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# Epistemological Differences in Tactical and Strategic Spatial Planning 

By Aynaz Lotfata<br>Technical University of Ankara, Turkey

Abstract - Purpose: In spatial processes, the terms strategy and tactic have frequently appeared without any clear distinguishing, whereas strategies and tactics have epistemologically characterized differently. Strategic knowledge have tries to defining visions of urban space through answering "What" and "Why" questions and its knowledge is the abstract knowledge, while tactical knowledge is the experiential knowledge via answering "How" question. Strategy and tactics are both terms from a military context where strategy has referred to long-term war planning in contrast to tactic as short-term flexible battle planning. Strategy has worked from the position of power that is in a place to force its opponents to accept its conditions. The strategic conventional ideologies empty of tactical policies have destroyed built spaces memories to organize urban society according to elite's tendencies. The Equivalent of strategy in urban planning is Master plan.

Tactics have not operated such dictated forces. Tactics are bottom-up spatial practices. Developing bottom-up dynamics have caused to flexibilities of the prevailed ideologies of the upper policies. Hayden calls short-small actions (Tactics) "power of places" to challenge homogenous urban planning. Homogenous urban planning has planned urban spaces in a frozen platform of time. Another important purpose of this study has been organized to expand "public policy time".

Keywords : strategy, tactic, synergy, imitating, empowered.
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# Epistemological Differences in Tactical and Strategic Spatial Planning 

Aynaz Lotfata


#### Abstract

Purpose: In spatial processes, the terms strategy and tactic have frequently appeared without any clear distinguishing, whereas strategies and tactics have epistemologically characterized differently. Strategic


 knowledge have tries to defining visions of urban space through answering "What" and "Why" questions and its knowledge is the abstract knowledge, while tactical knowledge is the experiential knowledge via answering "How" question. Strategy and tactics are both terms from a military context where strategy has referred to long-term war planning in contrast to tactic as short-term flexible battle planning. Strategy has worked from the position of power that is in a place to force its opponents to accept its conditions. The strategic conventional ideologies empty of tactical policies have destroyed built spaces memories to organize urban society according to elite's tendencies. The Equivalent of strategy in urban planning is Master plan.Tactics have not operated such dictated forces. Tactics are bottom-up spatial practices. Developing bottom-up dynamics have caused to flexibilities of the prevailed ideologies of the upper policies. Hayden calls short-small actions (Tactics) "power of places" to challenge homogenous urban planning. Homogenous urban planning has planned urban spaces in a frozen platform of time. Another important purpose of this study has been organized to expand "public policy time".

Findings: Thereby, strategic spatial planning without tactics has justly characterized as an abstract phenomenon. Time and space co-existence policies have gotten its legitimacy via witnessing spatial tactics. The tactics developed by ordinary people are at root attempts to negotiate power relationships, discourses and representations of identity. To develop the empowered spatial planning, the synergic relations amid localities tactics and strategies have to implement for tackling with the stochastic world. And the arguments have orderly developed on permanent and temporary identities of spatial strategic and tactics

Results: The paper has aimed to solve the problem of misunderstandings in tactics and strategies definitions and applications in urban planning. Additionally through explanations of strategies and tactics differences in spatial planning, the project has tries to argue that strategy of locality cannot be duplicated like spatial tactics imitating all over the world. Localities got used to dismantling other localities strategies and tactics to enhance their situation in the competition platform. However, a strategy is hard to duplicating such tactics.

To sum up, strategies and practices (Tactics) have shaped the everyday life of inhabitants and urban planning should make balance in utilizing both. Additionally locality

[^1]should not imitate spatial tactics and strategies of other localities. Otherwise, it has reified spatial tactics and strategies. Every locality has own priorities to consider in urban planning.

Originality: In planning literature, implementing spatial strategies have not been the recent phenomena. The differences have reverted to the deficiency of synergic relations amid tactics and strategy. The conventional regulated spatial planning has generally formulated without spatial tactics to reach spatial goals while to tackle the real world future, the reciprocal connections of tactics and strategy have gotten priorities. In other words, planning has to move on toward an experimental science of planning.

With considering the novel re-configuring urban planning, the paper has tries to shed light based on simulating urban planning via "Artificial Intelligence" achievements. This will support arguments on systematic planning definitions to control the uncertain world. In moving form toy-world domains that characterized early conventional planning, we are looking at a wide range of issues, including reasoning in uncertain worlds, interacting with processes and events beyond the agent's direct control and controlling systems in real non-linear time. The disciplinary background of the paper is philosophical-epistemological. The enquiry is conceptual.
Keywords : strategy, tactic, synergy, imitating, empowered.
I. InTRODUCTION-DYNAMIC AND

Uncertain Domains; Planning With Stochastic Actions

One of the main concerns of socio-spatial policy makers all over the world is to improve their ability to anticipate and control the future. Designing human futurity, whether long or short-term is not a simple matter. The sophistication involved in dealing with ongoing fundamental changes in modern societies challenges the ability to control human futurity and to sustain continuity. Here our concerns in the following exploration are time perception and time management in public policy. Time related public policy literature is generally farmed in terms of long term and short term policy. This study suggests juxtaposing "tactical policy time" and "strategic policy time". Tactical policy time is defined as "taking a specific time-related plan or action aimed at achieving a defined policy result". Tactical policy time has applied in the case of short time tables. Strategic policy time has defined as "taking a specific time-related plan or action with the aim of coping better with uncertainty in the future". These arguments have invited attentions on time-related
planning or action aiming to achieve a defined policy result or cope better with uncertainty in the future". The mapping of time management in public policy generally indicates two main trends: 1- a pragmatic trend-shortterm policy has based on the response-oriented policy (Tactic) and 2- a normative trend-long-term policy inspired by "the voice of the future" to avoid uncertainty" (Strategy).

In other words, "Why do we plan?" Planning is to respond necessities of real world. To control real world, there are two focus points: coping with uncertainties and real time planning. The planning knowledge is incomplete whereas that is the process. The process definition of planning has gotten back ton on-predicted events in the world by which control and pre-determination of domains have not been done completely. There is the world of uncertainties. The planning has to discovery new approaches of intervention in the world such reactive planning, tactical planning and conditional planning.

The arguments have supported that the planning process has not only defined due to theoretical discussions but also that has included the practical exercises. Relying justly on reactive, tactical and conditional planning with the practical essence has not improved the controlling uncertainties. The planning process requires mutual connections of theory and practice. In reality, tactical, reactive and conditional planning has justly supported the incremental practical planning. However, to control the world with stochastic actions where linear and universal plans have not functioned any more, incremental and conventional spatial practices combinations have insistently emphasized.

Therefore, planning in realistic domains has forced us to confront two main issues: uncertainty and urgency. Uncertainty arises since the planner is neither omnipotent, omniscient one nor alone in the world to control stochastic actions. The paper aims to consider spatial planning as the automatic planning by which planning has prepared to any stochastic actions of the world in which has witnessed the social, economic, environmental and politic upheavals. Thereby the conventional traditional planning should re-modify to achieve goals of planning with high probability. That does not mean, refuting result rationality of conventional planning in which its rationality measures how efficiently the plan achieves its specified objectives. Planner should re-construct planning with making balance between result rationality of conventional planning and process rationality of tactical planning.

Therefore, the lost and disregarded part of planning in dynamic and uncertain world has characterized via tactical planning. Planning has been a process changing its long term focus point toward short term planning. To control uncertain and dynamic world, planners should be familiar with reactive planning.

Nilsson has proposed the concepts of actions networks for reactive planning/tactical planning. Actions networks differ from universal plans in that they allow the formation of action hierarchies (Hanks, 1990). This supports argument that we view planning as the process and planning has been converted from long term prospects into short term tasks. That does not mean that process has thrown out the strategic planning and justly focused on tactical planning. This process must consider both the strategic and tactical aspects of planning. Tactical or incremental planning has emphasized on tasks/ actions which achieve short term goals. Purely strategic planning cannot immediately react to a changing world while tactical planning can answer changes quickly. The traditional planning logic is Boolean logic where the values of variable are the truth values, truth and false, usually denoted 1 and 0.

However planning is the process and it has formulated in between 1 and 0 . The deductive knowledge of Boolean planning has distrusted on urban society with stochastic actions. The figure 1 has simulated spatial planning with intervention of Artificial Intelligence (Al) in the sphere of urban planning to emphasize on importance of tactics in controlling the stochastic world. There is a Robotic motion planning that explicitly considers actions (Tactics) to control probable uncertainties, avoid collisions and successfully reaching a goal. To reduce system failures, Markov decision process formulates dynamic planning to optimize Robotic motion in the selected path to achieve its goals.


Figure 1: From an initial configuration (solid square) to a goal (open circle) - Source: (Alterovitz, 2007)

The remainder of paper is organized as follows; section 2 explanation on non-linear world and the world of cause and effect to declare necessity of dynamic planning, section 3 discussion on planning re-cognition names "empowered planning", "synergic phenomena" and "strategic and tactic imitation", section 4 discusses result and future work.

## II. Planning as Temporal Reasoning; Necessity of Dynamic Planning

We have invited attentions on modeling dynamic planning rather static traditional planning due
to realities of the non-linear real world. Traditional conventional planning has been a model of planning with certain goals whereas in a-changing world, witnessing planning with certain goals has not been the possible phenomenon. The linear world and the perception of cause and effect is simply a trick of the mind to create the illusion of predictability and control. Thereby, tactical spatial planning which has characterized as a short range planning emphasizing on the current operations of various parts of the spatial complex and non-linear system has not been ignorable anymore. Short range has defined as a period of time extending about one year or less in the future. Figure 2 has discussed on the time non-linearity amid events in spatial system. Inhabitants often claim that it is easy to see how the events unfolded with hindsight in linear time. However, it is often possible to understand events reasons with foresight. Additionally events can happen simultaneously instead of the linear pre-determined perspectives and the spatial layouts have been witnessed hidden and complex non-linear causes and effects.


Figure 2 : Dashed line: linear time, filled points: events in spatial layout- Source: by Author

In the real-world framework, there is not any linear reality. The complex spatial system has embedded with pluralities of actions by which the urban system has directed to complexities of causes and effects. The spatial temporal actions have taken place on self-emergencies and planned bottom-up activities. Figure 3 has explained realities of real world where actions have made influences upon each-others and created complex non-linear systems.


Figure 3 : Cause and effects dynamics of bottom-up spatial activities- Source: by Author
The bottom-up planned actions are spatial relationship between space and time has been tactics/reactions by which time and space co-existence have implemented. And tempo-spatial co-existence policies can immediately re-act urban society's upheavals. In figure 4, the incremental tempo-spatial formulated in the most diverse planning theories and has fascinated mankind from the beginning until the conventional planning strategy by which real-time tends to zero.


Figure 4 : Co-existence of time and space in tactical planning; 1-The efficiency: is the operational level of planning via asking" how can we best deploy and control resources?" 2- The effectiveness: is the tactical level of planning via asking" how can we best organize ourselves to reach success?" 3-The competitiveness: is he strategic level of planning via asking" what are our aims and what are marketable to do global competitiveness?" - Source: by Author

The next section of the paper has discussed on re-formulating planning named "empowered planning" through integrate tactical spatial practices in conventional classical planning to configure planning system.

## iil. Plan Recognition; Empowered <br> Planning

Strategic planning has emphasized on the analyzing future and tactical planning has functioned on controlling everyday life. Despite their differences, tactical and strategic planning is internally related. System without strategy only based on tactics leads to
shooting in dark. Sun Tzu innovation on "The Art of War" has taught the strategy such the timeless lesson as humans' nature. Strategy and tactics have depended on each other. Goldratt has defined "Strategy" as, simply, the answer to the question: "What for?" (The answer is the objective of a proposed change). "Tactic" is defined as, simply, the answer to the question "How to?" (The answer is the details of the proposed change). From these definitions, it is clear that every Strategy (What for?) should have an associated Tactic (How to?) and therefore Strategy and Tactic must always exist in "pairs" and must exist at every level of the organization (Figure 5).


Figure 5 : Every level of organization such Municipal level has composed of strategic vision and relational spatial tactics which has connections with the upper plans such regional levels orderly - Source: by Author

Tactical planning should focus on what to do in short term to contribute the spatial organization achieving the long term objectives determined by strategic planning. The short term tactical policies are more common in the political competitive sphere where citizens involvement in public sphere management. In the area of planning, there has been considerable debate about whether top-down or bottom-up planning is best spatial practice. However the empowered planning model has combined and made balance between long term and short term planning. Foucault's (1991) notion of 'govern mentality' has also composed of active tactics and strategies by governments and agents. Strategy without tactics is the slowest route to victory. Tactics without strategy is the noise before defeat.

The conventional instrumental planning has modeled relied on the rational calculation is also the strategic challenge apart from tactical policies. However, great upheavals in uncertain world have led to the lack of trust on rational calculation empty of spatial tactics to control the stochastic actions. To support the argument,

Friedman (1987) said that municipal level of the spatial development cannot justly answer local spatial dynamics via upper policies strategies, but it has to consider the local bottom-up knowledge and plan spatial tactics to reach strategic goals of the locality. In planning literature, it is time to integrate tactical spatial practices in conventional strategic planning. In this sense, the planning organization has simulated the novel "process policy" on spatial planning which Habermas (1995) has put forward that on "communicative action theory". Generally, "strategy" is really at the highest level of spatial systems by which the directions of all activities are dictated and "tactics" are lower down in spatial systems and define the activities that are needed to implement the Strategy, then where does "Strategy" end in which do "Tactics" begin.

