

# Efficient Solution of a Class of Location-Allocation Problems with Stochastic Demand and Congestion

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## Abstract

We consider a class of location-allocation problems with immobile servers, stochastic demand and congestion that arises in several planning contexts: location of emergency medical clinics; preventive healthcare centers; refuse collection and disposal centers; stores and service centers; bank branches and automated banking machines; internet mirror sites; web service providers (servers); and distribution centers in supply chains. The problem seeks to simultaneously locate service facilities, equip them with appropriate capacities, and allocate user demand to these facilities such that the total cost, which consists of the fixed cost of opening facilities with sufficient capacities, the access cost of users' travel to facilities, and the queuing delay cost, is minimized. Under Poisson user demand arrivals and general service time distributions, the problem is set up as a network of independent M/G/1 queues, whose locations, capacities and service zones need to be determined. The resulting mathematical model is a non-linear integer program. Using simple transformation and piecewise linear approximation, the model is linearized and solved to  $\epsilon$ -optimality using a constraint generation method. Computational results are presented for instances up to 400 users, 25 potential service facilities, and 5 capacity levels with different coefficient of variation of service times and average queuing delay costs per customer. The results indicate that the proposed solution method is efficient in solving a wide range of problem instances.

*Keywords:* Location-Allocation; Service System Design; Queueing; Stochastic Demand; Congestion; Constraint Generation Method

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## 1. Introduction

Problems arising in several planning contexts require deciding: (i) the location of service facilities and their capacities; and (ii) allocation of service zones to the located service facilities. Examples include location of emergency service facilities such as medical clinics and preventive health care facilities (Zhang et al., 2009, 2010, 2012); stores and service centers; bank branches and automated banking machines (Wang et al., 2002; Aboolian et al., 2008; Castillo et al., 2009); automobile emission testing stations (Castillo et al., 2009); web service providers' facilities (Aboolian et al., 2009); proxy/mirror servers in communication networks (Wang et al., 2004) and distribution centers in supply chains (Huang et al., 2005; Vidyarthi et al., 2009). All the above examples are characterized by servers (medical clinics, bank branches, distribution centers, etc.) that are immobile in that the customers need to travel to the service facilities to avail of their services, as opposed to the servers travelling (mobile servers) to the customers' site in response to calls for their services. Such problems are generally also characterized by random nature of service calls (demand arrivals) and their service requirements (service times). These problems are commonly known in the literature as facility location problems with immobile servers, stochastic demand and congestion (Berman and Krass, 2004). They are also termed as service system design problems with stochastic demand and congestion (Amiri, 1997, 1998, 2001; Elhedhli, 2006). Literature review for this class of problems is provided by Berman and Krass (2004) and Boffey et al. (2007).

For facility location problems with stochastic demand and congestion, the following two factors are important: (i) the costs of providing service; and (ii) the quality of service, with an objective generally requiring a balance between the two. The costs of providing service are related to the fixed cost of opening/operating the service facilities and the cost of accessing these facilities by the users. The service quality, on the other hand, is often measured in terms of: (i) the average number of users waiting for service; (ii) average waiting time per user; or (iii) the probability of serving a user within a time limit (Elhedhli, 2006). Balance between service costs and service quality is commonly achieved in the literature using a combination of the total cost of opening and accessing facilities and the cost associated with waiting customers, which is minimized in the objective function (Amiri, 1997, 1998; Wang et al., 2002; Elhedhli, 2006; Castillo et al., 2009). Others in the literature minimize the cost of providing service subject to a minimum threshold on the service quality, where the service quality may be defined in one of the ways described above (Marianov and Serra, 1998, 2002; Silva and Serra, 2008).

In this paper, we use the former of the two approaches described above, i.e., we consider minimization of the total cost, which includes the cost of opening and accessing facilities and the cost associated with waiting customers. It is worth noting that due to the complex-

1 ity of the underlying problem, most papers in this category make assumptions such as: (i)  
2 either the number or capacity of the facilities (or both) are fixed; (ii) the demand arrival  
3 process is Poisson; and (iii) the service times follow an exponential distribution (see Amiri,  
4 1997; Marianov and Serra, 2002; Wang et al., 2002; Elhedhli, 2006; Aboolian et al., 2008,  
5 and references therein). Despite these simplifying assumptions, the techniques proposed to  
6 date to solve the problem, with the exception of Elhedhli (2006), are either approximate or  
7 heuristic based.

8  
9 The contribution of this paper is twofold. First, by assuming a general distribution for  
10 the service times at facilities, as opposed to exponential distribution, we present a more  
11 generalized model of the problem than available in the extant literature. More specifically,  
12 our proposed model seeks to determine the minimum cost system-optimal configuration (lo-  
13 cation of service facilities and their capacity levels as well as the allocation of service zones  
14 to these facilities) of a service system under Poisson arrivals and general service time distri-  
15 bution, where the total cost consists of the costs of opening and accessing service facilities  
16 and the cost associated with waiting customers. As discussed above, the problem, even with  
17 the simplifying assumption of exponential service time distribution, is too difficult to solve  
18 using exact methods. The proposed model, with general service time distribution, is even  
19 more challenging to solve. So, our second contribution lies in the exact ( $\epsilon$ -optimal) solution  
20 method that we propose to solve our model. Our proposed solution method is based on  
21 a simple transformation and piecewise linearization of our non-linear integer programming  
22 (IP) model, which is solved to optimality (or  $\epsilon$ -optimality) using a constraint generation  
23 algorithm.

24  
25 The remainder of the paper is organized as follows. In Section 2, we describe the problem  
26 setting, followed by its non-linear IP model. Section 3 describes the transformation and the  
27 piecewise linearization approach for the non-linear IP model. To solve the linearized model,  
28 we present a constraint generation based solution approach in Section 4. Computational  
29 results are reported in Section 5. Section 6 concludes with some directions for future research.

## 30 **2. Problem Formulation**

31 Consider a set of user nodes, each indexed by  $i \in I$  whose demand for service occurs  
32 continuously over time according to an independent Poisson process with rate  $\lambda_i$ . We con-  
33 sider a *directed choice* environment, where users are assigned to facilities, each indexed by  
34  $j \in J$ , by a central decision maker. This is applicable, for example, in the case of a “virtual  
35 call center” consisting of geographically dispersed telephone call centers, routing of calls to  
36 which is centrally determined (see Castillo et al., 2009, and references therein). Directed  
37 choice model is also applicable in the case of medical clinics and preventive health care  
38 facilities; automobile emission testing stations; and distribution centers in supply chains, if

1 users' choice can be influenced through imposition of tolls or differential service fees. Later,  
2 we show how our model can be adapted to the *user choice* environment where the choice  
3 of service facility is not dictated or influenced by the central authority but exercised solely  
4 by the users. Recent studies of models with directed choice settings include Aboolian et al.  
5 (2012) whereas models with user choice settings can be found in Baron et al. (2008), Zhang  
6 et al. (2012), and references therein.

7  
8 We assume that users from any node are entirely assigned to a single service facility,  
9 where each facility operates with an infinite buffer to accommodate users waiting for ser-  
10 vice. If  $x_{ij}$  is a binary variable that equals 1 if the demand for service from user node  $i$  is  
11 satisfied by facility  $j$ , and 0 otherwise, then the aggregate demand arrival rate at facility  
12  $j$ , as a result of the superposition of Poisson processes, also follows a Poisson process with  
13 mean  $\Lambda_j = \sum_{i \in I} \lambda_i x_{ij}$  (Gross and Harris, 1998).

14  
15 There are two approaches to model the capacity of a service facility (Baron et al., 2008;  
16 Aboolian et al., 2012). One is to model the given service facility as a single server with  
17 flexible service capacity  $\mu$ , which can be adjusted either continuously or in discrete steps.  
18 The second approach is to assume multiple parallel servers, each with a given single capacity  
19 level  $\mu$ . In this case, the decision variable is the appropriate number of servers to be installed  
20 at the given service facility. In the case of call centers, automated banking machines, auto-  
21 mobile emission testing stations, or distribution centers, where adding capacity would imply  
22 adding a call center employee, a banking machine, a testing station, or a loading/unloading  
23 dock respectively, multiple server model is more appropriate. However, in cases where it  
24 is not clear what a "server" represents (e.g. hospitals or emergency medical clinics), and  
25 the capacity can be increased in a variety of ways (by improving patient flow or technology;  
26 adding nurses, doctors, support staff or examination rooms, etc.), single server model would  
27 be suitable. In this paper, we adopt the former approach, and model each facility as a single  
28 server with multiple capacity levels, from which one capacity level is to be selected, if the  
29 facility is opened. We take this approach primarily for tractability of the resulting model.  
30 However, a single server model may still be a good approximation of a multi-server facility  
31 if the utilization of the service facility is reasonably high. This is because under reasonably  
32 high system utilization, a system with  $s$  parallel servers, each with capacity  $\mu$ , is known to  
33 perform similar to a single server with capacity  $s\mu$ .

34  
35 For each service facility, we allow the option of selecting one of the several capacity levels  
36  $\mu_{jk}, k \in K$  with fixed cost  $f_{jk}$  (amortized over the planning period). Let  $y_{jk}$  be a binary  
37 variable that equals 1 if facility at site  $j$  is open and equipped with a capacity level  $k \in K$ , 0  
38 otherwise. Further, assume that the service times at any facility  $j$  are independent and iden-  
39 tically distributed with a mean  $1/\mu_{jk}$  and variance  $\sigma_{jk}^2$  if it is equipped with a capacity level  
40  $k$ . Each facility  $j$  is thus modeled as an  $M/G/1$  queue with a service rate  $\mu_j = \sum_{k \in K} \mu_{jk} y_{jk}$

1 and variance in service times given by  $\sigma_j^2 = \sum_{k \in K} \sigma_{jk}^2 y_{jk}$ . Thus, the service system design  
 2 problem is modeled as a network of independent M/G/1 queues.

