A GENERALIZED, STEADY-STATE SIMULATION

 \mathbf{OF}

WASTEWATER TREATMENT PLANTS

A GENERALIZED, STEADY-STATE SIMULATION

OF

WASTEWATER TREATMENT PLANTS

by

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A Thesis

Submitted to the Faculty of Graduate Studies In Partial Fulfillment of the Requirements

For the Degree

MASTER OF ENGINEERING

McMaster University

January, 1972

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MASTER OF ENGINEERINGMcMASTER UNIVERSITY(Chemical Engineering)Hamilton, Ontario

TITLE: A Generalized, Steady-State Simulation of Wastewater Treatment Plants

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NUMBER OF PAGES:

SCOPE AND CONTENTS:

Laboratory and field measurements were combined with theory in developing models for the various wastewater treatment processes. Some assumptions were made, both in an attempt to simplify the complex mechanisms encountered and to explain known experiences and observations.

Models for a primary settling tank, a secondary settling tank, a generalized aerobic biological sludge treatment scheme, an anaerobic digestion system and several other unit processes have been developed. The generalized aerobic biological sludge treatment scheme is an attempt to describe in one physical package, all the various activated sludge process modifications.

The executive system used is GEMCS, together with its library of subroutines, such as CONTL1, SEPAO1, JUNCO1 and

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and SETSPI, some of which have been modified slightly.

The models have been tested on the Drury Lane and the Skyway, Burlington, Sewage Treatment Plants, and have been found to agree very well with plant data. The former has a conventional activated sludge process, whilst the latter employs the extended aeration modification.

An inter-active version has been developed, which requires the minimum knowledge of GEMCS for its use. This can be applied as a training and teaching tool for plant operators and students in the field.

ACKNOWLEDGEMENTS

The author wishes to express his sincere gratitude to:

1. Dr. J. D. Norman, for his many ideas and contributions to the project, and for the interest and guidance he has shown throughout.

2. Dr. K. L. Murphy and Dr. A. Benedek for sharing their vast experience and knowledge.

3. Mr. K. Cronin, a fellow graduate student, for assistance in the field studies as well as in the laboratory.

4. Miss C. P. Lee, for help in the analyses and in the drawings and graphs for the thesis.

5. Miss I. Ellis for her patience in typing the manuscript.

6. Mr. C. Fiddy, Mr. G. Brewer, and all the operating personnel at the Dundas, Drury Lane and Skyway Water Pollution Control Plants for their kind co-operation.

7. The National Research Council of Canada for financial assistance provided during the course of this study.

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CHAPTER 1

1. INTRODUCTION

Although waste treatment practices have been known ever since the turn of the century, it was not until about the late 1940's that an effort was made to provide firm theoretical bases for design. Significant advances have been made in recent years, resulting in a better understanding of the complex variables affecting the operation of the various unit processes. It is now possible, where it would be beneficial, to bring together the models of the various waste treatment processes, to simulate the operation of full-scale plants, on a computer.

A successful simulation can have great potential applications, a few of which are:

- The performance of the individual units and of the plant as a whole can be predicted over a wide range of feed flows and operating conditions,
- (2) The computer simulation can help in the selection of an optimal design for a plant to treat a given waste to a satisfactory level, at the least cost,
- (3) It can be used effectively in the training of plant operators and as a valuable educational tool for students in the field, and
- (4) It can be used for on-line process control purposes.

Several computer simulations have been attempted, both in the United States and in Canada, examples of which include those by Smith (46, 47), Silveston (40, 41), and by Fan and Erickson (10, 11). However, there is a severe limitation in their models for the activated sludge system. They have based their kinetics either on the entire sludge mass or on the volatile suspended matter in the sludge mass, when it is known that microorganisms are the only agents capable of stabilizing the waste. It is for this reason that their models break down when they are applied to the various modification of the conventional activated sludge process, as the composition of the sludge then changes.

One of the major objectives of this work is to develop a means by which the microbial mass may be estimated. The kinetic expressions can then be formulated around the physiological activities of the microorganisms, leading to a more meaningful model for the activated sludge process. This generalized model can then be applied to the various modifications of the activated sludge process.

As a great variety of raw wastes are encountered in wastewater treatment practice, it would be useful to develop generalized models that can handle all of them. However, the parameters pertaining to each type of waste will be different. Consequently, another objective of this work is to compile a series of experiments that can be used to

characterize a given wastewater and its treatability parameters. Although the work in this thesis is based primarily on domestic sewage, the same experiments used here may be performed on other wastewaters.

It is also realized that the complexity of a simulation system is discouraging its use by most plant operators or engineers. Furthermore, there is a tendency for interested parties to create their own simulation system rather than to try to understand and use other systems. It is for this reasons that we feel it necessary to develop an interactive version of the computer simulation which would require the minimum knowledge of the simulation system, for its use.

CHAPTER 2

2. LITERATURE REVIEW

The model for the primary settler proposed by R. Smith (46), was derived by plotting the data for suspended solids removal versus overflow rate for full-sized plants, as reported in "Sewage Treatment Plant Design", (38). He obtained the relationship

 $FRPS = 0.82 \exp(-GPS/2780)$ (2.1)

where FRPS = fraction of solids removed

GPS = overflow rate usgpd/ft.²

The fractional removal was applied to all classes of suspended solids, which may not be a valid assumption.

P. Silveston (41), tested the model on two Ontario plants and found that the effluent suspended solids predicted were about 25% higher than that actually observed. This led him to suggest that actual plant data have to be used to achieve a reasonable simulation. He proposed the relationship

FRPS = 1 - exp(-k/GPS) (2.2)

where k = constant fitted from plant data.

Both models, however, do not account for the effect of turbulence on the settling process. The constant k in Silveston's model could be interpreted to include both the effects of turbulence and the settling characteristics of the feed solids.

Camp (4) has developed a mechanistic model for discrete particle sedimentation in a continuous-flow tank. He first introduced the concept that a real settling tank may be divided into an inlet zone, a settling zone and an outlet zone. The actual sedimentation occurs in the middle zone, whilst turbulence at both the inlet and outlet zones prevent any further sedimentation from occuring. He also introduced the use of batch settling curves to predict the performance of a continuous-flow sedimentation tank.

Most of the models for the activated sludge process have evolved from the complete mixing model first presented by McKinney (25). The original paper was a little vague, and in a later discussion, Washington (47) corrected and clarified several points. However, his model can be interpreted as follows, noting that all the various forms of organic carbon are in their oxygen equivalents:

In the presence of unlimited food the rate of synthesis of the active solids is limited only by the concentration of the active solids, i.e.,

$$\frac{d(MLVSS)}{dt} = k_{s}'^{(MLVSS)}$$
(2.3)

But in the case of food-limiting situation, the rate becomes

$$\frac{d(MLVSS)}{dt} = k_{s}C \qquad (2.4)$$

where MLVSS = oxygen equivalent of active solids C = oxygen equivalent of substrate k, k = rate constants.

He assumed that the volatile solids in the sludge is entirely active. He also proposed that the rate of substrate removal is proportional to the rate of synthesis of active solids, i.e ,

$$\frac{dC}{dt} = -(1 + k_a) \frac{d(MLVSS)}{dt}$$
(2.5)

The basal metabolism rate was also assumed to be proportional to the concentration of active solids, i.e.,

$$\frac{d(MLVSS)}{dt} = -k_{d}(MLVSS)$$
(2.6)

By performing mass balances around the stirred tank reactor, he obtained, assuming a food-limiting situation:

$$C_{0} = \frac{C_{1}}{1 + (1 + k_{2}) k_{2} \bar{t}}$$
 (2.7)

and

and
$$M_0 = \frac{k_s C_0}{k_d + x}$$

 \overline{t}
where $C_i = \text{influent substrate concentration}$
 $C_0 = \text{effluent substrate concentration}$
 $M_0 = \text{effluent active solids concentration}$
 $\overline{t} = \text{detention time}$
and $x = \text{fraction of sludge mass wasted.}$

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(2.8)

Estimated values of his constants were given as: $k_a = 0.5$ $k_d = 0.006 \text{ hr.}^{-1}$ $k_s' = 15 \text{ hr.}^{-1}$ (minimum) and $k_s = 7 \text{ hr.}^{-1}$ (minimum)

R. Smith (46) in his model, assumed that the rate of substrate removal is proportional to both the concentration of active solids and the substrate concentration, i.e.,

$$\frac{dC}{dt} = -k_{s}^{"} C (MLSS) \qquad (2.9)$$

as contrasted to equations (2.3) and (2.4). k_s " is a rate constant given by:

$$k_{s}'' = 0.0001 (1.047)^{T-20} (hr)^{-1} (mg/1)^{-1}$$
 (2.10)

Erickson, Ho and Fan (11) assumed Michaelis-Menten kinetics with the expression

$$U = \frac{1}{MLSS} \frac{dC}{dt} = - \frac{U_{max}C}{MLSS}$$
(2.11)
where U_{max} = maximum unit growth rate
 C^* = value of C where $U = \frac{1}{2} U_{max}$
 Y = yield factor.

It may be noted that equation (2.11) reverts to the form of equation (2.3) when C >> C*, and to the form of equation (2.9) when C << C*.

Little quantitative information exists on the operation of the secondary settler. The models proposed by Ott and Bogan (29), Erickson and Fan (10, 11), and Silveston (41), have assumed that a constant fraction β , of the input solids escapes over the wier and that the underflow solids are compacted by a ratio C. R. Smith (46) proposed a model, based on the data of Villiers (51) where

$$\beta = \frac{556 (\text{GPS})^{0.494}}{\text{MLSS}^{1.82} (\text{TA})^{0.439}}$$
(2.12)

GPS is the settler overflow rate (USgpd/ft.²), MLSS is the inlet mixed liquor suspended solids (mg/l), and TA the aeration time (hours). For some reason he later reverted back to the simple splitter model.

Anaerobic digestion is a complex operation involving several reaction stages and different microorganisms. An accurate description for the process is difficult to obtain. McCarty (24) using Michaelis-Menten kinetics and a completely stirred tank model, arrived at the relationship:

$$\frac{k_{\text{max}} F}{F^* + F} = \frac{1}{\bar{t}}$$
 (2.13)
F* + F

where k_{max} = maximum unit growth rate constant F* = value of F at which unit growth rate is $\frac{1}{2} k_{max}$

F = concentration of biodegradable carbon
in effluent

and \overline{t} = digester residence time.

The reported values of k_{max} and F* are: $k_{max} = 0.28 \exp \left[-0.036 (35 - T)\right]$ (2.14) and F* = 700 exp $\left[0.10 (35 - T)\right]$ (2.15) where T = temperature of digestion, ^OC.

His relationship implies that the effluent degradable carbon is independent of the influent value, but depends only on the detention time. He also suggested an expression to estimate the organic carbon of the volatile acids (VAC) in the effluent as:

$$VAC = \frac{200 \exp \left[0.12 (35 - T)\right]}{1 + k_{max} \bar{t}}$$
(2.16)

Extensive cost information is available in the literature. R. Smith (45, 46) presented a thorough investigation of the capital and operating costs of domestic waste treatment unit processes. The capital and operating costs of a unit are calculated as a function of its most significant capacity variable. Michel et al. (26) reported on the total operating and maintenance costs of a treatment plant as a function of the waste flow and also of the population equivalent. Barnard and Eckenfelder (2) included industrial waste treatment costs in their report. A more complete list of references may be obtained from the above reports. Several papers have been published on the optimization studies of wastewater treatment processes. Erickson and Fan (10) studied the optimization of the hydraulic regime of activated sludge systems. Erickson, Ho and Fan (11) later investigated the modelling and optimization of the step aeration processes, which consists of a series of completely mixed tanks. Shih and Krishnan (39) reported on their studies on the dynamic optimization of industrial waste treatment plant designs. Realizing that many design parameters are not accurate, Berthouex and Polkowski (3) studied the effect of such uncertainties on the optimal design of a treatment plant.

Further references from the literature will be made, as the thesis is developed.

CHAPTER 3

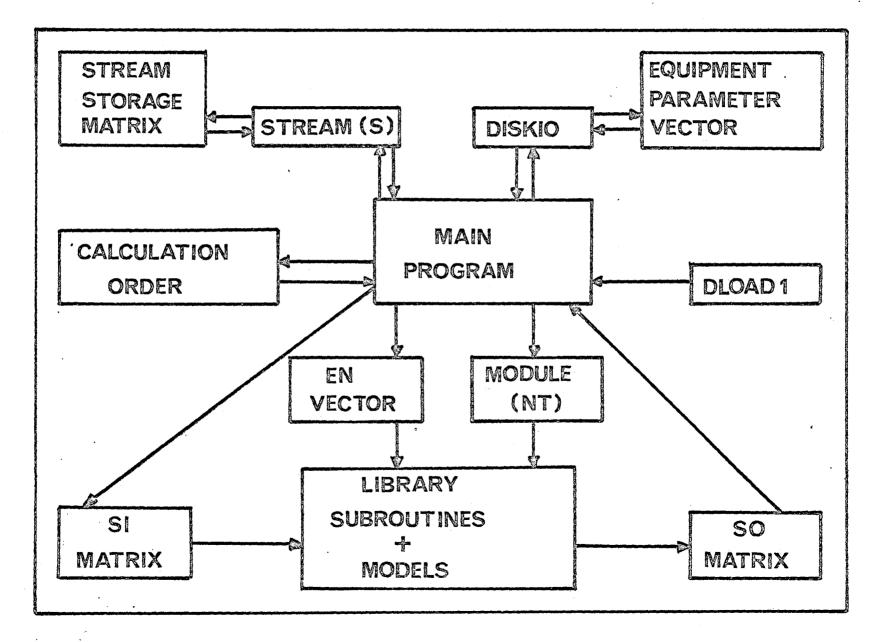
3. THE SIMULATION SYSTEM

3.1 The GEMCS Executive

Any processing or manufacturing plant can be broken down into its component unit processes. The unit process is a basic operation within the framework of the plant, and can usually be identified with a distinct physical piece of equipment, examples of which are primary sedimentation, chlorination, drying, etc. The mathematical model of a unit process is known as a unit computation.

In a simulation system, it is necessary to have a master-mind to execute each unit computation in its logical sequence. Such is the purpose of the executive program. It should also supervise the flow of information between storage locations and the unit computations.

The Chemical Engineering Department of McMaster University, in co-operation with the Canadian General Electric Company, has developed a simple but yet very flexible executive program known as GEMCS (the General Electric-McMaster Simulator) for the simulation of steady-state processes. It has been used successfully in the simulation of several chemical process industries. The extensive work done is well summarized in the book "Chemical Plant Simulation" (5). Only a very brief description of the executive program will be attempted here.



 $(\mathbf{1}, \mathbf{1})$

Figure 3.1.1. Information Flow Within the GEMCS Executive

The aid of Figure 3.1.1 is invaluable. The executive program first calls an assistant subroutine DLOAD1 which reads in input streams, equipment parameters and an order of calculation in which to execute the unit computations. The information is then stored for subsequent use. The main program then prepares to execute the first unit computation in the plant. It retrieves the equipment parameters for the unit from the storage location, the EEN vector, via a linking subroutine DISKIO. It then places the equipment parameters (this is done directly by DISKIO) into a working equipment vector, the EN vector. The main program then retrieves from the storage location for the streams, the SN matrix, the input streams flowing into the unit computation. Subroutine STREAM(S) acts as the link between the SN matrix and the main The input streams are then placed in the working program. The unit computation is then called matrix, the SI matrix. and executed using information provided in the SI and EN via the subroutine MODULE (NT) locations. Any output streams generated by the unit computation are stored in the working matrix, the SO matrix, and which are then transferred by the main program into the SN storage matrix.

The other unit computations in the calculation order are similarly executed. In the case where there is a recycle of streams, a control subroutine must be placed in the calculation order to direct the main program to recalculate all the

unit computations within the recycle loop, until convergence is reached, within a specified tolerance.

It should be noted that the unit computations deal directly only with the working equipment vector and input and output stream matrices. This creates a uniform pattern whereby the unit computations may be constructed.

3.2 The Stream List

In every process simulation a list of components that will adequately describe any stream within the process has to be selected. Crowe et al (5) have listed several criteria for the selection of stream variables. They include

- The variables should be useful to satisfy the objectives of the simulation,
- (2) All variables in recycle streams are usually important, and
- (3) The unit computations must be able to manipulate the variables to an accuracy consistent throughout the whole system.

These criteria were used as an aid in the selection of the stream variables for the present generalized wastewater treatment plant simulation. In addition, the best features of the stream lists used by Smith (46) and Hoffman (20) were selected and in some cases modified. The stream list used is shown in Table 3.2.1.

No.	Phase	Stream	Variable	Units
NO.	Flidbe	Deream	AUTUDIE	
1		Stream number	2	-
2		рН		-
3		Total volumet	cric flow	igph
4		Temperature		°c
5	Short	Head		ft. of water
6	List	Water		lbs/hr
7		Dissolved oxy	gen	lbs/hr
8		Total Organic sum 15-22)	c Carbon (TOC =	lbsC/hr
9		Dissolved Orc sum 20-22)	ganic Carbon (DOC =	lbsC/hr
10		Biochemical (Dxygen Demand (BOD)	lbs/hr
11		Total Suspend	led Solids (TSS)	lbs/hr
12		Volatile Suspended Solids (VSS)		lbs/hr
13		Particulate	settleable	lbs/hr
14	Suspended	Inorganics	non-settleable	lbs/hr
15	Solids	Microorganism	Microorganisms	
16	Phase	-	settleable, bio- degradable	lbsC/hr
17		Particulate	settleable, non- biodegradable	lbsC/hr
1.8		Organics	scum (floatable)	lbsC/hr
			non-settleable	lbsC/hr

No.	Phase	Stre	Stream Variable		
20		Purgeable, bio- degradable		lbsC/hr	
21		Dissolved	Non-purgeable, bio- degradable	lbsC/hr	
22		Organics -	Non-purgeable, non- biodegradable	lbsC/hr	
23		Alkalinity, as CaCO ₃		lbs/hr	
24		-			
25					
26					
27		N in NH ₃ (c	N in NH3 (dissolved)		
28		N in NO ₂ and NO ₃ (dissolved)		lbs/hr	
29		Dissolved P in all forms of phosphates		lbs/hr	

Table 3.2.1 Stream List for Wastewater Simulation

It may be noticed that the stream list is divided into three parts. The first part is a list of traditional variables commonly used in wastewater treatment practice. The second part covers the water-insoluble components normally found in wastewaters and the last category includes the water-soluble components. The second and third parts may be further sub-divided into an inorganic and an organic fraction.

The traditional variables in elements 8-12 are not fundamental components in the sense that they may be derived from the finer classification provided in elements 13-22. However, these are the variables that most operators and environmental engineers understand and hence are invaluable in our communication with them. Another reason for retaining these variables is that several of the correlations used in our simulation are expressed in terms of them. A simple simulation can be effected by the use of only these first twelve variables.

Suspended solids can either be organic or inorganic in nature. The inorganic portion is inert and does not take part in any biological processes, although chemically, it may be altered by a changing pH. However, we have to distinguish between a colloidal inorganic solid, which remain in suspension and a settleable inorganic solid which is removed in physical sedimentation.