The figure 6 has represented differences on strategic and tactic perspectives in detail by which the paper next argument has clarified via declaring difficulties on imitating spatial strategies rather sociospatial tactics.

| Strategy | Tactic |
| :--- | :--- |
| Future (Longer Term ) | Now (in the moment) |
| Preparing and Planning | Doing activities |
| A journey | A trip |
| Broad perspective | Narrow perspective |
| A purpose | A task |
| Anticipation | Reaction |
| Risk | Caution |
| Important | urgent |
| Difficult to copy | Easy to copy |
| Large scale | Small scale |

Figure 6 : Strategic and tactics differences-source: by Author

## a) Strategic and tactical imitations

Imitation strategy is the strategy that mimics the strategy of other territories. Territories have performed this kind of the imitation strategy to attract global capitals. This strategy is an illegal and unethical activity on condition that territories inner dynamics have refuted (Figure 7).


Figure 7 : The bottom-up ethical process- Source:
(Nielsen, 1994)

The more interesting argument is duplicating spatial tactics without paying attention on territories authenticities and dynamic bottom-up knowledge. In reality, tactics vary with circumstances and, especially, technology. Alan Emrich says, "If I were to teach you how to be a soldier during the American Revolution, you would learn how to form and maneuver in lines, perform the 27 steps in loading and firing a musket, and how to ride and tend to a horse. Naturally, yesterday's tactics won't win today's wars - but yesterday's strategies still win today's wars... and will win them tomorrow and into the future. Therefore, strategy and tactics require a different focus." After debating on necessity of strategic and tactical planning authenticity to dismantle empowered spatial planning, it will be more interesting to concentrate on in what manner spatial tactics have integrated in urban planning through "synergic planning".

## b) Synergic Planning

Synergy comes from the Greek word synergia, meaning joint work and cooperative action. Synergy is when the result is greater than the sum of the parts. Synergy has been created when things work in concert together to create an outcome that is in some way of more value than the total of what the individual inputs is. The synergic phenomena have supported "What to change, but more importantly, what not to change and especially How to implement the changes and Why." Empower planning has ethically planned socio-physicalspatial changes owing to making synergy amid different spatial localities of urban systems such synergy in between two localities strategies and tactics towards planning overlapping to reach mega goals of a territory (Figure 8).
have combined and utilized in balance. This research has tries to introduce a new mode of intervention in planning since the empowered planning is the subordinate system theories framework. System theories focus on complexity and system inter-dependencies. The followers of the system theory in the field of sociology also give light to what is happening in the socio-spatial context in cities. Among them, Nikolas Luhmann argues the significance of the continuity of social processes and inter-activities among parts in such processes.

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## A Proposed SAT Algorithm

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Abstract - This paper reviews existing SAT algorithms and proposes a new algorithm that solves the SAT problem. The proposed algorithm differs from existing algorithms in several aspects. First, the proposed algorithm does not do any backtracking during the searching process that usually consumes significant time as it is the case with other algorithms. Secondly, the searching process in the proposed algorithm is simple, easy to implement, and each step is determined instantly unlike other algorithms where decisions are made based on some heuristics or random decisions. For clauses with three literals, the upper bound for the proposed algorithm is $\mathrm{O}\left(1.8171^{\mathrm{n}}\right)$. While some researchers reported better upper bounds than this, those upper bounds depend on the nature of the clauses while our upper bound is independent of the nature of the propositional formula.

Keywords : propositional satisfiability, NP-complete, complexity, complete algorithms.
GJCST-D Classification: F.2.1

Strictly as per the compliance and regulations of:


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# A Proposed SAT Algorithm 

Bagais A. ${ }^{\alpha}$, Junaidu S. B. ${ }^{\sigma}$ \& Abdullahi M. ${ }^{\rho}$

Abstract - This paper reviews existing SAT algorithms and proposes a new algorithm that solves the SAT problem. The proposed algorithm differs from existing algorithms in several aspects. First, the proposed algorithm does not do any backtracking during the searching process that usually consumes significant time as it is the case with other algorithms. Secondly, the searching process in the proposed algorithm is simple, easy to implement, and each step is determined instantly unlike other algorithms where decisions are made based on some heuristics or random decisions. For clauses with three literals, the upper bound for the proposed algorithm is $\mathrm{O}\left(1.8171^{n}\right)$. While some researchers reported better upper bounds than this, those upper bounds depend on the nature of the clauses while our upper bound is independent of the nature of the propositional formula.
Keywords : propositional satisfiability, NP-complete, complexity, complete algorithms.

## I. Introduction

Propositional satisfiability (SAT) is one of the classical problems in Computer Science. The importance of SAT comes from the fact that a large class of real-world problems can be expressed in terms of a SAT instance and that it was the first problem proven to be NP-Complete (Cook, 1971). The SAT problem has a wide range of practical real world applications (Barbour, 1992; Crawford \& Baker, 1994; Devadas, 1989; Kauts \& Selman, 1992; Larrabee, 1992). Many algorithms, categorized into complete and incomplete algorithms, were proposed to solve this problem efficiently over the last decades.

Complete algorithms can state whether a SAT instance is satisfiable giving the satisfying assignments
or unsatisfiable giving a 'no' answer. Incomplete algorithms can only give an answer of 'yes' for satisfiable SAT instances only but cannot give an answer for unsatisfiable instances.

This paper proposes a new complete algorithm that differs from the ones in the literature in the following aspects:

- No backtracking during the searching process that usually consumes significant amount of time.
- Has a simple, deterministic and easy to implement search process, unlike other algorithms where decisions are either made randomly or based on some heuristics.

The remainder of the paper is structured as follows. Section 2 describes the proposed algorithm with the aid of an example. Section 3 captures the algorithm in pseudo code while Section 4 presents the complexity analysis of the algorithm. We present related work in Section 5. Sections 6 and 7 summarize and provide references, respectively.

## II. Illustrating the Proposed Algorithm

Unlike other algorithms that make a decision on a single value (true/false) for a variable $x$, the proposed algorithms takes into consideration all satisfying assignments for a clause C and use them for the next clauses so that backtracking is avoided.

$$
\text { Consider the following formula: } F=\left(x_{1} \vee x_{3} \vee \bar{x}_{4}\right) \wedge\left(\bar{x}_{1} \vee x_{2} \vee \bar{x}_{5}\right) \wedge\left(x_{2} \vee \bar{x}_{3} \vee x_{4}\right)
$$

The first clause can be satisfied by any of the following assignments $x_{1}=$ true, $x_{3}=$ true, $x_{4}=$ false. The algorithm tries to find assignments for all variables in clause while preserving at least one of the given assignments for $x_{1}, x_{3}$, or $x_{4}$ in the first clause.

In general, the process starts from the first clause $C_{1}$ and produces the set of assignments that satisfy $C_{1}$ which obviously are the literals in that clause.

[^3]If the clause has $k$ literals, then $k$ assignments can satisfy it (as in the previous formula, the first clause has three assignments). In the next step, the set of assignments that satisfy the set of previous clause(s) are checked with all the literals of the next clause. The process continues until all the clauses in the formula are covered, after which the resulting set of assignments each satisfies the formula.

When a set of assignments from previous clause(s) is checked with the literals of the current clause, each literal may agree, disagree or be neutral to the assignment. A literal agrees with an assignment when the assignment includes the literal. A literal disagrees with an assignment when the assignment includes a negation of the literal. A literal is neutral to an
assignment when the assignment neither agrees nor disagrees with the literal.
$F=\left(x_{1} \vee x_{3} \vee \bar{x}_{4}\right) \wedge\left(\bar{x}_{1} \vee x_{2} \vee \bar{x}_{5}\right) \wedge\left(x_{2} \vee \bar{x}_{3} \vee x_{4}\right)$


Figure 1 : Assignment Production

In the first step, the satisfying assignments for the first clause are its literals. The assignments produced for the first clause are shown in the top-left rectangle in Figure 1. Each of these assignments is checked with the literals of the second clause, $\left(\bar{x}_{1} \vee x_{2} \vee \bar{x}_{5}\right)$. The assignment of $x_{1}$ disagrees with the first literal of the second clause, $\bar{x}_{1}$ resulting in no assignment produced. The same assignment, $x_{1}$ is checked with the second literal, $x_{2}$. Since this literal is neutral to $x_{1}$, a new assignment is produced by combining $x_{1}$ and $x_{2}$, as shown in the middle rectangle in Figure 1. Next, $x_{1}$ is checked with $\bar{x}_{5}$, giving $x_{1} \bar{x}_{5}$, since $\bar{x}_{5}$ is neutral to $x_{1}$. Similarly, the assignments $x_{3}$ and $\bar{x}_{4}$ are checked with the literals of the second clause leading to six additional assignments as shown in the middle rectangle of Figure 1. To complete this example, the literals of the third clause are checked with these eight assignments producing the 18 new assignments in the right-most rectangle of Figure 1. Note that each of these 18 assignments satisfies the given formula.

Note that when an assignment agrees with the clause in consideration, the process might produce shorthand for $x_{1} \vee x_{2}$ etc. We will illustrate this with the pair of clauses:

$$
\begin{aligned}
& \left(x_{1} \vee x_{2} \vee x_{3}\right) \\
& \left(x_{1} \vee x_{4} \vee x_{5}\right)
\end{aligned}
$$

The satisfying assignments for this pair of clauses are:

$$
\begin{array}{lll}
x_{1} x_{1}\left(\text { or } x_{1}\right) & x_{2} x_{1} & x_{3} x_{1} \\
x_{1} x_{4} & x_{2} x_{4} & x_{3} x_{4} \\
x_{1} x_{5} & x_{2} x_{5} & x_{3} x_{5}
\end{array}
$$

From this group, it can be seen that the assignments $\left\{x_{1} x_{4}, x_{1} x_{5}, x_{2} x_{1}, x_{3} x_{1}\right\}$ are subsumed in the first assignment $x_{1}$. This is because each of these assignments produces the same result as $x_{1}$.

Thus, these assignments can be dropped to avoid redundancy. Therefore, Figure 1 can now be redrawn without the subsumed assignments as shown in Figure 2.


Figure 2: Satisfying assignments without redundancies

Since the subsumed assignments are produced from clauses that have a literal in common, the proposed algorithm starts by extracting all clauses that do not share a literal. For a satisfiability formula with $n$ literals each clause containing exactly $k$ literals, the
minimum number of clauses in which no two clauses have a common literal is $\frac{2 n}{k}$.

$$
F=\left(x_{1} \vee x_{2} \vee \bar{x}_{4}\right) \wedge\left(\bar{x}_{1} \vee x_{3} \vee x_{4}\right) \wedge\left(x_{1} \vee \bar{x}_{3} \vee x_{6}\right) \wedge\left(x_{5} \vee \bar{x}_{2} \vee \bar{x}_{6}\right) \wedge\left(x_{4} \vee \bar{x}_{5} \vee x_{2}\right) \wedge\left(x_{3} \vee \bar{x}_{6} \vee \bar{x}_{5}\right)
$$

For example, we need at least 4 clauses to have the 12 literals in the following formula. But because of the distribution of literals, we need 5 for that purpose. Therefore, the algorithm will extract the clauses that do not have common literals. There are two advantages in doing so:

1. The algorithm will save the time to check the existence of subsumed assignments which is a process that consumes an amount of time equal to the number of assignments.
2. The time complexity of the algorithm becomes easier to prove (see Section 4).
Theorem 1
Consider a satisfiability formula with $m$ clauses each of $k$ literals. An agreement between an assignment and a literal in the $i^{\text {th }}$ clause produces at least $\left\{\begin{array}{c}2(k-1) k^{m-i} ; i=2 \\ (k-1) k^{m-i} ; 3 \leq i \leq m\end{array}\right.$ redundant assignments.

Proof: (By induction).

## Base Case

The base case is when $i=m$ and the total number of redundant assignments will be $(k-1) k^{m-m}=(k-1) k^{0}=(k-1) . \quad$ Clearly, the

## Inductive Hypothesis

Suppose the theorem holds for $i=2,3,4, \ldots, p$ for some clause $2 \leq p<m$. The total redundant assignments will be $\left\{\begin{array}{c}2(k-1) k^{m-p} ; p=2 \\ (k-1) k^{m-p} ; 3 \leq p<m\end{array}\right.$. If a literal with which an assignment agrees is in $p+1$ clause, then the total redundant assignments will be

$$
\left\{\begin{array}{l}
\frac{2(k-1) k^{m-p}}{k}=2(k-1) k^{m-p-1}=2(k-1) k^{m-(p+1)} ; i=2 \\
\frac{(k-1) k^{m-p}}{k}=(k-1) k^{m-p-1}=(k-1) k^{m-(p+1)} ; 3 \leq i \leq m
\end{array} .\right.
$$

That is, the theorem holds for $p+1$. By induction on $p$, the theorem is true for all values of $i$.