3  
 4 Under steady state conditions ( $\Lambda_j/\mu_j < 1$ ), first-come-first-serve (FCFS) queuing disci-  
 5 pline, and infinite buffers to accommodate users waiting for service, the expected waiting  
 6 time (including the time spent in service) of users at facility  $j$  is given, by the Pollaczek-  
 7 Khintchine formula (Gross and Harris, 1998), as:

$$E[w_j] = \left( \frac{1 + Cv_j^2}{2} \right) \frac{\tau_j \rho_j}{1 - \rho_j} + \tau_j = \left( \frac{1 + Cv_j^2}{2} \right) \frac{\Lambda_j}{\mu_j(\mu_j - \Lambda_j)} + \frac{1}{\mu_j} \quad (1)$$

where  $\tau_j = 1/\mu_j$  is the average service time at facility  $j$ ,  $\rho_j = \Lambda_j/\mu_j$  is the average utilization  
 of facility  $j$ , and  $Cv_j = \sigma_j \mu_j$  is the coefficient of variation of service times at facility  $j$ .  $E[w_j]$   
 can be written in terms of location and allocation variables ( $y_{jk}$  and  $x_{ij}$ ) as:

$$E[w_j(\mathbf{x}, \mathbf{y})] = \frac{(1 + \sum_{k \in K} Cv_{jk}^2 y_{jk}) \sum_{i \in I} \lambda_i x_{ij}}{2 \sum_{k \in K} \mu_{jk} y_{jk} (\sum_{k \in K} \mu_{jk} y_{jk} - \sum_{i \in I} \lambda_i x_{ij})} + \frac{1}{\sum_{k \in K} \mu_{jk} y_{jk}} \quad (2)$$

8 The expected number of users in service or waiting for service at facility  $j$  is given, using  
 9 Little's law, as  $\Lambda_j E[w_j]$ . If  $d$  denotes the average waiting time cost per customer (henceforth  
 10 called unit queuing delay cost), then the *total delay/congestion cost* in the network can  
 11 be expressed as  $d \sum_{j \in J} \Lambda_j E[w_j(\mathbf{x}, \mathbf{y})] = \sum_{j \in J} \sum_{i \in I} \lambda_i x_{ij} E[w_j(\mathbf{x}, \mathbf{y})]$ . We assume there is  
 12 a variable access cost  $c_{ij}$  of providing service to users at node  $i$  from facility at site  $j$ .  
 13 The problem is to simultaneously determine: (i) the locations of the service facilities and  
 14 their corresponding capacity levels; (ii) the assignment of users to located service facilities,  
 15 such that the total system-wide cost, consisting of cost of opening service facilities with  
 16 appropriate capacities, cost of accessing service facilities by users and cost associated with  
 17 customers' waiting, is minimized.

1 To model the problem, we first summarize the notations:

2

3 *Indices, Sets and Parameters:*

$i$  : Index for user nodes,  $i \in I$

$j$  : Index for potential facility sites,  $j \in J$

$k$  : Index for capacity levels at facility sites,  $k \in K$

$f_{jk}$  : Fixed set up cost of locating a facility with capacity level  $k$  at site  $j$

$c_{ij}$  : Cost of providing service to user node  $i$  from facility at site  $j$

$\lambda_i$  : Mean demand rate of service requests from user node  $i$

4

$\mu_{jk}$  : Mean service rate at facility site  $j$  with capacity level  $k$

$\sigma_{jk}^2$  : Variance of *service times* at facility site  $j$  with capacity level  $k$

$d$  : Unit queueing delay cost per user

$D_{ij}$  : Distance between user node  $i$  and service facility  $j$

$C_{ij}$  : Set of all service facility locations that are closer to user node  $i$  than the facility located at  $j$ ;  $C_{ij} = \{l | D_{il} < D_{ij}\}$

5 *Decision Variables:*

$y_{jk} = 1$  if facility at site  $j$  is open and equipped with a capacity level  $k \in K$ ,  
0 otherwise

6

$x_{ij} = 1$  if the demand for service from user node  $i$  is satisfied by facility  $j$ , 0 otherwise

7 *Derived Variables:*

$\mu_j$  : Mean service rate at facility site  $j$ ;  $\mu_j = \sum_{k \in K} \mu_{jk} y_{jk}$

$\Lambda_j$  : aggregate demand arrival rate at facility  $j$ ;  $\Lambda_j = \sum_{i \in I} \lambda_i x_{ij}$

8

$Cv_j$  : Coefficient of variation of service times at facility site  $j$ ;  $Cv_j = \sigma_j / \mu_j$

$\rho_j$  : Average utilization of service facility at site  $j$ ;  $\rho_j = \Lambda_j / \mu_j$

9

$E[w_j]$  : Expected waiting time (including time spent in service) at facility  $j$

The resulting non-linear integer program (IP) model of the problem is as follows:

$$[P] : Z(\mathbf{x}, \mathbf{y}) = \min_{\mathbf{x}, \mathbf{y}} \sum_{j \in J} \sum_{k \in K} f_{jk} y_{jk} + \sum_{i \in I} \sum_{j \in J} c_{ij} x_{ij} + d \sum_{j \in J} \sum_{i \in I} \lambda_i x_{ij} E[w_j(\mathbf{x}, \mathbf{y})] \quad (3)$$

$$\text{s.t.} \quad \sum_{i \in I} \lambda_i x_{ij} \leq \sum_{k \in K} \mu_{jk} y_{jk} \quad \forall j \quad (4)$$

$$\sum_{j \in J} x_{ij} = 1 \quad \forall i \quad (5)$$

$$\sum_{k \in K} y_{jk} \leq 1 \quad \forall j \quad (6)$$

$$x_{ij}, y_{jk} \in \{0, 1\} \quad \forall i, j, k \quad (7)$$

10 The first term in the objective function (3) is the amortized cost of opening facilities at  
11 appropriate capacity levels. The second term is the cost of accessing service facilities, while

1 the third term captures the cost of users' waiting at facilities. The expression for  $E[w_j(\mathbf{x}, \mathbf{y})]$   
2 in the objective function is given by (2). Constraint set (4) ensures that the total demand  
3 allocated to any service facility is within its capacity. Note that constraint set (4) will be  
4 non-binding at optimality, else the term  $E[w_j(\mathbf{x}, \mathbf{y})]$  in the objective function goes to infin-  
5 ity. This ensures the stability of the queueing system ( $\rho_j = \Lambda_j/\mu_j < 1$ ) at each facility  $j$ .  
6 Constraint set (5) ensures that each user node is assigned to only one of the open facilities  
7 for its service. Constraint set (6) states that at most one of the multiple capacity levels  
8 is selected at a facility. Constraint set (7) imposes binary restrictions on the location and  
9 allocation variables.

10  
11 (3)-(7) models a directed choice environment. This can also be adapted to the user choice  
12 environment by explicitly specifying the decision model for the users' choice of facilities. For  
13 this, the most common assumption used in the literature is that users always choose the  
14 closest service facility. Under this assumption, the users' choice of facilities is generally  
15 modelled using the following set of constraints, called the *closest assignment constraints*  
16 (originally proposed by Rojeski and ReVelle, 1970):

$$x_{ij} \geq \sum_{k \in K} y_{jk} - \sum_{l \in C_{ij}} \sum_{k \in K} y_{lk} \quad (8)$$

17 where  $C_{ij} = \{l | D_{il} < D_{ij}\}$  is the set of all service facility locations that are closer to user  
18 node  $i$  than the facility located at  $j$ . Alternative ways of modelling the closest assignment  
19 constraints are discussed by Berman et al. (2006) and Espejo et al. (2012).

20  
21 The presence of the non-linear term  $\sum_{i \in I} \sum_{j \in J} \lambda_i x_{ij} E[w_j(\mathbf{x}, \mathbf{y})]$  in the objective function  
22 makes  $[P]$  challenging to solve. In the next section, we first present an approach to linearize  
23 the expression for the total waiting time spent by the users at a facility. We then present an  
24 exact solution procedure, based on a constraint generation algorithm, to solve the linearized  
25 model. We present the proposed solution method only with respect to the directed choice  
26 environment. However, the same solution method is also applicable to the user choice  
27 method with the addition of constraint (8).

### 28 3. Model Linearization

The non-linear term in the objective function of  $[P]$  can be written, using (1), as:

$$\Lambda_j E[w_j] = \left( \frac{1 + Cv_j^2}{2} \right) \frac{\Lambda_j^2}{\mu_j(\mu_j - \Lambda_j)} + \frac{\Lambda_j}{\mu_j} = \left( \frac{1 + Cv_j^2}{2} \right) \frac{\rho_j^2}{(1 - \rho_j)} + \rho_j \quad (9)$$

To linearize (9), it can be rewritten, upon rearranging its terms, as:

$$\Lambda_j E[w_j] = \frac{1}{2} \left\{ (1 + Cv_j^2) \frac{\rho_j}{1 - \rho_j} + (1 - Cv_j^2) \rho_j \right\} \quad (10)$$

1 Let us define a set of nonnegative auxiliary variables,  $U_j$ , such that:

$$U_j = \frac{\rho_j}{1 - \rho_j} = \frac{\Lambda_j}{\mu_j - \Lambda_j} = \frac{\sum_{i \in I} \lambda_i x_{ij}}{\sum_{k \in K} \mu_{jk} y_{jk} - \sum_{i \in I} \lambda_i x_{ij}} \quad (11)$$

2 which implies:

$$\rho_j = \frac{U_j}{1 + U_j} \quad (12)$$

Using  $\rho_j = \Lambda_j / \mu_j$ , the total demand  $\Lambda_j$  at facility  $j$  can be expressed as:

$$\Lambda_j = \sum_{i \in I} \lambda_i x_{ij} = \rho_j \mu_j = \rho_j \sum_{k \in K} \mu_{jk} y_{jk} = \sum_{k \in K} \mu_{jk} z_{jk}, \quad \text{where } z_{jk} = \begin{cases} \rho_j, & \text{if } y_{jk} = 1 \\ 0, & \text{otherwise} \end{cases}$$

Hence, the total demand at any facility  $j$  can be expressed as:

$$\sum_{i \in I} \lambda_i x_{ij} = \sum_{k \in K} \mu_{jk} z_{jk} \quad \forall j$$