For the organic suspended solids, we have to distinguish between the non-living organic objects from the living biomass which is active in bringing about the stabilization of the waste-load. The major portion of the non-living organic solids is biodegradable, while the remainder may be relatively non-biodegradable. The inertness may be ^{aue} to several factors, one of which is a high degree of polymeric crosslinking. It is impossible to distinguish between the biodegradable portion from the non-biodegradable portion, and the distinction, although provided in the stream list is not used.

Physically, the non-living organic solids can be differentiated by their settling characteristics. We can distinguish a colloidal fraction, which remains in suspension, a settleable fraction and a scum (or floating) fraction.

Organics in solution can also be classified as biodegradable and non-biodegradable. Physically, certain organics in solution may be volatilized by purging with air. These are the solvents and other low molecular weight compounds. The non-biodegradable fraction, being largely composed of high molecular weight compounds, is relatively non-purgeable.

Soluble inorganics of interest include the nutrients phosphorus and nitrogen, and the alkalinity causing ion. Inorganic nitrogen is further subdivided into ammonia and the sum of nitrite and nitrate ions. Several blank spaces

are left in the soluble inorganics section to facilitate the addition of variables, should the simulation be extended to include tertiary treatment.

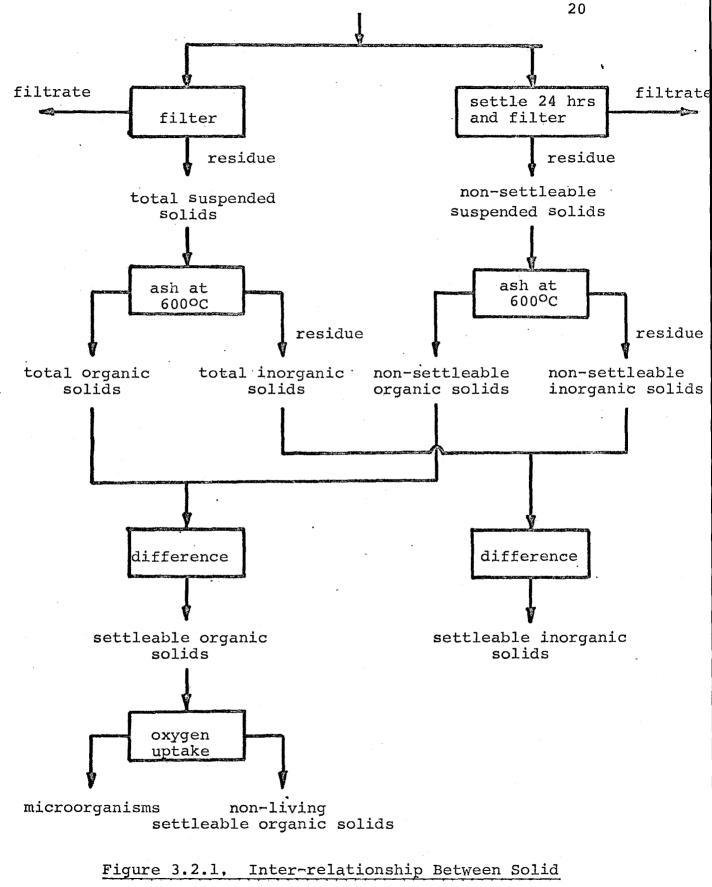
The units used are also shown in Table 3.2.1. It should be noted that concentration units facilitate the splitting of streams, while mass flow units facilitate the mixing of streams. A disadvantage of concentration units is that while concentrations based on unit volume are usually measured, the simulation system invariably has to use concentrations based on unit mass. Volumetric and mass concentrations are approximately equivalent under 1%, but become increasingly inaccurate above that. To avoid this complication, mass flow units are chosen.

As in Hoffman et al (20), the organic components are all measured in terms of the carbon content. This greatly facilitates the arithmetic of the biological processes.

The interplay being the variables is schematically represented in Figures 3.2.1 and 3.2.2, together with the analytical procedure followed. This is essentially the outline given by Hoffman et al with slight modifications. Further analytical details are given in "Standard Methods" (48).

The conversion of variables within our "finer" classification to the traditional variables is described in Section 9.1.

wastewater



Components

filtrate

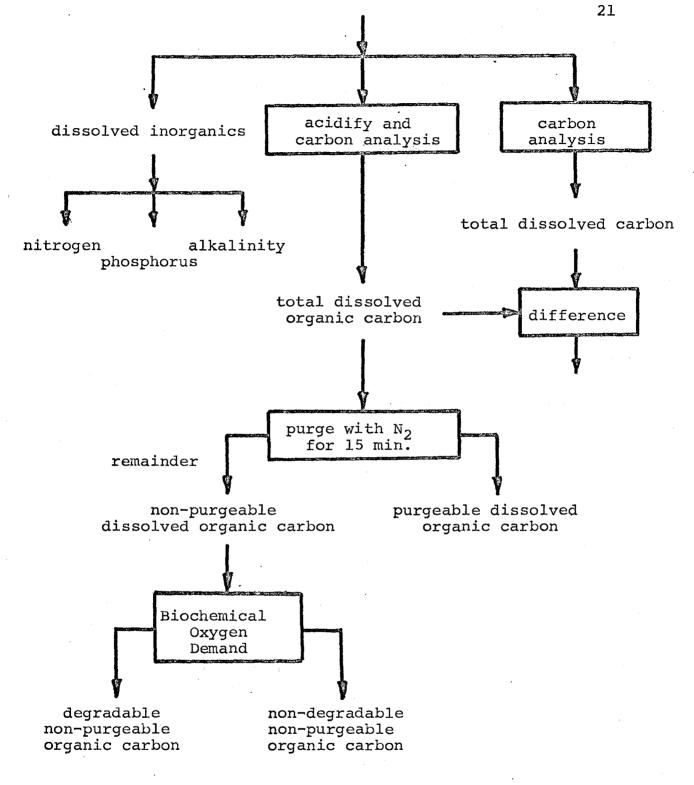


Figure 3.2.2. Inter-relationship Between Dissolved

Components

CHAPTER 4

4. PRIMARY SEDIMENTATION

The solid particles carried in suspension by wastewater flows can be partially removed by gravity settling in a relatively quiescent tank. The process, often called primary sedimentation, is used to reduce the sludge load downstream of the units. In the case where no further treatment is provided, primary sedimentation reduces the formation of sludge banks in the receiving waters. In most cases, it is used as a preliminary step ahead of biological treatment, to reduce the amount of solids being carried around in the latter system.

4.1 Theory

During sedimentation, the solid particles in a dilute suspension, may retain their individual shapes and sizes, and settle out as discrete particles. However, many other solids tend to flocculate (coalesce into a larger particle upon contact with each other), as they settle. This increases their settling velocities and hence the flocculation of the particles is a desirable property from this view-point.

The simpler case of discrete particle sedimentation will be dealt with first, and then it will be expanded to include the case of flocculent sedimentation.

A discrete particle, settling through a fluid, quickly reaches a terminal velocity. At this point, the gravitational force is exactly balanced by the bouyant and drag forces acting on the particle. This leads to the classical equation of discrete particle sedimentation

$$U_{t} = \begin{bmatrix} \frac{4}{9} & (\frac{P_{s} - P}{p}) & D_{p} \end{bmatrix}^{\frac{1}{2}}$$
(4.1.1)
where U_{t} = terminal velocity (ft/sec)
 g = acceleration due to gravity (ft/sec²)
 P_{s} = specific gravity of the solid particle
 P = specific gravity of the fluid
 C_{D} = drag coefficient
and D_{p} = diameter of the particle (ft)

The drag coefficient C_D is a function of the particle Reynolds Number and of the sphericity of the particle. This can be obtained from most standard texts on Fluid Mechanics.

Using the fact that a discrete particle quickly reaches a terminal velocity, during sedimentation, we can calculate the removal obtained in a batch settling test.

Consider the sketch of a batch settling column in Figure 4.1.1. If a dilute suspension of discrete particles occupies the entire volume uniformly, at the start of the test, then if a sample is drawn off at a depth h at time t, it will not contain any particles with terminal velocities greater

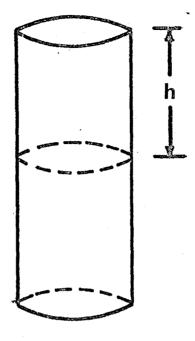
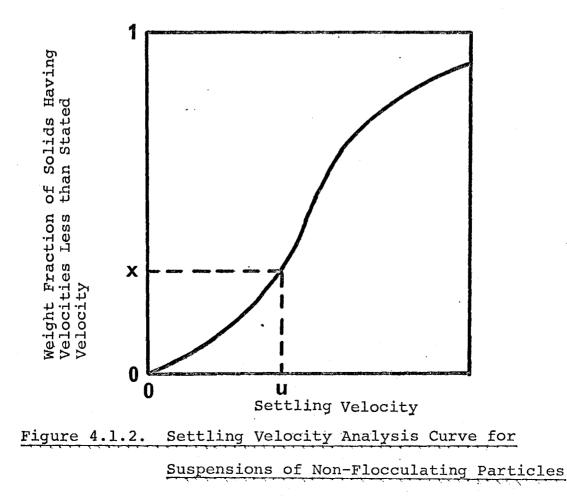


Figure 4.1.1. Settling Column



than U = h/t. The fraction of the solids concentration in the sample to the original solids concentration will be the weight fraction of particles having terminal velocities $U \leq h/t$. Let this weight fraction be represented by x, and x = f(u) - may be used to represent the settling characteristics of the particles. Figure 4.1.2 shows a typical settling velocity analysis curve for a suspension of non-flocculating particles.

If the liquid above a depth N_o were to be decanted, at time t, it will only contain particles with terminal velocities $\leq h_o/t = U_o$. But particles with terminal velocities $U \leq U_o$ will have fallen through a distance h = Ut, and hence only a fraction $(h_o - h)/h_o = (U_o - U)/U_o$ of such particles will remain in the decanted liquid.

Hence the weight fraction of the solids in the decanted (or clarified) liquid will be

$$W_{c} (U_{o}) = \int_{0}^{x_{o}} (U_{o} - U) / U_{o} dx$$

= $x_{o} - \int_{0}^{x_{o}} U / U_{o} dx$ (4.1.2)

where W_c = weight fractions of solids in decanted liquid x_o = weight fraction of particles having terminal velocities $U \leq U_o$.

and U = sometimes referred to as overflow velocity.

The weight fraction of solids not in the decanted liquid (or the weight fraction removal) is then given by

$$W_r (U_o) = 1 - W_c (U_o) = (1 - x_o) + \int_0^{x_o} U/U_o dx$$
 (4.1.3)

The integral may be approximated by a summation term and hence equations 4.1.2 and 4.1.3 may be rewritten as

$$W_{c}(U_{o}) = x_{o} - \sum_{x=0}^{x=x_{o}} U/U_{o} \Delta x$$
 (4.1.4)

and
$$W_r(U_o) = (1 - x_o) + \sum_{x=0}^{x=x_o} U/U_o \Delta x$$
 (4.1.5)

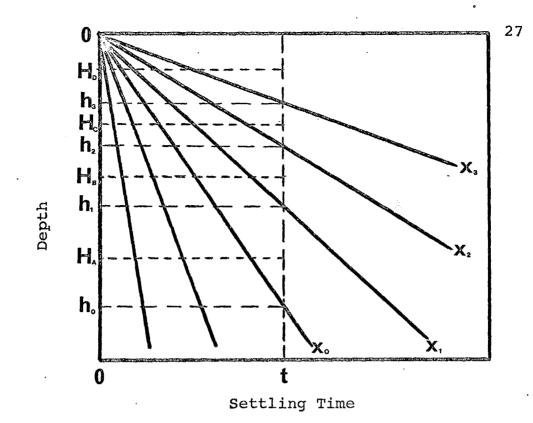
The integral or the summation term can be evaluated with the use of Figure 4.1.2.

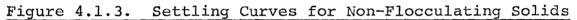
If, during the batch settling test, samples are withdrawn at various depths and elapsed times, Figure 4.1.3 will result. The lines shown join points of equal weightfraction. The slopes of the lines will give the associated terminal velocities.

The weight fraction of solids removed can also be evaluated from Figure 4.1.3. Equation 4.1.5 may be rewritten, for that purpose, as

$$W_r (U_o) = (1 - x_o) + \sum_{x=0}^{x=x_o} h/h_o \Delta x$$
 (4.1.6)

From Figure 4.1.3, if h_A is the average of h_o and h_1 , h_β the average of h_1 and h_2 , and so on, then the summation term can be evaluated as





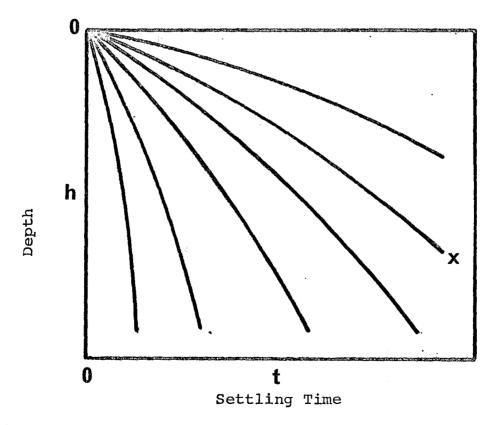


Figure 4.1.4. Settling Curves for Flocculating Solids

$$\sum_{x=0}^{x=x_{0}} h/h_{0} \Delta x = \frac{1}{h_{0}} \left\{ h_{A}(x_{0} - x_{1}) + h_{\beta}(x_{1} - x_{2}) + H_{C}(x_{2} - x_{3}) + h_{D}(x_{3} - 0) \right\}$$
(4.1.7)

In the case where the particles tend to flocculate upon contact with each other, the settling velocities of the particles are increased, as sedimentation proceeds. Hence there is an added effect of the settling time.

The effect of flocculation is best seen in Figure 4.1.4, which is obtained in the same manner as Figure 4.1.3. The lines joining points of equal weight-fraction are no longer straight, as in the case of discrete particle sedimentation, but curved. The curvature reflects the increasing settling velocity, as the sedimentation proceeds.

Equation 4.1.2 to 4.1.6 still holds and may be used to evaluate the solids removal. Also, a similar procedure, as the one used in obtaining Equation 4.1.7, may be used as a graphical solution.

The batch settling test can be directly applied to a continuous flow primary settling tank, assuming plug flow conditions exist in the tank. The depth h_o refers directly to the depth of the tank and the time t is the residence time of the fluid in the tank. Deviations from plug flow conditions are treated in section 4.3.

4.2 The Correlation of Batch Settling Data

Raw sewage from the Drury Lane, Burlington and Dundas Sewage Treatment Plants were used in the settling experiments. The settling column is 8 in. in diameter and 8 ft. in height. The column has five sampling points, equally spaced 1 ft. apart.

The raw sewage was introduced into the column and allowed to equilibrate to room temperature for at least four hours. This was done to minimize thermal convection currents. Then the suspension was stirred up by blowing air into the sewage for one minute.

Samples were withdrawn initially, just after the mixing, and at regular time intervals from the five sampling ports. Prior to sampling, the sampling lines were flushed to wash out any accumulated solids. The column was also left for 24 hours, and the suspended solids determined. This represented the non-settleable or colloidal portion of the waste. The suspended solids of the samples were determined by vacuum filtration on 0.45 micron membrane filter papers. The papers were dried in an oven set at 103^OC for at least half an hour. A Mettler balance with an accuracy of 0.1 mg was used for the weighing. All determinations were done in duplicate, and averaged.

The suspended solids data are given in Appendix A. Two previous runs, performed in 1969 and 1970, as part of a

wastewater course, using raw sewage from the Dundas Sewage Treatment Plant are also included. Figure 4.2.1 to 4.2.4 show the weight fraction of settleable solids in the samples withdrawn as a function of time and depth. Equation 4.1.6 was used to calculate the weight fraction of settleable solids removal as a function of depth and time, and the results are tabulated in Appendix A.

In the absence of theoretically derived expressions, empirical correlations had to be derived. However, we note that at constant h_0 ,

if $t \rightarrow \infty$, then $W_c \rightarrow 0$ or $W_r \rightarrow 1$. and if $t \rightarrow 0$, then $W_c \rightarrow 1$ or $W_r \rightarrow 0$. (4.2.1)

Also that at constant t,

if $h_0 \rightarrow \infty$, (or $U_0 \rightarrow \infty$), then $W_c \rightarrow 1$ or $W_r \rightarrow 0$ and if $h_0 \rightarrow 0$ (or $U_0 \rightarrow 0$), then $W_c \rightarrow 0$ or $W_r \rightarrow 1$ (4.2.2)

This suggests a relationship of the form

$$W_{c} = e^{-kt^{\circ}/h_{o}\beta} \qquad (4.2.3)$$

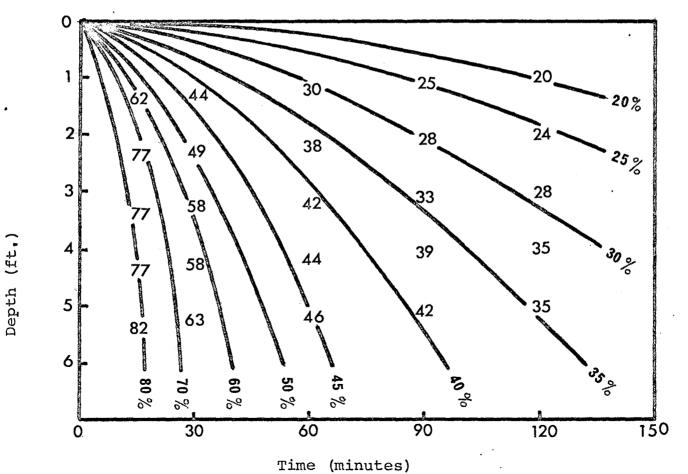
or
$$W_r = 1 - W_c = 1 - e^{-kt^{\alpha}/h_0^{\beta}}$$
 (4.2.4)

By comparing Equations 4.2.4. a d 4.2.2 we note the similiarty between our model and Silveston's model. However, Silveston has assumed that $\alpha = \beta = 1$ (note: GPS $\alpha \xrightarrow{h_0}$).

