	2432	2178
Fack	2802	2462	2897	2131	2376
Sack1	2269	2416	2201	2554	2082

Table 4 : Number received packet for various TCP variant with respect to threshold for simulation time 270s

TCP variants	15	20	25	30	35
Reno	3142	3403	3312	3323	2902
Newreno	3383	3220	3204	3265	2928
Vegas	2624	2749	2778	2538	2799
Fack	3545	3088	2856	2681	4298
Sack1	3888	3216	3051	3232	3409

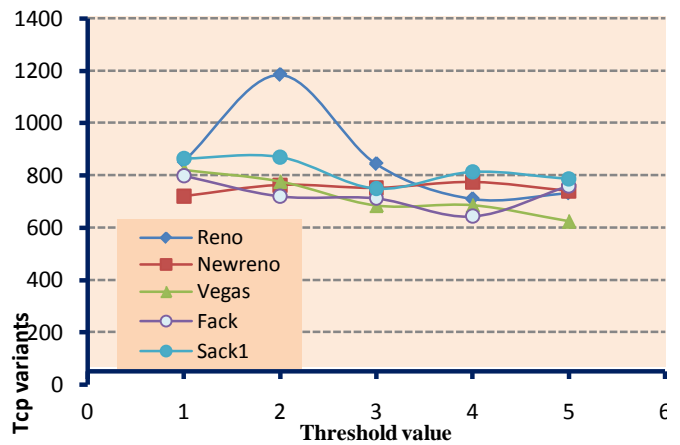


Figure1 : Graph of received packet for various TCP variants with respect to threshold for simulation time 70s

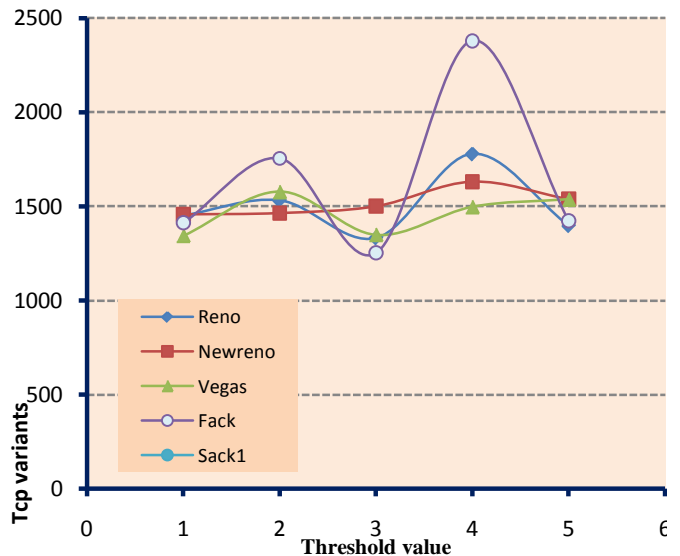


Figure2 : Graph of received packet for various TCP variants with respect to threshold for simulation time 140s

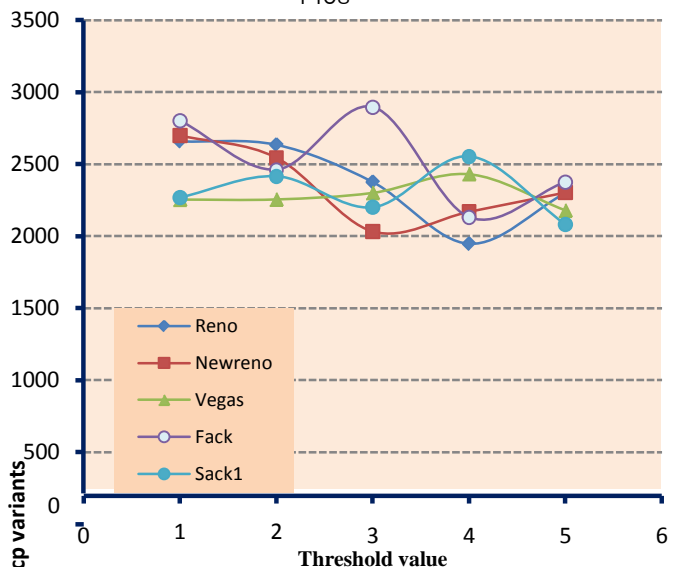


Figure3 : Graph of received packet for various TCP variants with respect to threshold for simulation time 210s