Given that any facility can have at most one capacity level, i.e., there exists at most one  $k = k'$  such that  $y_{jk'} = 1$ , while  $y_{jk} = 0 \forall k \neq k'$ . Further,  $\rho_j < 1$ . Using this information,  $z_{jk}$  can alternatively be expressed using the following set of constraints:

$$\begin{aligned} z_{jk} &\leq y_{jk} && \forall j, k \\ \sum_{k \in K} z_{jk} &= \rho_j && \forall j \\ z_{jk} &\geq 0 && \forall j, k \end{aligned}$$

With the above substitutions in (10), the expression for  $\Lambda_j E[w_j]$  reduces to:

$$\begin{aligned} \Lambda_j E[w_j] &= \frac{1}{2} \left\{ \left( 1 + \sum_{k \in K} Cv_{jk}^2 y_{jk} \right) U_j + \left( 1 - \sum_{k \in K} Cv_{jk}^2 y_{jk} \right) \rho_j \right\} \\ &= \frac{1}{2} \left( U_j + \sum_{k \in K} Cv_{jk}^2 w_{jk} + \rho_j - \sum_{k \in K} Cv_{jk}^2 z_{jk} \right), \quad \text{where } w_{jk} = \begin{cases} U_j, & \text{if } y_{jk} = 1 \\ 0, & \text{otherwise} \end{cases} \end{aligned}$$

Again, using the fact that there exists at most one  $k = k'$  such that  $y_{jk'} = 1$ , while



$y_{jk} = 0 \forall k \neq k'$ ,  $w_{jk}$  can alternatively be expressed using the following set of constraints:

$$\begin{aligned} w_{jk} &\leq M y_{jk} && \forall j, k \\ \sum_{k \in K} w_{jk} &= U_j && \forall j \\ w_{jk} &\geq 0 && \forall j, k \end{aligned}$$

1 where  $M$  is a large number (Big-M). We now state a Lemma that helps us linearize the  
2 above non-linear model  $[P]$ .

3

4 **Lemma 1:** *The function  $\rho_j(U_j) = \frac{U_j}{1+U_j}$  is concave in  $U_j \in [0, \infty)$ .*

5 **Proof:**

6 Differentiating  $\rho_j$  w.r.t.  $U_j$ , we get the first derivative,  $\frac{\delta \rho_j}{\delta U_j} = \frac{1}{(1+U_j)^2} > 0$ , and the second  
7 derivative,  $\frac{\delta^2 \rho_j}{\delta U_j^2} = \frac{-2}{(1+U_j)^3} < 0$ , which proves that the function  $\rho_j(U_j)$  is concave in  $U_j$ . ■

8

Lemma 1 implies that for a given set of points indexed by  $h$ ,  $h \in H$ , the function  $\rho_j(U_j)$   
can be approximated arbitrarily close by a set of piecewise linear functions that are tangent  
to  $\rho_j$  at points  $\{U_j^h\}_{h \in H}$ , such that:

$$\rho_j = \min_{h \in H} \left\{ \frac{1}{(1+U_j^h)^2} U_j + \frac{(U_j^h)^2}{(1+U_j^h)^2} \right\} \quad (13)$$

This is equivalent to the following set of constraints:

$$\rho_j \leq \frac{1}{(1+U_j^h)^2} U_j + \frac{(U_j^h)^2}{(1+U_j^h)^2}, \quad \forall j, h \in H \quad (14)$$

9 provided  $\exists h \in H$  such that (14) holds with equality.

Using the above substitutions result in the following linear mixed integer program (MIP)

reformulation of [P]:

$$[P(H)] : \min \sum_{j \in J} \sum_{k \in K} f_{jk} y_{jk} + \sum_{i \in I} \sum_{j \in J} c_{ij} x_{ij} + \frac{d}{2} \sum_{j \in J} \left\{ U_j + \rho_j + \sum_{k \in K} C v_{jk}^2 (w_{jk} - z_{jk}) \right\} \quad (15)$$

s.t. (5) – (7), (14)

$$\sum_{i \in I} \lambda_i x_{ij} - \sum_{k \in K} \mu_{jk} z_{jk} = 0 \quad \forall j \quad (16)$$

$$z_{jk} \leq y_{jk} \quad \forall j, k \quad (17)$$

$$\sum_{k \in K} z_{jk} = \rho_j \quad \forall j \quad (18)$$

$$w_{jk} \leq M y_{jk} \quad \forall j, k \quad (19)$$

$$\sum_{k \in K} w_{jk} = U_j \quad \forall j \quad (20)$$

$$0 \leq z_{jk}, \rho_j \leq 1; \quad w_{jk}, U_j \geq 0 \quad \forall j, k \quad (21)$$

1 The linear MIP model  $[P(H)]$  has  $2(|J| + |J| * |K|)$  additional continuous variables com-  
 2 pared to the non-linear IP model  $[P]$ . Further,  $[P(H)]$  has  $(|I| + 4 * |J| + 2 * |J| * |K| + |J| * |H|)$   
 3 constraints, as opposed to only  $(|I| + 2 * |J|)$  constraints in  $[P]$ . Hence, the non-linearity of  
 4  $[P]$  is eliminated at the expense of having to deal with a large number of additional variables  
 5 and constraints in  $[P(H)]$ .

6  
 7 Equivalence between  $[P]$  and  $[P(H)]$  requires that  $\exists h \in H$  such that (14) holds with  
 8 equality. Proposition 1 states that this condition will always be satisfied at optimality.

9  
 10 **Proposition 1:** *In the linearized model  $[P(H)]$ , at least one of the constraints in (14) will*  
 11 *be binding at optimality.*

**Proof:**

Upon rearranging the terms, (14) can be rewritten as:

$$U_j \geq (1 + U_j^h)^2 \rho_j - (U_j^h)^2, \quad \forall j, h \in H \quad (22)$$

Since  $U_j$  appears in the objective function of  $[P(H)]$  with a positive coefficient,  $[P(H)]$  attains its minimum value only when  $U_j$  is minimized. This implies that  $\forall j \in J, \exists h \in H$  such that (22) holds with equality if  $(1 + U_j^h)^2 \rho_j - (U_j^h)^2 \geq 0$ , else  $U_j = 0$  if  $(1 + U_j^h)^2 \rho_j - (U_j^h)^2 < 0$ .

Further,

$$\begin{aligned}
0 &\leq (1 + U_j^h)^2 \rho_j - (U_j^h)^2 \\
&= (\rho_j - 1)(U_j^h)^2 + 2\rho_j U_j^h + \rho_j \\
&\Leftrightarrow U_j^h \in \left[0, \frac{\rho_j + \sqrt{\rho_j}}{1 - \rho_j}\right] \quad \forall j \in J, h \in H \text{ (since } \rho_j \leq 1, U_j \geq 0)
\end{aligned}$$

Thus, to prove that  $\forall j \in J, \exists h \in H$  such that (22) holds with equality, it is sufficient to show that  $U_j^h \in \left[0, \frac{\rho_j + \sqrt{\rho_j}}{1 - \rho_j}\right]$ . Since  $U_j^h$  is an approximation to  $U_j$ , we obtain:

$$\begin{aligned}
0 \leq U_j^h &\approx U_j = \frac{\rho_j}{1 - \rho_j} \\
&\leq \frac{\rho_j + \sqrt{\rho_j}}{1 - \rho_j}
\end{aligned}$$

1 This proves that  $\forall j \in J, \exists h \in H$  such that (22) holds with equality. ■

2

### 3.1. Special Cases

In many cases, the service at facilities involve repeated steps without much variation, i.e.,  $Cv_{jk} = 0$  (such that each service facility is modeled as an  $M/D/1$  queuing system). For such deterministic service times, the users' expected waiting time at facility  $j$  is given by:  $\Lambda_j E[w_j] = \frac{1}{2} \left( \frac{\Lambda_j}{\mu_j - \Lambda_j} + \frac{\Lambda_j}{\mu_j} \right) = \frac{1}{2}(U_j + \rho_j)$  and the resulting linear MIP model is as follows:

$$\begin{aligned}
[P(H)_{Cv=0}] : \min & \quad \sum_{j \in J} \sum_{k \in K} f_{jk} y_{jk} + \sum_{i \in I} \sum_{j \in J} c_{ij} x_{ij} + \frac{d}{2} \sum_{j \in J} (U_j + \rho_j) \\
\text{s.t.} & \quad (4) - (7), (14) \\
& \quad 0 \leq \rho_j \leq 1; \quad U_j \geq 0 \quad \forall j
\end{aligned}$$

For exponentially distributed service times at the facilities, i.e.,  $Cv_{jk} = 1$  ( $M/M/1$  case), the expression is given by:  $\Lambda_j E[w_j] = \frac{\rho_j}{1 - \rho_j} = U_j$ , and the linear model reduces to:

$$\begin{aligned}
[P(H)_{Cv=1}] : \min & \quad \sum_{j \in J} \sum_{k \in K} f_{jk} y_{jk} + \sum_{i \in I} \sum_{j \in J} c_{ij} x_{ij} + d \sum_{j \in J} U_j \\
\text{s.t.} & \quad (4) - (7), (14) \\
& \quad 0 \leq \rho_j \leq 1; \quad U_j \geq 0 \quad \forall j
\end{aligned}$$

## 4. Solution Approach

5 We state the following two propositions, which are used in the development of the solu-  
6 tion algorithm for  $[P(H)]$ .