Figure 4.2.1. Fraction of Settleable Solids Versus Depth and





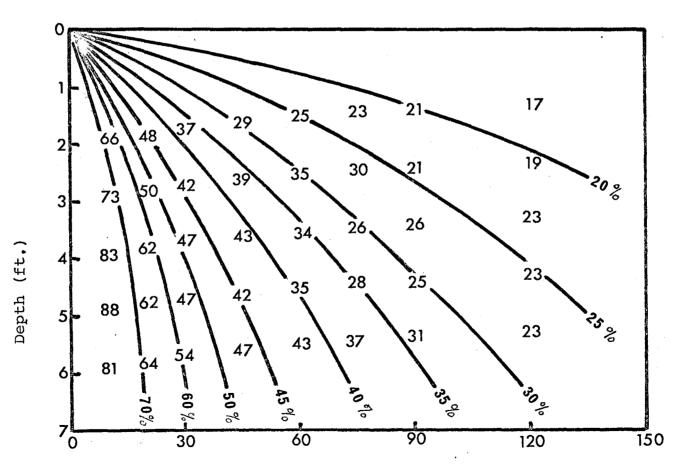


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Figure 4.2.2. Fraction of Settleable Solids Versus Depth and





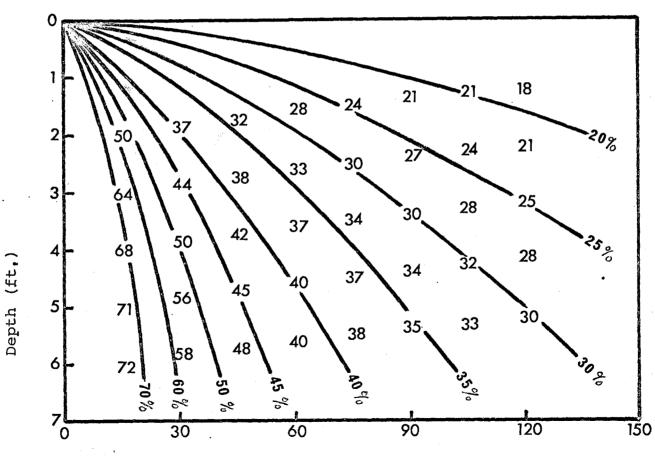


'Time (minutes)

Figure 4.2.3. Fraction of Settleable Solids Versus Depth and

Time

RUN A.3



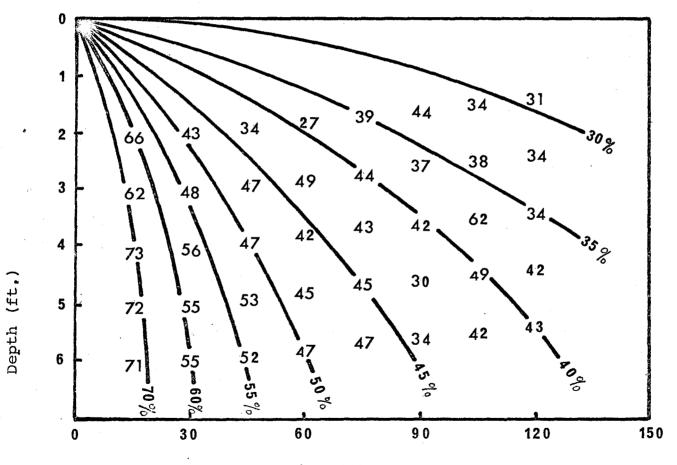
Time (minutes)

ω

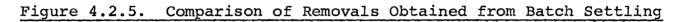
Figure 4.2.4. Fraction of Settleable Solids Versus Depth and

Time

RUN A.4



Time (minutes)

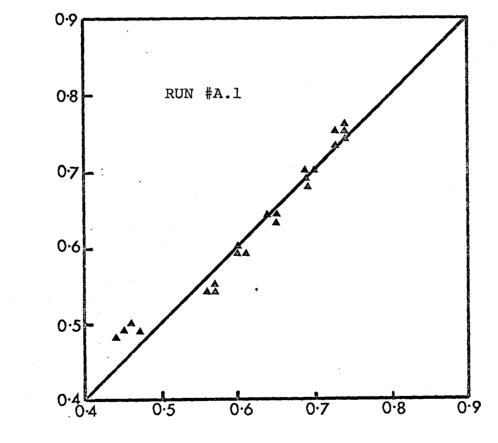


and from Correlation

from Correlation

Calculated

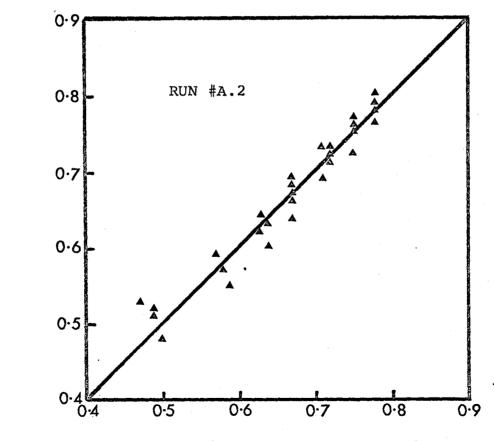
Removal



Removal Obtained from Batch Settling

ω 5

Figure 4.2.6. Comparisons of Removals Obtained from Batch Settling and from Correlation



from Correlation

Calculated

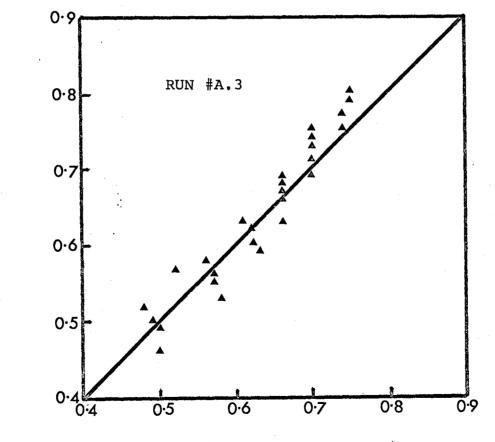
Removal

Removal Obtained from Batch Settling

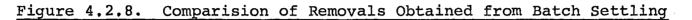
Figure 4.2.7. Comparison of Removals Obtained from Batch Settling

and Correlation

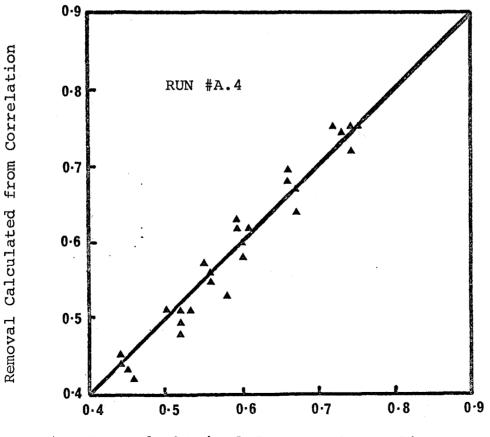
Removal Calculated from Correlation



Removal Obtained from Batch Settling



and Correlation



Removal Obtained from Batch Settling

A three parameter search, using a SIMPLEX routine was then used to determine k, \propto and β . The objective function uses a least sum of squares criterion, the sum of squares being $\geq (W_r \text{ experimental} - W_r \text{ correlation})^2$. The values of k, \propto and β are summarized in Table 4.2.1 and the correlated data are shown in Figures 4.2.5 to 4.2.8.

Run No.	Source of Raw Waste	k	œ	β	Correlation Coefficient
A-1	Dundas	0.215	0.473	0.233	0.95
A-2	Burlington	0.239	0.515	0.252	0.94
A-3	Dundas	0.238	0.513	0.251	0.94
A-4	Dundas	0.213	0.273	0.221	0.94

TABLE 4.2.1. Parameters for Quiescent Settling Model (Note: t is in minutes and h is in feet.)

The correlations obtained are good and the closeness in the values of the parameters indicate that the raw waste from the two sources are not too different. The average values of k, \propto and β obtained are 0.226, 0.494, and 0.239.

4.3 Deviations from Ideal "Plug Flow" Settling

Conditions in a primary settling tank are seldom of the ideal "plug flow" nature. Turbulence and shortcircuiting in the flow can be caused by several factors, and their effects on sedimentation have to be accounted for.

Short-circuiting in the flow results mainly from density currents, which are bulk movements of fluids caused by density variations. Temperature differences between the feed and the water in the tank, or the presence of high suspended solids in the feed, will cause these density differences to occur.

Due to poor inlet design, the kinetic energy of the incoming flow can be quite considerable. The resulting energy dissipation can cause turbulent conditions to exist in the inlet zone. Inlet baffles, if properly designed, can reduce the kinetic energy to a large extent.

Several investigators have examined the effects of turbulence on settling. A very good summary of the various approaches used can be found in Fitz (15). Ensign Dobbins (6), starting from the Continuity Equation, made several assumptions and arrived at a simplified expression, for the case of discrete particle sedimentation

$$\frac{dC}{dt} = e \frac{d^2C}{dy^2} + U \frac{dC}{dy}$$
(4.3.1)
dt $dy^2 dy$

where C = concentration of solids having settling
 velocity U

- U = settling velocity
- y = depth
- t = time

and e = eddy diffusivity.

Dobbins has obtained a solution for Equation 4.3.1, but it is too complex for use in a simulation model. A graphical solution of Equation 4.3.1 is available (4), and could be used in conjunction with the batch settling test as outlined in Section 4.2.

Another approach to the problem would be to obtain information on the flow patterns in the tank from a residence time distribution analysis. Fitz (15) used mixed models of combinations of stirred tanks, plug flow volumes and dead spaces to fit the residence time distributions thus obtained, to predict solids removal. His results were not too encouraging.

A dispersion model also does not appear to be promising as it rarely fits the observed exit age distributions. This is to be expected as the dispersion model assumes only relatively small deviations from plug flow with no gross short-circuiting or density currents.

In an attempt to estimate the degree of turbulence in the primary tanks of the Dundas and Drury Lane Water Pollution Control Plants, tracer studies were carried out, using Rhodamine B dye and a Fluorometer, a Turner and Associates Model III. The experimental technique is well described by Wilson (49). The exit age distributions were followed for at least two residence times. Samples of the influent and effluent were collected during the tracer studies and analyzed for suspended solids.

Eight runs were made, four at Dundas and four at Drury Lane. Due to pump and recorder difficulties, two runs had to be discarded, one from each of the two plants. The exit age distributions are given in Appendix B. The tracer recoveries were poor and were approximately 80-85%. The calculated residence times from the exit age distributions were much lower than the actual fluid residence times. The tracer results are too unreliable for any use.

However, there seemed to be a consistent efficiency factor for each of the two plants, where the efficiency factor is defined as

Ø = observed % of settleable solids removed
theoretical (batch) % of settleable solids removed

(4.3.2)

A high efficiency factor will indicate a low degree of turbulence or short-circuiting, while a low effiency factor indicates a highe level of turbulence or short-circuiting.

The theoretical or batch removal is calculated from the equation developed in Section 4.2. The efficiency factors obtained are summarized in the table below:

Plant	Plant Flow (migd)	t (mins)	ø
Dundas	2.1	66 ·	0.62
(two tanks in	2.3	60	0.54
parallel)	1.8	77	0.56
Drury Lane	2.2	84	0.83
(two tanks in	2.0	93	0.88
parallel)	1.6	116	0.86
I			

TABLE 4.3.1. Efficiency Factors for the Primary Sedimentation Tanks at the Dundas and Drury Lane Treatment Plants.

The average efficiency factor for the primary settlers at Dundas is 0.57 whilst that at Drury Lane is 0.85. The high degree of turbulence in the primary settling tank at Dundas is reflected in the low efficiency factor and is quite evident visually from the effluent turbidity.

Our overall correlation for the primary settler is then described by

$$W_{\mathbf{r}} = \emptyset \left\{ |-\exp(-kt^{\alpha}/H^{\beta}) \right\}$$
(4.3.3)

4.4 The Primary Settling Tank Model - PRISET1

A listing of the program is given in Appendix K. The equipment vector for the model is reproduced below.

EN VEC	
	STANDARD FORMAT
4.	NUMBER OF TANKS IN PARALLEL HEAD LOSS THROUGH TANKS (FT_OF_WATER)
2.	ETDST INDUT STREAM (RAW WASHE SEREAM)
8.	SECOND INDUIT STREAM ISHPERNALE EKUM THE DIGESTERS AT ANY A
9.	THIRD INPUT STREAM (WASTE ACTIVATED SLUDGE - IF ANY) FIRST OUTPUT STREAM (EFFLUENT)
13.	SECOND OUTPUT STREAM (UNDERFLOW)
5 7 8 9 12 13 14	THIRD OUTPUT STREAM (SCUM TROUGH FLOW)
16.	LENGTH OF TANK (FEET) = 0.0 IF TANK IS CIRCULAR
17.	WIDTH OF TANK (FEFT)
_	= RADIUS OF TANK IF CIRCULAR (FEET)
18.	DEPTH OF TANK (FEET) NOT CURRENTLY USED
19• 20•	SOLIDS CONCENTRATION OF SCUM FLOW (MG/L)
21.	EFFICIENCY FACTOR FOR LANKS
22.	FRACTION OF FLOCCULATED SOLIDS FROM WASTE ACTIVATED SLUDGE AND DIGESTER SUPERNATE THAT IS RESUSPENDED
23.	NUMBER OF UNDERFLOW SLUDGE PUMPS
24.	CARACITY OF PUMPS IGPH FACH
25.	FRACTION OF CAPACITY AT WHICH PUMPS ARE RUN

The computations carried out by the program can be explained in the following steps:

- (1) The pH of the output streams is initially calculated.
- (2) Input streams 2 and 3 are sludge streams carrying solids which have much higher settling velocities than the solids in the raw waste. Hence, the removal of the sludge solids is essentially complete. EN(22) provides

for the small fraction of the sludge solids which do escape over the wiers due to short-circuiting or turbulence.

- (3) The removal of the solids in the raw waste is calculated from Equation 4.3.3.
- (4) The underflow is calculated from pump capacities. The concentration of the underflow solids is then calculated, for which a maximum value of 6% is assumed. If the pump capacity is insufficient, excess solids over 6% is carried over to the effluent.
- (5) The scum flow is then calculated, assuming 90% removal of scum solids. The solids concentration of the scum flow is specified by EN(20).
- (6) Soluble components are then split according to the split in water flow among the three output streams.

CHAPTER 5

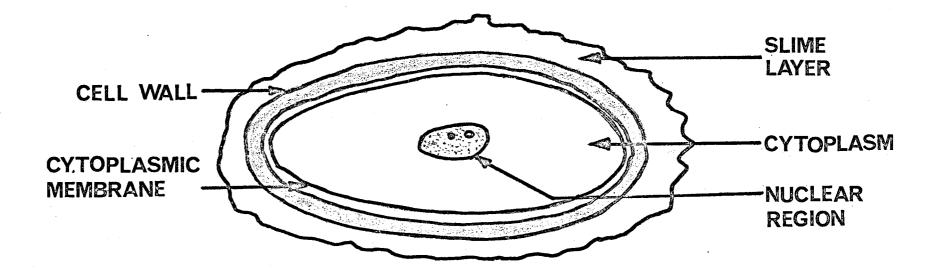
5. AEROBIC BIOLOGICAL SLUDGE PROCESSES

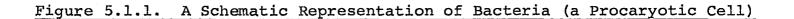
The theory here represents an attempt to model all aerobic biological sludge processes in one general package. Currently, all the various modifications of the activated sludge process, such as extended aeration, step aeration, contact stabilization and aerobic digestion, have been considered as separate processes. However, a bacterial cell does not know what hydraulic regime it is in, nor does it know what process it is supposed to work under. The kinetics describing the activity of the bacteria should be similar in all cases. The differences in rates observed with the various processes can only be due to differences in composition of the activated sludge mass.

5.1 The Bacteria and its Activities

The morphology of the bacteria is well described in any standard textbook on microbiology. A very brief description will be given here, while the physiological functions of the bacteria and their corresponding kinetic expressions will be described in more detail.

Bacterial cells are small and have sizes of the order of one micron. The cell shape may be spherical (coccus), rod-like (bacillus), or spirally-shaped (spirillum), and is defined to a large part by a relatively tough outer cell wall.





Inside the cell wall is a semi-permeable cytoplasmic membrane, which is responsible for a selective transfer of nutrients and waste products into and out of the cell. The region within the cytoplasmic membrane includes the cytoplasm and the nuclear region. The cytoplasm is characterized by a relatively high concentration of ribonucleic acid (RNA), which is associated with protein synthesis. The nuclear region contains a high concentration of deoxyribonucleic acid (DNA). The genetic code is stored in the DNA. Some bacteria secrete slimy materials on their surfaces, creating a slimy layer outside the cell wall. A rough sketch of a bacterial cell is shown in Figure 5.1.1.

The activitiy of the bacteria will be described in the following four sections.

(a) Metabolism:

Bacterial cells, like all living cells are capable of feeding and growing. Soluble organic matter (substrate) diffuses into the bacterial cell where they are metabolized and converted partly into new cell matter. The remainder of the organic substrate is respired to provide energy for the cell. The respiration products are mainly carbondioxide and water.

Schematically, this may be represented as:

The yield factor, Y, is defined as the fraction of soluble carbon metabolized which goes into new cell growth. The fraction (1-Y) which is respired will require an equivalent amount of oxygen.

The rate of soluble carbon degradation can be written as

rate =
$$\frac{dC}{dt}$$
 = - UM (5.1.1)

where C = concentration of soluble degradable carbon

$$U = f(c) = unit growth rate$$

and M = concentration of microorganisms

(as equivalent carbon)

For the case of a single substrate and a pure culture, the functional dependence of u is described by Michaelis-Menton kinetics as

 $U = \frac{U_{max} C}{C^* + C}$ (5.1.2)

where U_{max} = maximum growth rate

and C^* = constant = value of C at which U = $\frac{1}{2}U_{max}$

However, in waste treatment, mixed substrates are generally encountered. Pure cultures are also difficult to maintain and is often less efficient than mixed cultures. Consequently, there is moneed to restrict the description of U by Michaelis-Menton kinetics. The measurements of U and Y as described in Section 5.2 show that a linear correlation

$$U = k_{\Delta}C \tag{5.1.3}$$

is a better fit. It should also be noted that the Michaelis-Menton equation becomes linear when C << C*.

(b) Endogenous Respiration

Bacterial cells can also utilize stored food products (bacterial fat) to provide energy for its various physiological activities, especially in perids of low external food supplies. This is known as endogenous respiration or sometimes as basal metabolism.

Schematically, this may be represented as

stored "bacterial fat" + $0_2 \longrightarrow \text{Respiration products}$ (5.1.4)

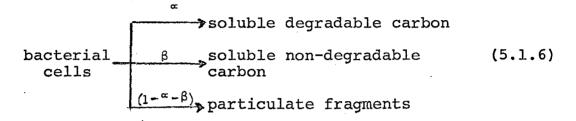
The rate of endogenous respiration should be proportional to the number of bacterial cells present, or is first order with respect to the bacterial concentration, namely,

rate =
$$\frac{dM}{dt}$$
 = - k₁M (5.1.5)

where $k_1 =$ endogenous respiration rate constant.

(c) Death and Cell Lysis

In cell lysis, the inner protoplasm of the cell is released to the surroundings through a breach in the cell wall. Part of the protoplasm is available as food for other microorganisms, while the remainder consists of relatively non-biodegradable high molecular weight matter, schematically, this may be shown as



 \propto is the fraction of soluble degradable carbon, and β is the fraction of soluble non-degradable carbon released during cell lysis. The remaining fraction $(1-{}^{\alpha}-{}^{\beta})$ is particulate matter consisting of cell walls, slime layers, various membranes and so on, with varying degrees of biodegradability.

This process explains the build-up of soluble nondegradable carbon during the prolonged aeration of activated sludge.

The rate of death and subsequent cell lysis should also be proportional to the number of bacterial cells present:

rate =
$$\frac{dM}{dt}$$
 = - k_2^M (5.1.7)

where $k_2 = 1$ ysis rate constant.

(d) Solubilization of Particulate Organics

In general, solid organic matters have to be solubilized before they can be utilized by the bacteria. This process is speeded up by the secretion of exo-enzymes by the bacterial cells. The solubilized organic matter then diffuses back into the bacterial cell as substrate. The rate of solubilization should be proportional to the concentration of the organic solids. It should also be proportional to the rate of release of exo-enzymes or equivalently, to the number of bacterial cells present:

rate =
$$\frac{dP}{dt}$$
 = $-k_3 PM$ (5.1.8)

where P = solid organics concentration

(as equivalent carbon),

and k₂ = solubilization rate constant.