## Theorem 2

Consider a satisfiability formula with $m$ clauses each of $k$ literals. A disagreement between an assignment and a literal in the $i^{\text {th }}$ clause reduces the number of assignments by at least by $k^{m-i} ; 2 \leq i \leq m$. theorem holds for $i=m$.

Proof: (By induction)

## Base Case

The base case is when $i=m$ and the total number of assignments will be reduced by $k^{m-m}=k^{0}=1$. Clearly, the theorem holds for $i=m$.

Inductive Hypothesis
Suppose the theorem holds for $i=2,3,4, \ldots, p$ for some clause $2 \leq p<m$. The total assignments will be reduced by $k^{m-p}$. If a literal with which an assignment agrees with is in $p+1$ clause, then the total assignments will be reduced by $\frac{k^{m-p}}{k}=k^{m-p-1}=k^{m-(p+1)}$. That is, the theorem holds for $p+1$. There by induction on $p$, the theorem is true for all values of $i$.

## iII. The Proposed Algorithm Pseudocode

The most important step in any complete or incomplete SAT algorithm is the decision over the value

## The Algorithm

Input: $F[m]$; //formula with $m$ clauses
Output : A[k $\left.k^{m}\right]$; //Possible assignment satisfying m clauses.

## 1. getDistinctClauses(F[m]);


For $j=1$ to $k / / k$ is the number of literals in a clause
LIT[i][j] := disticntclauses[i];
End for
End for
3. For $i=1$ to $k$

A[i] := LIT[1][i]; //literals of the first clause(initial set of satisfying substitutions)
End for
4. For $i=2$ to disticntclauses.length;//number of distinct clauses

For $j=1$ to $k$
generateAssignment(LIT[i][j], A[], temp[]);
//A[] contains the set of satisfying substitutions from previous clauses
//temp[] contains assignments formed by combining assignments in A[] with a literal LIT[i][j]
End for
$A[]:=A[]+$ temp[];
End for

For $j=1$ to $k / / k$ is the number of literals in a clause
LIT[i][j] := nondistinctclauses[i];
End for
End for
For $i=$ distinctclauses.length to $m$
For $j=1$ to $k$
generateAssignment(LIT[i][j], A[], temp);
End for
removeSubsumedAssignments(tempassignments[], arraysubsumed[]);
A[] := A[] + temp;
End for

## 6. If A[] is empty

Output "the formula is unsatisfiable";
Else
Output the assignments in A[] as the satisfying assignments for the formula F.

```
Procedure getDistinctClauses(F[m])
Input: Formula with \(m\) clauses
Output: arrayofdistinctclauses and arrayofnondistinccaluses
distinctclauses[1] = clause[1];
n1:=0;
n2: \(=1\);
distinct \(=\) true;
for \(i=2\) to \(m\)
    for \(j=1\) to distinctclauses.length -1
        if (distinctclause[j] intersection clause[i] != empty)
                        nondistinctclauses[n1++] = clause[i];
                distinct = false;
                        break;
            Endif
        Endfor
        If (distinct \(==\) true)
            disticntclauses[n2++] = clause[i];
        Endif
Endfor
```

Procedure: removeSubsumedAssignments(tempassignments[], arraysubsumed[])
Input: list of assignments containing subsumed assignments and list of assignments subsuming the subsume assignments. Output: list of assignments without subsumed assignments.

```
n:=0;
For i = 0 to tempassignments.length - 1
    For j = 0 to arraysubsumeb.length - 1
            If (arraysubsumed[j] is not contained in tempassignents[i])
                    arrayassignments[n++] = tempassignment[i] ;
        Endfor
Endfor
Return arrayassignments[];
```

Procedure: generateAssignment(lit, A[], temp[]);
Input: a literal in a clause and a list of assignments in A[].
Output: a list of assignments stored in temp[] produced by combining lit with A[].

```
For i=1 to A.length
    If lit did not conflict with the assignment then
                Combine the lit and the assignment;
            Add the combination in temp[];
    elseif lit agrees with the assignment then
            Add the assignment in temp[];
            Add the assignment in arraysubsumed[];
        Endif
Endfor
Return temp[];
```


## IV. Time Complexity of the Algorithm

The first three steps of the algorithm take polynomial time of number of clauses. Steps four and five are clearly the main contributors to the time complexity of the whole algorithm. These two steps rely on the number of assignments generated in each iteration of the for-loop. For step four, that number is determined by the clauses in CLS and for step five, that number is determined by the end of step four. Therefore, let us start with step four.

## a) Finding number of assignments

Whenever a clause is considered in the for-loop, the number of assignments is multiplied by $k$ (in the worst case). The first clause initializes A with $k$ assignments (the literal in that clause). Then, the second clause will produce at most $k^{2}$ assignments, and the third clause may generate as $k^{3}$ assignments and so on. That means the number of the assignments is $\leq k^{m}$ where $m$ is some number of clauses. In step four, clauses in CLS could either be:

1. $\frac{2 n}{k}$ clauses (worst case).
2. or more than $\frac{2 n}{k}$ clauses (as explained in Section 2).

$$
N\left(P_{1}, P_{2}, P_{3}, \ldots, P_{n}\right)=\sum_{1 \leq i \leq n}\left|A_{i}\right|-\sum_{1 \leq i<j \leq n}\left|A_{i} \cap A_{j}\right|+\sum_{1 \leq i<j<k \leq n}\left|A_{i} \cap A_{j} \cap A_{k}\right|-\ldots+(-1)^{n+1}\left|A_{1} \cap A_{2} \cap A_{3} \cap \ldots \cap A_{n}\right|
$$

The proof of the principle can be found in (Rosen, 1999).
If $P_{i}$ is the assignment where $x_{i}$ and $\bar{x}_{i}$ appear for $i=1,2,3, \ldots, \alpha$ where $\alpha \leq n$, then the exact number of assignments for case 1 is $k^{\frac{2 n}{k}}-N\left(P_{1}, P_{2}, P_{3}, \ldots, P_{\alpha}\right)$

For any satisfiability instance, the previous quantity cannot be found. That is because unlike the example given earlier, the arrangement of variables or literals differs from one instance to another. However, there is an arrangement that will produce the highest number of variables.

## b) The upper bound

At this point, we need to prove two theorems. One that states case 1 is the worst case and the other states the arrangement that will produce the highest number of assignments.

## Theorem 3

In step 5 of the algorithm, generating assignments with the least number of clauses $\left(\frac{2 n}{k}\right)$ that include 2 n literals is the worst case.

Case 1 is the worst because if more than $\frac{2 n}{k}$ clauses are needed, then we must have repeated literals. This can be shown easily as follows: If we have $\frac{2 n}{k}+1$ clauses, then the number of literals is $\left(\frac{2 n}{k}+1\right) k$ which gives us $2 n+k$ literals. That means we have $k$ repeated literals in these clauses.

Because of the existence of repeated literals in Case 2, Case 1 will produce the maximum number of assignments (see Theorem 3).

We now determine the number of possible assignments, $A(n)$, in the worst case. If the clauses in CLS have conflicting literals, $A(n) \leq k^{m}$.

In this case, a literal in one clause will not be combined with a literal $\bar{x}_{1}$ in another clause. The number of substitutions to be eliminated is shown by Theorem 2.

To count the exact number of assignments, the principle of inclusion-exclusion is used. The principle states that the number of elements that have property 1 , property 2 , property $3, \ldots$, or property $n$ is found by the summation.

## Proof

If more than $\frac{2 n}{k}$ clauses are needed to include the $2 n$ literals then we must have literals that are repeated. If we have one additional clause, then there must be $k$ literals repeated and this will make the set of assignments to be excluded more than n. Having a repeated literal means that we have three clauses of this form: $x_{1} \vee x_{2} \vee x_{3} \quad x_{1} \vee x_{4} \vee x_{5} \quad \bar{x}_{1} \vee x_{6} \vee x_{7}$.

The two clauses that have the repeated literal $x_{1}$ will produce the unnecessary assignments. These assignments are generated when the repeated literal is combined with the $(k-1)$ literals of the other clause. This means that the assignments that include $\left\{x_{1} x_{4}, x_{1} x_{5}, x_{1} x_{2}, x_{1} x_{3}\right\}$ are unnecessary. The only useful assignment is $x_{1}$ produced from $\left(x_{1} x_{1}\right)$. This indicates that $2(k-1)$ sets of assignments should be discarded. In addition to these assignments, the two repeated literals when combined with $\bar{x}_{1}$ will produce two sets of assignments of the form $\left(x_{1} \bar{x}_{1}\right)$ that are
also discarded from the total number of assignments when we count them using the inclusion exclusion principle. Therefore, a repeated literal will result to discard $2(k-1)+1$ additional sets excluded.

Writing the inclusion exclusion series with $n$ sets plus $k(2(k-1)+1)$ sets is hard because there will be
approach to show that $2 n / k$ is the worst case is to exclude the additional sets first from the total number of assignments and compare that with the worst case. The number of assignments of the additional sets can be counted by: many possibilities for the intersection of sets. The

$$
\begin{aligned}
& A=(2(k-1)+1) * C(k, 1) * k^{\frac{2 n}{k}-1}-(2(k-1)+1)^{2} * C(k, 2) * k^{\frac{2 n}{k}-3} \\
& +(2(k-1)+1)^{3} * C(k, 3) * k^{\frac{2 n}{k}-5}-\ldots+(-1)^{k+1}(2(k-1)+1)^{k} * C(k, k) * k^{\frac{2 n}{k}-2 k+1} \\
& =\sum_{i=1}^{k}(-1)^{i+1}(2(k-1)+1)^{i} C(k, i) k^{\frac{2 n}{k}-2 i+1}
\end{aligned}
$$

Excluding this from the total assignments

$$
\begin{aligned}
& N=k^{\frac{2 n}{k}+1}-\sum_{i=1}^{k}(-1)^{i+1}(2(k-1)+1)^{i} C(k, i) k^{\frac{2 n}{k}-2 i+1} \\
& N=k^{\frac{2 n}{k}-2 k+1}\left(k^{2 k}-\sum_{i=1}^{k}(-1)^{i+1}(2(k-1)+1)^{i} C(k, i) k^{2(k-i)}\right)
\end{aligned}
$$

Evaluating $\left(k^{2 k}-\sum_{i=1}^{k}(-1)^{i+1}(2(k-1)+1)^{i} C(k, i) k^{2(k-i)}\right)$ for values of $k$ gives quantity less than $k^{2 k-1}$ and result to a number of assignments less than $k^{\frac{2 n}{k}}$ and excluding the $n$ sets of the form (v-v) from N gives a value that is less than the one in the worst case.

$$
\begin{gathered}
k^{\frac{2 n}{k}} \operatorname{excld}(\mathrm{n} \text { sets })>k^{\frac{2 n}{k}-2 k+1}\left(k^{2 k}-\sum_{i=1}^{k}(-1)^{i+1}(2(k-1)+1)^{i} C(k, i) k^{2(k-i)}\right) \quad \text { excld(n sets) because } \\
k^{\frac{2 n}{k}}>k^{\frac{2 n}{k}-2 k+1}\left(k^{2 k}-\sum_{i=1}^{k}(-1)^{i+1}(2(k-1)+1)^{i} C(k, i) k^{2(k-i)}\right)
\end{gathered}
$$

This is for one additional clause. For $i$ intersected with the maximum possible number of other additional clauses the limit of the summation is to $i k$ and this also will give the same result.

Theorem 3 tells us that step six will not generate assignments that are more than step five. This should make step 5 the dominant factor for time complexity.

## Theorem 4

For the worst case, the upper bound is $(k(k-1))^{\frac{n}{k}}$

## Proof

The inclusion-exclusion principle takes care of assignments that are counted more than once by considering the intersections between the n sets to be excluded as seen in the summation. Therefore, the least value of $N\left(P_{1}, P_{2}, P_{3}, \ldots, P_{n}\right)$ indicates the maximum possible number of assignments generated by the algorithm. This occurs when each set $x, \bar{x}$ is
sets. For example consider two clauses that has $x_{1}$ and $\bar{x}_{1}$ literals:

$$
\begin{array}{lll}
x_{1} & x_{2} & x_{3} \\
\bar{X}_{1} & & x_{4}
\end{array}
$$

The assignments that include $x_{1}$ and $\bar{x}_{1}$ can never occur with assignments that include $x_{2}$ and $\bar{x}_{2}$, $x_{3}$ and $\bar{x}_{3}, x_{4}$ and $\bar{x}_{4}, x_{5}$ and $\bar{x}_{5}$ literals. Therefore, there is no intersection between $x_{1} \bar{x}_{1}$ assignment set and 4 sets of assignments. The least intersection $(k-1)$ happens if both clauses of $x_{5}$ and $\bar{x}_{1}$ have literals of the same variables. For the previous example the two clauses should look like this
$x_{1} \vee x_{2} \vee \bar{x}_{3} \quad \bar{x}_{1} \vee \bar{x}_{2} \vee x_{3}$ to make the quantity $N\left(P_{1}, P_{2}, P_{3}, \ldots, P_{n}\right)$ the least. If this happens with all
variables, the following arrangement will produce the maximum number of assignments.

$$
x_{1} \vee x_{2} \vee \bar{x}_{3}, \bar{x}_{1} \vee \bar{x}_{2} \vee x_{3}, \bar{x}_{4} \vee \bar{x}_{5} \vee x_{6}, x_{4} \vee x_{5} \vee \bar{x}_{6}, x_{7} \vee \bar{x}_{8} \vee x_{9}, \bar{x}_{7} \vee x_{8} \vee \bar{x}_{9}, \cdots
$$

The number of assignments between clauses of conflicting literals is $k(k-1)$. Since we need $\frac{2 n}{k}$ clauses to consider $n$ variables and each 2 clauses have $k(k-1)$ assignments, then the number of assignments
will be $k(k-1)^{\frac{n}{k}}$.