7

**Proposition 2:** For any given subset of points  $\{U_j^h\}_{H^q \subset H}$ , (23) provides a lower bound on the optimal objective function value of  $[P]$ , where  $v(\bullet)$  is the objective function value of the problem  $(\bullet)$  and  $(\mathbf{x}^p, \mathbf{y}^q, \rho^q, \mathbf{w}^q, \mathbf{z}^q, \mathbf{U}^q)$  is the optimal solution to  $[P(H^q)]$ .

$$LB = v(P(H^q)) = \sum_{j \in J} \sum_{k \in K} f_{jk} y_{jk}^q + \sum_{i \in I} \sum_{j \in N} c_{ij} x_{ij}^q + \frac{d}{2} \sum_{j \in J} \left\{ U_j^q + \rho_j^q + \sum_{k \in K} C v_{jk}^2 (w_{jk}^q - z_{jk}^q) \right\} \quad (23)$$

1 **Proof:**

2 Since  $[P(H^q)]$  is a relaxation of the full problem  $[P(H)]$ , the objective function value of  
3  $[P(H^q)]$ , given by (23), provides a lower bound on the optimal objective function value of  
4  $[P(H)]$ , and hence on the optimal objective function value of  $[P]$ .  $\blacksquare$

5

**Proposition 3:** For any given subset of points  $\{U_j^h\}_{H^q \subset H}$ , (24) provides an upper bound on the optimal objective function value of  $[P]$ , where  $(\mathbf{x}^p, \mathbf{y}^q)$  is the optimal solution to  $P[(H^q)]$ .

$$UB = Z(\mathbf{x}^q, \mathbf{y}^q) = \sum_{j \in J} \sum_{k \in K} f_{jk} y_{jk}^q + \sum_{i \in I} \sum_{j \in J} c_{ij} \lambda_i x_{ij}^q + d \sum_{j \in J} \left\{ \frac{(1 + \sum_{k \in K} C v_{jk}^2 y_{jk}) (\sum_{i \in I} \lambda_i x_{ij})^2}{2 \sum_{k \in K} \mu_{jk} y_{jk} (\sum_{k \in K} \mu_{jk} y_{jk} - \sum_{i \in I} \lambda_i x_{ij})} + \frac{\sum_{i \in I} \lambda_i x_{ij}}{\sum_{k \in K} \mu_{jk} y_{jk}} \right\} \quad (24)$$

6 **Proof:**

7 For any subset of points  $\{U_j^h\}_{H^q \subset H}$ , the optimal solution  $(\mathbf{x}^p, \mathbf{y}^q)$  to  $[P(H^q)]$  is also a fea-  
8 sible solution to  $[P]$  as all the constraints of  $[P]$  are also contained in  $[P(H^q)]$ . Hence, the  
9 objective function of  $[P]$  evaluated at  $(\mathbf{x}^p, \mathbf{y}^q)$ , which is given by (24), provides an upper  
10 bound on the optimal objective of  $[P]$ .  $\blacksquare$

11

#### 12 4.1. Solution Algorithm

13 Although there are a large number of constraints/cuts (14) in the linear MIP model  
14  $[P(H)]$ , it is not necessary to generate all of them. Instead, it suffices to start with a  
15 subset  $H^1 \subset H$  of these cuts, where  $H^1$  may be empty or chosen a priori, and generate  
16 the rest as needed. Our preliminary computational experiments, presented in Table 1,  
17 suggest a much faster convergence of the algorithm when  $H^1$  is non-empty. We, therefore,  
18 use a carefully chosen subset  $H^1$  of initial cuts in all our subsequent experiments. The  
19 subset of points  $\{U_j^h\}_{H^1 \subset H}$ , required to obtain the initial subset of cuts, is generated to  
20 approximate the function  $\rho_j(U_j) = \frac{U_j}{1+U_j}$  using its tangents  $\hat{\rho}_j(U_j)$  at these points such that  
21 the approximation error, measured as  $\hat{\rho}_j(U_j) - \rho_j(U_j)$ , is at most 0.001 (Elhedhli, 2005).  
22 The resulting  $[P(H^1)]$  is solved, giving a solution  $(\mathbf{x}^1, \mathbf{y}^1, \rho^1, \mathbf{w}^1, \mathbf{z}^1, \mathbf{U}^1)$ . The lower bound  
23  $(LB^1)$  and upper bound  $(UB^1)$  are computed using (23) and (24) respectively. If  $UB^1$   
24 equals  $LB^1$  within some accepted tolerance  $(\epsilon)$ , then  $(\mathbf{x}^1, \mathbf{y}^1)$  is an optimal solution to  $[P]$ ,

1 and the algorithm stops. Otherwise, a new set of points  $\{U_j^h\}$  is generated using the current  
2 solution as  $U_j^{h_{new}} = \frac{\sum_{i \in I} \lambda_i x_{ij}^1}{\sum_{k \in K} \mu_{jk} y_{jk}^1 - \sum_{i \in I} \lambda_i x_{ij}^1} \quad \forall j$ . A new set of constraints/cuts of the form (14)  
3 is generated at these new points, which are appended to  $[P(H^1)]$  to give  $[P(H^2)]$ . Then,  
4  $[P(H^2)]$  is solved to yield a solution  $(\mathbf{x}^2, \mathbf{y}^2, \rho^2, \mathbf{w}^2, \mathbf{z}^2, \mathbf{U}^2)$  and a lower bound  $LB^2$ . Since  
5 the upper bound, as given by (24), changes non-monotonically, the new upper bound  $UB^2$   
6 is retained as  $\min\{UB^1, Z(\mathbf{x}^2, \mathbf{y}^2)\}$ . If  $UB^2$  equals  $LB^2$  within the accepted tolerance ( $\epsilon$ ),  
7 then the algorithm terminates with  $(\mathbf{x}^2, \mathbf{y}^2)$  as an optimal solution. Otherwise, the above  
8 process is repeated until  $UB^q$  equals  $LB^q$  within ( $\epsilon$ ) for some iterations  $q$ . The steps of the  
9 algorithm are outlined below:

---

**Algorithm 1** Constraint Generation Algorithm for  $[P(H)]$

---

- 1:  $q \leftarrow 1; UB^{q-1} \leftarrow +\infty; LB^{q-1} \leftarrow -\infty$ .
  - 2: Choose an initial set of points  $\{U_j^h\}_{h \in H^q}$  to approximate the function  $\rho_j(U_j) = U_j/1+U_j$ .
  - 3: **while**  $(UB^{q-1} - LB^{q-1})/UB^{q-1} > \epsilon$  **do**
  - 4:   Solve  $P(H^q)$ , and obtain its optimal solution  $(\mathbf{x}^q, \mathbf{y}^q, \rho^q, \mathbf{w}^q, \mathbf{z}^q, \mathbf{U}^q)$ .
  - 5:   Update the lower bound:  $LB^q \leftarrow v(P(H^q))$  using (23).
  - 6:   Update the upper bound:  $UB^q \leftarrow \min\{UB^{q-1}, Z(\mathbf{x}^q, \mathbf{y}^q)\}$  using (24).
  - 7:   Generate a new set of points  $U_j^{h_{new}} = \frac{\sum_{i \in I} \lambda_i x_{ij}^q}{\sum_{k \in K} \mu_{jk} y_{jk}^q - \sum_{i \in I} \lambda_i x_{ij}^q} \quad \forall j$  using (11).
  - 8:    $H^{q+1} \leftarrow H^q \cup \{h_{new}\}$ .
  - 9:    $q \leftarrow q + 1$
  - 10: **end while**
- 

10 **Proposition 4:** *The constraint generation algorithm to solve  $[P(H)]$  is finite.*

11 **Proof:**

12 Since  $x_{ij}, y_{jk} \in \{0, 1\} \quad \forall i, j, k$  and  $U_j = \frac{\sum_{i \in I} \lambda_i x_{ij}}{\sum_{k \in K} \mu_{jk} y_{jk} - \sum_{i \in I} \lambda_i x_{ij}}$ ,  $U_j$  can take only a finite set of  
13 values. Therefore, in order to prove the finiteness of Algorithm 1, it is sufficient to prove  
14 that the generated values of  $U_j^h$  are not repeated.

15

Consider an iteration  $q$ , wherein the Algorithm 1 has not yet converged, that is  $UB^q > LB^q$ . Further, suppose  $(\mathbf{x}^2, \mathbf{y}^2, \rho^2, \mathbf{w}^2, \mathbf{z}^2, \mathbf{U}^2)$  is a solution to  $[P(H^q)]$ . The new points  $U_j^{h_{new}}$  generated at iteration  $q$  are given by:

$$U_j^{h_{new}} = \frac{\sum_{i \in I} \lambda_i x_{ij}^q}{\sum_{k \in K} \mu_{jk} y_{jk}^q - \sum_{i \in I} \lambda_i x_{ij}^q} \quad \forall i, j, k$$

16 Suppose the values of  $U_j^{h_{new}}$  are already generated at iteration  $q^0 < q \quad \forall j \in J$ . Then,

17 We have  $U_j^{h_{new}}$  generated in iteration  $q$  is given by :  $U_j^{h_{new}} = \frac{\sum_{i \in I} \lambda_i x_{ij}^q}{\sum_{k \in K} \mu_{jk} y_{jk}^q - \sum_{i \in I} \lambda_i x_{ij}^q}$

$$(14) \Rightarrow \frac{U_j^{h_{new}}}{1 + U_j^{h_{new}}} \leq \frac{1}{1 + U_j^{h_{new}}} U_j^q + \left( \frac{U_j^{h_{new}}}{1 + U_j^{h_{new}}} \right)^2 \quad \forall j$$

$$\Rightarrow U_j^{h_{new}} \leq U_j^q \quad \forall j$$

We now have:

$$\begin{aligned}
LB^q &= \sum_{j \in J} \sum_{k \in K} f_{jk} y_{jk}^q + \sum_{i \in I} \sum_{j \in J} c_{ij} x_{ij}^q + \frac{d}{2} \sum_{j \in J} \left\{ U_j^q + \rho_j^q + \sum_{k \in K} C v_{jk}^2 (w_{jk}^q - z_{jk}^q) \right\} \\
&\geq \sum_{j \in J} \sum_{k \in K} f_{jk} y_{jk}^q + \sum_{i \in I} \sum_{j \in J} c_{ij} x_{ij}^q + \frac{d}{2} \sum_{j \in J} \left\{ U_j^{h_{new}} + \rho_j^q + \sum_{k \in K} C v_{jk}^2 (w_{jk}^q - z_{jk}^q) \right\} \\
&= \sum_{j \in J} \sum_{k \in K} f_{jk} y_{jk}^q + \sum_{i \in I} \sum_{j \in J} c_{ij} x_{ij}^q + \frac{d}{2} \sum_{j \in J} \left\{ \frac{\sum_{i \in I} \lambda_i x_{ij}^q}{\sum_{k \in K} \mu_{jk} y_{jk}^q - \sum_{i \in I} \lambda_i x_{ij}^q} + \rho_j^q + \sum_{k \in K} C v_{jk}^2 (w_{jk}^q - z_{jk}^q) \right\} \\
&= \sum_{k \in K} f_{jk} y_{jk}^q + \sum_{i \in I} \sum_{j \in J} c_{ij} x_{ij}^q + d \left\{ \frac{(1 + \sum_{k \in K} C v_{jk}^2 y_{jk}^q) (\sum_{i \in I} \lambda_i x_{ij}^q)^2}{2 \sum_{k \in K} \mu_{jk} y_{jk}^q (\sum_{k \in K} \mu_{jk} y_{jk}^q - \sum_{i \in I} \lambda_i x_{ij}^q)} + \frac{\sum_{i \in I} \lambda_i x_{ij}^q}{\sum_{k \in K} \mu_{jk} y_{jk}^q} \right\} \\
&= Z(\mathbf{x}^q, \mathbf{y}^q) \\
&\geq \min\{UB^{q-1}, Z(\mathbf{x}^q, \mathbf{y}^q)\} \\
&= UB^q
\end{aligned}$$