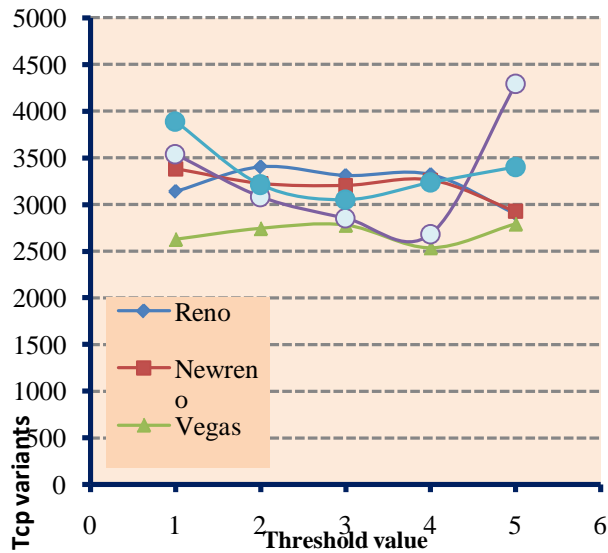


Figure 4 : Graph of received packet for various TCP variants with respect to threshold for simulation time 280s

b) Performance Comparison

We that when threshold increase then variation course in received among various TCP variants and all arriving packets are received when average queue size exceeds max threshold or less than minimum threshold then packets are dropped which is shown in above all tables and corresponding figure. We found that Newreno TCP variants is the best because mean number of received packet is high mean number of dropped packet is low.

c) Comparison of TCP and UDP

i. Comparison of Received Packet

Table 5 : Comparison of received packet between UDP and TCP

Times		70s	140s	210s	280s
U	15	675	1294	1996	2586
	20	797	1222	1803	2694
D	25	758	1187	2127	2633
	30	795	1484	2085	2794
P	35	749	1336	1963	2783
T	15	566	1352	2725	2457
	20	665	1606	2374	3284
C	25	637	1438	2425	3694
	30	548	1656	2247	2832
P	35	834	1614	2413	3438

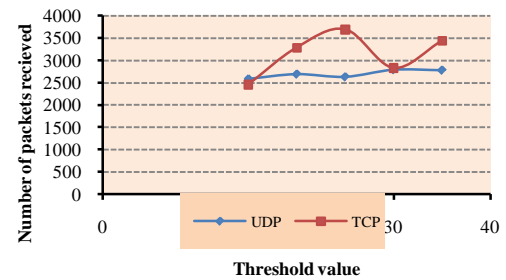


Figure 5 : comparison graph of received packet between UDP and TCP for simulation time 280s

ii. Comparison of Dropped Packet

Table 6 : Comparison of received packet between UDP and TCP

Times	70s	140s	210s	280s	
U	15	25	126	246	374
	20	67	104	113	354
D	25	24	53	426	696
	30	135	113	162	344
P	35	36	34	433	357
T	15	0	26	37	73
	20	0	14	33	46
C	25	0	5	36	43
	30	0	8	23	17
P	35	0	4	14	12

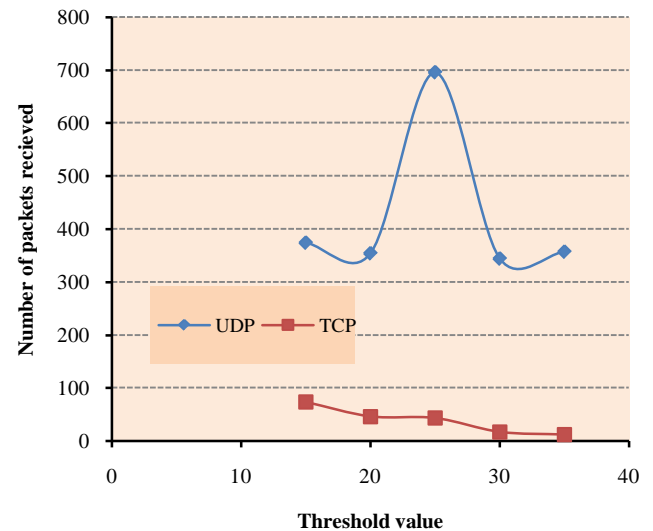


Figure 6 : comparison graph of dropped packet between UDP and TCP for simulation time 280s

IV. CONCLUSION

From the aforementioned comparison of the performance it is found that **TCP** is better than **UDP** because packet received is higher in it with respect to **UDP**. That is why packet loss is lower in **TCP**. In case of packet drop, it is clear those packet drop is higher in **UDP** than **TCP** and also occur more congestion in it. It is possible to control congestion in **TCP** using **RED** model.