## c) Related Work

Complete algorithms for SAT satisfiability problems include those algorithms that can state whether or not a SAT instance is satisfiable, giving a 'yes' answer together with a satisfying assignment or a 'no' answer as the case may be. The first complete algorithm is the Davis Putnam procedure (Davis \& Putnam, 1960). This procedure is based on resolution rule to eliminate variables one by one till the formula is satisfied. When a variable is eliminated in each iteration, all resolvents are added to the set of the clauses. This algorithm requires polynomial space. It handles CNF formulas and it is one of the efficient SAT algorithms. (Davis, Logemann, \& Loveland, 1962) Developed a divide-and-conquer algorithm that enhances on (Davis \& Putnam, 1960). This improved algorithm is the main procedure for most state-of-the-art SAT solvers today. The search space of DPLL could grow as large as $2^{n}$ which is the worst case for any complete algorithm. Due to the possibility of consuming huge amount of time, researchers have been focusing on mechanisms to reduce that and came up with more reasonable time complexities. These improvements usually come in two aspects: the decision to branch to next literal and the backtracking mechanism if a solution is not found in the chosen branch. The achievements accomplished in improving SAT algorithm in these two aspects show that the complexity could be reduced significantly.

## i. Branching Decisions

DPLL procedure chooses any literal for branching and goes down that region in the search space. The procedure will spend time searching for a solution and if it discovers that the branch is not successful, it backtracks to the other branch and continues searching. Choosing the next literal for branching more carefully will allow the algorithm to save time exploring a region where a satisfying assignment cannot be found at all and direct the searching to regions where a solution is likely to be found. In order to accomplish this, several heuristics have been proposed and the most effective ones can be found in (Bruni \& A., 2003; Freeman, 1995; Hooker \& Vinay, 1994; Jeroslow \&

Wang, 1990; Li \& Anbulagan, 1997; Moskewicz, Madigan, Zhao, Zhang, \& Malik, 2001; Pretolani, 1993).

## ii. Backtracking Mechanisms

When the algorithm fails to find an answer or an empty clause (contradiction) appears down the path of the search tree, it backtracks to a certain point and continues searching in another part of the tree. The DP backtracks to the most recently untoggled (complemented) literal and tests its complement branch. As mentioned earlier this will cost a lot of time for DP to discover that this part of the search space does not have a solution and search for a solution elsewhere. For backtracking in the DP procedure, much work has not been done as compared to branching decision. This is due to the fact that backtracking is an essential step in any algorithm to prove its completeness. Nevertheless, there are a number of proposals to improve the backtracking in the DP procedure. (Lynce \& MarquesSilva, Building State-of-The-Art SAT Solver, 2002) tested different backtracking strategies and the most effective ones can be found in (Lynce \& Marques-Silva, 2002; Stallman \& Sussman, 1977).

## iii. Upper Bounds

The improvements made in backtracking and branching heuristics are of practical interests. However, the experimental analysis of these improvements indicates that satisfiability could be solved in time less than $2^{n}$. A number of people gave lower bounds for this problem but most of them rely on a certain structure or property that exists in the formula. The following are some of the achievements made to find an upper bound that is better than the trivial one.

## a. Autarkness Principle

The first attempt to achieve a non-trivial upper bound for SAT was done by (Monien \& Speckenmeyer, 1985). They introduced the notion of autarks which are partial assignments of variables. If all clauses that include the variables in the assignment are satisfied, then that assignment is an autark. They proved that the time complexity of their algorithm is $\mathrm{O}\left(2^{n \log \alpha_{k}}\right)$.

## b. 2-clause

When dealing with 3-SAT problem, the clauses with 2 literals help in reducing the search space. Schiermeyer was the first to make use of the number of clauses with 2 literals after the resolution step is made (Schiermeyer, 1993). He said that for the next branch, a 2-clause is chosen such that it produces at least one new 2-clause in every branch that follows. With the help
of these reduced clauses, he proved an even lower bound for 3 -SAT with time complexity $\mathrm{O}\left(1.579^{n}\right)$. (Kullmann, 1999) showed that the algorithm of Schiermeyer can perform better through a new concept called blocked clauses. A clause C is blocked for a literal $l$ if every clause $C^{\prime}$ containing $l$ has also another literal that is complemented with C. By making use of these blocked clauses, Kullmann proved that the algorithm in (Schiermeyer, 1993) can have a time complexity of $\mathrm{O}\left(1.504^{n}\right)$.

## c. Saisfiability Coding Lemma

This lemma is based on isolated assignments which are satisfying assignments to the formula where a change of one value of any variable will make it dissatisfying. The lemma states that such assignments can be encoded in a message of length ( $n-\frac{n}{k}$ ) and this is where the complexity comes from. (Marques-Silva \& Sakallah, 1999) shows that through satisfiability coding lemma their algorithm finds an answer in $\mathrm{O}\left(2^{n-\frac{n}{2 k}}\right)$.
d. P-literal
(Hirsch, Two New Upper Bounds for SAT, 1998) presented two algorithms that rely on P-literal notion. This notion says that if a literal occurs exactly 2 times in the clause set and at least 3 times in its negation form, then it is P -literal. He used these special literals to simplify the formula and came up with two algorithms with time complexity $\mathrm{O}\left(2^{0.3089 m}\right)$ and $\mathrm{O}\left(2^{0.10537 \mathrm{~L}}\right)$ respectively where m is the number of clauses and $L$ is the length of the formula. An improvement was made to the second algorithm in (Hirsch, 2000) to become O( $\left.2^{0.10299 \mathrm{~L}}\right)$.
e. Covering Codes
(Danstin, et al., 2002) proposed a deterministic algorithm that is based on covering codes. This algorithm can be seen as a derandomization of (Schoning, 1999) algorithm that uses random walk model. The search space is divided into group of assignments say balls of some radius r. Each group or ball represents some assignment $a$ and all assignments that differ with it in $r$ variables. The algorithm checks in each ball if there is a satisfying assignment and if there is none in any ball then the formula is unsatisfied. The authors of (Danstin, et al., 2002) showed that the time complexity of this $\mathrm{O}\left(2-\frac{2}{k+1}\right)$ for k-SAT. For 3 -SAT, they managed to further improve the algorithm by identifying useless branching and reduce the search space to come up with running time $\mathrm{O}\left(1.481^{n}\right)$.

## V. Conclusion and Future Work

The proposed does not require the clauses or the formula to have any specific structure to achieve a competitive upper bound which is a significant advantage over the existing algorithms in the literature where they derive their time complexity based on a property that must exist in the formula. The algorithm gives a new insight towards solving SAT. Most of the other algorithms are based on the classical rule of splitting the search space into regions and search for a solution in each one. The new perspective of the algorithm has the potential to design further effective SAT algorithms that outperforms the existing ones in theory and practice.

The implementation of the proposed algorithm will be considered in future work. The algorithm proposed here can also be improved. The time complexity of the proposed algorithm is based on preprocessing of clauses in the formula. This arrangement is so unlikely to exist in all clauses considered. That means that there exists a tighter upper bound for the algorithm but to achieve that the order in which clauses are considered should be more intelligent. To show that such an upper bound exists, many cases have to be covered and counted. Parallelisation of the proposed algorithm is also a potential future work.

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# Efficient V-B Block Designs for CDC Method 4 

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Summary - Some optimal incomplete block designs for complete diallel cross method 4 are known in literature. These designs require several replications for each cross and thus consume more resources such as experimental units, experimental material, time etc. So, there is a need to evolve designs which require minimum possible replications of parental lines. In this paper a method of construction of these designs is proposed by using mutually orthogonal Latin squares. These designs are connected for cross effects and perform well when compared to connected and not connected optimal designs reported by Dey and Midha (1996), Chai and Mukerjee (1999) and Gupta and Kageyama (1994), respectively.

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GJCST-D Classification: F.2.m

Strictly as per the compliance and regulations of:


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# Efficient V-B Block Designs for CDC Method 4 

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Summary - Some optimal incomplete block designs for complete diallel cross method 4 are known in literature. These designs require several replications for each cross and thus consume more resources such as experimental units, experimental material, time etc. So, there is a need to evolve designs which require minimum possible replications of parental lines. In this paper a method of construction of these designs is proposed by using mutually orthogonal Latin squares. These designs are connected for cross effects and perform well when compared to connected and not connected optimal designs reported by Dey and Midha (1996), Chai and Mukerjee (1999) and Gupta and Kageyama (1994), respectively.

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## I. Introduction

0rthogonal Latin squares are used for construction of Graeco Latin square, balanced incomplete block designs and square lattice designs. A set of $p-1$ orthogonal Latin square of side $p$ can always be constructed if $p$ is a positive prime or power of a positive prime.

If $p=4 t+2$ and $\mathrm{t}>1$, then there exits pairs of mutually orthogonal Latin squares of order p (Bose, Shrikhande and Parker (1960)). From a practical view point, mutually orthogonal Latin squares are important and an exhaustive list of these squares is available in Fisher and Yates (1963). In this paper we use mutually orthogonal Latin squares in construction of mating designs for the diallel cross method 4 referred to Griffing (1956).

A diallel cross is a type of mating design used in plant breeding and animal breeding to study the genetic properties and potential of inbred lines or individuals. Let $p$ denote the number of lines and let a cross between lines $i$ and $j$ be denoted by $i \times j$, where $i<j=0,1, \ldots, p-1$ and $p(p-1) / 2$ possible crosses. Among the four types of diallel discussed by Griffing (1956), method 4 is the most commonly used diallel in plant breeding. This type of diallel crossing includes the genotypes of one set of $\mathrm{F}_{1}{ }^{\mathrm{S}}$ means of the type $(i \times j)=$ $(j \times j)$, but neither the parents nor the reciprocals with all possible $\mathrm{v}=p(p-1) / 2$ crosses. This is sometimes referred to as the modified diallel. We shall refer to it as a complete diallel cross (CDC).

The problem of finding optimal mating designs for complete diallel cross experiments has received

[^5]attention in recent years; see Gupta and Kageyama (1994), Dey and Midha (1996) and Chai and Mukerjee (1999). Most of the results on optimal block designs for diallel crosses have been derived for the general combining ability (gca) under the assumptions that the model does not include parameters representing the specific combining ability (sca) Gupta and Kageyama (1994) and Dey and Midha (1996) but with few exceptions Chai and Mukerjee (1999) and Choi et al. (2002). The designs of these authors can be used to estimate specific combining ability (sca) but they demand more resources in terms of experimental units and experimental material. In such a situation there is need for designs which require minimum possible number of experimental units in conducting CDC experiments and are equally efficient in comparison to optimal block designs and randomized block designs when the model, in addition to the block effects and general combining ability, includes specific combining ability.

In the present paper we are proposing efficient variance balanced incomplete block designs for CDC experiments through mutually orthogonal Latin squares under the assumption that the model includes the parameter of specific combining ability.

## iI. Method of Design Construction

It is known that when $p$ is a prime positive integer or a power of prime positive integer, it is possible to construct ( $p-1$ ) orthogonal Latin squares in such a way that they differ only in a cyclical interchange of the rows from $2^{\text {nd }}$ to $\mathrm{p}^{\text {th }}$. Such squares are taken for the construction of incomplete block designs for diallel crosses. For $p=6$, such squares cannot be constructed.

Assume that there are $p$ inbred lines and it is desired to find an incomplete block design for a mating design involving $p(p-1) / 2$ crosses. Out of $(p-1)$ mutually orthogonal Latin square (MOLS), consider any two MOLS of semi-standard form of order $p$ and superimposed one square over the other. We obtain one Graeco Latin square in which each cell contains ordered pairs of integers ( $i, j$ ) taking values from 0 to $p-1$. These ordered pairs of integers occur once in a square. From Graeco Latin square remove the pairs of the type with $i$ $=j$ and considering other ordered pairs of integers as crosses between lines $i$ and $j$ and the columns as blocks. By doing so we get an incomplete block design d for diallel cross experiment method 4 with parameters $\mathrm{v}=p(p-1) / 2, \mathrm{~b}=p, \mathrm{k}=p-1$, and $\mathrm{r}=2$. The total
number of experimental units to be allotted to $\mathrm{v}=p$ ( $p$ $1) / 2$ is $n=p(p-1)$. Henceforth $d(v, b, k)$ will denote the class of all block designs with $v$ treatments, $b$ blocks and block size k .