1

2 This contradicts are initial assumption  $UB^q > LB^q$ . Therefore, at any given iteration,  
3  $U_j^{h_{new}}$  will be different from the previously generated points at least for any one  $j$ . Fur-  
4 thermore, the number of values that  $U_j^h$  can take is finite. Hence, the algorithm should  
5 terminate in a finite number of iterations. ■

6

## 7 5. Computational Results

8 We present our computational experiments with the solution approach proposed in Sec-  
9 tion 4. The solution procedures are coded in C++ (Visual Studio 2010), while  $[P(H^q)]$  at  
10 every iteration  $q$  is solved using IBM ILOG CPLEX 12.4 on a personal laptop with Intel  
11 Core i5-3230M, 2.60 GHz CPU; 4 GB RAM; and Windows 64-bit operating system. The  
12 test instances are generated using a combination of the schemes used by Amiri (1998) and  
13 Holmberg et al. (1999), described in detail in Section 5.1. We generate four sets of test prob-  
14 lems by varying the combination of the number of user nodes and the number of potential  
15 facility locations ( $|I|, |J|$ ) as (100, 10), (200, 15), (300, 20), and (400, 25). The number of  
16 capacity levels ( $|K|$ ) is set at 5, and the tolerance level for optimality  $\epsilon$  is set at  $10^{-5}$  in all  
17 the experiments.

### 18 5.1. Data Generation

19 The co-ordinates of the nodes representing user nodes are randomly generated using a  
20 uniform distribution  $U \sim (10, 300)$ . The mean demand rate at any user node  $i$  is ran-  
21 domly generated as  $\lambda_i \sim U(10, 50)$ , following. The locations of the potential facilities are  
22 obtained from the solution to a  $p$ -median facility location problem. The other parameters  
23 are generated as follows:

- 1 • *Capacity Levels* ( $\mu_{jk}$ ) are set as:  $\mu_{j1} = 0.50 * \mu_{j3}$ ;  $\mu_{j2} = 0.75 * \mu_{j3}$ ;  $\mu_{j4} = 1.25 * \mu_{j3}$ ;  
2  $\mu_{j5} = 1.50 * \mu_{j3}$ , where  $\mu_{j3} = (1.25 \sum_{i \in I} \lambda_i) / (|J| * LR)$ . The Load Ratio (LR), which  
3 is defined as the average facility utilization if all the potential facilities are assigned  
4 capacity level 3 ( $k = 3$ ) to satisfy 125% of the total demand in the network, is set at  
5 0.60.
- 6 • *Fixed Costs* ( $f_{jk}$ ) are generated as:  $f_{j3} = FCR * E_{jj_0}$ , where  $E_{jj_0}$  is the Euclidean  
7 distance between the facility site  $j$  and a fixed point  $j_0 = (155, 155)$ , which is the center  
8 of the box within which the coordinates of user nodes are generated. FCR, called the  
9 Fixed Cost Ratio, is a constant set at 40.  $f_{j1} = 0.60 * f_{j3}$ ,  $f_{j2} = 0.85 * f_{j3}$ ,  $f_{j4} = 1.15 * f_{j3}$ ,  
10  $f_{j5} = 1.35 * f_{j3}$ . The chosen values of fixed cost for different capacity levels exhibit  
11 both an underlying economy as well as diseconomy of scale. For example, for a 25%  
12 increase in capacity (corresponding to  $\mu_{j4}$  over  $\mu_{j3}$ ), the capacity cost increases only  
13 15%. However, for the a 50% increase in capacity (corresponding to  $\mu_{j5}$  over  $\mu_{j3}$ ), the  
14 capacity cost increases 35%.
- 15 • *Service Costs* ( $c_{ij}$ ) are generated as:  $c_{ij} = 5 * E_{ij}$ , where  $E_{ij}$  is the Euclidean distance  
16 between user node  $i$  and the facility site  $j$ .
- 17 • *Unit Queueing Delay Cost* ( $d$ ) is assumed to be one of the values from the set  $\{1, 10,$   
18  $25, 50, 100, 250, 500, 1000, 5000\}$ .

## 19 5.2. Analysis of Computational Results

20 Table 1 compares the performance of the proposed solution approach with a priori cuts  
21 versus without a priori cuts for test instances with 100 user nodes and 10 facilities. The  
22 table reports the number of iterations required (#Iter) and the computation time in seconds  
23 (CPU) for different values of coefficient of variation of service times ( $Cv$ ) and the unit queu-  
24 ing delay cost ( $d$ ). The table also reports the percentage reduction in computation time as a  
25 result of adding a priori cuts, which is computed as:  $(\text{CPU time without a priori cuts} - \text{CPU}$   
26  $\text{time with a priori cuts}) \times 100 / (\text{CPU time without a priori cuts})$ . As described in Section 4.1,  
27 a priori cuts for the function  $\rho(U) = U / (1 + U)$  are generated based on the piecewise linear  
28 approximation  $\hat{\rho}(U)$  of the function  $\rho(U)$  such that the approximation error (measured by  
29  $\hat{\rho}(U) - \rho(U)$ ) is at most 0.001 (Elhedhli, 2005). Figure 1 shows for ( $|I| = 100, |J| = 10$ ) the  
30 percentage reduction in the number of iterations and the CPU times as a result of addition  
31 of a priori cuts for different values of  $d$  and  $Cv$ .

32  
33 Following observations can be made from Table 1 and Figure 1. The results in Table  
34 1 show that without a priori cuts, the problem takes, on an average, 949 seconds and 9  
35 iterations to solve. The addition of a priori cuts reduces these numbers to less than 4  
36 seconds of CPU time and less than 3 iterations. The percentage reduction in CPU time due  
37 to the addition of a priori cuts varies between 24.21% to 99.97%, with an average reduction

1 of 75.25%. Plots in Figure 1 further show that the benefits of a priori cuts, in terms of %  
2 reduction in CPU time and number of iterations, increase significantly with an increase in the  
3 unit queuing delay cost ( $d$ ). This is because, as Table 1 suggests, without a priori cuts, the  
4 computation time and the number of iterations required by the proposed solution approach  
5 increases significantly with an increase in  $d$ . However, with the addition of a priori cuts, the  
6 computation time and the number of iterations required are almost *insensitive* to  $d$ . Tables  
7 2-5 summarize the computational results for four sets of instances ( $|I| = 100, |J| = 10$ ;  
8  $|I| = 200, |J| = 15$ ;  $|I| = 300, |J| = 20$ ;  $|I| = 400, |J| = 25$ ), for different values of the unit  
9 queuing delay cost ( $d = 1, 10, 25, 50, 100, 250, 500, 1000, 5000$ ) and coefficient of variation  
10 of service times ( $Cv = 0, 1, 0.5, 1.5, 2, 2.5$ ). In solving each of the problem instances, we  
11 use a priori set of 32 cuts, which corresponds to a maximum approximation error of 0.001  
12 in the linear approximation of  $\rho(U)$ . The tables report for each problem instance the total  
13 cost (TC); fixed cost (FC), access cost (AC), and delay cost (DC), expressed as a percentage  
14 of the total cost; computation time in seconds (CPU); number of iterations for convergence  
15 (#Iter); number of facilities opened (#Facility); and the minimum, maximum, and average  
16 facility utilizations among the open facilities. Figure 2 shows the effect of varying  $d$  on FC,  
17 AC, DC, and TC for different values of  $Cv$ . Figure 3 shows the effect of varying  $d$  on the  
18 maximum and average facility utilizations. Following observations can be made from the  
19 figures:

- 20 • An increase in  $d$  or  $Cv$  increases TC, as expected. An increase in  $d$  also results in a  
21 higher DC, which is expected, provided the expected total number of users waiting  
22 at different service facilities in the network ( $\sum_j \Lambda_j E[w_j]$ ) remains unchanged with  
23 an increase in  $d$ . However, an increase in  $d$  implies a larger penalty for congestion  
24 (waiting customers), which forces the system to either attain more uniform utilization  
25 ( $\rho_j = \Lambda_j / \mu_j$ ) among the open facilities or to install a larger total service capacity in the  
26 network (either by opening more facilities or by installing larger capacities at fewer,  
27 more or the same number of opened facilities). In either case, the maximum facility  
28 utilization decreases and the average facility utilization either decreases or remains  
29 constant, as evident from Figure 3. This results in a decrease in the expected total  
30 number of users waiting in the network ( $\sum_j \Lambda_j E[w_j]$ ). We observe from our results  
31 that the percentage decrease in the expected total number of waiting users in the  
32 network is smaller compared to the percentage increase in  $d$ , such that the net effect  
33 is always an increase the total delay cost (DC).
- 34 • For a fixed network configuration (location and allocation of service facilities), an  
35 increase in  $Cv$  is expected to increase the expected total number of waiting users  
36 in the network, and hence increase DC. However, when the location and allocation  
37 of service facilities are also decision variables, as they are in the current problem,  
38 Figure 2 suggests that an increase in  $Cv$  may sometimes result in a decrease in DC,  
39 which initially appears to be counter intuitive. However, this can be justified as follows.