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Clustering Method for categorical and Numeric Data sets

By Simmi Bagga, Dr. G.N. Singh

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Abstract - Many issues concerned with clustering process are due to large datasets involves. In clustering computation become expensive when there are large data sets involved and work efficiently when there is limited number of cluster with relatively small data set. This paper will present a new technique for clustering for large datasets. That will work efficiently equally with large data set as well as with small data sets. The main idea behind this method is to divide the whole process in two steps. The first step uses a cheap approximate distance measure that divide the data into overlapped subsets we call it stubs. Then in second step clustering is performed for measuring exact distances only between points that occur in common stubs. The stub based clustering approach reduces computation time over a traditional clustering and also increases its efficiency.

Keywords : *Clustering, stubs, centroid, K-Means Clustering.*

GJCST-C Classification : *1.5.3*



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Clustering Method for categorical and Numeric Data sets

Simmi Bagga^α, Dr. G.N. Singh^Ω

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Keywords: *Clustering, Stubs, Cetroid K-Means Clustering.*

I. INTRODUCTION

From the last few years, the development of Information technology become very fast. So there has been widespread change in the adoption and utilization of new technologies in every field. That generated huge amount of data in the various fields. So there has been widespread change in the adoption and utilization of new technologies in every field that generated huge amount of data in the various fields.

To handle such large amount of data having complex structure effectively and efficiently for decision making, we use the concept of Data Mining. Data Mining is the process of identifying valid, useful and understandable patterns in data. There are various techniques have been developed and used in Data Mining including association, classification, clustering, prediction and sequential patterns etc. Clustering techniques are used for combining group that are similar to each other. Each cluster should be different from other clusters.

Clustering is one of the oldest and effective techniques of Data Mining. Basically clustering algorithms are divided in to two types these are Partitional and Hierarchical.

a) Partitional clustering algorithms

In Partitional clustering algorithm firstly they compute partitions of the data based on similarity of data, and then they chose one that optimizes the criteria. These kinds of algorithms are highly complexity.

b) Hierarchical Clustering Algorithms

Hierarchical clustering algorithm creates hierarchical decomposition of the objects set. They are either agglomerative or divisive:

- (a) Agglomerative algorithm starts with object as a separate cluster and then merges group according to a distance measure. These algorithm stops when all object converts in to single object or at the point where user wants to stop. These algorithms follow bottom-up merging.
- (b) Divisive algorithms works opposite to agglomerative strategy. They start with one group of all objects and then split groups into smaller ones. This process repeats until each object falls in one cluster, or according to the user desire. These types of algorithm follows top down merging.

Further there are various categories of clustering algorithm. These categories are mainly focused on specific kind of data set or with some specific problems.

i. Density-Based Clustering:

Density Based clustering algorithms group objects according to the functions that deal with density objectives. Density is defined as number of objects resides near the data objects. In this approach cluster grows longer as the density increases i.e. number of object in neighborhood increases. These are mainly hierarchical in nature.

ii. Grid-Based Clustering:

Grid based Clustering algorithms deals with the spatial data i.e. about the objects related to space. Spatial data includes structure of objects in space, its relationships and properties. In these types of algorithm, we quantize data in to cells. Then we with only with those objects that's belongs to cell.

iii. Model-Based Clustering:

These algorithm deals with the approximations of model i.e. deals with the various parameters of the model that best fit the data. These types of methods can either be partitional or hierarchical depends on the

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model. They are also very closer to density-based algorithms.

iv. *Categorical Data Clustering:*

These algorithms are mainly developed for data where numerical-oriented, distance measures cannot be applied. These are very close to both partitional and hierarchical methods.