## Example1:-

Let us consider the mating design for CDC experiment method 4 for $p=5$ parents. Consider two mutually orthogonal Latin squares $L_{1}$ and $L_{2}$ of semistandard form of order 5 . Superimposing one over the other square we get Graeco Latin square.

| 0 | $\mathrm{~L}_{1}$ | 2 |  |  |  |  |  |  |  |  |
| :--- | :--- | :--- | :--- | :--- | :--- | :--- | :--- | :--- | :--- | :--- |
| 1 | 2 | 3 | 3 | 4 |  | 0 | 1 | 2 | 3 | 4 |
| 2 | 3 | 4 | 4 | 0 |  | 2 | 3 | 4 | 0 | 1 |
| 3 | 4 | 0 | 1 | 2 |  | 4 | 0 | 1 | 2 | 3 |
| 4 | 0 | 1 | 2 | 3 |  | 3 | 2 | 3 | 4 | 0 |

After superimposition $L_{2}$ over $L_{1}$ and removing cross of the type $\mathrm{i}=\mathrm{j}$ and considering columns as blocks, we obtain design d as given below:

Design d

| $\mathrm{B}_{1}$ | $\mathrm{~B}_{2}$ | $\mathrm{~B}_{3}$ | $\mathrm{~B}_{4}$ | $\mathrm{~B}_{5}$ |
| :---: | :---: | :---: | :---: | :---: |
| $1 \times 2$ | $2 \times 3$ | $3 \times 4$ | $4 \times 0$ | $0 \times 1$ |
| $2 \times 4$ | $3 \times 0$ | $4 \times 1$ | $0 \times 2$ | $1 \times 3$ |
| $3 \times 1$ | $4 \times 2$ | $0 \times 3$ | $1 \times 4$ | $2 \times 0$ |
| $4 \times 3$ | $0 \times 4$ | $1 \times 0$ | $2 \times 1$ | $3 \times 2$ |

iII. Analysis

For the analysis of data obtained from design d we will follow Singh and Hinkelmann- (1998) two stage procedures for estimating gca and sca effects. The first stage is to consider the proposed designs to estimate cross effects, say,

$$
\boldsymbol{\tau}=\left(\tau_{01}, \tau_{02}, \ldots, \tau_{(p-2)(p-1) / 2}\right) \text { for design } d \text { by the }
$$

following model.

$$
\begin{equation*}
\mathbf{y}=\mu \mathbf{1}+\mathbf{X} \boldsymbol{\tau}+\mathbf{D} \boldsymbol{\beta}+\mathbf{e} \tag{3.1}
\end{equation*}
$$

Where y is an $\mathrm{n} \times 1$ vector of observations, 1 is the $n \times 1$ vector of ones, $X$ is the $n \times v$ design matrix for treatments and D is an $\mathrm{n} \times \mathrm{b}$ design matrix for blocks, that is, the $(h, u)^{\text {th }}\left((h,)^{\text {th }}\right)$ element of X (respectively, of D ) is 1 if the $l^{\text {h }}$ observation pertains to the $u^{\text {h }}$ cross ( to $l^{h}$ block), and is zero otherwise ( $h=1, \ldots, h ; u=1, \ldots, v$; and $1, \ldots, b), \mu$ is a general mean, $\tau$ is a $v \times 1$ vector of treatment parameters, $\boldsymbol{\beta}$ is $\mathrm{ab} \times 1$ vector of block parameters and e is an $\mathrm{n} \times 1$ vector of residuals. It is assumed that vector $\boldsymbol{\beta}$ is fixed and e is normally distributed with $E(e)=0, V(e)=\sigma^{2} I$ and $\operatorname{Cov}\left(\boldsymbol{\beta}, e^{\prime}\right)$ $=(0)$, , where I is the identity matrix of conformable

Following Tocher (1952), Raghavarao (1971) and Dey (1986), the least square method for the analysis of a proposed designs leads to the following reduced normal equations for the model (3.1).

$$
\begin{equation*}
\mathrm{C}_{\mathrm{d}} \mathbf{\tau}=\mathrm{Q}_{\mathrm{d}} \tag{3.2}
\end{equation*}
$$

Where $C_{d}=r^{\delta}-\mathrm{Nk}^{-1} \mathrm{~N}^{\prime}$ and $\mathrm{Q}_{\mathrm{d}}=\left(\mathrm{Q}_{1 \mathrm{~d}}, \ldots\right.$, $\left.Q_{\mathrm{vd}}\right)=\mathrm{T}-\mathrm{Nk}^{-5} \mathrm{~B}$

In the above expressions above $\mathrm{r}^{\delta}$ and $\mathrm{k}^{\delta}$ are diagonal matrices of order $v \times v$ and $b \times b$ with elements 2 and $p$, respectively of design d. $N=X^{\prime} D$ is the $v \times b$ incidence matrix of the design $d ; T=X^{\prime} y$ and $B=D^{\prime} y$ are vector of cross totals and block totals of order $\mathrm{v} \times 1$ and $\mathrm{b} \times 1$ for design d , respectively.
Hence a solutions to (3.2) is given by

$$
\begin{equation*}
\hat{\boldsymbol{\tau}}=\mathrm{C}_{\mathrm{d}}^{-} \mathrm{Q}_{\mathrm{d}} \tag{3.3}
\end{equation*}
$$

Where $\mathrm{C}_{\mathrm{d}}{ }^{-}$is a generalized inverses of $\mathrm{C}_{\mathrm{d}}$ with property $\mathrm{C} \mathrm{C}^{-} \mathrm{C}=\mathrm{C}$. The sum of squares due to crosses are $Q^{\prime}{ }_{d} C_{d}{ }^{-} Q_{d}$ with degrees of freedom (d.f.) $=$ rank $\left(\mathrm{C}_{\mathrm{d}}\right)$ for design d and expectation and variance $Q_{d}$ is as

$$
\begin{equation*}
E\left(Q_{d}\right)=C_{d} \mathbf{\tau} \text { and } V\left(Q_{d}\right)=\sigma^{2} C_{d} \tag{3.4}
\end{equation*}
$$

Now we will utilize the above equations to estimate the genetic parameters in the proposed design. The second stage is to utilize the fact that the cross effects can be expressed in terms of gca and sca effects. So we can write

$$
\begin{equation*}
\boldsymbol{\tau}_{i j}=g_{i}+g_{j}+s_{i j} \tag{3.5}
\end{equation*}
$$

Where $\mathrm{g}_{\mathrm{i}}\left(\mathrm{g}_{\mathrm{j}}\right)$ is the gca for the $\mathrm{i}^{\text {th }}\left(\mathrm{j}^{\text {th }}\right)$ parent, $\mathrm{s}_{\mathrm{i}} \mathrm{j}$ $\left(s_{i j}=s_{j}\right)$ is the sca for the cross between the $j^{\mathrm{j}}$ and the $j^{\text {th }}$ parent $(i<j=0,1, \ldots, p-1)$. In matrix notation equation (3.5) can be written as

$$
\begin{equation*}
\tau=Z \mathrm{~g}+\mathrm{s} \tag{3.6}
\end{equation*}
$$

Where $Z=\left(z_{u i}\right)(u=1,2, \ldots, n: i=0,1, \ldots$, $p-1$ ) is the cross and gca relation matrix.
$z_{u i}=2$, if the $u^{\text {th }}$ cross has both parents i.
$=1$, if the $U^{\text {th }}$ cross has only one parent $i$.
$=0$, otherwise.
Following the approach used in Kempthorne and Curnow (1961), equation (3.2) can then be written as

$$
\begin{gather*}
C_{d} \tau=C_{d} Z g+C_{d} s \\
\text { or } E\left(Q_{d}\right)=C_{d} Z g+C_{d} s \tag{3.7}
\end{gather*}
$$

Since the matrix $C$ is singular, we use the unified theory of least square due to Rao (1973). So we get estimator of g as order.

$$
\begin{equation*}
\hat{\mathbf{g}}=\left(Z^{\prime} C_{d} C_{d}{ }^{-} C_{d} Z\right)^{-} Z^{\prime} Q_{d}=\left(Z^{\prime} C_{d} Z\right)^{-} Z^{\prime} Q_{d} \tag{3.8}
\end{equation*}
$$

Here the matrix $\left(Z^{\prime} \mathbf{C}_{\mathrm{d}} \mathbf{Z}\right)=2 p(p-3) /(p-1)\left[1_{p}-\frac{1}{p} 1 p 1^{\prime} p\right]$.

So $\hat{\mathbf{g}}=\left(Z^{\prime} C_{d} Z\right)^{-} Z^{\prime} C_{d} \mathbf{\tau}$
Hence $\hat{\mathbf{g}}=\mathrm{H}_{1} \mathbf{T}$, where $\mathrm{H}_{1}=\left(\mathbf{Z}^{\prime} \mathrm{C}_{\mathrm{d}} \mathbf{Z}\right)^{-} \mathbf{Z}^{\prime} \mathrm{C}_{\mathrm{d}}$
Now $\operatorname{Cov}(\hat{\mathbf{g}})=\mathrm{H}_{1} \mathrm{C}_{\mathrm{d}} \mathrm{H}^{\prime}{ }_{1} \sigma 2=\sigma^{2}(p-1) / 2 p(p-3) \mathrm{I}_{p}$

Since the covariance matrix of $\hat{\mathbf{g}}$ is a constant times the identity matrix, therefore the proposed design d is variance-balanced for general combining ability effects. We thus have the following results.
always exist variance- balanced incomplete block design for CDC experiment method 4 .

Now substituting the estimate of g in equation (3.6), we obtain the estimator of $\boldsymbol{s}$.

## Theorem

For a positive prime $p>3$, if there exits a mutually orthogonal Latin square of order $p$, then there

$$
\begin{align*}
\hat{\mathbf{s}} & =\left(C_{\mathrm{d}}^{-}-(p-1) / 2 p(p-3) Z Z^{\prime}\right) Q_{d}=\left(C_{d}^{-}-(p-1) / 2 p(p-3) Z Z^{\prime}\right) C_{d} \tau \\
& =H_{2} \tau \tag{3.11}
\end{align*}
$$

Where

$$
\begin{align*}
& \mathrm{H}_{2}=\left(\mathrm{C}_{\mathrm{d}}{ }^{-}-(p-1) / 2 p(p-3) \mathbf{Z} \mathbf{Z}^{\prime}\right) \mathrm{C}_{\mathrm{d}} \\
& \operatorname{Var}(\hat{\mathbf{s}})=\mathrm{H}_{2} \mathrm{C}_{\mathrm{d}} \mathrm{H}^{\prime}{ }_{2} \sigma^{2} \tag{3.12}
\end{align*}
$$

Since $H_{1} 1_{v}=0, H_{2} 1_{v}=0, H_{1} H_{2}{ }^{\prime}=0$, rank $\left(H_{1}\right)=p-1$ and rank $\left(H_{2}\right)=v-p$.

It follows that g and s represented by treatment contrasts that carry $p-1$ and $v-p$ degrees of freedom respectively and that contrasts representing g are orthogonal to those representing $\mathbf{s}$. It means the proposed design d allows for gca and sca effects to be estimated independently.

The sum of squares due to gca and sca for d are given by

$$
\begin{gather*}
S S(\mathrm{gca})=\mathrm{Q}^{\prime}{ }_{\mathrm{d}} Z\left(Z^{\prime} \mathrm{C}_{\mathrm{d}} Z\right)^{-} Z^{\prime} \mathrm{Q}_{\mathrm{d}}  \tag{3.13}\\
\mathrm{SS}(\mathrm{sca})=\mathrm{Q}_{\mathrm{d}}{ }^{\prime}\left(\mathrm{C}_{\mathrm{d}}{ }^{-}-(p-1) / 2 p(p-3) Z^{\prime}\right) \mathrm{Q}_{\mathrm{d}} \tag{3.14}
\end{gather*}
$$

The ANOVA is then given in Table 1.
Table 1 : Analysis of variance for design d

| Source of variation | Degrees of Freedom | Sum of squares |
| :---: | :---: | :---: |
| Block | $p-1$ | $\mathrm{B}^{\prime} \mathrm{B} / \mathrm{p}-\mathrm{G}^{2} / p(p-1)$ |
| Crosses (adjusted for blocks) | rank ( $\mathrm{C}_{\mathrm{d}}$ ) | Q ${ }_{d} \mathrm{C}_{\mathrm{d}}{ }^{-} \mathrm{Q}_{\mathrm{d}}$ |
| gca | rank ( $\mathrm{H}_{1}$ ) | $\mathrm{Q}^{\prime}{ }_{\mathrm{d}} \mathbf{Z}\left(\mathrm{Z}^{\prime} \mathrm{C}_{\mathrm{d}} \mathbf{Z}\right)^{-} \mathrm{Z}^{\prime} \mathrm{Q}_{\mathrm{d}}$ |
| sca | rank ( $\mathrm{H}_{2}$ ) | $\mathrm{Q}_{\mathrm{d}}{ }^{\prime}\left(\mathrm{C}_{\mathrm{d}}{ }^{-}-(p-1) / 2 p(p-3) Z \mathrm{Z}^{\prime}\right) \mathrm{Q}_{\mathrm{d}}$ |
| Residual | $\begin{gathered} (\mathrm{n}-1)-\operatorname{rank}\left(\mathrm{C}_{\mathrm{d}}\right)-\operatorname{rank} \\ \left(\mathrm{H}_{1)}-\operatorname{rank}\left(\mathrm{H}_{2}\right)\right. \end{gathered}$ | $\begin{gathered} \mathrm{y}^{\prime} \mathrm{y}-\mathrm{G}^{2} / p(p-1)-\mathrm{B}^{\prime} \mathrm{B} / \mathrm{p}- \\ \mathrm{Q}^{\prime}{ }_{\mathrm{d}} \mathrm{C}_{\mathrm{d}}^{-} \mathrm{Q}_{\mathrm{d}} \end{gathered}$ |
| Total | n -1 | $\mathrm{y}^{\prime} \mathrm{y}-\mathrm{G}^{2} / p(p-1)$ |