The four reactions proposed above involves principally five components, namely the microorganisms (M), the particulate organics other than microorganisms (P), soluble degradable organic carbon (SDOC), soluble non-degradable organic carbon (SNDOC) and carbon dioxide. However, carbon dioxide is partially purged from the system by the air and hence is difficult to monitor. But, it can always be obtained from a mass balance on carbon.

The total soluble organic carbon, SDOC+SNDOC, can be determined by means of the carbon analyzer. The soluble degradable organic carbon can be obtained from the Biochemical Oxygen Demand of the filtrate, using the relationship

SDOC = BOD (filtrate)/1.9 (5.1.9) This is discussed in Section 9.1. The SNDOC is then obtained by difference. The total concentration of microorganisms and particulate solids, M + P, can be obtained from the volatile suspended solids using the relationship (see also Section (9.1)

$$M + P = VSS/2.1$$
 (5.1.10)

The concentration of microorganisms alone can be indirectly measured from its oxygen uptake rate. In addition to the oxygen required for the carbon respired, oxygen is also consumed in the nitrification of ammonia-nitrogen associated with the sludge. A chemical analysis of activated sludge reports an average composition of $C_5H_7O_2N$ (54). The oxidation reaction can then be written as:

 $C_5H_7O_2N + 7O_2 \longrightarrow 5CO_2 + 3H_2O + HNO_3$ (5.1.11) Hence each gram of carbon lost through respiration requires 3.73 grams of oxygen or

OUR = 3.73
$$\begin{bmatrix} k_1 + (1-Y)U \end{bmatrix}$$
 M (5.1.12)

therefore,

$$M = \frac{OUR}{3.73 \left[k_{1} + (1-Y)U \right]}$$

Hence, our aerobic biological system involves four simultaneous, kinetically independent reactions, five components and seven constants. The constants involved are k_1 , k_2 , k_3 , k_4 , Y, α and β .

5.2 Chemostat Experiments

The purpose of this series of experiments is to evaluate the functional dependence of the unit growth rate on the soluble degradable carbon concentration. A chemostat is essentially a biological reactor that acts as a continuous-flow, completely stirred tank reactor. The defining equations are easily derived from a mass balance around the reactor

In Figure 5.2.1, Q is the flow rate through the reactor and V the volume of the reactor. The concentration of soluble degradable organic carbon in the feed is C_0 and that in the effluent is C. The concentration of micro-organisms in the feed is assumed to be zero while that in the vessel or effluent is M (as equivalent carbon).

Performing a mass balance on the soluble carbon, we have at steady-state,

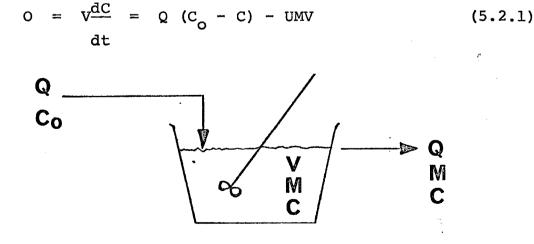


FIGURE 5.2.1. A Schematic Representation of a Chemostat, and Operating Variables.

where U = unit growth rate (see Section 5.1)

Hence U = f(c) =
$$\frac{(C_0 - C)}{M_{+}^{+}}$$
 (5.2.2)

where $\bar{t} = V/Q$ = residence time. Also from definition, the yield factor

 $= Y = M/(C_0 - C)$ (5.2.3)

In our experiments, settled raw sewage was filtered through a two-inch thick layer of glass wool and used as the feed. This was stored in a constant head bottle and allowed to drip into an aerated 6 litre vessel (see Figure 5.2.2). Seeding was accomplished by the addition of about 100 ml of activated sludge to the reaction vessel. After two detention times, the effluent was sampled and analyzed regularly, until a steady-state was reached.

The results from seven runs are summarized in Appendix C. The average value of the yield factor Y obtained is 0.57 with a standard deviation of 0.04. the functional dependence of U is found to be linear with respect to the soluble degradable carbon concentration, in the range studies (25 - 50 mg/l). This is plotted in Figure 5.2.3. The correlation obtained is

 $U = 0.0010 C = k_4 C \qquad (5.2.4)$ with a correlation coefficient of 0.85.

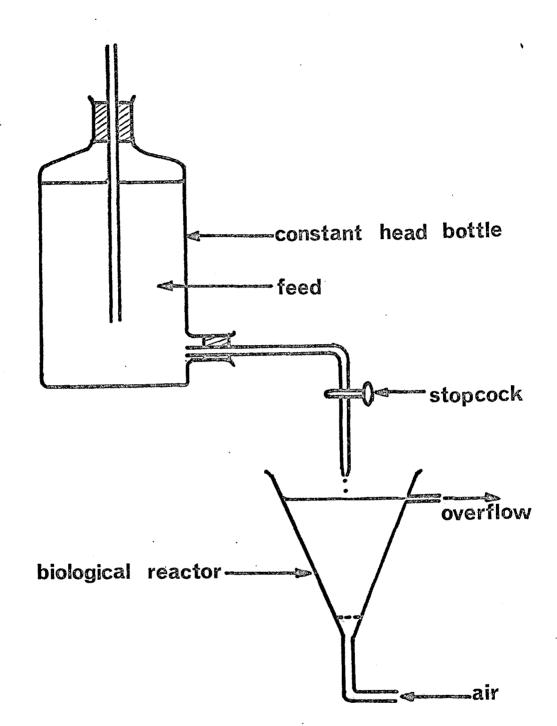
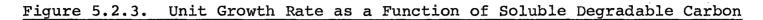
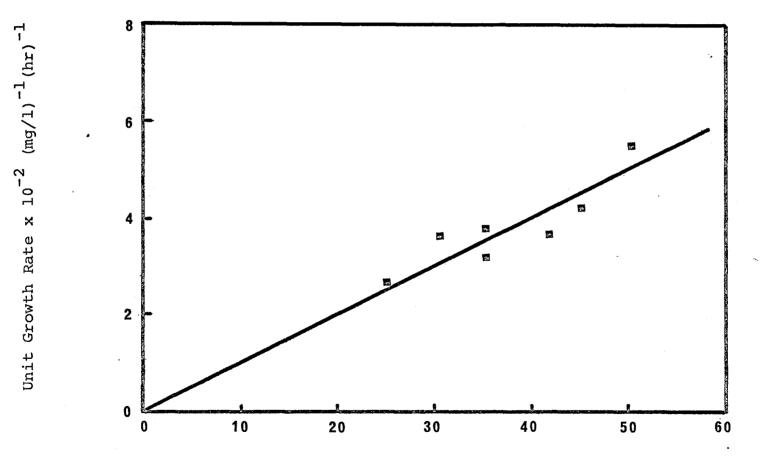


Figure 5.2.2. Experimental Set-Up for the Chemostat



Concentration



Soluble Degradable Carbon (mg/l)

We have assumed that the solids in the feed is not significantly solublized in the reactor.

To compare the results obtained we have to convert Equation 5.1.2 to traditional units:

We have
$$\frac{dC}{dt} = \frac{d BOD_5}{dt} \times \frac{1}{1.9}$$

It is estimated that microorganisms account for approximately 30% of the Mixed Liquor Suspended solids. Hence $M \ge 2.1 = 0.3$ (MLSS)

Therefore, Equation 5.1.2 becomes

$$\frac{1}{1.9} \times \frac{\text{dBOD}_5}{\text{dt}} = -k_4 \times \frac{\text{BOD}_5}{1.9} \times \frac{0.3}{2.1} \text{ (MLSS)}$$

or $\frac{\text{dBOD}_5}{\text{dt}} = -k_4 \frac{0.3}{2.1}$ (BOD₅) (MLSS)

$$= - K_{s}^{"} (BOD_{5}) (MLSS)$$
(5.2.5)
or $K_{s} = k_{4} \frac{0.3}{2.1}$

= 0.00014 (hr)⁻¹ (mg/1)⁻¹

Eckenfelder (2) reports a value of

$$K_s = 0.00010 - 0.00013 (hr)^{-1} (mg/1)^{-1}$$

Keshavan et al (22) reports a value of

 $K_s = 0.00016 (hr)^{-1} (mg/1)^{-1}$

Hence, our results are approximately in agreement with the literature values.

5.3 Aeration Studies of Activated Sludge

As some of the rate proposed in Section 5.1 are expected to be slow, batch studies of aerated activated sludge have to be carried out over an extended period. Return activated sludge is a convenient "reaction mixture" to use, as it contains all the components in our reaction set.

Return activated sludge from the Drury Lane plant was used, and the aeration was carried out in a 6-litre vessel, for twenty five days. Samples were taken periodically, and the following analyses performed:

(1) Biochemical Oxygen Demand of filtrate,

(2) Dissolved Organic Carbon,

(3) Suspended Solids, and

(4) Oxygen Uptake rate.

Inorganic solids were determined both initially and at the end.

Two runs were made with different samples of return activated sludge from the Skyway Plant, Burlington. The results obtained are tabulated in Appendix D. The plots of (P + M), OUR, SDOC and SNDOC versus time are shown in Figures 5.3.1 - 5.3.6. The volatile suspended solids and the oxygen uptake rate were observed to drop with length of aeration. The soluble degradable organic carbon dropped slightly and then remained fairly constant. There was a marked increase in soluble non-degradable organic carbon with aeration in both cases.

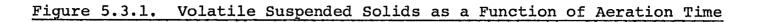
To obtain an estimate of the variances in the measurements, a third run was started and allowed to run for a week. Then five samples were withdrawn simultaneously from the reaction vessel and analyzed immediately. The results are also tabulated in Appendix D, together with the variancecovariance matrix and its inverse matrix.

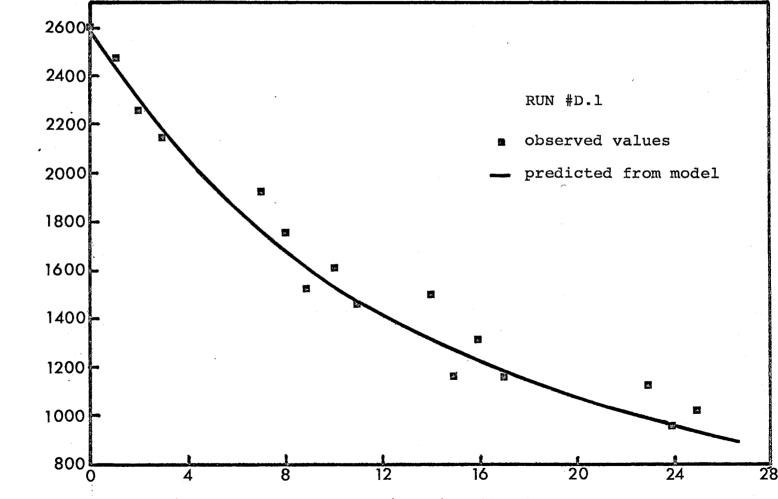
Before the parameter search was performed, an eight constant Q_0 was introduced and it is defined as the ratio of the initial concentration of microorganisms, M_0 , to that of the total volatile solids, namely,

$$Q_0 = \frac{M_0}{(P + M)_0}$$
 (5.3.1)

where the subscript $_{O}$ refers to initial values. Otherwise, we have to calculate M_{O} from the intial value of the oxygen uptake rate, and as M_{O} is a critical value, an error in the oxygen uptake rate can have a marked effect on the search.

A fourth order Runge Kutta was used to calculate the values of the four components, P, M SNDOC, and SDOC. The



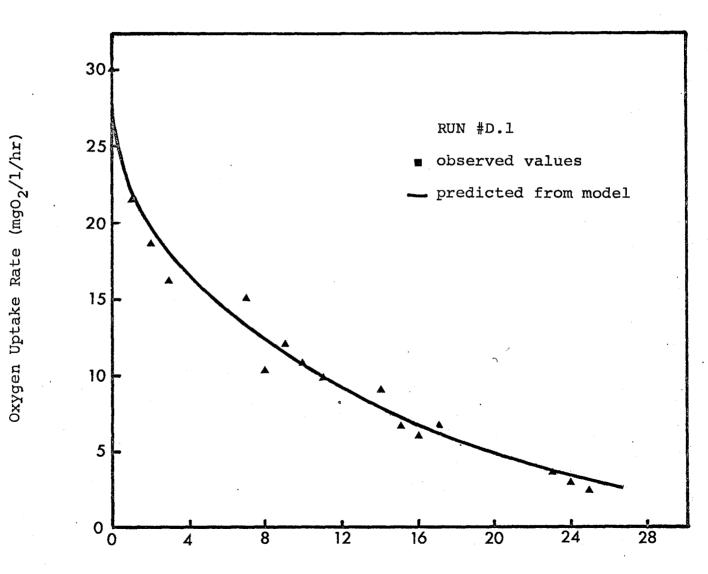


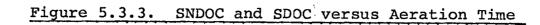
Solids (mgC/1)

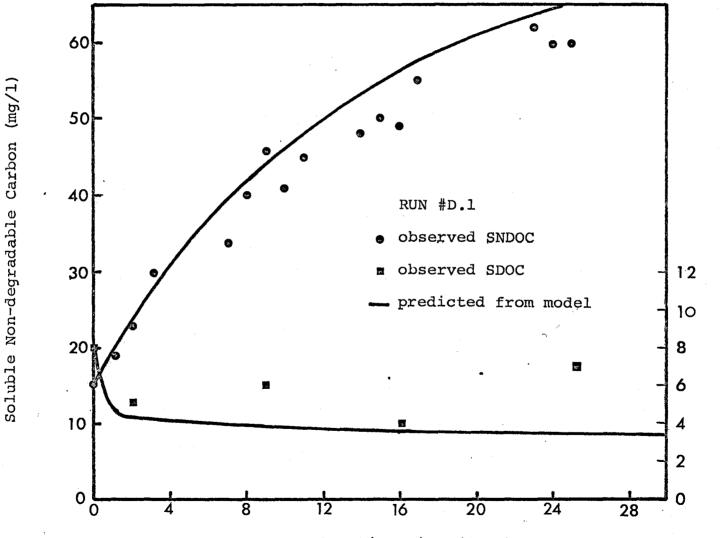
Suspended

Volatile

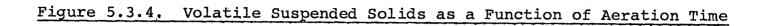
Aeration Time (days)

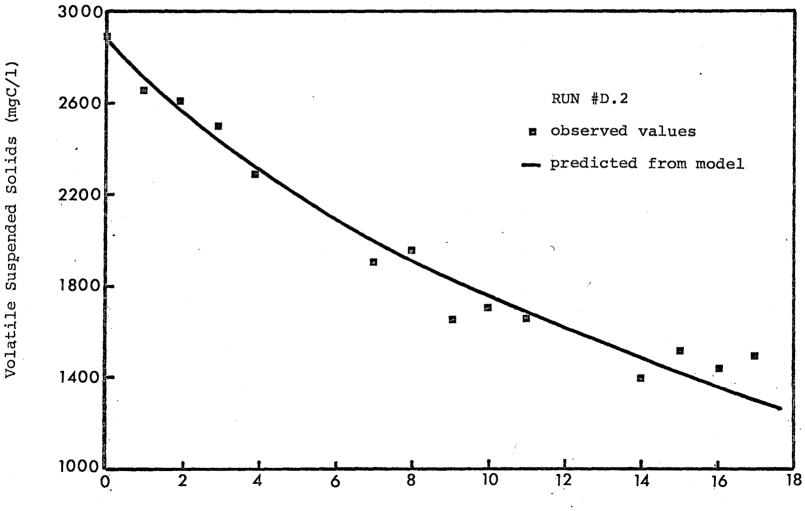


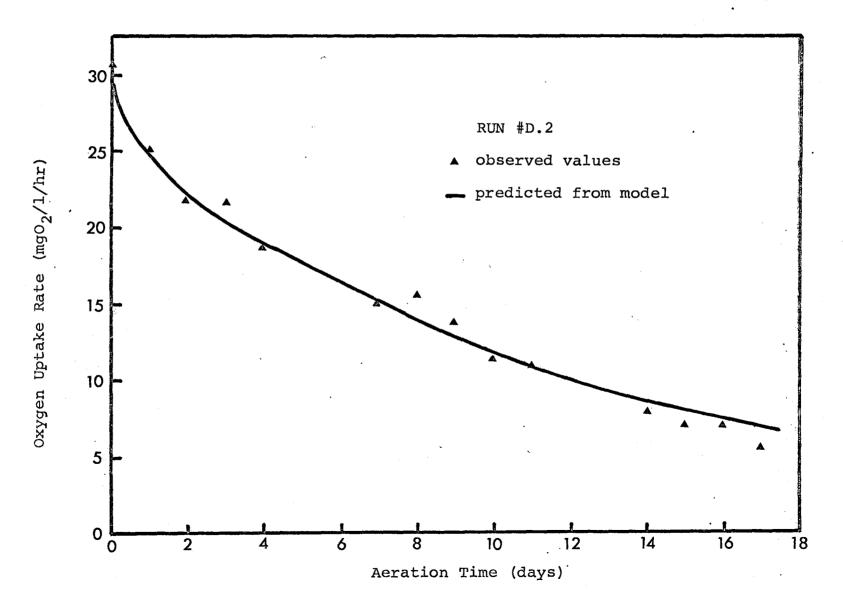




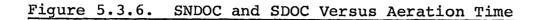
Soluble Degradable Carbon (mg/l)





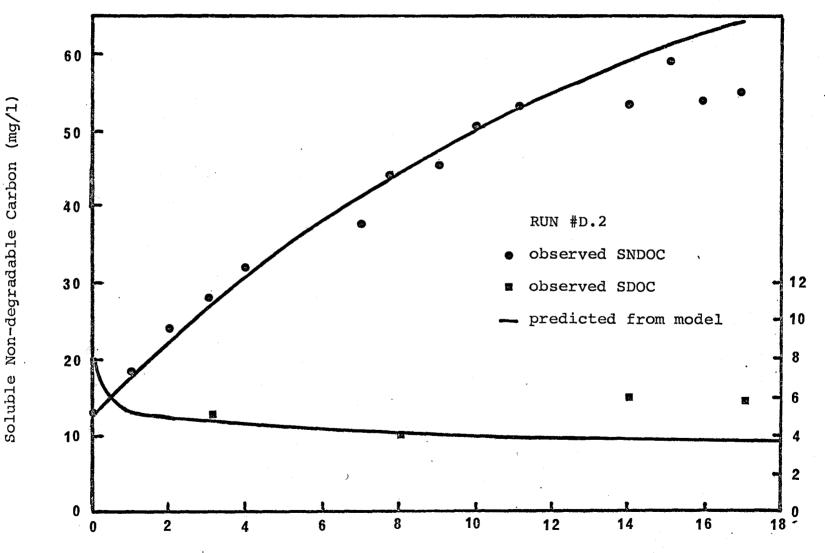


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Aeration Time (days)

Soluble Degradable Carbon (mg/l)

set of differential equations used were

$$\frac{dP}{dt} = \begin{bmatrix} -k_3P + (1-\alpha-\beta)k_2 \end{bmatrix} M$$

$$\frac{dt}{dt} \quad \text{with an initial value of } (1-Q_0)(P+M)_0 \quad (5.3.2)$$

$$\frac{dM}{dt} = \begin{bmatrix} k_4 \text{ (SDOC)}Y - (k_1 + k_2) \end{bmatrix} M$$

$$\frac{dt}{with an initial value of Q_0(P+M)_0} (5.3.3)$$

$$\frac{dSNDOC}{dt} = \beta k_2^{M}$$

$$\frac{dt}{dt} \qquad \text{with an initial value of (SNDOC)}_{O} \qquad (5.3.4)$$

and
$$\frac{dSDOC}{dt} = \begin{bmatrix} k_3^P - k_4(SDOC) + k_2^{\alpha} \end{bmatrix}^M$$
.
dt with an initial value of (SDOC) (5.3.5)

A SIMPLEX optimization routine was used and the objective function to be minimized is (18)

$$\phi = \sum_{i=1}^{n} \sum_{r=1}^{4} \sum_{s=1}^{4} \sigma^{rs} (y_{ri} - \eta_{ri}) (y_{si} - \eta_{si}) \quad (5.3.6)$$

 y_{ri} = the ith observation of the rth response and η_{ri} = the ith prediction of the rth response. The four responses used are:

- (1) Volatile suspended solids (mgC/l), with the predicted value calculated from P + M
- (2) Oxygen uptake rate $(mgO_2/1/hr)$, with the predicted value calculated from 3.7M $k_1 + (L-Y)k_4$ (SDOC)
- (3) Soluble non-degradable organic carbon, SNDOC, (mgC/1), and
- (4) Soluble degradable organic carbon, SDOC, (mgC/1)

A schematic representation of the parameter search is shown in Figure 5.3.7.