1 To counter the increase in congestion at a higher value of  $Cv$ , the system chooses to  
2 either attain more uniform utilization among the open facilities or to install a larger  
3 total service capacity in the system, thereby resulting in a decrease in the expected  
4 total number of waiting users in the network, and hence a decrease in DC.

- 5 • The fixed cost (FC) changes non-monotonically with an increase in  $d$  or  $Cv$ , which  
6 also seems counter intuitive since to counter the effect of increase in  $d$  or  $Cv$ , the  
7 system is expected to either attain more uniform utilization among the open facilities  
8 among the open facilities or to install a larger total service capacity in the system,  
9 neither of which should decrease FC. In case the system chooses to redistribute the  
10 total service capacity more uniformly among the open facilities to uniformize their  
11 utilizations, some facilities may exhibit economies of scale while others may exhibit  
12 diseconomies in scale in fixed costs (as described in Section 5.1), which may result  
13 in either an increase or decrease in FC depending on the net effect of economies and  
14 diseconomies of scale. Moreover, since the fixed cost of opening a service facility with a  
15 given capacity level ( $f_{jk}$ ) depends on its distance from a fixed point ( $j_0$ ), an increase in  
16  $d$  or  $Cv$  may result in an increase or decrease in FC depending on whether the system  
17 chooses to open service facilities closer to or farther from  $j_0$  at the higher value of  $d$   
18 or  $Cv$ . AC may also change non-monotonically with increase in  $d$  or  $Cv$  for similar  
19 reason.
- 20 • Comparison of CPU times across Tables 2-5 shows, as expected, that the computation  
21 time increases as the number of user nodes and potential facilities ( $|I|, |J|$ ) increases.
- 22 • The proposed solution approach succeeds in finding exact (within an optimality gap  
23 of  $10^{-5}$ ) solutions to the four sets of problem instances ( $|I| = 100, |J| = 10; |I| = 200,$   
24  $|J| = 15; |I| = 300, |J| = 20; |I| = 400, |J| = 25$ ) within an average CPU time of  
25 4, 29, 100 and 323 seconds, respectively, while the maximum CPU times for these  
26 sets are 17, 64, 668, and 1585 seconds, respectively. The average number of iterations  
27 are 2, 3, 3, and 3 whereas the maximum number of iterations are 3, 3, 3, and 6,  
28 respectively for the four sets of problem instances. This reveals the efficiency of the  
29 solution approach where the number of iterations/cuts imply that only a fraction of  
30 the exponential number constraints is required.

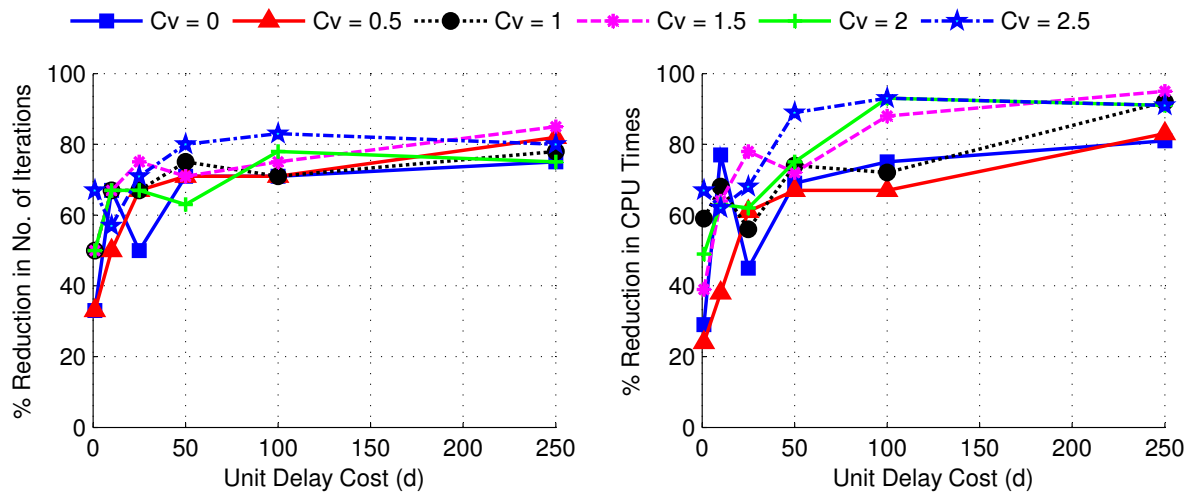


Figure 1: Effect of Adding a-priori Cuts on the Number of Iterations and the CPU Times of the Algorithm

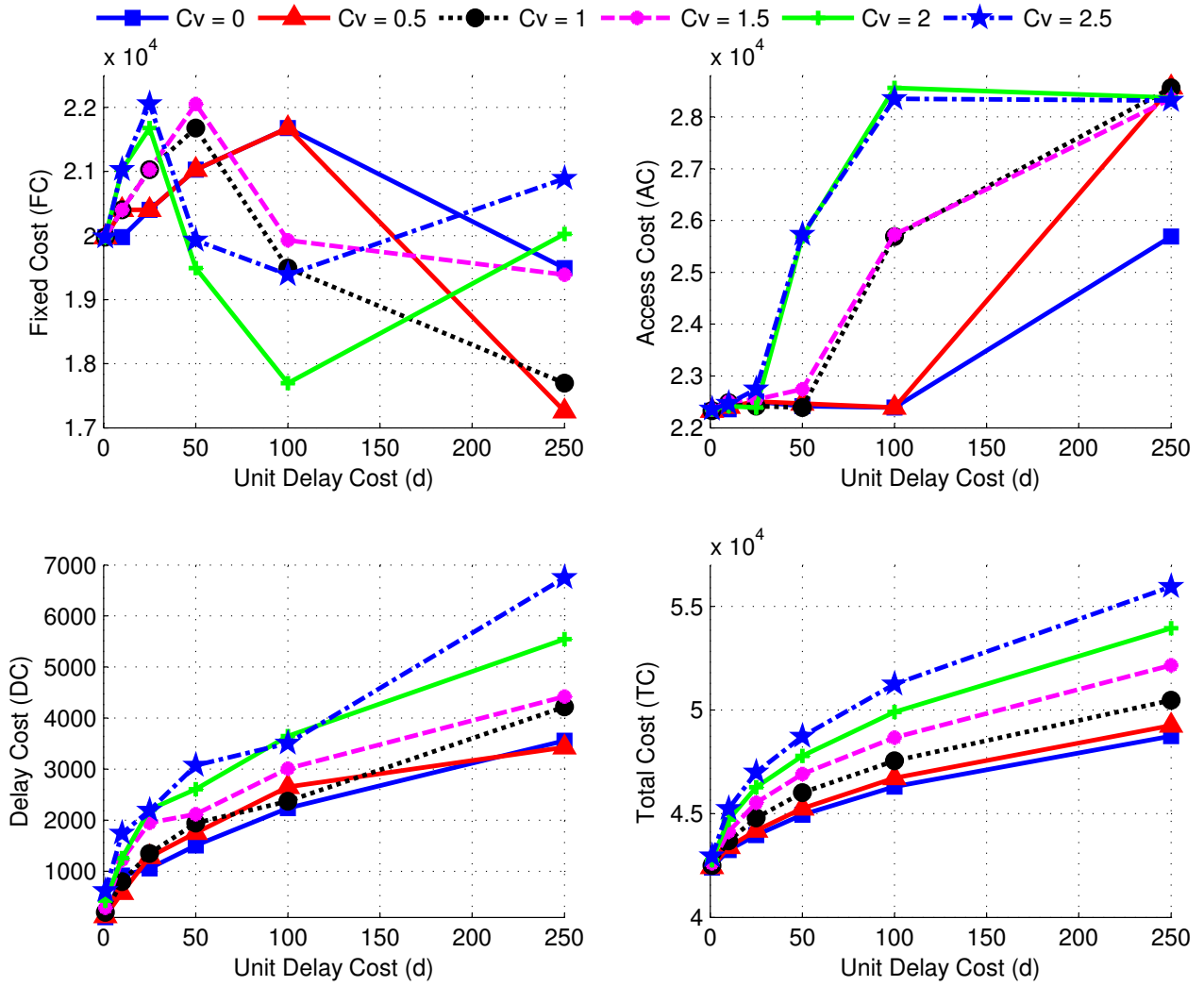


Figure 2: Effect of Varying Unit Delay Cost on the Fixed Cost, Access Cost, Delay Cost, and the Total Cost

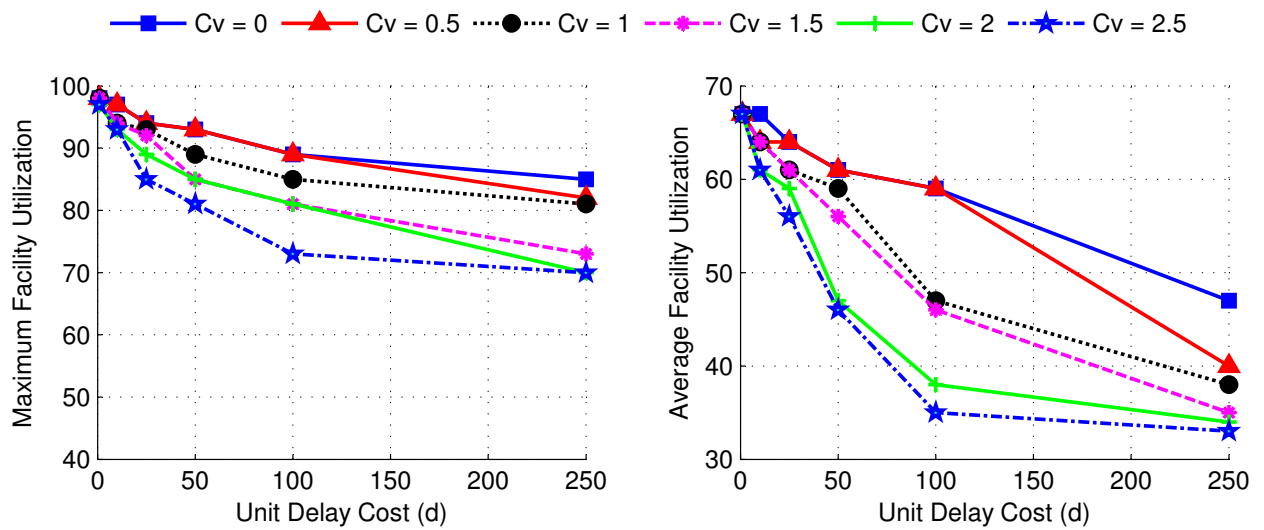


Figure 3: Effect of Varying Unit Delay Cost on Maximum and Average Facility Utilization