## IV. Efficiency Factor

If instead of the proposed design d , one adopts a randomized complete block design with 2 blocks and each block contains $p(p-1) / 2$ crosses, the $\mathrm{C}_{\mathrm{R}^{-}}$matrix can easily shown to be

$$
\begin{equation*}
\mathrm{C}_{\mathrm{R}}=2(p-2)\left(\mathrm{I}_{p}-1 / p \mathrm{~J}_{p}\right) \tag{4.1}
\end{equation*}
$$

Where $I_{p}$ is a identity matrix of order $p$ and $J_{p}$ is a matrix of $1^{\text {ss }}$. So that the variance of best linear unbiased estimate (BLUE) of any elementary contrast among the gca effects is $\sigma_{1}{ }^{2} /(p-2)$, where $\sigma_{1}{ }^{2}$ is the per observation variance in the case of randomized block experiment. It is clear from (3.10) that using design d each BLUE of any elementary contrast among gca effects is estimated with variance $\sigma^{2}(p-1) / p(p-3)$. . Hence efficiency factor E of design d as compared to randomized block design under the assumption of equal intra block variances is

$$
\begin{equation*}
E=\left(\sigma_{1}{ }^{2}=\sigma^{2}\right) \text { is } p(p-3) /(p-1)(p-2) \tag{4.2}
\end{equation*}
$$

In Tables 2, 3, and 4, we are presenting the efficiency factors of CDC by Gupta and Kageyama (1994), universally optimal and efficient block designs reported by Dey and Midha (1996) and design d in relation to randomized block design, respectively.

Table 2 : Efficiency of GK designs and designs din comparison to RBD

| S.No. | p | n | r | 2 k | $\mathrm{E}_{\mathrm{GK}}$ | $\mathrm{E}_{\mathrm{d}}$ |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: |
| 1 | 4 | 6 | 3 | 4 | 1.00 | 0.66 |
| 2 | 5 | 10 | 4 | 4 | 0.83 | 0.83 |
| 3 | 7 | 21 | 6 | 6 | 0.93 | 0.93 |
| 4 | 8 | 28 | 7 | 8 | 1.00 | 0.95 |
| 5 | 8 | 28 | 7 | 4 | 0.66 | 0.95 |
| 6 | 9 | 36 | 8 | 8 | 0.96 | 0.96 |
| 7 | 9 | 36 | 8 | 6 | 0.85 | 0.96 |
| 8 | 10 | 45 | 9 | 10 | 1.00 | 0.98 |
| 9 | 10 | 45 | 9 | 6 | 0.83 | 0.98 |


| 10 | 11 | 55 | 10 | 10 | 0.97 | 0.98 |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: |
| 11 | 12 | 66 | 11 | 12 | 1.00 | 0.97 |
| 12 | 12 | 66 | 11 | 6 | 0.80 | 0.98 |
| 13 | 12 | 66 | 11 | 4 | 0.60 | 0.98 |
| 14 | 13 | 78 | 12 | 12 | 0.98 | 0.99 |
| 15 | 13 | 78 | 12 | 6 | 0.78 | 0.99 |
| 16 | 13 | 78 | 12 | 4 | 0.59 | 0.99 |
| 17 | 14 | 91 | 13 | 14 | 1.00 | 0.99 |
| 18 | 15 | 105 | 14 | 14 | 0.98 | 0.99 |
| 19 | 15 | 105 | 14 | 6 | 0.77 | 0.99 | the efficiencies of GK, DM and design d in comparison of RCBD, respectively.

## V. Discussion

In Table 2, we find that for $p=4,5,8,9,10$, 11, 12, 13, and 15 parental lines, the design d perform well in comparison to optimal diallel cross Gupta and Kageyama (1994). In Table 3, for $p=5,7$ and 9 the performance of design $d$ is more or less same in comparison to optimal design Dey and Midha (1996). In Table 4, for $p=5,7,8$ and 10 the design perform well in comparison to efficient designs. Since design d requires minimum possible experimental units, therefore, design d can be used in place of GK and DM designs for estimating gca and sca effects.

## VI. Illustration

We show the essential steps of analysis of a diallel cross experiment, using an incomplete block design proposed in this paper. For this purpose, we take data from an unpublished experiment conducted by Dr. Terumi Mukai on Drosophila melanogaster Cockerham and Weir (1977) on page 203. For the purpose of illustration, we take data of relevant crosses from this experiment. Each cross is replicated twice. The layout and observations in parentheses are given below.

Table 3 : Efficiency of Optimal DM designs and designs d in comparison to RBD

| S.No. | Ref. <br> No. | p | $\mathrm{n}_{1}$ | $\mathrm{E}_{\mathrm{DM}}$ | $\mathrm{n}_{2}$ | $\mathrm{E}_{\mathrm{d}}$ |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: |
| 1 | T 2 | 5 | 30 | 0.83 | 20 | 0.83 |
| 2 | T 3 | 5 | 60 | 0.83 | 20 | 0.83 |
| 3 | T 4 | 5 | 90 | 0.83 | 20 | 0.83 |
| 4 | T 8 | 7 | 210 | 0.70 | 42 | 0.93 |
| 5 | T 22 | 7 | 210 | 0.93 | 42 | 0.93 |
| 6 | T 40 | 8 | 280 | 1.00 | 56 | 0.95 |
| 7 | T 41 | 9 | 252 | 0.96 | 72 | 0.96 |
| 8 | T 54 | 10 | 315 | 1.00 | 90 | 0.98 |

Table 4 : Efficiency of DM efficient designs and designs d in comparison to RBD

| S.No. | Ref. | p | $\mathrm{n}_{1}$ | $\mathrm{E}_{\mathrm{DM}}$ | $\mathrm{n}_{2}$ | $\mathrm{E}_{\mathrm{d}}$ | S.No. | Ref | p | $\mathrm{n}_{1}$ | $\mathrm{E}_{\mathrm{DM}}$ | $\mathrm{n}_{2}$ | $\mathrm{E}_{d}$ |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: |
| 1 | T 12 | 5 | 60 | 0.84 | 20 | 0.83 | 9 | T 58 | 5 | 60 | 0.84 | 20 | 0.83 |
| 2 | T 13 | 5 | 90 | 0.92 | 20 | 0.83 | 10 | T 60 | 5 | 60 | 0.97 | 20 | 0.83 |
| 3 | T 33 | 5 | 40 | 0.94 | 20 | 0.83 | 11 | T 94 | 7 | 210 | 0.84 | 42 | 0.93 |
| 4 | T 34 | 5 | 80 | 0.80 | 20 | 0.83 | 12 | T 95 | 7 | 210 | 0.91 | 42 | 0.93 |
| 5 | T 37 | 5 | 100 | 0.87 | 20 | 0.83 | 13 | T 77 | 8 | 196 | 0.98 | 56 | 0.95 |
| 6 | T 44 | 5 | 30 | 1.00 | 20 | 0.83 | 14 | T 85 | 9 | 252 | 1.00 | 72 | 0.96 |
| 7 | T 45 | 5 | 60 | 0.84 | 20 | 0.83 | 15 | T 91 | 10 | 405 | 0.92 | 90 | 0.98 |
| 8 | T 57 | 5 | 30 | 0.84 | 20 | 0.83 |  |  |  |  |  |  |  |

DM denotes Dey and Midha, Ref means the design number reported by Dey and Midha(1996), $n_{1}$ and $n_{2}$ are number of experimental units required by Dey and Midha's designs and design $d$, respectively. $E_{G K}, E_{D M}$ and $E_{d}$ are

| $\mathrm{B}_{1}$ | $\mathrm{~B}_{2}$ | $\mathrm{~B}_{3}$ | $\mathrm{~B}_{4}$ | $\mathrm{~B}_{5}$ |
| :---: | :---: | :---: | :---: | :---: |
| $1 \times 2$ | $2 \times 3$ | $3 \times 4$ | $4 \times 0$ | $0 \times 1$ |
| $(21.0)$ | $(16.8)$ | $(13.8)$ | $(18.8)$ | $(16.5)$ |
| $2 \times 4$ | $3 \times 0$ | $4 \times 1$ | $0 \times 2$ | $1 \times 3$ |
| $(15.2)$ | $(16.2)$ | $(12.2)$ | $(31.8)$ | $(13.0)$ |
| $3 \times 1$ | $4 \times 2$ | $0 \times 3$ | $1 \times 4$ | $2 \times 0$ |
| $(11.4)$ | $(15.4)$ | $(17.8)$ | $(13.6)$ | $(30.4)$ |
| $4 \times 3$ | $0 \times 4$ | $1 \times 0$ | $2 \times 1$ | $3 \times 2$ |
| $(15.2)$ | $(14.6)$ | $(15.4)$ | $(23.0)$ | $(16.3)$ |

The following are the vector of treatment total, block total and adjusted treatment total, respectively.
$\mathrm{T}=(31.9,62.2,34.0,33.4,44.0,24.4,25.8,33.10,30.6$, 29.0) '
$B=(62.8,63.0,59.2,87.2,76.2)^{\prime}$
$Q=(-1.95,21.35,3.45,-4.15,6.50,-10.35,-10.80,-$ 1.70, -0.85, -1.50) '

ANOVA, estimates of gca, and sca along with their standard errors are shown in Tables 5, 6 and 7.

Table 5 : Analysis of variance of the data

| Source | D.F | Sum of <br> squares | Mean <br> sum of <br> square | $F$ |
| :---: | :---: | :---: | :---: | :---: |
| Blocks | 4 | 137.81 |  |  |
| Crosses | 9 | 418.92 | 46.54 | 53.70 |
| g.c.a | 4 | 341.70 | 85.42 | 98.56 |
| s.c.a | 5 | 77.20 | 15.44 | 17.81 |
| Intra block error | 6 | 5.2 | 0.86 |  |
| Total | 19 | 561.93 |  |  |

Table 6 : Estimates of the general combining ability and their estimated standard error

| Parent | Estimates of (gca) | $\pm$ S E |
| :---: | :---: | :---: |
| 0 | -1.24 | 0.4147 |
| 1 | -0.65 | 0.4147 |
| 2 | -2.16 | 0.4147 |
| 3. | 2.58 | 0.4147 |
| 4. | 1.47 | 0.4147 |

Table 7 : Estimates of sca effects and their estimated standard error

| SCA | Estimate of (sca) | $\pm$ S E | SCA | Estimate of (sca) | $\pm$ S E |
| :---: | :---: | :---: | :---: | :---: | :---: |
| $\mathrm{s}_{01}$ | -0.63 | 0.4818 | $\mathrm{~s}_{13}$ | -5.29 | 0.4818 |
| $\mathrm{~s}_{02}$ | 8.65 | 0.4818 | $\mathrm{~s}_{14}$ | -5.61 | 0.4818 |
| $\mathrm{~s}_{03}$ | 3.13 | 0.4818 | $\mathrm{~s}_{23}$ | -1.26 | 0.4818 |
| $\mathrm{~S}_{04}$ | -3.04 | 0.4818 | $\mathrm{~S}_{24}$ | 0.52 | 0.4818 |
| $\mathrm{~s}_{12}$ | 2.58 | 0.4818 | $\mathrm{~S}_{34}$ | 0.95 | 0.4818 |

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# A Simple Neural Network Approach to Software Cost Estimation 

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Abstract - The effort invested in a software project is one of the most challenging task and most analyzed variables in recent years in the process of project management. Software cost estimation predicts the amount of effort and development time required to build a software system. It is one of the most critical tasks and it helps the software industries to effectively manage their software development process. There are a number of cost estimation models. Each of these models have their own pros and cons in estimating the development cost and effort. This paper investigates the use of Back-Propagation neural networks for software cost estimation. The model is designed in such a manner that accommodates the widely used COCOMO model and improves its performance. It deals effectively with imprecise and uncertain input and enhances the reliability of software cost estimates. The model is tested using three publicly available software development datasets. The test results from the trained neural network are compared with that of the COCOMO model. From the experimental results, it was concluded that using the proposed neural network model the accuracy of cost estimation can be improved and the estimated cost can be very close to the actual cost.