Clustering algorithms become computationally expensive when the data set is to be large. There are mainly three reasons in which the data set can be large:

- When large number of elements in the data set involved
- When data element can have many features.
- When many clusters to be discover from the data set.

Recent cluster algorithms have focused on the efficiency issues. K-means clustering algorithm is very efficient when we start by finding good initial points, but is not efficient when the size of cluster become large. There is no such algorithm works efficiently when any of above three kinds of data is there in data set because in that case we have millions of data element with many features and cluster.

In this paper we introduce a new clustering technique that work well even in the case of large data set and large clusters. The main idea of this new technique is to perform clustering in two steps. In first step, divide the data into overlapping subsets called stubs, then in final step in which expensive distance measured only among points that occur in a common stub. First step can be performed very quickly and roughly where as second step is little rigorous. During the first step we mainly built stub by using approximate distance measure, the second step can be performed by any standard algorithm of clustering. The computation time and complexity is saved by the approximate distance measure used to create stubs. This computation is saved by eliminating all of the distance comparisons among the data set. We have found small accuracy increases due to the usage of two different distance measures. Clustering based on stubs applied to many different standard clustering algorithms, like K-means, Greedy Agglomerative Clustering and Expectation-Maximization.

II STUB BASED CLUSTERING

The basic idea behind the stub based algorithm is that one can reduce the number of distance computations for clustering. The first step cheaply partitioning the data set into overlapping subsets called stubs and then in second step the distance is measuring only among the data points that belong to a common subset. The stub based technique uses two different sources to cluster items: a cheap and approximate similarity measure. For example, in first

step we just measure approximate distances among the data set like proportional comparison between data set and calculate the similarities between the data set that will reduce the size of the data set up to some extent. This step can be performed cheaply and in little time. In the final step, the more accurate similarity measures are performed that is more expensive in nature. In this step detailed distance measurement is performed.

We divide this clustering process into two steps. In the first step we use the short distance measure to create some number of overlapping subsets, called stubs. These are calculated by the proportionate measure. Stub is just a subset of the data elements find by approximate measure of its similarities. These distances are the distance performed from a central point. A data element can appear in more than one subset or stub but every data element must belong to at least one stub. Stubs are created with the intention that data element appearing in common stub may be far different that they could not possibly be in the same cluster. The method used for calculating the distance to create stubs is approximate. There exist many overlapping stubs in the data set, because we choose a large enough distance to ensure each and every data element should belong to any stub. These stubs are just made by measuring approximate distance and it's a very cheap method of calculating and it reduces the size of data set.

In the second step, we perform traditional clustering algorithm on that filtered stubs, such as K-means, Greedy Agglomerative Clustering or by using any accurate distance measure algorithm. The main restriction impose in this method is that we do not calculate the distance between two points that do not belong to same stub. This restriction is imposed because we assume the distance between the two different stubs to be infinite. The expensive distance measurements will only be made between the same stubs. This is will overall reduce the number of calculation.

If the first step is not properly performed that is if stubs are not properly made then it degrade the performance of the second step also. So stubs should be created carefully. If stubs are not too large and do not overlap much, then we cannot avoid expensive computation for clustering. The constraints imposed on the clustering imposed by the stubs may not lose accuracy but will increase computation efficiently. If distance to a cluster is measured to the centroid of the cluster, then clustering accuracy will be preserved exactly.

For every cluster, there exists a set of stubs. Expensive distance measurements will only made between pairs of data points in the same stubs.

a) *Creating Stubs*

In the case of stub based clustering, user will be able to focus domain-specific features in order to design a short distance measures. User efficiently creates stubs using these measures. For example, if we have large data of patients of number of hospitals that contain information of diagnosis, treatments and payment histories. A cheap measure calculates the similarity between diagnoses of the patients. Result might be 1, if they have the similarities and 0 if they do not have any similarity. In this case stub creation is small and the common diagnosis results fall in the same stub. If the same patient falls in multiple diagnoses then he will fall into multiple stubs and also some stubs will overlap. The small number features are sufficient to built stub, even if the data item may have thousands of features.

b) *Cheap Distance Measurement*

There are various methods to calculate the cheap distance measure. One of the methods for distance measure for text is based on the inverted index. An inverted index is in the form of sparse matrix in which, each word can directly access the list of documents containing that word. When we want to find all documents according to some, we need not to measure the distance to all documents, but only have to examine the list of documents associated with each word in the query. The documents which have no words in common with the query will never be considered. The inverted index can efficiently calculate a distance metric.