Table 1: Effect of Adding a-priori Cuts on the Performance of the Solution Method: Instances with 100 User Nodes and 10 Potential Facilities

$Cv$	$d$	Without a-priori cuts		With a-priori cuts		% Reduction in CPU times
		# Iter	CPU	# Iter	CPU	
0	1	3	2.24	2	1.58	29.45
	10	6	7.97	2	1.80	77.47
	25	6	5.36	3	2.97	44.64
	50	7	12.14	2	3.75	69.14
	100	7	13.54	2	3.42	74.73
	250	8	10.97	2	2.04	81.44
	500	8	12.06	2	1.59	86.85
	1000	12	21.67	3	2.53	88.32
	5000	20	24.78	2	2.48	90.01
0.5	1	3	2.07	2	1.57	24.21
	10	6	5.89	3	3.65	38.13
	25	6	6.14	2	2.40	60.95
	50	7	10.54	2	3.48	66.97
	100	7	12.34	2	4.02	67.42
	250	11	13.83	2	2.33	83.19
	500	12	19.13	3	2.85	85.10
	1000	14	22.14	3	2.26	89.81
	5000	18	28.03	2	1.81	93.54
1	1	4	3.16	2	1.31	58.54
	10	6	7.07	2	2.23	68.45
	25	6	8.67	2	3.80	56.18
	50	8	15.98	2	4.09	74.43
	100	7	12.06	2	3.33	72.39
	250	9	16.68	2	1.35	91.93
	500	16	22.20	3	2.38	89.30
	1000	13	24.42	2	2.26	90.77
	5000	15	38.65	3	8.25	78.65
1.5	1	4	2.63	2	1.61	39.00
	10	6	6.42	2	2.33	63.79
	25	8	14.73	2	3.18	78.43
	50	7	14.83	2	4.16	71.99
	100	8	17.07	2	1.98	88.39
	250	13	34.21	2	1.75	94.88
	500	13	25.71	2	2.16	91.60
	1000	11	29.30	2	3.20	89.07
	5000	12	60.59	3	16.79	72.29
2	1	4	2.80	2	1.42	49.45
	10	6	9.12	2	3.38	62.98
	25	6	11.73	2	4.42	62.30
	50	8	15.75	3	3.92	75.10
	100	9	19.78	2	1.43	92.76
	250	12	25.27	3	2.29	90.96
	500	12	22.45	2	2.86	87.26
	1000	12	38.50	3	4.33	88.75
	5000	12	162.50	3	15.60	90.40
2.5	1	6	5.19	2	1.73	66.63
	10	7	12.16	3	4.68	61.51
	25	7	12.23	2	3.91	68.05
	50	10	23.52	2	2.53	89.27
	100	12	31.28	2	2.05	93.46
	250	10	25.39	2	2.32	90.88
	500	11	26.78	3	4.90	81.72
	1000	11	49.96	3	4.78	90.44
	5000	14	50,154.70	3	13.14	99.97
	Min.	3	2.07	2	1.31	24.21
	Avg.	9	948.78	2	3.56	75.25
	Max.	20	50,154.70	3	16.79	99.97

Table 2: Computational Performance: Instances with 100 User Nodes and 10 Potential Facilities

$Cv$	$d$	$TC$	FC (%)	AC (%)	DC (%)	# Facility	% Utilization			# Iter	CPU
							Min.	Max.	Avg.		
0	1	42,405	47	53	0	7	89	98	67	2	2
	10	43,248	46	52	2	7	95	97	67	2	2
	25	43,962	46	51	2	7	89	94	64	3	3
	50	44,955	47	50	3	7	83	93	61	2	4
	100	46,308	47	48	5	7	81	89	59	2	3
	250	48,741	40	53	7	6	74	85	47	2	2
	500	51,438	34	56	10	5	72	81	38	2	2
	1,000	55,569	36	51	13	5	65	72	34	3	3
	5,000	79,578	36	32	32	6	48	56	32	2	2
0.5	1	42,429	47	53	0	7	89	98	67	2	2
	10	43,392	47	52	1	7	89	97	64	3	4
	25	44,186	46	51	3	7	89	94	64	2	2
	50	45,251	46	50	4	7	83	93	61	2	3
	100	46,720	46	48	6	7	81	89	59	2	4
	250	49,255	35	58	7	5	75	82	40	2	2
	500	52,233	37	54	9	5	69	73	35	3	3
	1,000	56,507	35	50	14	5	65	72	34	3	2
	5,000	81,897	35	31	34	6	48	56	32	2	2
1	1	42,501	47	53	0	7	89	98	67	2	1
	10	43,684	47	51	2	7	89	94	64	2	2
	25	44,802	47	50	3	7	83	93	61	2	4
	50	46,013	47	49	4	7	81	89	59	2	4
	100	47,555	41	54	5	6	74	85	47	2	3
	250	50,479	35	57	8	5	72	81	38	2	1
	500	53,859	37	53	10	5	65	70	34	3	2
	1,000	59,040	35	48	17	5	61	70	33	2	2
	5,000	88,546	41	25	34	7	42	52	32	3	8
1.5	1	42,615	47	52	1	7	95	98	67	2	2
	10	44,135	46	51	3	7	89	94	64	2	2
	25	45,525	46	50	4	7	83	92	61	2	3
	50	46,920	47	48	5	7	74	85	56	2	4
	100	48,671	41	53	6	6	72	81	46	2	2
	250	52,162	37	54	8	5	69	73	35	2	2
	500	56,163	37	50	12	5	61	70	33	2	2
	1,000	62,778	35	45	20	5	62	66	32	2	3
	5,000	97,044	38	23	39	7	42	48	32	3	17
2	1	42,763	47	52	1	7	95	97	67	2	1
	10	44,711	47	50	3	7	83	93	61	2	3
	25	46,278	47	48	5	7	81	89	59	2	4
	50	47,791	41	54	5	6	74	85	47	3	4
	100	49,903	35	57	7	5	72	81	38	2	1
	250	53,941	37	53	10	5	65	70	34	3	2
	500	59,023	35	48	17	5	61	70	33	2	3
	1,000	66,506	42	38	20	6	49	59	33	3	4
	5,000	107,202	41	19	40	8	36	44	32	3	16
2.5	1	42,954	47	52	1	7	95	97	67	2	2
	10	45,238	46	50	4	7	83	93	61	3	5
	25	46,993	47	48	5	7	74	85	56	2	4
	50	48,735	41	53	6	6	72	81	46	2	3
	100	51,249	38	55	7	5	69	73	35	2	2
	250	55,950	37	51	12	5	61	70	33	2	2
	500	61,983	43	41	16	6	48	60	34	3	5
	1,000	70,952	40	36	23	6	48	56	32	3	5
	5,000	118,106	44	15	41	9	34	39	32	3	13
Min.		42,405	34	15	0	5	34	39	32	2	1
Avg.		54,645	42	48	10	6	72	80	48	2	4
Max.		118,106	47	58	41	9	95	98	67	3	17

Table 3: Computational Performance: Instances with 200 User Nodes and 15 Potential Facilities

$Cv$	$d$	$TC$	FC (%)	AC (%)	DC (%)	# Facility	% Utilization			# Iter	CPU
							Min.	Max.	Avg.		
0	1	64,897	55	44	0	14	91	99	90	2	18
	10	66,409	55	43	2	14	76	98	87	2	18
	25	67,976	54	43	3	14	78	95	84	2	23
	50	69,594	55	41	4	14	71	93	80	2	20
	100	72,116	49	46	5	12	67	90	66	3	60
	250	76,189	42	51	7	9	62	82	47	2	26
	500	80,387	42	49	9	9	56	76	42	2	17
	1,000	86,593	43	45	12	9	45	72	37	2	14
	5,000	120,794	36	33	32	9	48	60	32	3	28
0.5	1	64,961	55	44	1	14	91	99	90	2	16
	10	66,685	55	43	2	14	88	96	87	2	22
	25	68,372	55	42	3	14	76	94	82	2	23
	50	70,108	55	41	5	14	71	93	80	2	19
	100	72,765	49	45	6	12	70	87	66	2	31
	250	76,982	43	51	7	9	62	82	45	3	40
	500	81,364	43	48	9	9	50	76	40	2	15
	1,000	87,860	42	44	13	9	50	72	37	3	10
	5,000	124,123	40	29	31	10	41	55	32	5	57
1	1	65,136	55	44	1	14	91	99	90	2	19
	10	67,401	55	42	3	14	88	96	87	4	54
	25	69,295	55	41	3	14	71	93	80	2	19
	50	71,555	54	41	5	14	75	89	77	3	48
	100	74,309	45	50	5	10	67	83	52	2	32
	250	78,810	43	50	7	9	56	76	42	2	23
	500	83,774	44	46	10	9	46	72	38	2	17
	1,000	91,294	41	43	15	9	46	68	36	3	17
	5,000	132,533	38	27	35	10	43	51	32	4	35
1.5	1	65,396	55	44	1	14	92	98	90	2	22
	10	68,258	55	42	3	14	76	94	82	2	23
	25	70,557	54	41	5	14	75	93	79	2	24
	50	73,197	49	46	5	12	67	87	64	3	59
	100	76,017	45	48	6	10	64	82	51	3	35
	250	81,115	43	48	9	9	51	76	40	2	20
	500	86,919	43	45	12	9	50	70	37	3	17
	1,000	95,844	42	41	17	9	45	65	34	3	23
	5,000	144,737	39	24	38	11	38	48	32	4	53
2	1	65,675	56	43	1	14	76	98	87	2	16
	10	69,116	56	41	3	14	71	93	80	2	21
	25	72,011	54	41	6	14	75	89	77	2	36
	50	74,649	45	50	5	10	67	83	52	3	42
	100	77,852	43	50	7	9	56	79	43	2	23
	250	83,654	44	46	9	9	50	72	37	3	15
	500	90,632	43	44	13	9	45	67	35	3	22
	1,000	101,278	46	36	18	10	41	56	34	5	52
	5,000	159,170	39	20	40	12	37	44	32	4	43
2.5	1	65,957	56	43	1	14	76	98	87	2	20
	10	70,031	55	41	4	14	73	93	80	2	20
	25	73,347	49	45	5	12	70	86	64	3	53
	50	76,108	45	48	6	10	64	80	51	3	40
	100	79,628	44	49	7	9	50	76	40	2	24
	250	86,404	44	45	11	9	50	68	37	2	16
	500	94,597	42	42	16	9	49	65	34	4	32
	1,000	106,997	45	34	22	10	44	55	33	3	34
	5,000	175,392	40	17	43	13	33	41	32	6	64
Min.		64,897	36	17	0	9	33	41	32	2	10
Avg.		84,015	48	42	10	11	62	80	57	3	29
Max.		175,392	56	51	43	14	92	99	90	6	64