Keywords : artificial neural networks, back-propagation networks, COCOMO model, project management, soft computing techniques, software effort estimation.
GJCST-D Classification: B.2.m

Strictly as per the compliance and regulations of:


[^6]
# A Simple Neural Network Approach to Software Cost Estimation 

Anupama Kaushik ${ }^{\alpha}$, A.K. Soni ${ }^{\sigma}$ \& Rachna Soni ${ }^{\rho}$


#### Abstract

The effort invested in a software project is one of the most challenging task and most analyzed variables in recent years in the process of project management. Software cost estimation predicts the amount of effort and development time required to build a software system. It is one of the most critical tasks and it helps the software industries to effectively manage their software development process. There are a number of cost estimation models. Each of these models have their own pros and cons in estimating the development cost and effort. This paper investigates the use of BackPropagation neural networks for software cost estimation. The model is designed in such a manner that accommodates the widely used COCOMO model and improves its performance. It deals effectively with imprecise and uncertain input and enhances the reliability of software cost estimates. The model is tested using three publicly available software development datasets. The test results from the trained neural network are compared with that of the COCOMO model. From the experimental results, it was concluded that using the proposed neural network model the accuracy of cost estimation can be improved and the estimated cost can be very close to the actual cost.


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## I. INTRODUCTION

Software cost estimation is one of the most significant activities in software project management. It refers to the predictions of the likely amount of effort, time and staffing levels required to build a software system. The effort prediction aspect of software is made at an early stage during project development, when the costing of the project is proposed for approval. It is concerned with the prediction of the person hour required to accomplish the task. However, estimates at the early stages of the development are the most difficult to obtain because very little is known about the project and the product at the beginning. So, estimating software development effort remains a complex problem and it continues to attract research attention. There are several cost estimation techniques proposed and they are grouped into two major categories: (1) Parametric models or Algorithmic models, which uses a mathematical formula

[^7]to predict project cost based on the estimates of project size, the number of software engineers, and other process and product factors [1]. These models can be built by analysing the costs and attributes of completed projects and finding the closest fit formula to actual experience. (2) Non Parametric models or Non algorithmic models which are based on fuzzy logic (FL), artificial neural networks (ANN) and evolutionary computation (EC). In this paper, we focus on non parametric cost estimation models based on artificial neural networks, and particularly Back-Propagation networks. Neural networks have learning ability and are good at modelling complex nonlinear relationships. They also provide more flexibility to integrate expert knowledge into the model. There are many software cost estimation models that have been developed using neural networks over the years. The use of radial basis function neural networks for software effort estimation is well described by many researchers [2, 3 and 4]. The clustering algorithms used in those designs are the conventional algorithms.
K. Vinay Kumar et al. [5] Uses wavelet neural networks for predicting software development cost. B. Tirimula Rao et al. [6] provided a novel neural network approach for software cost estimation using functional link artificial neural network. G. Witting and G. Finnie [7] uses back propagation learning algorithms on a multilayer perceptron in order to predict development effort. N. Karunanitthi et al. [8] reports the use of neural networks for predicting software reliability including experiments with both feed forward and Jordan networks. N. Tadayon [9] also reports the use of neural network with a back propagation learning algorithm. However it was not clear how the dataset was divided for training and validation purposes. T.M. Khoshgoftaar et al.[10] presented a case study considering real time software to predict the testability of each module from source code static measures. Ch. Satyananda Reddy and KVSVN Raju [11] proposed a cost estimation model using multi layer feed forward neural network. Venkatachalam [12] also investigated the application of artificial neural network (ANN) to software cost estimation.

Artificial neural networks are the promising techniques to build predictive models. So, there is always a scope for developing effort estimation models with better predictive accuracy.

## il. Overview of the Models and Techniques Used

## a) COCOMO // Model

The COCOMO model, is the best known algorithmic cost model published by Barry Boehm in 1981 [1]. It was developed from the analysis of sixty three software projects. It is a hierarchy of software cost estimation models, which includes Basic, Intermediate and Detailed sub models. It was the most cited and plausible of all the traditional cost estimation models. COCOMO II is the revised version of the original COCOMO and is tuned to the life cycle practices of the $21^{\text {st }}$ century. It also provides a quantitative analytic framework, and set of tools and techniques for evaluating the effects of software technology improvements on software life cycle costs and schedules. It consists of three sub models and they are:

- Application Composition Model: This model is suitable for quickly developed applications using interoperable components like components based on GUI builders and is based on new object point's estimation.
- Early Design Model: This model is used in the early stages of a software project and can be used in Application Generator, System Integration, or Infrastructure Development Sector. It uses Unadjusted Function Points (UFP) as the measure of size.
- Post Architecture Model: This is the most detailed of the three and is used after the overall architecture for the project has been designed. One could use function points or LOC as size estimates with this model. It involves the actual development and maintenance of a software product.

COCOMO II describes 17 cost drivers and 5 scale factors that are used in the Post Architecture model. The cost drivers for COCOMO II are rated on a scale from very low to extra high. Their product is used to adjust the nominal effort. Table 1 lists COCOMO II cost drivers along with their multipliers. Scale factor is a particular characteristic of the software development that has an exponential effect of increasing or decreasing the amount of development effort and they are Precedentness, Development flexibility, Architecture/Risk resolution, Team cohesion and Process maturity. These factors are rated on a six point scale i.e., very low, low, nominal, high, very high and extra high as given in Table 2.
COCOMO II post architecture model is given as:

$$
\mathrm{PM}=\mathrm{A} \times\left[\begin{array}{ll}
\text { size } \tag{1}
\end{array}\right]^{1.01+\sum_{i=1}^{5}} \times \prod_{\mathrm{i}=1}^{17} \mathrm{E}
$$

Where PM is the effort expressed in person months, A is a multiplicative constant, size is the projected size of the software project expressed in thousands of lines of code KLOC, $\mathrm{EM}_{\mathrm{i}}(\mathrm{i}=1,2 \ldots .17)$ are
effort multipliers and $\mathrm{SF}_{\mathrm{i}}(\mathrm{i}=1,2 \ldots . .5)$ are exponent scale factors.

## b) Artificial Neural Networks

An artificial neural network (ANN) is an efficient information processing system which resembles in characteristics with a biological neural network. ANN's possess large number of highly interconnected processing elements called neurons. Each neuron is connected with the other by a connection link. Each connection link is associated with weights which contain information about the input signal. This information is used by the neuron net to solve a particular problem. Each neuron has an internal state of its own. This internal state is called the activation level of neuron, which is the function of the inputs the neuron receives. There are a number of activation functions that can be applied over net input such as Gaussian, Linear, Sigmoid and Tanh. It is the Sigmoid function that is the most frequently used in neural nets. Thus, the models of ANN are specified by the three basic entities namely [13]:

1. The model's synaptic interconnections;
2. The training or learning rules adopted for updating and adjusting the connection weights;
3. Their activation functions.

The neural network process starts by developing the structure of the network and establishing the technique used to train the network using an existing data set. Neural network architectures are divided into two groups:

1. Feed forward networks where no loops in the network path occur.
2. Feedback networks that have recursive loops.

Table 1 : COCOMO II cost drivers with multipliers

| S.No | Cost <br> Driver | Very <br> Low | Low | Nominal | High | Very <br> High | Extra <br> High |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: |
| 1 | RELY | 0.75 | 0.88 | 1.00 | 1.15 | 1.39 | -- |
| 2 | DATA | -- | 0.93 | 1.00 | 1.09 | 1.19 | -- |
| 3 | CPLX | 0.75 | 0.88 | 1.00 | 1.15 | 1.30 | 1.66 |
| 4 | RUSE |  | 0.91 | 1.00 | 1.14 | 1.29 | 1.49 |
| 5 | DOCU | 0.89 | 0.95 | 1.00 | 1.06 | 1.13 |  |
| 6 | TIME | -- | -- | 1.00 | 1.11 | 1.31 | 1.67 |
| 7 | STOR | -- | -- | 1.00 | 1.06 | 1.21 | 1.57 |
| 8 | PVOL | -- | 0.87 | 1.00 | 1.15 | 1.30 | -- |
| 9 | ACAP | 1.50 | 1.22 | 1.00 | 0.83 | 0.67 | -- |
| 10 | PCAP | 1.37 | 1.16 | 1.00 | 0.87 | 0.74 | -- |
| 11 | PCON | 1.24 | 1.10 | 1.00 | 0.92 | 0.84 | -- |
| 12 | AEXP | 1.22 | 1.10 | 1.00 | 0.89 | 0.81 | -- |
| 13 | PEXP | 1.25 | 1.12 | 1.00 | 0.88 | 0.81 | -- |
| 14 | LTEX | 1.22 | 1.10 | 1.00 | 0.91 | 0.84 | -- |
| 15 | TOOL | 1.24 | 1.12 | 1.00 | 0.86 | 0.72 | -- |
| 16 | SITE | 1.25 | 1.10 | 1.00 | 0.92 | 0.84 | 0.78 |
| 17 | SCED | 1.29 | 1.10 | 1.00 | 1.00 | 1.00 | -- |

Table 2 : COCOMO II Scaling Factors

| Scaling Factors | Very <br> Low | Low | Nominal | High | Very <br> High | Extra <br> High |
| :--- | :---: | :---: | :---: | :---: | :---: | :---: |
| Precedentness | 6.20 | 4.96 | 3.72 | 2.48 | 1.24 | 0.00 |
| Development Flexibility | 5.07 | 4.05 | 3.04 | 2.03 | 1.01 | 0.00 |
| Architecture/Risk <br> Resolution | 7.07 | 5.65 | 4.24 | 2.83 | 1.41 | 0.00 |
| Team Cohesion | 5.48 | 4.38 | 3.29 | 2.19 | 1.10 | 0.00 |
| Process Maturity | 7.80 | 6.24 | 4.68 | 3.12 | 1.56 | 0.00 |

The most common architecture of neural networks which is used in software cost estimation is the Back-Propagation trained Feed Forward networks [14, 15]. The training algorithm of back propagation involves four stages:

1. Initialization of weights
2. Feed forward
3. Back Propagation of errors
4. Updation of the weights and biases

## III. Proposed Work

The performance of a neural network depends on its architecture and their parameter settings. There are many parameters governing the architecture of the neural network including the number of layers, the number of nodes in each layer, the transfer function in each node, learning algorithm parameters and the weights which determine the connectivity between nodes. There is no rule which determines the ideal parameter settings but even a slight parameter changes can cause major variations in the results of almost all networks. This property of the neural network is captured in the present work for predicting the software costs. The neural network model proposed is based on multi layer feed forward neural network and it uses the architecture given by Ch. Satyananda Reddy and

KVSVN Raju [11]. The model accommodates the COCOMO II model.

The aim of this work is to evaluate the results of software cost estimation using COCOMO II by varying the activation functions at the input, hidden and the output layers. The model proposed uses the identity function at the input layer which is defined by $f(x)=x$. The hidden and the output layer uses unipolar sigmoid function defined by $\mathrm{f}(\mathrm{x})=\frac{1}{1+\mathrm{e}^{-x}}$.

This function is especially advantageous to use in neural networks trained by back-propagation algorithms. Because it is easy to distinguish, and this can interestingly minimize the computation capacity for training.

## a) Architecture of the Neural Network Mode/

The proposed structure of the neural network accommodates the COCOMO II post architecture model given by Eq. 1. The use of neural network to estimate PM (person months) in Eq. 1 requires twenty four input nodes in the input layer which corresponds to seventeen EM's, five SF's and two bias values. The COCOMO model which is a non linear model is transformed into a linear model using natural logarithms as shown in Eq. 2.

$$
\begin{equation*}
\operatorname{Ln}(\mathrm{PM})=\ln (\mathrm{A})+\ln \left(\mathrm{EM}_{1}\right)+\ln \left(\mathrm{EM}_{2}\right)+\ldots \ldots+\ln \left(\mathrm{EM}_{17}\right)+\left[1.01+\mathrm{SF}_{1}+\ldots \ldots+\mathrm{SF}_{5}\right]^{*} \ln (\text { size }) \tag{2}
\end{equation*}
$$

The above equation becomes :

$$
\begin{equation*}
C_{P M}=\left[b_{1}+x_{1}{ }^{*} z_{1}+x_{2}{ }^{*} z_{2}+\ldots \ldots+x_{17}{ }^{*} z_{17}\right]+\left[b_{2}+z_{18}+\ldots \ldots . .+z_{22}\right]^{*}\left[y_{i}+\ln (\text { size })\right] \tag{3}
\end{equation*}
$$

Where,
$\mathrm{C}_{\mathrm{PM}}=\operatorname{In}(\mathrm{PM})$;
$\mathrm{z}_{1}=\ln \left(\mathrm{EM}_{1}\right) ; \mathrm{z}_{2}=\ln \left(\mathrm{EM}_{2}\right) ; \ldots \ldots ; \mathrm{z}_{17}=\ln \left(\mathrm{EM}_{17}\right) ;$
$\mathrm{z}_{18}=\mathrm{SF}_{1} ; \ldots \ldots, \ldots \mathrm{z}_{22}=\mathrm{SF}_{5}$;
$b_{1}$ and $b_{2}$ are the biases and the coefficients $x_{i}$ and $y_{i}$ are the additional terms used in the model which act as the weights from the input layer to the hidden layer.