Using the above distance measure stub can creates as follows. Start with a list of the data items, and with two distance thresholds, lets say T_1 and T_2 , where $T_1 > T_2$. Pick a data point from the list and approximately measure distance to all points. Put all similar data points to the distance threshold T_1 into a stub. Remove from the list all points that are within distance threshold T_2 . Repeat until the list is empty. The inverted index can be applied to real-valued data item.

c) *K-Means Stubs*

One can also use the stubs idea to speed up prototype based clustering methods like K-means. K-Mean is well known partitioning based method of clustering. It is simple and iterative method works around one artificial point which represent the average location of the cluster is called Centroid.

This algorithm takes an input a number of clusters that is the k from. Means is an average location of all the members of a cluster. In this algorithm we have to partition n object set to k clusters. Cluster is measured on the basis of its mean or average location. The basic idea behind this algorithm is to first randomly select k of the object that is centroid of cluster. Using this k of clusters, we optimize intra cluster similarity and inter-cluster dissimilarity. Each remaining object, the most similar object is assigned to cluster based the distance between object and centroid or cluster mean.

Then we compute new mean. This process repeats until the whole function overages. This is an iterative method in which we always redefine the center point or centroid until cluster detection is finished.

This approach is basically based on prototypes that are associated with the stubs that contain them. The prototypes are only influenced by data inside the associated stubs. After creating the stubs, we decide how many prototypes will be created for each stub. Then we place prototypes into each stub. For each prototype, we find the stubs that contain in it (computed by using cheap distance measures) and then calculate distances from that prototype to points within those stubs.

K-means algorithm not gives just similar results for stub. In K-means each data point is assigned to a single prototype. As long as the cheap and expensive distance measures are sufficiently similar that the nearest prototype is within the boundaries of the stubs that contain that data point, then the same prototype will win.

III. COMPUTATIONAL COMPLEXITY

We can simply say that computational time is saving using stub based method. This technique has two step and mainly done two types of comparisons. One is relative fast step where stubs are created. The other one is a slow clustering process in which we apply K-Mean clustering algorithm. If we create stubs by using inverted index method then there is no need to perform pair wise distance comparison.

In the case of K-means or Expectation-Maximization, clustering without stubs requires $O(nk)$ distance comparisons per iteration of clustering. Consider the K-mean method with stubs where each cluster belongs to one or more stubs. Assume that clusters have the same overlap factor f as data points do. Then, each cluster needs to compare itself to the fn/c points in f different stubs.

IV. CONCLUSION

Clustering large data sets having large cluster is a very tedious task. It is very expensive and inefficient to deal with large data set having large number of clusters using traditional clustering methods. The goal of this paper is to describe a new stub based method for clustering that takes relatively less computation time and performs result more effectively in the case of large data set. In this paper we described how can we create stubs and how can we apply traditional cluster methods like K-mean clustering method to perform effective clustering.



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Comparison of Time Taken and Compression Efficiency for Different Sizes of Databases

By Dr. Mrs Pushpa Suri, Mrs. Meenakshi Sharma

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Abstracts - Data compress for object oriented data warehousing. A data warehouse is an essential component to the decision support system. The traditional data warehouse provides only numeric and character data analysis. But as information technologies progress, complex data such as semi-structured and unstructured data become vastly used. Data Compression is of interest in business data warehousing, both because of the cost saving it offers and because of the large volume of data manipulated in many business application.[3],[5]. The entropy is used in many areas such as image processing, document images. But in our research we used the entropy in object oriented data warehousing. Creation of different sizes of databases in oracle. Employment of object oriented programming for compression using Datawarehousing.

Keywords : *Data warehousing, Data compression, Object oriented, Entropy.*

GJCST-F Classification : *C.2.4, E.4*



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Dr. Mrs Pushpa Suri^α, Mrs. Meenakshi Sharma^Ω

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Keywords : Data warehousing, Data compression, Object oriented, Entropy.