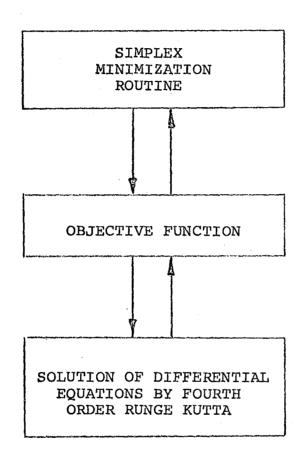


FIGURE 5.3.7. Schematic Representation of Parameter Search

The values of the constants obtained are shown in Table 5.3.1. The fitted curves are shown in Figures 5.3.1 to 5.3.6, together with the observed points.

Run Number	k _l	^k 2	^k 3	^k 4	Y	œ	β	Θο
D-1	0.0029	0.0015	2.0x10 ⁻⁶	6.0×10^{-4}	0.55	0.41	0.082	0.56
D-2	0.0029	0.0014	1.8x10 ⁻⁶	5.6x10 ⁻⁴	0.57	0.44	0.089	0.56

Table 5.3.1. Parameters for the Aerobic Biological Sludge Process Model

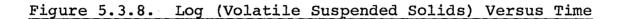
The agreement in the values of the parameters in both runs are very good, considering the analytical errors involved. From the Chemostat experiments, the values of k_4 and Y obtained are 1.0 x 10^{-3} and 0.57. The agreement in the yield factor is good, but the metabolism rate factor is much lower in the "endogenous phase" of the extended aeration studies. The results imply that the metabolism rate is retarded under conditions of prolonged starvation.

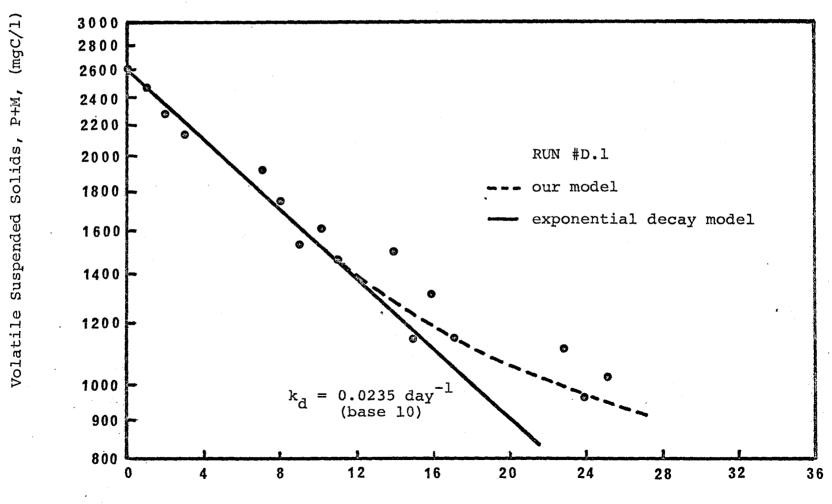
Sedivy (37) in his thesis on residual organics found β to be about 1-5%, using glucose as substrate. Our value is a bit high. The value of θ_0 obtained suggests that only about half of the volatile suspended solids in the return sludge from the Skyway Plant, is actually "active mass".

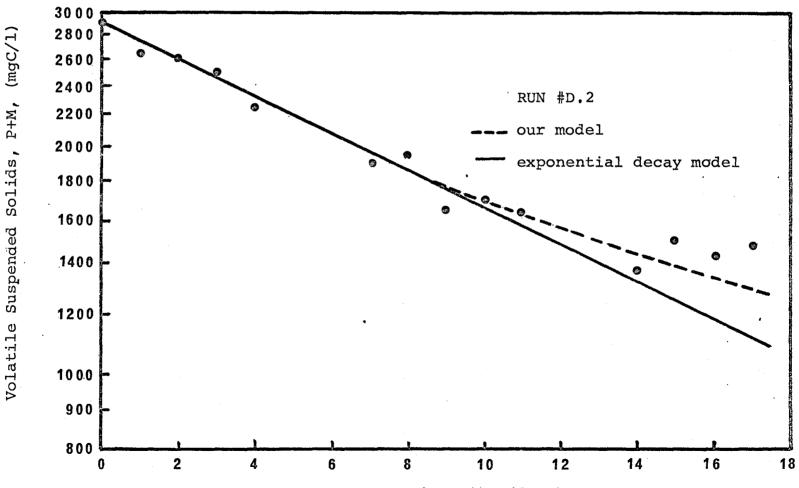
No equivalent values for the other constants exist in the literature. Note that the basal metabolism rate constant $\mathbf{k}_{\mathbf{d}}$ proposed by other workers ia actually measured from the rate of decrease of volatile suspended solids and which in our case is a combination of k_1 , k_2 and k_3 . To illustrate that our reaction scheme fits the rate of decrease of volatile suspended solids better than the simple exponential decay, semi-log plots are shown in Figure 5.3.8 and 5.3.9, of the volatile solids versus time. In both runs, our model agrees with the exponential decay model up to about an aeration time of ten days. Beyond that the exponential decay model predicts much lower solids than is actually observed. Our model gives a much better fit beyond ten days, although it still predicts lower solids than is actually observed. This is probably because our model still lacks the ability to predict the concentration of particulate non-degradable organic carbon, which would build up with extended aeration.

As a matter of interest, the k_d value from our data, which is 0.055 day⁻¹ (base e) agrees with Pipes and Meade (31) very well. Their data showed a range of k_d between 0.025 to 0.078 day⁻¹ with an average value of 0.053 day⁻¹. Eckenfelder (2) reports a value of 0.048 day⁻¹.

As the statistical calculations are quite difficult, the estimation of confidence limits for the various constants







will not be attempted. Instead, the average sum of squares deviation from the regression line will be compared to the variance obtained in the replicate samples. This is presented in Table 5.3.2.

	Number of	Component					
Run	Points or Samples	Volatile Suspended Solids		Soluble Non-degradable Organic Carbon			
D-1 D-2	16 14	8000 8400	1.2 1.0	18.5 13.6	1.3 (only 5 points) 2.3 (only 5		
Repli- cates	5	6450	2.0	18.5	points) 1,2		

Table 5.3.2.Comparison of Average Sum of Squares Deviationfrom Regression Line, with the Variance from
Replicate Samples

The average sum of squares from the regression line compares quite well with the variance from the replicate analysis. This indicates that the model proposed is significant.

No measurements were made on the effect of temperature on the various rates. The above two runs were made at $23^{\circ}C \pm 2^{\circ}C$. A temperature correction factor of 1.047 will be used, as proposed by several authors (2, 47). Hence,

> $k_1(T) = 0.0026 (1.047)^{T-23}$ $k_2(T) = 0.0015 (1.047)^{T-23}$

÷4

 $k_3(T) = 2.0 \times 10^{-6} (1.047)^{T-23}$ $k_4(T) = 0.0010 (1.047)^{T-23}$ or $k_4(T) = 0.0006 (1.047)^{T-23}$ in food limiting conditions.

(5.3.7)

5.4 The Stripping Rate of Purgeable* Soluble Organics

Purgeable soluble organics tend to be stripped off . in the aeration process. Hence their removal can occur both by biological action and by physical stripping due to the passage of air (16, 17).

According to mass transfer principles, the mass transfer will depend on the first order of the driving force, or concentration difference and also on the surface area of the air bubbles, namely,

$$\frac{dSPOC}{dt} = -k_{L} \frac{A}{V} (SPOC - SPOC_{b})$$
(5.4.1)
dt V

where SPOC = soluble purgeable organic carbon in solution $SPOC_b$ = soluble purgeable organic carbon in bubble k_L = a mass transfer coefficient A = surface area of bubbles and V = volume of reactor

 ^{*} Waste treatment terminology reserves the use of the word "volatile" for volatiles at 600°C. The term "purgeable" is then used here to define the soluble organics which can be physically removed by purging with a gas, at normal temperatures.

But
$$A/V = a_{vq}$$
 (5.4.2)

where a_V = surface area to yolume ratio of air bubbles
and q = volumetric rate of air per unit volume of
reactor.

Hence,

$$\frac{d(SPOC)}{dt} = -k_L a_V q. (SPOC) \qquad (5.4.3)$$

$$= -k_5.q.(SPOC)$$
 (5.4.4)

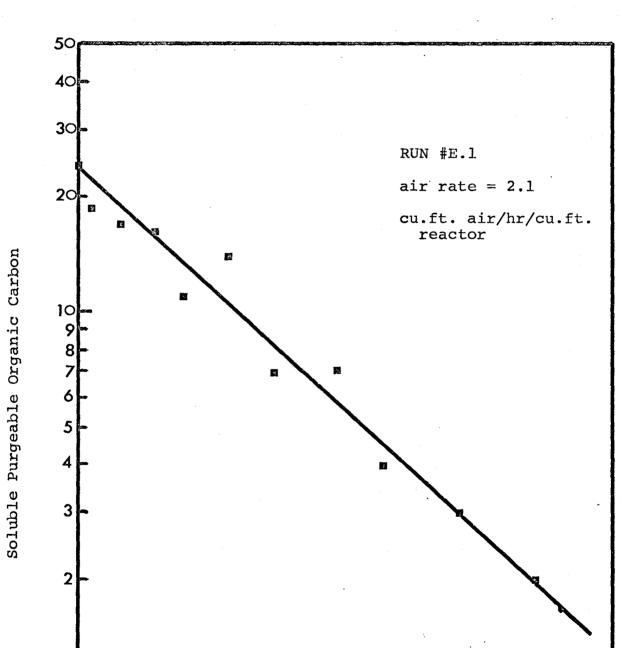
Solving,

$$\ln \frac{(\text{SPOC})}{(\text{SPOC})} = -k_5 \text{ qt} \qquad (5.4.5)$$

where (SPOC) = initial value of SPOC.

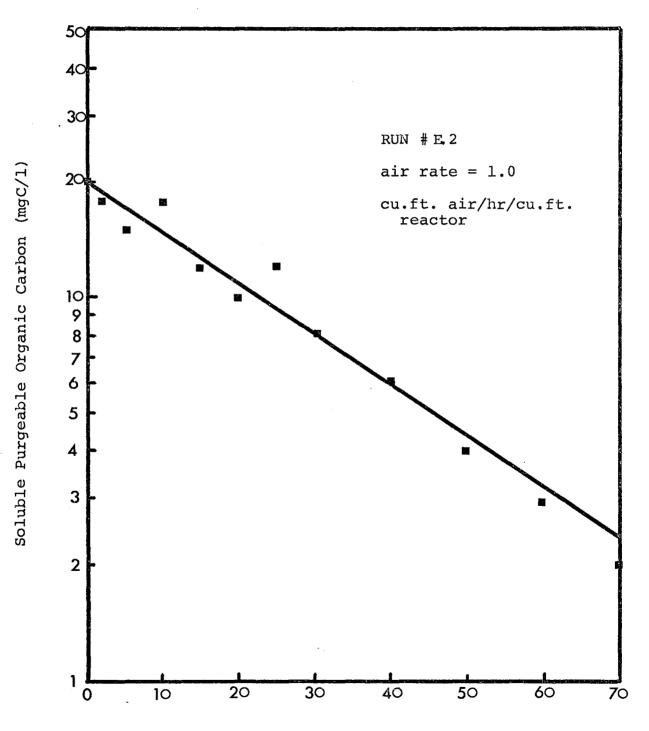
To estimate the rate of stripping, settled raw sewage was filtered through a two-inch layer of glass wool and aerated at a fixed rate in a 2 litre vessel (no activated sludge was added). The soluble organic carbon was measured at regular intervals, until no further drop was observed. The remaining soluble organics is non-purgeable. The difference between the soluble organic carbon and the ultimate value (or the non-purgeable organic carbon) is the soluble purgeable organic carbon (SPOC).

Four air rates of 0.5, 1.0, 1.5 and 2.1 cu. ft. air/ hr/cu. ft. reactor were used and the results are tabulated in



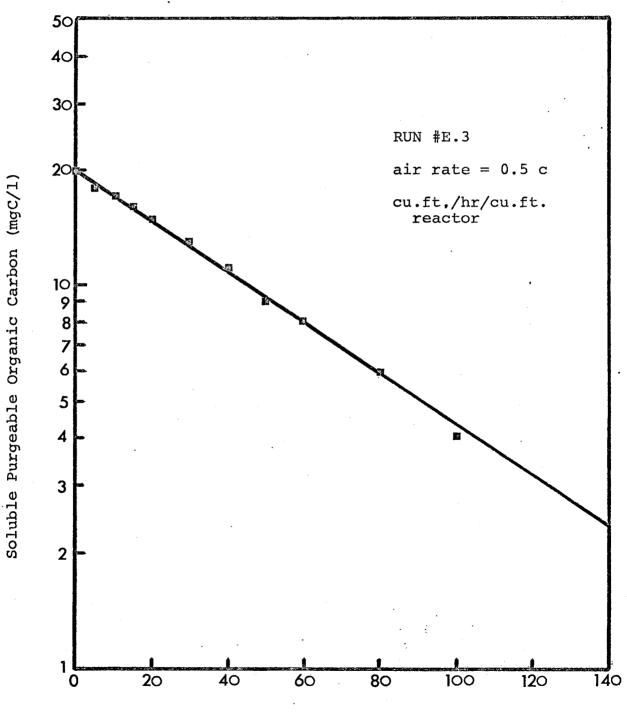
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Purging Time (minutes)

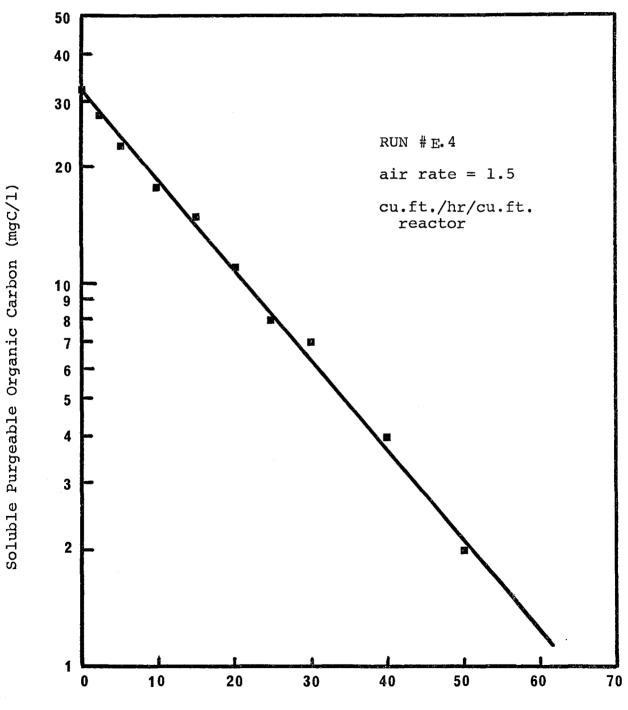


Time (minutes)

. .



Time (minutes)



Time (minutes)

Appendix E. Semi-log plots of SPOC versus time are shown in Figures 5.4.1 to 5.4.4. The slopes of the lines will be $-k_5q$. Table 5.4.1. shows the values of k_5 obtained from the four runs. The average value of k_5 is 2.06 with a standard deviation of 0.27.

Run Number	E-1	E-2	E-3	E-4	Average
q cu.ft.air/hr cu.ft. reactor volume	2.1	1.0	0.5	1.5	
$slope = k_5q hr^{-1}$	2.18	0.80	0.40	1.40	:
k ₅ (base e)	2.40	1.85	1.85	2.15	2.06

5.5 Oxygen Transfer Efficiencies

The rate of oxygen transfer can be found by using the same equation as that for the stripping of purgeable organics, namely

$$\frac{d [o_2]}{dt} = k'_L \frac{A}{V} ([o_2]_{sat} - [o_2])$$
(5.5.1)
$$\frac{dt}{V} = V$$

where $\begin{bmatrix} 0_2 \end{bmatrix}$ = concentration of oxygen in liquid $\begin{bmatrix} 0_2 \end{bmatrix}$ sat = saturation concentration of oxygen in liquid k = liquid film mass transfer coefficient
A = surface area of air bubbles

and V = volume of aeration tank

The saturation concentration of oxygen in water at 1 atm can be calculated from the equation (9)

$$O_2$$
 sat., l atm. = 14.16 - 0.3943T + 0.007714T²
- 0.0000646T³ (5.5.2)

where $T = temperature in {}^{O}C$

 \cdot $k_{L}^{\dagger}A$ is a function of the aeration device, the air flow rate and of the depth of the deration tank. The functional relationship is

$$k_{1}A = N.G^{(1-n)}H^{(1-m)}$$
 (5.5.3)

where N = sometimes referred to as absorption number

G = total air rate, scfm.

H = depth, ft.

and $n_m = constant$

Table 5.5.1 shows the various values of the constants, for several aeration devices (9).

5.6 Nutrient Balance

The assumption made is that the nitrogen to carbon (N/C) and phosphorus to carbon (P/C) ratios in the raw waste is the same for both the particulate and soluble organic

Unit	N.	(l-n)	(l-m)	
Aloxite tubes	270	0.85	0.45	
Sparjer	210	0.86	0.78	
Seran tubes	275	0.8	(0.60)	
Carborundum plates	65	0.8	(0.75)	

Table 5.5.1. Oxygen Transfer Characteristics of Some Common Diffused Aeration Equipment.

components. This is supported by analyses made by the Ontario Water Resources Commission for the Drury Lane W.P.C.P. This is shown in the table below.