Table 4: Computational Performance: Instances with 300 User Nodes and 20 Potential Facilities

$Cv$	$d$	$TC$	FC (%)	AC (%)	DC (%)	# Facility	% Utilization			# Iter	CPU
							Min.	Max.	Avg.		
0	1	87,130	57	43	1	18	84	99	87	2	668
	10	89,241	56	42	2	18	91	97	84	3	52
	25	91,380	55	41	3	18	83	96	83	3	56
	50	93,915	54	41	5	18	85	92	81	3	71
	100	97,458	55	39	6	18	76	90	77	3	220
	250	103,111	47	45	8	14	61	84	54	2	75
	500	109,103	45	45	10	13	47	76	46	4	119
	1,000	117,835	42	46	12	11	41	72	36	3	72
	5,000	163,701	36	33	31	12	42	61	32	3	93
0.5	1	87,242	57	43	1	18	86	99	87	2	262
	10	89,583	56	42	2	18	91	97	84	2	38
	25	91,976	55	41	4	18	83	95	82	2	35
	50	94,704	55	41	5	18	76	92	79	2	48
	100	98,312	49	45	5	15	76	88	62	2	91
	250	104,193	46	47	7	13	56	79	48	3	91
	500	110,477	45	45	10	13	47	75	44	2	50
	1,000	119,613	41	46	13	11	46	71	36	4	93
	5,000	167,997	39	29	32	13	40	57	32	3	66
1	1	87,523	56	43	1	18	91	99	87	3	114
	10	90,593	55	42	3	18	91	96	84	2	59
	25	93,513	55	41	4	18	85	92	81	3	70
	50	96,690	55	39	6	18	76	90	77	3	119
	100	100,348	48	46	6	14	61	85	56	4	128
	250	106,782	45	46	8	13	51	78	47	4	98
	500	113,941	43	46	11	12	46	74	40	3	76
	1,000	124,301	40	45	14	11	41	67	34	2	47
	5,000	179,921	37	28	35	13	44	52	32	5	110
1.5	1	87,845	56	43	1	18	92	98	87	2	56
	10	91,847	55	41	4	18	83	95	82	2	38
	25	95,403	55	40	5	18	76	90	77	2	71
	50	98,800	50	45	5	15	61	84	60	3	115
	100	102,896	48	46	6	14	56	81	52	3	108
	250	110,165	45	45	9	13	51	74	44	3	71
	500	118,582	42	46	12	11	46	69	35	3	78
	1,000	130,498	42	41	17	12	42	64	34	4	120
	5,000	197,199	41	22	36	15	38	49	32	4	112
2	1	88,279	56	43	1	18	92	98	86	2	123
	10	93,271	55	41	4	18	85	92	81	3	74
	25	97,338	56	39	5	18	68	87	74	4	202
	50	100,919	49	44	6	15	68	83	59	3	107
	100	105,374	46	47	7	13	51	78	47	3	79
	250	113,993	44	46	10	12	43	70	38	2	45
	500	123,494	44	43	13	12	44	64	35	3	74
	1,000	137,761	46	36	18	13	40	59	33	3	77
	5,000	216,365	44	19	37	17	34	42	32	4	88
2.5	1	88,695	56	42	1	18	86	97	84	2	38
	10	94,674	55	41	5	18	76	92	79	3	83
	25	98,978	50	45	5	15	61	84	60	2	73
	50	102,958	48	46	6	14	56	80	51	2	71
	100	107,960	46	46	8	13	47	74	44	3	76
	250	117,909	46	43	12	13	44	68	39	2	51
	500	128,813	44	41	15	12	42	64	33	3	87
	1,000	145,395	45	34	21	13	42	57	32	5	116
	5,000	238,223	44	16	40	18	33	38	32	6	150
Min.		87,130	36	16	1	11	33	38	32	2	35
Avg.		113,782	49	41	10	15	62	79	58	3	100
Max.		238,223	57	47	40	18	92	99	87	6	668

Table 5: Computational Performance: Instances with 400 User Nodes and 25 Potential Facilities

$C_v$	$d$	TC	FC (%)	AC (%)	DC (%)	# Facility	% Utilization			# Iter	CPU
							Min.	Max.	Avg.		
0	1	108,302	53	46	0	19	97	99	74	3	807
	10	110,755	53	45	2	19	90	97	72	2	70
	25	113,050	52	45	3	19	87	97	70	2	110
	50	115,745	52	44	4	19	78	94	67	3	241
	100	119,588	53	42	6	19	73	92	65	2	396
	250	126,660	50	43	7	17	66	85	52	2	332
	500	134,432	45	45	10	15	53	80	43	3	206
	1,000	145,174	43	44	13	14	48	76	37	3	106
	5,000	203,766	38	30	32	15	45	64	32	4	242
0.5	1	108,430	53	46	1	19	97	99	74	3	988
	10	111,135	53	45	2	19	92	97	71	3	98
	25	113,694	53	44	3	19	82	96	69	3	150
	50	116,659	53	43	4	19	78	92	66	4	1,033
	100	120,705	52	43	6	18	78	90	60	2	511
	250	128,061	50	42	7	17	62	83	51	3	237
	500	136,165	45	45	10	15	54	79	42	4	192
	1,000	147,499	42	44	14	14	48	76	36	2	65
	5,000	209,065	41	27	31	16	40	56	32	4	149
1	1	108,768	53	46	1	19	86	98	73	2	558
	10	112,176	52	45	2	19	81	97	70	2	75
	25	115,324	52	44	4	19	78	94	67	3	229
	50	118,776	53	42	5	19	73	92	65	2	396
	100	123,345	50	44	6	17	68	87	54	4	945
	250	131,521	50	41	9	17	58	81	49	3	183
	500	140,513	46	43	11	15	48	75	39	3	112
	1,000	153,517	43	42	15	14	46	71	34	3	131
	5,000	224,302	41	25	34	17	39	53	32	5	335
1.5	1	109,115	53	46	1	19	86	98	73	2	93
	10	113,548	53	44	3	19	84	94	69	3	151
	25	117,369	54	42	4	19	73	92	65	4	707
	50	121,414	51	43	6	18	73	88	59	3	734
	100	126,409	50	43	7	17	62	83	51	4	335
	250	135,945	46	45	9	15	54	78	41	3	181
	500	146,324	44	44	12	14	48	74	35	5	208
	1,000	161,516	42	40	17	14	45	68	33	3	113
	5,000	245,741	43	21	36	19	37	51	32	4	175
2	1	109,592	53	46	1	19	88	98	73	2	100
	10	115,071	52	45	3	19	78	94	68	3	218
	25	119,552	53	42	5	19	78	88	64	3	771
	50	124,030	51	44	6	17	69	85	53	2	350
	100	129,675	50	42	8	17	58	80	49	3	162
	250	140,567	46	43	10	15	48	74	39	3	114
	500	152,752	44	42	13	14	46	69	33	5	240
	1,000	170,954	48	33	19	16	43	58	33	4	177
	5,000	269,628	42	18	40	20	35	45	32	4	122
2.5	1	110,130	53	45	1	19	82	98	72	2	181
	10	116,603	54	43	4	19	78	93	66	2	373
	25	121,739	51	43	5	18	73	88	58	5	1,585
	50	126,527	51	43	6	17	62	83	51	2	97
	100	133,103	50	41	9	17	57	78	47	2	82
	250	145,560	46	42	12	15	47	71	37	3	133
	500	159,695	46	39	15	15	45	66	34	6	369
	1,000	181,019	46	32	22	16	43	56	33	3	146
	5,000	298,560	43	16	41	22	32	40	32	5	621
Min.		108,302	38	16	0	14	32	40	32	2	65
Avg.		140,727	49	41	10	17	64	81	52	3	323
Max.		298,560	54	46	41	22	97	99	74	6	1,585

## 1 6. Conclusions

2 In this paper, we presented a class of location-allocation problems with immobile servers,  
3 stochastic demand and congestion. The model captures the trade-off among the fixed cost



1 of opening service facilities and equipping them with sufficient capacities, the access cost  
2 associated with users' travel to service facilities, and the queueing delay cost associated with  
3 customers waiting for service. Under the assumption that the customer demands follow a  
4 Poisson process and service times follow a general distribution, the facilities were modeled  
5 as a network of independent M/G/1 queues, whose locations, capacity levels and workload  
6 allocations are decision variables. We presented a non-linear IP formulation and a constraint  
7 generation based exact method to solve its linear MIP reformulation. The computational  
8 results indicate that the proposed approach provides optimal solution for a wide range of  
9 problem instances within reasonable computation times. Future research directions may  
10 include extending the proposed solution procedures to deal with systems with multiple  
11 servers (M/G/s) and general demand processes (G/G/s).

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