I. INTRODUCTION

One of the hottest topics in the industry today is data warehousing and on-line analytical processing (OLAP). Although, data warehousing has been around in some form or another since the inception of data storage, people were never able to exploit the information that was wastefully sitting on a tape somewhere in a back room. Today, however, technology has advanced to a point to make access to this information an interactive reality. Organizations across the country and around the world are seeking expertise in this exploding field of data organization and manipulation. It is not a surprise, really, that business users want to get a better look at their data. Today, business opportunities measure in days, instead of months or years, and the more information empowering an entrepreneur or other business person, the better the chances of beating a competitor to the punch with a new product or service. The task of transitioning from a procedural mindset to an object-oriented paradigm can seem overwhelming; however, the transition does not require developers to step into another dimension or go to Mars in order to grasp a new way of doing things. In many ways, the object-oriented approach to development more closely mirrors the world we've been living in all along: We each know quite a bit about

objects already. It is that knowledge we must discover and leverage in transitioning to object-oriented tools and methodologies.

A data warehouse is a mechanism for data storage and data retrieval. Data can be stored and retrieved with a multidimensional structure--hypercube or relational, a star schema structure or several other data storage techniques.

II. DATA COMPRESSION

Data compression is of interest in business data warehousing, both because of the cost savings it offers and because of the large volume of data manipulated in many business applications. The types of local redundancy present in business data files include runs of zeros in numeric fields, sequences of blanks in alphanumeric fields, and fields which are present in some records and null in others. Run length encoding can be used to compress sequences of zeros or blanks. Null suppression may be accomplished through the use of presence bits. Another class of methods exploits cases in which only a limited set of attribute values exist. Dictionary substitution entails replacing alphanumeric representations of information such as bank account type, insurance policy type, sex, month, etc. by the few bits necessary to represent the limited number of possible attribute values.

The problem of compressing digital data can be decoupled into two subproblems: modeling and entropy coding. Whatever the given data may represent in the real world, in digital form it exists as a sequence of symbols, such as bits. The modeling problem is to choose a suitable symbolic representation for the data and to predict for each symbol of the representation the probability that it takes each of the allowable values for that symbol. The entropy-coding problem is to code each symbol as compactly as possible, given this knowledge of probabilities. (In the realm of lossy compression, there is a third subproblem: evaluating the relative importance of various kinds of errors.)

For example, suppose if it is required to transmit messages composed of the four letters a, b, c, and d. A straightforward scheme for coding these messages in bits would be to represent a by \00", b by \01", c by \10" and d by \11". However, suppose if it is known that for any letter of the message (independent of all other letters), a occurs with probability .5, b occurs with probability .25, and c or d occur with probability

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.125 each. Then a shorter representation might be chosen for a, at the necessary cost of accepting longer representations for the other letters. a could be represented by \0", b by \10", c by \110", and d by \111". This representation is more compact on average than the first one; indeed, it is the most compact representation possible (though not uniquely so). In this simple example, the modeling part of the problem is determining the probabilities for each symbol; the entropy-coding part of the problem is determining the representations in bits from those probabilities; the probabilities associated with the symbols play a fundamental role in entropy coding.

One well-known method of entropy coding is Huffman coding, which yields an optimal coding provided all symbol probabilities are integer powers of .5. Another method, yielding optimal compression performance for any set of probabilities, is arithmetic coding. In spite of the superior compression given by arithmetic coding, so far it has not been a dominant presence in real data-compression applications. This is most likely due to concerns over speed and complexity, as well as patent issues; a rapid, simple algorithm for arithmetic coding is therefore potentially very useful.

An algorithm which allows rapid encoding and decoding in a fashion akin to arithmetic coding is known as the Q-coder. The QM-coder is a subsequent variant. However, these algorithms being protected by patents, new algorithms with competitive performance continue to be of interest. The ELS algorithm is one such algorithm.

The ELS-coder works only with an alphabet of two symbols (0 and 1). One can certainly encode symbols from larger alphabets; but they must be converted to a two-symbol format first. The necessity for this conversion is a disadvantage, but the restriction to a two-symbol alphabet facilitates rapid coding and rapid probability estimation.

The ELS-coder decoding algorithm has already been described. The encoder must use its knowledge of the decoder's inner workings to create a data stream which will manipulate the decoder into producing the desired sequence of decoded symbols.