<u>Element</u> Phase	Nitrogen	Phosphorus	Carbon	N/C	<u>₽</u> /C
Particulate Soluble	22 23	7 6	• •		0.078 0.075

Table 5.6.1.Average Nitrogen, Phosphorus and CarbonConcentrations in Raw Sewage from DruryLane W.P.C.P.The Nitrogen and PhosphorusValues are Obtained from 29 Analyses by theO.W.R.C.The carbon concentrations are

estimated from our analyses.

.

The composition of the bacterial cells is assumed to be $C_5H_7O_2N$ (54). This has a (N/C) ratio of 0.23. The (P/N) ratio in activated sludge is reported to be about 0.25 (9), hence (P/C) is approximately 0.06.

After the various effluent organic carbon components have been calculated, a mass balance over nitrogen and phosphorus is then performed. All excess nitrogen is assumed to be converted to ammonia and all excess phosphorus to inorganic phosphates.

Downing's data for the rate of nitrification will be used (7):

$$\frac{dNH_3}{dt} = - \frac{G. NH_3. X}{Y (K + NH_3)}$$
(5.6.1)

and

 $\frac{dx}{dt} = - x \frac{ds}{dt}$ (5.6.2)

where NH_3 = ammonia concentration (mgN/1) X = Nitrosomonas concentration (mg/l) G = 0.014 hr⁻¹ K = 1.0 mg/l

and $Y = 0.05 \text{ gms Nitrosomonas/gm NH}_3-N$

A slight modification will be made to Downing's equation. This is to account for a decrease in the rate due to an oxygen concentration of less than 1.0 mg/l. The retarding factor f is defined where

$$f = 1 \text{ for } [0_2] \ge 1.0 \text{ mg/l}$$

and $f = [0_2] \text{ for } 0.0 \le [0_2] \le 1.0 \text{ mg/l}$ (5.6.3)

i.e.,

$$\frac{dNH_3}{dt} = - \frac{f.G. NH_3.X}{Y (K + NH_3)}$$
(5.6.4)

The mass of Nitrosomonas is found iteratively and stored in the EN vector as a fraction of the mixed liquor suspended solids mass.

The oxygen requirement during nitrification can be obtained from the equation

 $NH_3 + 20_2 \rightarrow HNO_3 + H_2O$ (5.6.5)

Hence, for each gram of NH_3 -N converted to NO_3^- , 4.57 grams of oxygen are required.

5.7 The Aerobic Biological Sludge Process Model - ACTSL1 and CSTR2

CSTR 2 is a continuous-flow, completely-stirred tank reactor model of the aerobic biological sludge process. ACTSL1 is a stirred tanks in series model which calls the basic CSTR2 program. ACTSL1 is the program that is actually used as the unit computation. Listings of the two programs are given in Appendix K. The equipment vector for ACTSL1 is reproduced below, of which EN(16) - EN(32) are actually used by the CSTR2 routine.

EN VECTOR ***** STANDARD FORMAT NUMBER OF REACTORS IN PARALLEL TOTAL HEAD LOSS THROUGH ACT. SL. TAN NUMBER OF INPUT STREAMS (= 2.0) FIRST INPUT STREAM (= RETURN SLUDGE SECOND INPUT STREAM (= FEED STREAM) NUMBER OF OUTPUT STREAMS (= 1.0) OUTPUT STREAM NUMBER 1.-15. 4. 5. TANKS (FT OF WATER) 6 7 8 11 12 FLOW) NOTE EN(16.-32.) ARE USED BY CSTR2 LENGTH OF REACTOR (FEET) WIDTH OF REACTOR (FEET) DEPTH OF REACTOR (FEET) = 1.0 IF NOT USING A RESIDENCE TIME DISTRIBUTION MODEL = EN(34) IF USING A RTD MODEL LYSIS RATE (K2 /HR) BASAL METAB. RATE (K1 /HR) RATE OF SOLUBILIZATION OF DEGRADABLE SOLIDS (K3 /HR) STRIPPING RATE OF VOLATILE ORGANICS (K5 /HR) MICROORGANISMS YIELD FACTOR (Y DIMENSIONLESS) FRACTION OF LYSIS PRODUCTS TO SOLUBLE DEGRADABLE CARBON FRACTION OF LYSIS PRODUCTS TO SOLUBLE NONDEGRADABLE CAR RATE OF AIR ADDITION TO EACH REACTOR (SCF/HR) UNIT GROWTH RATE CONSTANT (K4 /HR) FRACTION OF MLSS THAT IS NITRIFIERS • N/C RATIO IN RAW FEED $16 \\ 17$ 18.19. 20. 21. 23456789 DEGRADABLE CARBON NONDEGRADABLE CAR CARBON 30. N/C RATIO IN RAW FEED P/C RATIO IN RAW FEED ABSORPTION NUMBER (FOR OXYGEN TRNSFER) 31 32 NOT PRESENTLY USED 33. NUMBER OF CSTR,S IN SERIES - EITHER AN RTD MODEL OR THE ACTUAL NUMBER OF TANKS IN A STEP AERATION PROCESS MAXIMUM OF TEN TANKS IN SERIES FRACTION OF FEED STREAM TO FIRST TANK FRACTION OF FEED STREAM TO SECOND TANK 34. 35. 36. 37. ETC 44 45 46 FRACTION OF FEED STREAM TO TENTH TANK (IF ANY) NUMBER OF BLOWERS USED CAPACITY OF EACH BLOWER SCF/HR

ACTSL1 may be used as the step aeration process model or as a model for the activated sludge process and its various modifications. For the contact stabilization process, ACTSL1 will have to be used twice.

The computations carried out by the CSTR2 program may be explained by the following steps:

. •

- Stream variables are first converted to concentration units,
- (2) The iteration then begins on the effluent concentrations of the reactive components. All inert components are passed through the model. The reaction rates used are:

(a)
$$\frac{dM}{dt} = (Yk_4C - k_1 - k_2)M$$

where $M = SO(1, 15)$
 $C = SO(1, 20) + SO(1, 21) = C' + C''$
 $k_4 = EN(28)$
 $k_1 = EN(21)$
and $k_2 = EN(20)$
(b) $\frac{dS'}{dt} = (-k_3S' + k_2(1-\alpha-\beta))M$
 dt
where $S' = SO(1, 16)$
 $\alpha = EN(25)$
 $\beta = EN(26)$
and $k_3 = EN(22)$
(c) $\frac{dS''}{dt} = -k_3S''M$
 dt
where $S'' = SO(1, 18)$ as well as $SO(1, 19)$

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(d)
$$\frac{dC'}{dt} = -(k_4M + k_5q)C'$$

dt
where C' = SO(1, 20)
and q = volume air used/hr/volume of reactor.

(e)
$$\frac{dC''}{dt} = (k_3 S + k_2 a - k_4 C'')M$$

dt
where $S = 5' + S''$
and $C'' = SO(1, 21)$

(f) $\frac{dSNDC}{dt} = k_2 \beta M$

where SNDC = SO(1, 22)

The concentration of ammonia nitrogen released and the subsequent nitrification to nitrates are also calculated, within the iteration loop. The rate of change of dissolved oxygen is next given by

(g)
$$\frac{dO_2}{dt} = k_L a (O_S - O_2) - \frac{32}{12} \left\{ (1-Y) k_4 C + k_1 \right\} M$$
$$- 4.57 (\frac{dN}{dt})$$
$$dt$$
where $O_2 = SO(1, 7)$
$$O_S = saturation oxygen concentration$$

 $k_L a = oxygen mass tranfer coefficient$ and $\frac{dN}{dt} = nitrification rate.$

- (3) A balance is then made on phsophorus,
- (4) The exit pH is next calculated assuming that the carbondioxide concentration is in equilibrium with air.

CHAPTER 6

6. SECONDARY SEDIMENTATION

In the activated sludge process the waste to be treated is mixed with a flocculent sludge containing microorganisms and other organic and inorganic solids, and aerated. The microorganisms are used to feed on the soluble portion of the waste and to convert them into more cell mass which may then be settled out. The level of suspended solids used in the activated sludge process range from 2000-4000 mg/l. As the treated water will constitute the plant effluent in most cases, it is important that the activated sludge is removed and the suspended solids concentration in the effluentbe reduced to the level of about 20 mg/l. Secondary sedimentation performs such a purpose.

As the settled sludge will be recycled to the activated sludge tanks, and partially "wasted", it is equally important that we keep the solids content of the sludge high. The benefits resulting from a "thick" sludge are:

- (a) It maintains a high level of mixed-liquor suspended solids in the activated sludge tanks,
- (b) It reduces pumping costs,
- (c) It improves digester operation on the waste sludge and hence requires smaller digester volumes,

- (d) It increases the economy of sludge dewatering systemssuch as centrifuges, vacuum filters, etc., and
- (e) It reduces the sludge volume for land or sea disposal.

6.1. Activated Sludge Characteristics

The clarification and thickening propoerties of an activated sludge have been related to its composition and character. A sludge with a relatively high content of inorganic solids (such as clay) will tend to be denser and hence more easily thickened. However, the character of the microorganisms in the sludge can determine whether a sludge will be flocculent and hence have good clarifying properties or whether the sludge will be compact and have good thickening properties.

The microbial population in activated sludge include the bacteria, fungi, protozoa and some rotifiers. The bacteria are the most predominant group, and as they are responsible for stabilizing the organic matter in the waste and in floc formation, they are the most important group as well. Numerous types of bacteria may be found in any sample of an activated sludge, but sometimes one or several genera may predominate, depending on the particular waste being treated.

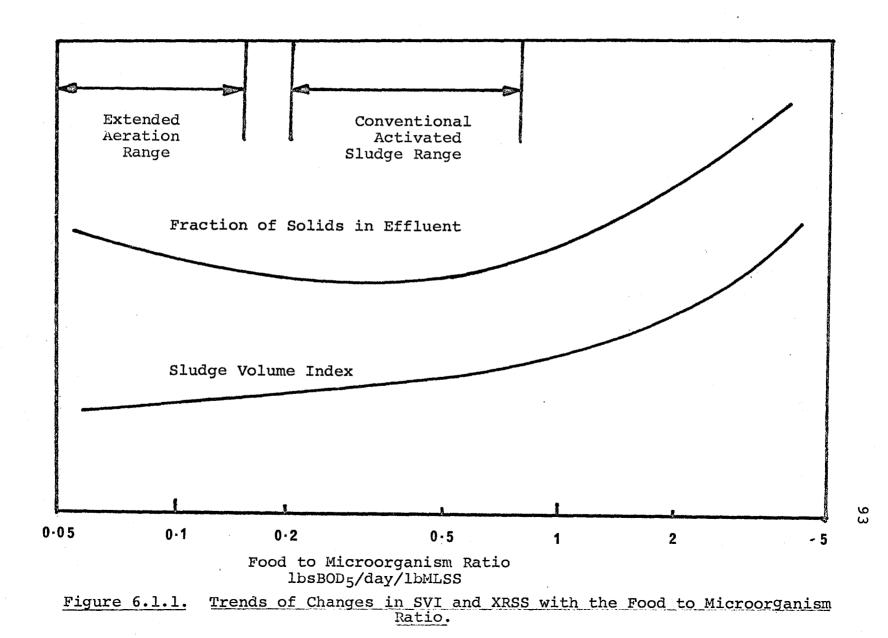
A brief description of new operating conditions may affect the character of the microorganisms in the sludge

will be presented here. Assuming that oxygen is supplied in sufficient quantities, a high food to microorganism ratio will lead to a rapid rate of bacterial metabolism and reproduction. The bacterial cells are highly energetic and will tend to stay as discrete entities. Another reason for the bacterial cells to stay apart, is the high surface area that each cell will enjoy, aiding in the transfer of food and oxygen. This leads to the formation of a dispersed or diffused sludge with poor clarifying and thickening properties.

At a lower food to microorganism ratio, the bacterial cells have less energy and flocs being to form when the cells are brought into contact with each other. The sludge also becomes more compact. The clarifying and thickening properties of the sludge improve.

At a very low food to microorganism ratio the bacterial cells are starved and many will die off. With less viable bacterial cells, the sludge will have a poorer flocculating ability, resulting in a poorer effluent. However, the sludge is still compact and will have good thickening characteristics. The above observations are illustrated in Figure 6.1.1.

In the case where the dissolved oxygen concentration in the activated sludge tanks is low, (0.5 mg/l) there is little oxygen penetration into the biological floc, resulting



in lowered bacterial growth. This encourages the growth of filamentous bacteria such as the Sphaerotilus, as they have a high surface area/volume ratio, and are more able to absorb the small amount of oxygen present. These filamentous growths render the sludge less compact and hence displays poor thickening properties. The effect of these filamentous growths on the flocculating and hence clarifying properties of the sludge is not too well know or reported.

Most fungi also have a filamentous structure and their growth tends to be stimulated by carbohydrate wastes, or conditions of low pH and nutrient deficiencies.

Prolonged periods of anaerobiasis in the secondary clarifier could lead to the production of gas, which when entrapped could cause the activated sludge flocs to rise, and escape over the effluent wiers. Denitrification could also contribute to the problem of a rising sludge.

6.2 Correlations for Activated Sludge Settling

From the above discussion, we realize that the twin functions of the secondary settling tanks, namely those of clarification and thickening, are very much related to the operating conditions in the activated sludge tanks. The level of clarification obtained can be described by the fraction of suspended solids escaping in the effluent, and the degree of compaction obtained, by the Sludge Volume Index.

The Sludge Volume Index is the volume in ml. occupied by 1 gm of the sludge after settling for half an hour.

Rex Chainbelt, Incorporated, has developed two correlations for this purpose (47):

$$XRSS = \frac{382 (GSS)^{0.12} (FM)^{0.27}}{(MLSS)^{1.35} (TA)^{1.03}}$$
(6.2.1)

and SVI = $56.1 + 113 (1.05)^{T-20}$ (FM) (6.2.2) where XRSS = fraction of solids escaping in the effluent GSS = overflow rate (usgpd/sq.ft.) MLSS = mixed liquor suspended solids (mg/l)

MLVSS = mixed liquor volatile suspended solids

TA = aeration time (hrs.)

SVI = sludge volume index (ml/gm)

and $T = temperature (^{O}C)$

The above correlations are based on 41 observations and the correlation coefficient for the first equation is 0.63 whilst that of the second is 0.78.

The maximum concentration of the underflow solids can be estimated by the value $(10^6/\text{SVI})$ mg/l. Although this represents the compaction obtained after half an hour of settling, it is a good estimate of the final compaction, as further changes in SVI are small, especially with mixed liquor suspended solids of less than 3000 mg/l. It should be noted that there is a wide misconception that the SVI represents the settling rate of an activated sludge. The half an hour of settling required by the test is not meant to be a measure of the rate, but merely provides a reasonable time for which the sludge to achieve close to the ultimate compaction.

R. V. Villiers (51) performed laboratory scale settling column studies and arrived at the correlation

$$XRSS = \frac{556 (GSS)^{0.494}}{(MLSS)^{1.82} (TA)^{0.439}}$$
(6.2.3)

This correlation does not include the effect of the food to microorganism ratio, as does the previous correlation by Rex Chainbelt Inc. However, there is quite a significant difference in the exponents on the variables, between the two studies.

In a preliminary study, P. Leung from the Chemical Engineering Department, McMaster University, using laboratory scale experiments, arrived at the correlation

$$XRSS = \frac{2560}{(MLSS)^{1.57} (TA)^{0.20}}$$
(6.2.4)

This correlation does not include either the food to microorganism ratio or the overflow rate. However, the exponent that he obtained for the effect of the mixed liquor suspended

solids is intermediate between those of the earlier studies, but he found a very much smaller effect of aeration time than did the other two investigators. His results are tabulated in Appendix F.

The disparity between the three correlations could be explained if the nature of the raw wastes entering the plants from which the sludges were taken, were somehow significantly different. It would seem necessary to obtain a correlation based on a sludge actually produced by the plant.

In the absence of a good correlation for the sludge produced in the Burlington plants we have to choose either Equations 6.2.1 or 6.2.3. It should be noted that the above equations are developed from quiescent batch column studies and to apply them to real settling tanks, a correction factor of 1.5 is suggested by most authors. We will use the operating data for the Drury Lane Plant, in 1969, as a comparison:

OR = 960 usgpd/sq.ft.

MLSS = 2200 mg/l

TA = 9.6 hrs.

F/M = 0.3 lbs BOD₅/lb MLVSS/day and the obtained XRSS = 0.090

Using Equation 6.2.1:

XRSS (calculated) = 0.019XRSS (corrected) = $0.019 \times 1.5 = 0.029$

Using Equation 6.2.3:

XRSS (calculated) = 0.054XRSS (corrected) = $0.054 \times 1.5 = 0.081$

The Rex Chainbelt correlation gives too good an effluent, whilst the Villiers' equation gives a result which is closer to the actual operating condition. Hence, Equation 6.2.3. will be selected for use in our program.

6.3 The Secondary Clarifier Model - SECLAR 1

The listing for the program is given in Appendix K. The equipment vector for the model is reproduced below.

EN VEC ****	***
115. 4. 5. 12. 13. 14. 16.	STANDARD FORMAT NUMBER OF CLARIFIERS IN PARALLEL HEAD LOSS THROUGH CLARIFIER (FEET OF WATER) FIRST OUTPUT STREAM (EFFLUENT) SECOND OUTPUT STREAM (UNDERFLOW) THIRD OUTPUT STREAM (SCUM TROUGH FLOW) MODE OF OPERATION = 1.0 FOR FIXED FRACTION OF RECYCLE = 1.0 FOR FIXED FRACTION OF RECYCLE
17. 18. 19. 20. 21. 22.	= 1.0 FOR MAXIMUM CONCENTRATION OF UNDERFLOW FRACTION OF RECYCLE IF MODE=1 SURFACE AREA OF CLARIFIER (PER TANK) - SQ.FT. SCUM FLOW CONCENTRATION (MG/L) NUMBER OF UNDERFLOW SLUDGE PUMPS CAPACITY OF PUMPS IGPH EACH FRACTION OF CAPACITY USED

The computations carried out by the program can be explained in the following steps:

- The Sludge Volume Index of the activated sludge is first calculated, based on operating conditions in the aeration tanks. Equation 6.2.2. is used.
- (2) The concentration of solids in the effluent, is then calculated from Equation 6.2.3.
- (3) Secondary clarifiers can be operated with either a constant fraction of recycle of the underflow or with a constant depth of sludge blanket. In the second case, the underflow solids is at its maximum concentration, which can be estimated from the Sludge Volume Index. In the first case, the sludge blanket

depth is not constant but could drop when the underflow pumps are set too high or could rise and finally escape over the wiers when the underflow pumps are set too low. When there is no sludge blanket, the concentration of solids in the underflow is below the maximum value, as calculated from the SVI. EN(16) determines the policy used in the plant, and the underflow is calculated accordingly.

- (4) The scum flow is then determined assuming 100% removal of all incoming scum. The scum solids concentration is determined by EN(19).
- (5) Dissolved components are split according to the water flows.

CHAPTER 7

7. ANAEROBIC DIGESTION

The "Sewage Treatment Plant Design Manual" (38) adequately describes the purpose of anaerobic digestion as follows:

"....A primary purpose of sludge digestion is to reduce the complex organic matter present in the raw sludge removed by sedimentation processes to a simpler, non-objectionable state. Digestion produces a sludge more amenable to dewatering without nuisance, and it renders the sludge fit for easy disposal by lagooning, dilution or similar means.....Digestors also reduce the volume of sludge and in so doing produce gas which can be utilized for heating purposes."