The COCOMO II model as given by Eq. 3 is shown in Fig.1. This network consists of two hidden layer nodes $\mathrm{C}_{\mathrm{EM}}$ and $\mathrm{C}_{\mathrm{SF}}$ that take into account the contribution of effort multipliers and scale factors. $\mathrm{C}_{\mathrm{PM}}$ is the node of the output layer where we get the value of $\ln (\mathrm{PM})$ which is the desired output of the model. In the above network all the original $E M_{\mathrm{i}}$ and $\mathrm{SF}_{\mathrm{i}}$ values of

COCOMO II are pre processed to $\ln \left(\mathrm{EM}_{\mathrm{i}}\right)$ and $\ln \left(\mathrm{SF}_{\mathrm{i}}\right)$ and used as input nodes. The two bias values are denoted by $b_{1}$ and $b_{2}$, which are $\ln (A)$ and 1.01 respectively. The size of the product is not considered as one of the inputs to the network but as a cofactor for the initial weights for scale factors (SF). The weights associated to the input nodes connected to the hidden layer are denoted by $\mathrm{x}_{\mathrm{i}}$ for $1 \leq i \leq 17$ for each input $\ln (E M i)$ and $b_{1}$. On the other hand, the weights associated to the hidden layer for each In (SFi) input nodes and $\mathrm{b}_{2}$ are $\mathrm{y}_{\mathrm{i}}+\mathrm{ln}$ (size) for $18 \leq i \leq 22$. These weights are initialized as $x_{i}=1$ and $y_{i}=0$. The weights from the hidden layer to the output layer are denoted by p and q and initialized as $\mathrm{p}=\mathrm{q}=1$.


Figure 1 : Neural Network Architecture

## b) Training Algorithm

The feed forward back propagation procedure is used to train the network by iteratively processing a set of training samples and comparing the network's prediction with the actual value. For each training sample, the weights are modified so as to minimize the error between the networks predicted value and the actual value. The following algorithm is used for training the proposed network and for calculating the new set of weights:

Step 1: Initialize the weights and learning rate $\alpha$ ( $0<a \leq 1$ ).

Step 2: Perform steps 3-10 when stopping condition is false.
Step 3: Perform steps 4-9 for each training pair.
Step 4: Each input unit receives input signal and sends it to the hidden unit.

Step 5: Each hidden unit $\mathrm{C}_{\text {EM }}$ and $\mathrm{C}_{\text {SF }}$ sums its weighted input signals to calculate net input given by:

$$
\begin{array}{ll}
C_{E M}=b_{1}+\sum z_{i}^{*} x_{i} & \text { for } i=1 \text { to } 17 \\
C_{S F}=b_{2}+\sum z_{i}^{*}\left(y_{i}+\ln (\text { size })\right) & \text { for } i=18 \text { to } 22
\end{array}
$$

Apply sigmoidal activation function over $\mathrm{C}_{\mathrm{EM}}$ and $\mathrm{C}_{\mathrm{SF}}$ and send the output signal from the hidden unit to the input of output layer units.

Step 6: The output unit $\mathrm{C}_{\mathrm{PM}}$, calculates the net input given by:

$$
\mathrm{C}_{\mathrm{PM}}=\mathrm{C}_{\mathrm{EM}}{ }^{*} \mathrm{p}+\mathrm{C}_{\mathrm{SF}}{ }^{*} \mathrm{q}
$$

Apply sigmoidal activation function over $\mathrm{C}_{P M}$ to compute the output signal $\mathrm{E}_{\text {est }}$.

Step 7: Calculate the error correction term as: $\delta=E_{\text {act }}-E_{\text {est }}$, where $E_{\text {act }}$ is the actual effort from the dataset and $E_{\text {est }}$ is the estimated effort from step 6.
Step 8: Update the weights between hidden and the output layer as:

$$
\begin{aligned}
& p(\text { new })=p(\text { old })+\alpha^{\star} \delta^{\star} C_{\text {EM }} \\
& q(\text { new })=q(o l d)+\alpha^{\star} \delta^{\star} C_{S F}
\end{aligned}
$$

Step 9: Update the weights and bias between input and hidden layers as:

$$
\begin{aligned}
& x_{i}(\text { new })=x_{i}(\text { old })+\alpha^{*} \delta_{\text {EE }}{ }^{\star} z_{\text {f }} \text { for } i=1 \text { to } 17 \\
& y_{i}(\text { new })=y_{i} \text { (old) }+\alpha^{*} \delta_{\text {SF }} z_{i} \text { for } i=18 \text { to } 22 \\
& \left.b_{1} \text { (new }\right)=b_{1} \text { (old) }+\alpha^{\star} \delta_{\text {EM }} \\
& \left.b_{2} \text { (new }\right)=b_{2}(\text { old })+\alpha^{\star} \delta_{\text {SF }}
\end{aligned}
$$

The error is calculated as

$$
\delta_{\mathrm{EM}}=\delta^{\star} \mathrm{p} ; \quad \delta_{\mathrm{SF}}=\delta^{\star} \mathrm{q} ;
$$

Step 10: Check for the stopping condition. The stopping condition may be certain number of epochs reached or if the error is smaller than a specific tolerance.

Using this approach, we iterate forward and backward until the terminating condition is satisfied. The variable $\alpha$ used in the above formula is the learning rate, a constant, typically having a value between 0 and 1 . The learning rate can be increased or decreased by the expert judgment indicating their opinion of the input effect. In other words the error should have more effect on the expert's indication that a certain input had more contribution to the error propagation or vice versa. For each project, the expert estimator can identify the importance of the input value to the error in the estimation. If none selected by the expert, the changes in the weights are as specified by the learning algorithm. The network should also be trained according to correct inputs. For example, if during estimation ACAP (Analyst Capability) is set as high but after the end of the project, the management realizes that it was nominal or low, then the system should not consider this as a network error and before training the system, the better values of cost factors should be used to identify the estimated cost.

## IV. Datasets and Evaluation Criteria

The data sets used in the present study comes from PROMISE Software Engineering Repository data
set [16] made publicly available for research purpose. The three datasets used are COCOMO 81 dataset, NASA 93 dataset and COCOMO_SDR.

The COCOMO 81 dataset consists of 63 projects which uses COCOMO model as described in section 2.1. Each project is described by its 17 cost drivers, 5 scale factors, the software size measured in KDSI (Kilo Delivered Source Instructions), the actual effort, total defects and the development time in months. The NASA 93 dataset consists of 93 NASA projects from different centres for various years. It consists of 26 attributes: 17 standard COCOMO-II cost drivers and 5 scale factors in the range Very_Low to Extra_High, lines of code measure (KLOC), the actual effort in person months, total defects and the development time in months.

The COCOMO_SDR dataset is from Turkish Software Industry. It consists of data from 12 projects and 5 different software companies in various domains. It has 24 attributes: 22 attributes from COCOMO II model, one being KLOC and the last being actual effort in man months.

The entire dataset is divided into two sets, training set and validation set in the ratio of 80:20 to get more accuracy of prediction. The proposed model is trained with the training data and tested with the test data.

The evaluation consists in comparing the accuracy of the estimated effort with the actual effort. A common criterion for the evaluation of cost estimation model is the Magnitude of Relative Error (MRE) and is defined as in Eq. 4.

## $M R E=\frac{\| \text { Actual Effort-Estimated Effc }}{\text { Actual Effort }}$

The MRE values are calculated for each project in the validation set, while mean magnitude of relative error (MMRE) computes the average of MRE over N projects.

$$
\begin{equation*}
M M R E=\frac{1}{N} \sum_{X=1}^{N} M 1 \tag{5}
\end{equation*}
$$

Another evaluation criterion is MdMRE, which measures the median of all MRE's. MdMRE is less sensitive to extreme values. It exhibits a similar pattern to MMRE but it is more likely to select the true model if the underestimation is served.

Since MRE, MMRE and MdMRE are the most common evaluation criteria, they are adopted as the performance evaluators in the present paper.

## V. Results and Discussion

This section presents and discusses the results obtained when applying the proposed neural network model to the COCOMO 81, NASA 93 and COCOMO_SDR datasets. The model is implemented in Matlab. The MRE, MMRE and MdMRE values are
calculated for the projects in the validation set for all the three datasets. These values are then compared with the COCOMO model.

Table 3 shows the results and comparison on COCOMO dataset. It also contain results given by Ch. Satyananda Reddy and KVSVN Raju [11] for the corresponding projects. For example, in the case of Project ID 5 it is 7.44 for COCOMO model, 5.08 for the model proposed by Ch. Satyananda Reddy and KVSVN Raju and 4.012 for the proposed model. The Mean Magnitude of Relative Error (MMRE) for the entire validation set is 15.938 for the COCOMO model, 8.745 for the model proposed by Ch. Satyananda Reddy and KVSN Raju and 3.546 for the proposed model. The Median of MRE (MdMRE) for the entire validation set is $12.4 \%$ for the COCOMO model, $9.73 \%$ for the model proposed by Ch. Satyananda Reddy and KVSN Raju and $3.67 \%$ for the proposed model. Fig. 2 shows the graphical representation of MRE values for the three models for COCOMO 81 dataset. There is a decrement in the relative error using the proposed model. The results obtained thus suggest that the proposed architecture can be applied for accurately predicting the software costs.

Table 4 shows the results and comparison on NASA 93 dataset. Here also, there is a decrease in the relative error using the proposed model. For example, the relative error calculated for Project ID 30 is 8.81 for COCOMO model, and 3.34 for our proposed model. The relative error calculated for Project ID 62 is 13.2 for

COCOMO model, and 5.00 for our proposed model. The Mean Magnitude of Relative Error (MMRE) for the entire validation set is 12.746 and 4.349 for the COCOMO model and our proposed model respectively. The MdMRE for the entire validation set is $13.43 \%$ for the COCOMO model and $4.46 \%$ for our proposed model. Fig. 3 shows the graphical representation of MRE values for the two models.

For COCOMO_SDR dataset, COCOMO ॥ model performs very poorly. For Project ID 1, it has estimated effort as 2241.4 whereas the actual effort is 1 and with our proposed model it is 1.24 . Similarly, for Project ID 2 COCOMO II effort is 901.6; its actual effort is 2 and the estimated is 1.95 . Table 5 shows the estimated effort and their MRE values using the proposed model on COCOMO_SDR dataset. MMRE value for the estimated effort is 6.34 . The MdMRE for the entire validation set is $4.62 \%$ for the proposed model. Fig. 4 shows the bar graph representation of actual effort values and estimated effort values with the proposed model for COCOMO_SDR. The bar graph shows that the estimated effort is very close to the actual effort.

The results obtained thus, suggest that the proposed model outperformed the COCOMO model and the model given by Ch. Satyananda Reddy and KVSN Raju in terms of all the discussed evaluation criteria i.e, MRE, MMRE and MdMRE. It can be applied for accurately predicting the software costs.

Table 3 : Comparison of MRE for the three models on COCOMO 81

| S.No | Project ID | MRE(\%) using <br> COCOMO model | MRE(\%) using <br> Model proposed by <br> Satyananda Reddy | MRE(\%) using <br> proposed model |
| :---: | :---: | :---: | :---: | :---: |
| 1 | 5 | 7.44 | 5.08 | 4.012 |
| 2 | 12 | 19.83 | 6.8 | 3.98 |
| 3 | 30 | 6.49 | 3.24 | 1.77 |
| 4 | 38 | 50.98 | 15.34 | 3.59 |
| 5 | 40 | 12.4 | 11.1 | 4.16 |
| 6 | 45 | 5.35 | 4.59 | 4.01 |
| 7 | 47 | 16.4 | 10.06 | 3.46 |
| 8 | 59 | 8.66 | 4.92 | 3.67 |
| 9 | 61 | 13.1 | 12.5 | 3.86 |
| 10 | 62 | 6.22 | 9.73 | 2.97 |
| 11 | 63 | 19.95 | 12.84 | 3.53 |

Table 4 : Comparison of MRE on NASA93

| S.No | Project <br> ID | MRE(\%) using <br> COCOMO <br> model | MRE(\%) using <br> proposed <br> model |
| :---: | :---: | :---: | :---: |
| 1. | 1 | 9.33 | 3.90 |
| 2. | 5 | 8.84 | 3.39 |
| 3. | 15 | 16.75 | 4.25 |
| 4. | 25 | 14.09 | 4.11 |
| 5. | 30 | 8.81 | 3.34 |
| 6. | 42 | 13.9 | 5.00 |
| 7. | 54 | 13.67 | 4.89 |
| 8. | 60 | 11.78 | 4.93 |
| 9. | 62 | 13.2 | 5.00 |
| 10. | 75 | 17.09 | 4.68 |



Figure 2 : Comparison of the Models on COCOMO dataset


Figure 4 : Comparison of Actual vs. Estimated effort on COCOMO_SDR dataset

## VI. Conclusion

Software development cost estimation is a challenging task for both the industrial as well as academic communities. The accurate predictions during the early stages of development of a software project can greatly benefit the development team. There are several effort estimation models that can be used in forecasting software development effort.

In the paper, Feed Forward Back Propagation model of neural network is used which maps the COCOMO model. The model used identity function at the input layer and sigmoidal function at the hidden and output layer. The model incorporates COCOMO dataset and COCOMO NASA 2 dataset to train and to test the network. Based on the experiments performed, it is observed that the proposed model outscored COCOMO model and the model proposed by Ch. Satyananda Reddy and KVSN Raju. Future research can replicate and confirm this estimation technique with other datasets for software cost estimation. Furthermore, the utilization of other neural networks architecture can also be applied for estimating software costs. This work can also be extended using Neuro Fuzzy approach.

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