As a practical matter, the encoder need not actually consider the entire coded data stream at one time. One can partition the coded data stream at any time into three portions; from end to beginning of the data stream they are: preactive bytes, which as yet exert no influence over the current state of the decoder; active bytes, which affect the current state of the decoder and have more than one consistent value; and postactive bytes, which affect the current state of the decoder and have converged to a single consistent value. Each byte of the coded data stream goes from preactive to active to postactive; the earlier a byte's position in the stream, the earlier these transitions occur.

A byte is not actually moved to the external file until it becomes postactive. Only the active portion of the data stream need be considered at any time. Since the internal buffer of the decoder contains two bytes, there are always at least two active bytes. The variable backlog counts the number of active bytes in excess of two. In theory backlog can take arbitrarily high values, but higher values become exponentially less likely.

III. RELATED WORK

"2-D Compression of ECG Signals Using ROI Mask and Conditional Entropy Coding," have given a novel 2-D compression scheme which employs 1-D discrete wavelet transform, the region of interest mask, and the conditional entropy coding based on context models. Experimental results on records selected from the Massachusetts Institute of Technology-Beth Israel Hospital arrhythmia database show that the proposed method outperforms some existing compression schemes [5]. "Lossless Compression Using Conditional Entropy-Constrained Subband Quantization," have proposed Lossless Compression Using Conditional Entropy-Constrained Subband Quantization [13]. Sang et al in their paper "A novel approach to scene change detection using a cross entropy," have shown that in huge video databases, an effective video indexing method is required. While manual indexing is the most effective approach to this goal, it is slow and expensive. Thus automatic indexing is desirable, and previously various indexing tools for video databases have been developed. For efficient video indexing and retrieval, the similarity measure is an important factor. This paper presents new similarity measures between frames and proposes a new algorithm to detect scene changes using a cross entropy defined between two histograms. Experimental results show that the proposed algorithm is fast and effective compared with several conventional algorithms to detect abrupt scene changes and gradual transitions including fade in/out and flash light scenes [12].