The anaerobic digestion process has several disadvantages which may discourage their use in a wastewater treatment plant. Firstly, a high capital outlay is required, amounting to about 25-35% of the total capital cost. Secondly, digester upsets are frequent, thus requiring constant attention, and thirdly, there is always the hazard of a gas explosion.

7.1 Theory

There are two distinct processes occuring in an anaerobic digester - liquifaction and gasification. Liquefaction of the sludge solids is brought about by the extracellular

enzymes secreted by the bacteria, which hydrolyses the complex solids. Cellulose and carbohydrates are converted to simple sugars, alcohols and fatty acids. Proteins are converted to amino acids, while fats and grease are converted to glycerol and fatty acids. The formation of organic acids could result in a depression of the pH, if the alkalinity in the digesting sludge is not sufficient for neutralization. This liquefaction stage of digestion, is also called the acid phase, for obvious reasons, and the bacteria responsible are usually referred to as acid-formers.

In the gasification process, the end products of liquefaction are further broken down to gaseous end products. This could involve the activity of the same bacteria in the first phase, but with the exception that they are not capable of utilizing their own acid end products. This requires the work of a second group of bacteria which metabolize the fatty acids to give methane and carbon dioxide, and convert amino acids to ammonia. The ammonia released tends to neutralize a portion of the acids remaining and raise the pH. The gasification phase is also sometimes referred to as the methane phase, and the bacteria involved as methane-formers.

The digestion process can be schematically represented as:

ORGANIC MATTER liquefaction ALDEHYDES ______ Gasification CARBONDIOXIDE ALCOHOLS AMMONIA

Figure 7.1.1. Schematic Representation of The Digestion Process

Since the digester is operated under anaerobic conditions, the bacterial population has to be composed of either the facultative bacteria or the strict anaerobes. The acid-formers are predominantly facultative and many of them originate from the activated sludge fed to the digester. The methane formers are strict anaerobes, and are a small specialized group of bacteria. They are highly sensitive to pH changes and have an optimum pH range of 6.4 to 7.2. When a digester goes "sour" (or acidic) the methane-formers die off, and the gasification phase is seriously retarded.

A properly operating digester should have the liquefaction and gasification processes proceeding at about the same rate. An upset can be produced by an increase in the liquefaction rate, which could be prompted by a sudden increase in the solids fed to the digester.

To provide a reasonable description of the digestion process we need an expression to account for the rate of liquefaction of the organic solids. If a steady-state operation

is reached, the rate of gasification should equal the rate of liquefaction. We would also need an expression to determine the level of volatile organic acids maintained in the digester. This is necessary since the pH and the BOD of the digesting sludge would depend on the amount of volatile acids present.

A model for the anaerobic digester was first proposed by Fair and Moore (13) in 1932. They followed the digestion process in batch reactors and concluded that the rate of gas production is proportional to the amount of "gas" remaining in the digesting sludge:

$$\frac{dy}{dt} = k (G-y)$$
 (7.1.1)

where y = amount of gas produced up to time t G = total amount of gas produced

and k = rate constant.

They found k to be 0.168 day⁻¹, at $95^{\circ}F$. Since then, other workers have also obtained first order rates. Schulze (36) determined k to be 0.14 day⁻¹ at $92^{\circ}F$., and Simpson (42) arrived at a k value of 0.3 day⁻¹.

Since it would be useful to our modelling studies to know the rate of liquefaction of the digesting solids, we have to rearrange Equation 7.1.1. If gas production is a measure of volatile solids destruction, then (G-y) is proportional to the amount of organic solids remaining.

ï.e.,

$$(G-y) = a S$$
 (7.1.2)

where S = amount of organic solids remaining

and a = proportionality constant.

The proportionality constant is the volume of gas produced per unit mass of organic solids. This has been reported to be approximately 16-18 cu. ft./lb. solids destroyed, for digesting sewage sludge.

Equation 7.1.1 then becomes

$$\frac{dS}{dt} = -KS \qquad (7.1.3)$$

Since most reactors are operated continuously or semi-continuously, rather than batchwise, we will assume a stirred tank model for the digester. The defining equation for the digester then is:

$$S = \frac{So}{1 + K\bar{t}}$$
 (7.1.4)

where S = volatile solids remaining

and \bar{t} = detention time

or

$$R = \frac{S}{So} = \frac{1}{1 + K_{t}^{2}}$$
(7.1.5)

where R = fraction of volatile solids remaining. Or

$$(1 - R)/R = K\bar{t}$$
 (7.1.6)

The relation here implies that the effluent volatile solids depend on both the influent volatile solids and the detention time, as contrasted to the theory of McCarty (see Chapter 2). This seems to be a more reasonable expression, and the importance of solids loading is supported by the observations of other workers (13, 19, 50).

Rankine (32) has presented data for the fraction of volatile solids destroyed as a function of the detention time in a conventional digester. His data is tabulated in Appendix G. A plot of (1 - R)/R versus \bar{t} is shown in Figure 7.1.2. K is found to be 0.025 day⁻¹, and the correlation coefficient is 0.65. The range of solids loading employed as 0.033 - 0.135 lbs VM/cu.ft./day.

For the high-rate digestion process, the data of Torpey (50), Roy and Sawyer (34) and Estrada (12) are used. Their data is also tabulated in Appendix G, and the plot of (1 - R)/R versus \overline{t} is shown in Figure 7.1.3. The reaction rate constant K is found to be 0.085 day⁻¹, and the correlation coefficient is 0.58. The range of solids loading employed here is 0.101 - 0.575 lbs/cu.ft./day.

Data on the dissolved biodegradable organic carbon in the supernatant is quite sparse. Hence, the equation

Figure 7.1.2. Conventional Digester Correlation

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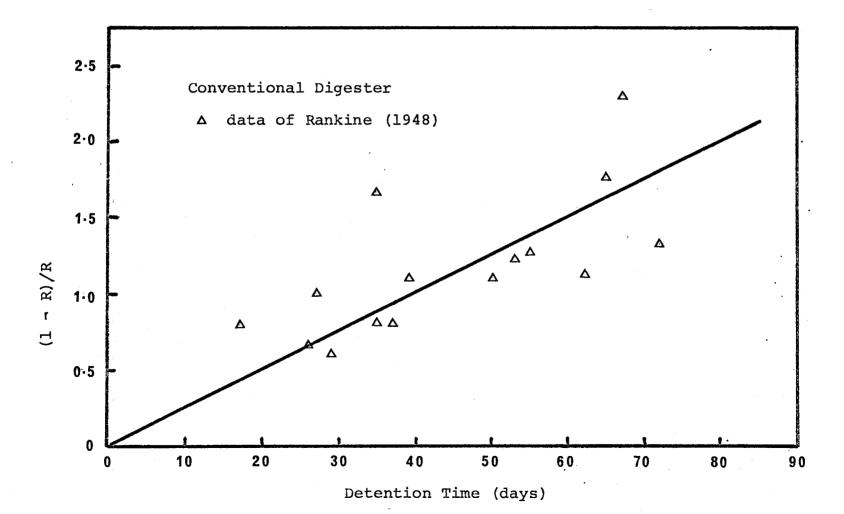
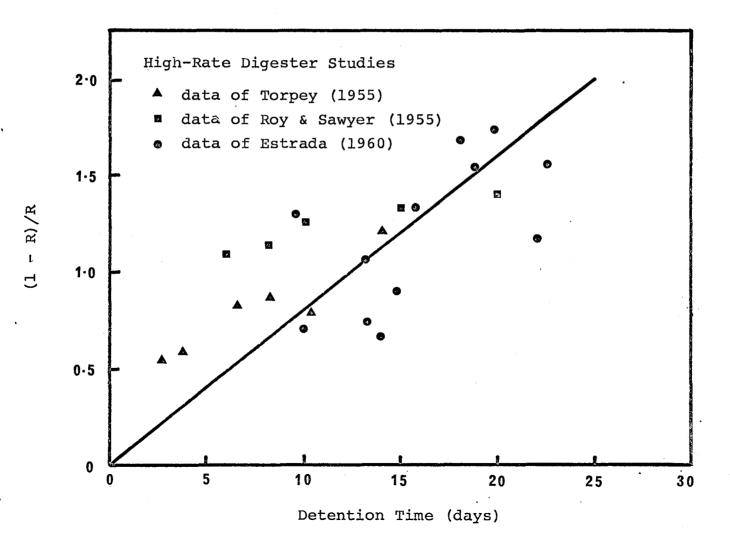


Figure 7.1.3. High-Rate Digester Correlation



proposed by McCarty (see Chapter 2) will be used. He also estimated that the dissolved organic carbon is mainly composed of volatile acids.

The acidity due to the volatile acids is estimated by assuming that acetic acid is the major component, i.e.,

(7.1.7)

The ammonia evolved tends to neutralize the volatile acids and raise the pH of the digester. Assuming that the nitrogen to carbon ratio in the feed sludge is approximately 0.23 (based on a sludge composition of $C_5H_7O_2N$), the ammonia nitrogen evolved is:

$$\Delta N = 0.23 \Delta C$$
 (7.1.8)

where $\Delta N =$ ammonia nitrogen evolved

and ΔC = organic carbon destroyed

Hence, the alkalinity due to the ammonia is given by

alkalinity = $\frac{\Delta N}{14}$ = $\frac{0.23\Delta C}{14}$ meq/l (7.1.9)

The effluent pH is then calculated by the module described in Section 9.2.

7.2 The Anaerobic Digestor Model - ANDIG1

A listing of the program is given in Appendix K. The equipment vector for the model is reproduced below:

EN: VEC1 ******* 115. 4. 5. 16. 17. 18. 19. 20. 21. 22.	<pre>*** STANDARD FORMAT NUMBER OF DIGESTERS IN PARALLEL HEAD LOSS IN DIGESTERS (FT OF WATER) DIAMETER OF DIGESTER (FEET) = 1.0 FOR CONVENTIONAL DIGESTER OPERATION = 2.0 FOR HIGH RATE DIGESTER OPERATION TEMPERATURE OF OPERATION FRACTION OF WATER TO FIRST OUTPUT STREAM (SUPERNATANT) = 1.0 IF FIRST STAGE OF A TWO-STAGE SYSTEM FRACTION OF SOLIDS TO FIRST OUTPUT STREAM (SUPERNATANT) (EXCLUDES SETTLEABLE INORGANIC SOLIDS) = 1.0 IF FIRST STAGE OF A TWO-STAGE SYSTEM FRACTION OF SETTLEABLE INORGANIC SOLIDS TO SUPERNATE = 1.0 IF FIRST STAGE OF A TWO-STAGE SYSTEM</pre>
22• 23• 24•	= 1.0 IF FIRST STAGE OF A TWO-STAGE STSTEM FRACTION OF SETTLEABLE INORGANIC SOLIDS TO SUPERNATE = 1.0 IF FIRST STAGE OF A TWO-STAGE SYSTEM STORAGE SPACE FOR GAS PRODUCED FRACTION OF VOLUME THAT IS EFFICIENTLY USED IN DIGESTION

The computations carried out by the program can be explained briefly by the following steps:

- (1) The fluid detention time is first calculated. EN(24) corrects for the fact that a fraction of the digester volume may be filled up with grit, etc., and hence is not available for digestion.
- (2) The fraction of volatile solids destroyed is calculated from Equation 7.1.5.
- (3) Based on a C:N:P ratio of 1.0:0.23:0.05, the amounts of nitrogen (as ammonia) and phosphorus released are then estimated.
- (4) The soluble organic carbon of effluent is then obtained from Equation 2.
- (5) The effluent pH is then calculated as discussed in Section 7.1.

(6) If the digester is used as the first stage of a twostage digester, the contents are normally sent to the second stage, with little or no phase separation. This is maintained by constant stirring. Phase separation is used in the second stage to separate a a clearer supernate from a concentrated solids underflow. There is no known correlation for the solids concentration of the supernate and hence the phase separations is specified by EN(20), EN(21), and EN(22). Soluble components are split according to the water flows.

CHAPTER 8

8. OTHER UNIT PROCESSES

Other unit processes commonly used in wastewater treatment include pretreatment, the use of biological filters (trickling filters) and chlorination. The primary purpose of pretreatment is to remove large and abrasive materials from the wastewater, to protect downstream equipment and pumps. It includes screening to remove the larger solid objects and grit removal.

Biological filters have been used as an alternative to the activated sludge process. In this process, the wastewater is passed over a biological growth, supported on a solid medium. The waste is absorbed by the biological growth and excess solids is sloughed off the "filter" and removed by sedimentation. The contact time is short, being of the order of a few minutes and the effluent is generally unsatisfactory.

Chlorination is practiced to prevent the transmission of pathogenic microorganisms to the receiving waters. It is generally applied to the secondary effluent and in cases where no secondary treatment is supplied, to the primary effluent. In the few cases where no treatment is practiced at all, the raw wastewater is chlorinated, prior to discharge.

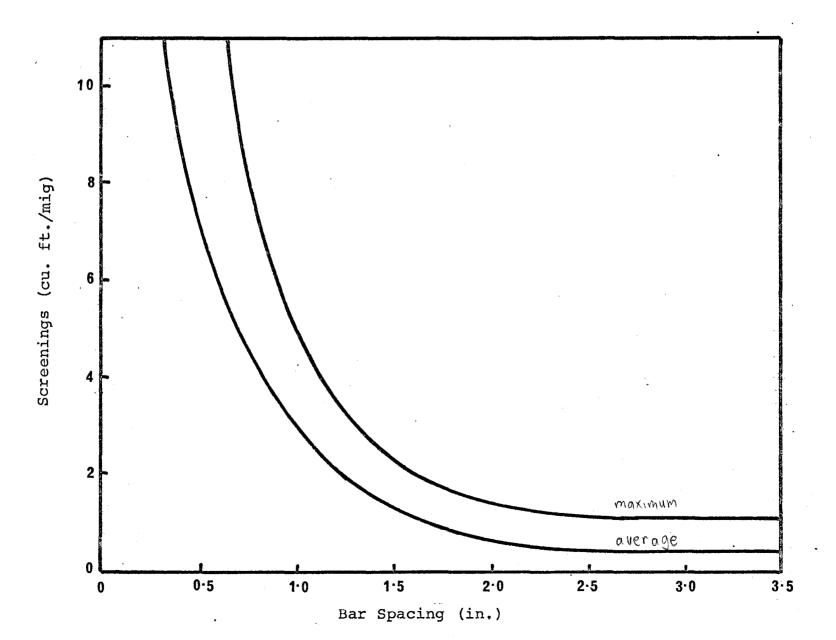
The thickening of waste sludges is also practiced. This reduces the volume of sludge to be handled, either for disposal or for further solids processing within the plant. Other methods of dewatering include vacuum filtration, centrifugation and air flotation.

8.1 Pretreatment

Screening devices are usually classified as fine or coarse screens. In the category of fine screens there are perforated plates, wire-meshes and closely spaced bars. The openings are generally 3/16 in. or less. Coarse screens include comminuting devices, bar screens and coarse wiremeshes. The openings may be as large as 3 in.

The choice for the size of the openings is dictated by the size of the largest object which may be allowed to pass. In the majority of cases, a 1 in. opening is quite satisfactory.

The volume of screenings removed is difficult to estimate and depends not only on the screen size but also on the nature of the waste received, and the velocity of flow through the screens. The Rex Chainbelt Company (33) has developed a graph from which the average and maximum amount of screenings may be expected as a function of screen opening. This is reproduced in Figure 8.1.1. The average curve is approximately described by:



$$V = 19 \times 10^{-0.8d}$$
(8.1.1)

where V = volume of screenings cu.ft./musg
and d = screen opening, inches

About 30 lbs. of dry solids may be expected per cu. ft. of screenings.

In grit removal, the object is to separate the nonputrescible solids from the waste flow with the minimum amount of entrained organic matter. This is achieved by differential sedimentation, which is made possible by the fact that the grit particles have subsiding velocities substantially greater than those of the organic solids. The flow-through velocities are also controlled to maintain the organic solids in suspension, by scouring the settled solids.

Velocity control can be maintained by the use of specially designed wiers, such as the Sutro Wier or the proportional wier. More recent developments in velocitycontrol devices include the use of compressed air to create a spiral current within the grit chamber. More details of the various types of grit chambers in use may be obtained from the "Sewage Treatment Plant Design Manual" (38).

As is the case with the quantity of screenings, the volume of grit to be expected at a plant cannot be confidently predicted. The quantities of grit received will depend on the area served by the sewers, the type of street and land

surfaces prevalent in the district and on the percentage of storm sewers feeding into the plant.

Due to the great variation in quantities of grit received, the model written for the grit chamber will require that we specify the volume of grit to be expected in cu.ft./mig. As a rough guide to use in Ontario, a survey of 28 plants having separate sewers gave an average grit volume of 2.4 cu.ft./mig, while 20 plants having partially combined sewers received on the average 3.3 cu.ft./mig.

8.2 Trickling Filters

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The theory proposed for the rapid rate of organic removal in a trickling filter is that the waste is biosorped and incorporated into the biological growth, supported on the solid media. As the wastewater passes down through the filter, progressive removal is achieved until the clear effluent emerges. Howland (21) and Schulze (35) have shown that at low concentrations, the rate of removal is of the first order with respect to the concentration of soluble organics remaining:

i.e.,
$$\frac{C}{c_0} = e^{-k} l^t$$
 (8.2.1)

where C = effluent concentration of soluble organics $C_{0} = influent$ concentration of soluble organics

k₁ = reaction rate constant
and t = mean contact time.

Schulze (35) and Howland (21) have also shown that the mean contact time in a trickling filter is directly proportional to the depth and inversely proportional to the hydraulic loading to a power, n, which depends on the solid media used.

i.e.,
$$t = k_2 D/Q^n$$
 (8.2.2)
where $t =$ mean contact time (min.)
 $D =$ depth of filter (ft.)
 $Q =$ hydraulic loading (usgpm/sq.ft.)
 $k_2 =$ proportionality constant

and n = constant.

Values of n range from 0.55 for spheres (55) to 0.66 for a screen filter (35). The n value for an increasingly popular media, the synthetic Dowpac Plastic Media, is 0.5.

It has also been shown that the reaction rate could decrease with depth. This could arise if the more easily removable portions of the waste are removed at the top of the filter, leaving a less readily degradable matter for the lower portion of the trickling filter. Another possibility exists whereby the microbial population in the filter is stratified with the more efficint bacteria and fungi at the top and progressing downwards to the predator microorganisms. A general relationship can be developed by combining Equations 8.2.1 and 8.2.2, and with a modification for the effect of depth on the reaction rate:

$$\frac{C}{c} = e^{-kD^{m}/Q^{n}}$$
(8.2.3)
$$C_{0}$$

where k = rate coefficient or treatability factor
m = constant.

In the case where there is no effect of the depth on the reaction rate, m = 1. This has been found to be the case for domestic sewage by several investigators. Schulze (35) determined k to be 0.020 m to be 1.0 and n to be 0.66 for a screen treating settled sewage. McDermott (55) concurs similarly, but his rate coefficient is slightly lower, at 0.018. Germain (56) found m to be 1.0, n to be 0.48 and k to be much higher at 0.088, for the case of the Dowpac Filter Media. However, his data is based on only a few points.