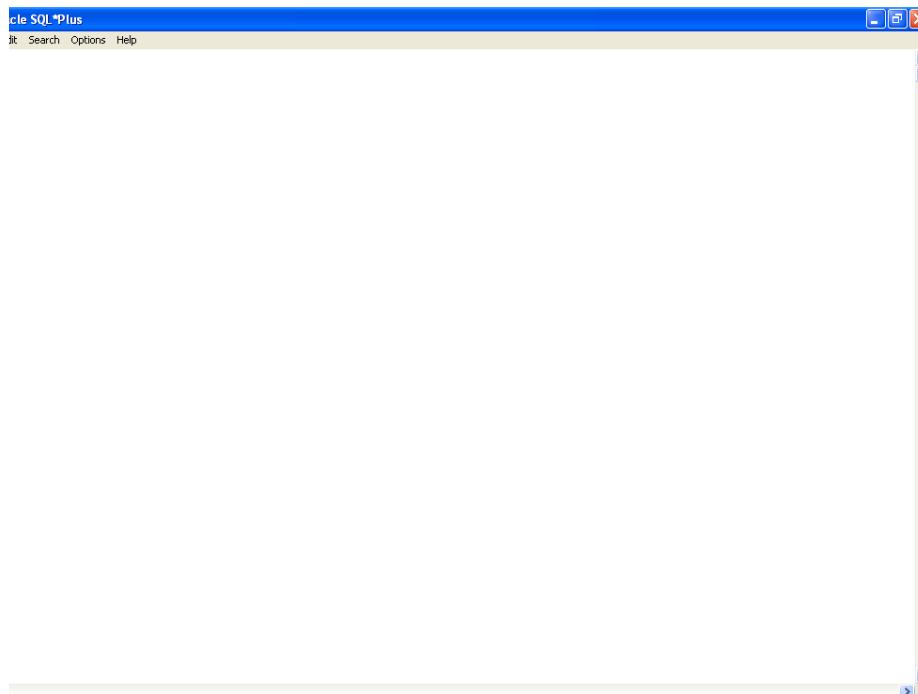
IV. OBJECTIVE

The objective of the present study is to

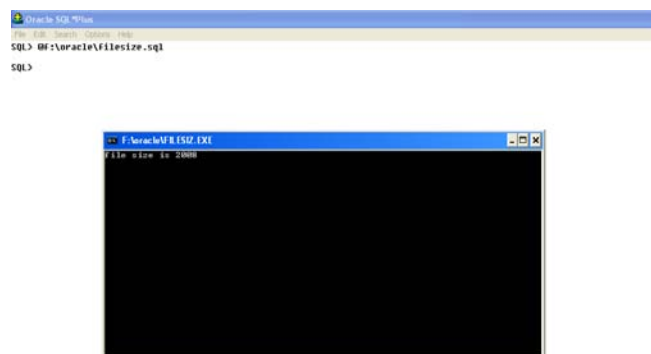
1. Develop data of compression for object oriented data warehousing.
2. Devise efficient compression algorithms in data warehousing to enhance the efficiency of the data warehousing packages so that less CPU time and less Memory is consumed.

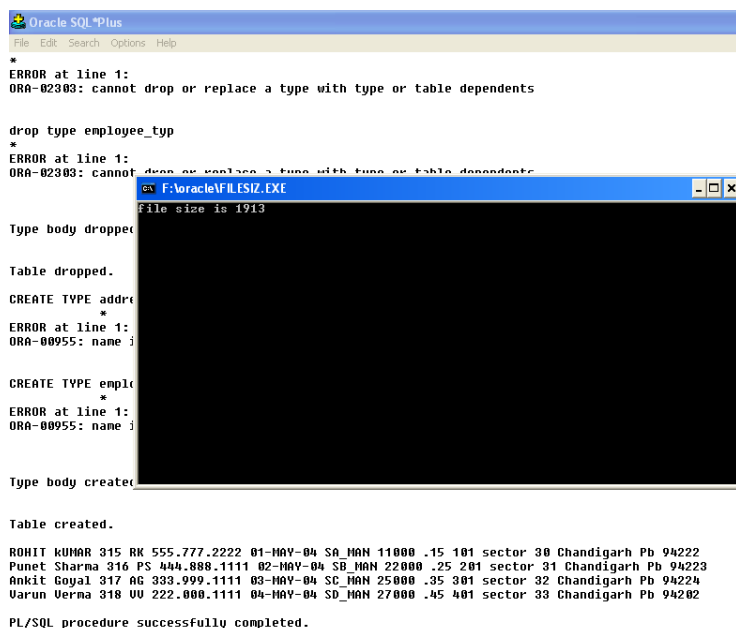
Implement compressor and expander using entropy algorithm and test its effectiveness on different sized databases

V. RESULTS



Main File size





```

Oracle SQL*Plus
File Edit Search Options Help

*
ERROR at line 1:
ORA-02303: cannot drop or replace a type with type or table dependents

drop type employee_type
*
ERROR at line 1:
ORA-02303: cannot drop or replace a type with type or table dependents

Type body dropped

Table dropped.

CREATE TYPE address
*
ERROR at line 1:
ORA-00955: name is already used for a type

CREATE TYPE employee
*
ERROR at line 1:
ORA-00955: name is already used for a type

Type body created

Table created.

ROHIT KUMAR 315 RK 555.777.2222 01-MAY-04 SA MAN 11000 .15 101 sector 30 Chandigarh Pb 94222
Punet Sharma 316 PS 444.888.1111 02-MAY-04 SB MAN 22000 .25 201 sector 31 Chandigarh Pb 94223
Ankit Goyal 317 AG 333.999.1111 03-MAY-04 SC MAN 25000 .35 301 sector 32 Chandigarh Pb 94224
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PL/SQL procedure successfully completed.

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VI. CONCLUSION

In this paper we have discuss the data compression and how the data is compresses in oracle 10g using object oriented language. Data Compression is of interest in business data warehousing, both because of the cost saving it offers and because of the large volume of data manipulated in many business application. The entropy is used in many areas such as image processing, document images. But in our research we used the entropy in object oriented data warehousing. Creation of different sizes of databases in oracle. Employment of object oriented programming for compression using data warehousing. Further compression of database .csv files using C++. Comparison of time taken and compression efficiency for different sizes of databases

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Approach:

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Approach

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References	Complete and correct format, well organized	Beside the point, Incomplete	Wrong format and structuring



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