8.3 Chlorination

The purpose of the model will be merely to calculate the chlorine requirements in a wastewater treatment plant. Since the effluent will not be directly used as a drinking water supply, but is discharged to a water-course, the criteria for chlorination is mot so strict. The chlorine dosage required depends on the degree of treatment the effluent has gone through. Raw sewage will require a higher chlorine dosage than the effluent from a secondary clarifier. The "Sewage Treatment Plant Design Manual" (38) gives the range of chlorine dosages commonly used, and this is reproduced in Table 8.3.1.

Stream	Chlorine Dosage mg/l
Raw sewage	6 - 12
Raw sewage (septic)	12 - 25
Settled sewage	5 - 10
Settled sewage (septic)	12 - 40
Chemical Precipitation effluent	3 - 10
Trickling filter effluent	3 - 10
Activated sludge effluent	2 - 8
Sand filter effluent	1 - 5

Table 8.3.1. Range of Chlorine Dosages Required for Disinfection.

8.4 Sludge Thickening and Dewatering

Sludge thickening performance is not very well correlated. Hence, only a simple mass balance model will

be used. This is a slight modification of the SEPAOL model available from the GEMCS library of routines. The fractions of water (and water soluble components) and solids to the supernatant is supplied as inputs to the program.

Other dewatering devices are also approximately described by the same model.

CHAPTER 9

9. MODULES FOR HANDLING STREAM VARIABLES

There are certain calculations which occur frequently in some or most of the simulation models used in this study. As mentioned in Section 3, the stream elements 13-22 form the basic components of a waste flow. The simulation models directly use these variables in their internal calculations. However, in wastewater treatment practice, the more commonly used variable are those of stream elements 8-12, which include the BOD, SS, VSS, TOC and DOC. Hence it is necessary to convert from the "working" variables to the commonly reported variables.

Another area where calculations are frequently needed is in the pH - alkalinity relationship. Also in the activated sludge model, it is more convenient to use concentration variables (i.e., mg/l) than it is to use flow variables such as lbs./hr. Hence a module is also available to convert component flows into component concentrations.

9.1 Inter-relationship of Stream Variables

A series of experiments were performed to determine the relationships between the stream elements 8-12 and those of 13-22. Section 3.2 gives the relevant defining equations.

The Total Organic Carbon (TOC), which includes both the particulate and dissolved organic carbon forms, was first determined. To ensure that the injection needle would not "filter" out any solids, the sample was subjected to ultra-frequency sonication for five minutes.

The Total Five-day Biochemical Demand (TBOD₅), which includes the BOD₅ of both the particulate and soluble components, was then determined using the original sample.

The sample was then filtered on two Whatman's #40 filter papers (ashless). One filter paper was then ashed in a muffle furnace at 600°C and the other was dried at 103°C. This will yield the volatile suspended solids (VSS) of the sample. The filtrate was then used to determine the Dissolved Organic Carbon (SOC) and the Dissolved Five-day Biochemical Oxygen Demand (DBOD₅). All the above mentioned analyses were performed according to "Standard Methods" (48).

The results are summarized in Appendix G. The average values obtained for the conversion factors are:

(a) FAC1 = $\frac{DBOD_5}{SOC}$ = 1.91

(b) FAC2 = $\frac{BOD_5}{OC}$ (particulate) = $\frac{TBOD_5 - DBOD_5}{TOC - SOC}$ = 1.24

with a standard deviation of 0.12

(c) FAC3 =
$$\frac{VSS}{OC}$$
 = $\frac{VSS}{TOC}$ = 2.09
OC (particulate) TOC - SOC
with a standard deviation of 0.11

Eckenfelder (8) derived the value of 1.85 for FAC1. Smith (46) used the value 1.87 for FAC1. These two values agreed with our data. Smith reported a value of 0.80 for FAC2, whilst our value of 1.24 is much higher. The value of 2.13 is used by Smith for FAC3, which is close to our value of 2.09.

9.2 pH - Alkalinity Relationships

The major contributors to the alkalinity of municipal wastewaters are the species in the carbonate buffer system and ammonia. To a lesser extent, the phosphates and borates could contribute to the alkalinity.

The equilibrium conditions existing in solution can be described by (49):

For the carbonate system

$$H_2 CO_3^* \rightleftharpoons H^+ + HCO_3^-$$
 (9.2.1)

 $HCO_3^- \rightleftharpoons H^+ + CO_3^-$ (9.2.2)

where $[H_2CO_3^*] = [CO_2 \text{ dissolved}] + [H_2CO_3^]$ (9.2.3)

For the ammonium system

$$\operatorname{NH}_4^+ \rightleftharpoons \operatorname{NH}_3^+ + \operatorname{H}^+$$
 (9.2.4)

Self-ionization of water

$$H_2^{0} \rightleftharpoons H^+ + 0H^-$$
 (9.2.5)

The equilibrium constants for the above equations

are:

$$k_{1} = \left[\frac{H^{+}}{H^{+}} \right] \left[\frac{HCO_{3}}{H_{2}CO_{3}} \right] = 10^{-6.3}$$
(9.2.6)

$$k_{2} = \left[\frac{H^{+}}{[H^{-1}]} = 10^{10.3} \right]$$
(9.2.7)

$$k_{n} = \frac{\left[H^{+}\right] \left[NH_{3}\right]}{\left[NH_{4}^{+}\right]} = 10^{-9.3}$$
(9.2.8)

and

k_w

=

$$\left[H^{+} \right] \left[OH^{-} \right] = 10^{-14.0}$$
 (9.2.9)

The definition for alkalinity can then be expressed as

$$\begin{bmatrix} ALK \end{bmatrix} = \begin{bmatrix} HW_3^{-} \end{bmatrix} + 2 \begin{bmatrix} CO_3^{+} \end{bmatrix} + \begin{bmatrix} NH_3 \end{bmatrix} + \begin{bmatrix} OH^{-} \end{bmatrix} \\ - \begin{bmatrix} H^{+} \end{bmatrix}$$
(9.2.10)
where $\begin{bmatrix} ALK \end{bmatrix} =$ alkalinity equivalents/litre
 $\begin{bmatrix} HCO_3 \end{bmatrix} =$ bicarbonate ion concentration moles/litre
 $\begin{bmatrix} CO_3 \end{bmatrix} =$ carbonate ion concentration moles/litre
 $\begin{bmatrix} NH_3 \end{bmatrix} =$ free ammonia concentration moles/litre
 $\begin{bmatrix} OH^{-} \end{bmatrix} =$ hydroxyl ion concentration moles/litre
and $\begin{bmatrix} H^{+} \end{bmatrix} =$ hydrogen ion concentration moles/litre

Due to the conservation of mass, we note that

$$\begin{bmatrix} H_2 CO_3^* \end{bmatrix} + \begin{bmatrix} HCO_3^- \end{bmatrix} + \begin{bmatrix} CO_3^- \end{bmatrix} = \text{constant} = \begin{bmatrix} C_T \end{bmatrix} \quad (9.2.11)$$

and $\begin{bmatrix} NH_3 \end{bmatrix} + \begin{bmatrix} NH_4^+ \end{bmatrix} = \text{constant} = \begin{bmatrix} C_N \end{bmatrix} \quad (9.2.12)$

Hence, from Equations 9.2.6 and 9.2.7, as well as from 9.2.11

$$\begin{bmatrix} C_T \end{bmatrix} = \begin{cases} \begin{bmatrix} \underline{H}^+ \end{bmatrix} + 1 + \frac{k_2}{[\underline{H}^+]} \end{bmatrix} \begin{bmatrix} HCO_3^- \end{bmatrix}$$

or
$$\begin{bmatrix} HCO_3^- \end{bmatrix} = \begin{bmatrix} C_T \end{bmatrix} \begin{cases} \frac{1}{[\underline{H}^+] + 1} + \frac{k_2}{[\underline{H}^+]} \end{cases}$$
(9.2.13)

Similarly

$$\begin{bmatrix} CO_3^{=} \end{bmatrix} = \begin{bmatrix} C_T \end{bmatrix} \left\{ \begin{array}{c} \frac{k_2}{\left[H^+\right]} \\ H^+ \end{bmatrix} \right\} \left\{ \begin{array}{c} \frac{1}{\left[H^+\right]} + 1 + \frac{k_2}{\left[H^+\right]} \\ k_1 & \begin{bmatrix} H^+ \end{bmatrix} \end{array} \right\}$$
(9.2.14)
and $\begin{bmatrix} NH_3 \end{bmatrix} = \begin{bmatrix} C_N \end{bmatrix} \left\{ \begin{array}{c} \frac{1}{1 + \begin{bmatrix} H^+ \end{bmatrix}} \\ 1 + \begin{bmatrix} H^+ \end{bmatrix} \\ K_n \end{array} \right\}$ (9.2.15)

Therefore

$$\begin{bmatrix} ALK \end{bmatrix} = \begin{bmatrix} C_T \end{bmatrix} \begin{cases} 1 + \frac{2k_2}{\begin{bmatrix} H^+ \end{bmatrix}} \end{cases} \begin{cases} \frac{1}{\begin{bmatrix} H^+ \end{bmatrix} + 1 + \frac{k_2}{\begin{bmatrix} H^+ \end{bmatrix}}} \end{cases}$$

$$+ \begin{bmatrix} C_{N} \end{bmatrix} \left\{ \begin{array}{c} \frac{1}{1 + \begin{bmatrix} H^{+} \end{bmatrix}} \\ k_{N} \end{array} \right\} + \begin{bmatrix} OH^{-} \end{bmatrix} - \begin{bmatrix} H^{+} \end{bmatrix}$$
(9,2.16)

This is the defining equation used to relate the and pH. Given three of ALK, $\left[C_{\mathrm{T}} \right]$, $\left[\mathrm{CN} \right]$ four variables the variables, the fourth may be calculated. Note also that if streams are mixed, and assuming a closed system, there must be conservation of ALK C_T and $\begin{bmatrix} C_N \end{bmatrix}$.

On expanding Equation 9.2.16, we obtain a polynomial of the fifth order in H⁺:)

F . 7

E

$$\begin{bmatrix} H^{T} \end{bmatrix}^{5} + \begin{bmatrix} H^{T} \end{bmatrix}^{4} \left\{ \begin{bmatrix} ALK \end{bmatrix} + k_{1} + k_{N} \right\}$$

$$+ \begin{bmatrix} H^{+} \end{bmatrix}^{3} \left\{ \begin{bmatrix} ALK \end{bmatrix} (k_{1} + k_{N}) + k_{1}k_{N} + k_{1}k_{2}-k_{w}-k_{1}\begin{bmatrix} C_{T} \end{bmatrix} - k_{N} \begin{bmatrix} C_{N} \end{bmatrix} \right\}$$

$$+ \begin{bmatrix} H^{+} \end{bmatrix}^{2} \left\{ k_{1}\begin{bmatrix} ALK \end{bmatrix} (k_{2}+k_{N}) + k_{1}k_{2}k_{N}-k_{N}k_{w}-k_{1}k_{N} \begin{bmatrix} C_{T} \end{bmatrix} \right\}$$

$$- k_{1}k_{w}-2k_{1}k_{2}\begin{bmatrix} C_{T} \end{bmatrix} - k_{1}k_{N} \begin{bmatrix} C_{N} \end{bmatrix} \right\}$$

$$+ \begin{bmatrix} H^{+} \end{bmatrix} \left\{ k_{1}k_{2}k_{N} \begin{bmatrix} ALK \end{bmatrix} - k_{1}k_{N}k_{w}-2k_{1}k_{2}k_{N} \begin{bmatrix} C_{T} \end{bmatrix} \right\}$$

$$- k_{1}k_{2}k_{w}-k_{1}k_{2}k_{N} \begin{bmatrix} C_{N} \end{bmatrix} \right\}$$

$$- k_{1}k_{2}k_{w}-k_{1}k_{2}k_{N} \begin{bmatrix} C_{N} \end{bmatrix} \right\}$$

$$- k_{1}k_{2}k_{N}k_{w} = 0 \qquad (9.2,17)$$

We could simplify the above equation by making some order of magnitude simplifications. Roughly, $\begin{bmatrix} ALK \end{bmatrix} \approx \begin{bmatrix} C \\ N \end{bmatrix} \approx \begin{bmatrix} C \\ T \end{bmatrix} \approx 10^{-3}$. This means that in the pH range of 5-9, the terms $\begin{bmatrix} OH^{-} \end{bmatrix} - \begin{bmatrix} H^{+} \end{bmatrix}$ $\ll \begin{bmatrix} ALK \end{bmatrix}$ in Equation 9.2.16. Hence we can drop out these two terms resulting in:

$$\begin{bmatrix} C_T \end{bmatrix} \begin{cases} 1 + \frac{2k_2}{[H^+]} \\ [H^+] \end{cases} \begin{cases} \frac{1}{[H^+]} + 1 + \frac{k_2}{[H^+]} \end{cases} = \begin{bmatrix} ALK \end{bmatrix} - \begin{bmatrix} C_N \end{bmatrix} \begin{cases} \frac{1}{[H^+]} \\ \frac{1}{[H^+]} \\ K_N \end{cases}$$

On expansion, a third order polynomial in $[H^+]$ results:

$$\begin{bmatrix} H^{+} \end{bmatrix}^{3} [ALK] + [H^{+}]^{2} \left\{ k_{N} \left([ALK] - [C_{N}] \right) + k_{1} \left([ALK] - [C_{T}] \right) \right\}$$

$$+ \begin{bmatrix} H^{+} \end{bmatrix} \left\{ k_{1} k_{N} \left([ALK] - [C_{N}] - [C_{T}] \right) + k_{1} k_{2} \left([ALK] - 2[C_{T}] \right) \right\}$$

$$+ \begin{bmatrix} k_{1} k_{2} k_{N} \left\{ [ALK] - [C_{N}] - 2[C_{T}] \right\} = 0$$

Equation 9.2.19 has been found for the several cases tried to give only one real root and two imaginary roots. This is very fortunate and helps simplify the solution of the cubic equation, for which a simple interval bisection technique is used. The bisection subroutine uses the geometric mean of the left and right estimates of the solution, in its convergence.

9.3 Inter-conversion Between Mass Flow and Concentration

Although the units used for the stream variables are for mass flow, it is frequently useful to convert them into concentration units. The model for the aerobic biological processes has its rate equations all in terms of component concentrations. The summary report generated has all the values expressed in concentrations, which is more meaningful than mass flows.

Due to the analytical procedures used in wastewater treatment, soluble components are usually expressed as milligrams per litre of water, while solid components are expressed in milligrams per litre of sample. In the case where the concentrations involved are less than 1000 mg/1 (0.1%) the difference between the two is negligible. However, since sludges of up to 80,000 mg/1 (8%) are handled in our system, the distinction should be made.

For soluble components we then use

$$C_{i} = \frac{F_{i}}{W} 10^{-6}$$
 (.3.1)

where C_i = concentration of soluble component i (mg/l)
F_i = mass flow of soluble component i (lbs/hr)
and W = mass flow of water (lbs/hr)

For particulate components we use

$$C_{i} = \frac{F_{i}}{(-1)} \times 10^{-5}$$
 (9.3.2)
Q

where C_i = concentration of particulate component i (mg/l) F_i = mass flow of particulate component i (lbs/hr) and Q = total volumetric flow of stream (gal/hr)

CHAPTER 10

10. WASTEWATER TREATMENT COSTS

Besides the technical aspects of wastewater treatment, we have to concernourselves with the costs that such a program would entail. In the expectation of increased activity in the area of pollution abatment, it is becoming more essential that we have accurate cost estimates to base our designs on. A large part of current practice has been to use "rules of thumb" which have fairly large safety factors built into them. An optimal design for a treatment plant, subject to the requirements of the regulatory bodies, can be arrived at by combining the technical section of our simulation with the corresponding cost estimates.

For each unit process in wastewater treatment, capital costs as well as operating and maintenance costs are available as a function of the most prominent capacity factor. For example, the capital cost of the activated sludge tanks is derived as a function of its volume. Most of the correlations are based on the same idea as the six-tenths capacity factor used in Chemical Engineering.

The bulk of the correlations come from the work of Russell and Swanson as reported in Smith's paper in 1969 (46), and from Eckenfelder and Barnard (2). Industrial waste treatment costs are also given in the latter report.

The capital costs are all updated to 1969 costs by the use of the Engineering News Record Index of 1120. This index, being based on changes in the price of steel, cement, lumber and common labour, is most appropriate for wastewater treatment process equipment. The capital cost correlations used are summarized in Table 10.1.

The total cost of the unit processes (installed) is then the sum of the separate costs. It is assumed that the duplication of a unit merely doubles the cost for the unit. This does not take into account that there may be savings due to shared pipings or channels or other factors.

The Physical Plant Cost can then be found by adding the cost of the control house and site improvements on to the total unit processes cost. The Engineering Costs added to the Physical Plant Cost will give the Direct Plant Cost. The Engineering Cost factor is obtained as a fraction of the Physical Plant Cost. As given in Smith (46)

 $CENG = 8.0 (1,000,000)^{0.146} (10.1)$ Physical Plant Cost

where CENG = Engineering Cost factor.

The Fixed Capital Cost can then be computed by adding the Contractor's fee (10% of Direct Plant Cost) and the contingency costs (15% of Direct Plant Cost) to the Direct Plant Cost. These figures are suggested by Smith (46) and

Unit	Capacity Variable	Capital Cost \$
Pretreatment	Design flow, Q, migd	21800 (A) ^{0.63}
Primary Sedimentation	Surface Area, SA, thousand	17300 (SA) + 6700 (SA) ^{0.1}
Activated Sludge Tanks	sq. ft. Volume, V, mig	27000 (V) + 67000
Air Blowers	Blower capacity, C, thousand	13600 + 7600 (C)
Final Sedimentation	cfm Surface area, SA, thousand	$16200 (SA) + 6900/(SA)^{0.13}$
Return Sludge Pumps	sq. ft. Pump capacity, PC, migh	4700 + 1700 (PC)
Anaerobic Digester	Volume, DV, thousand cu. ft.	1340 (DV) + 1 800 (V) $^{0.13}$
Chlorinator	Design flow, Q, migd	12600 (Q) 0.47
Vaccum Filter	Filter area, FA, hundred	16500 + 48000 (FA)
Sludge Incinerator	sq. ft. Sludge handing capacity, S,	7.1 (S) + 0.3 (S) 1.61
Sludge Drying Beds	lbs/day Surface area, A, sq. ft.	2.23 (A)
Sludge Thickener	Surface area, SA, thousand sq. ft.	(SA) (24200 + 11700/exp(SA/13.3)
Trickling Filter (Dowpac)	Volume, FV, thousand cu. ft.	66000 (FV)0.6

Table 10.1. Capital Cost Correlations

FIXED CAPITAL ESTIMATE

DRURY LANE SEWAGE TREATMENT PLANT - BURLINGTON. CAPITAL COST IN 1962

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PRETREATMENT	\$ 28567.21	
PRIMARY SETTLING TANKS	32399.94	
ACTIVATED SLUDGE TANKS	115112.80	
HAIR BLOWERS	36785.71	
SECONDARY SETTLING TANKS	56952.29	
SLUDGE RETURN PUMPS	6658.21	
ANAEROBIC DIGESTERS	118303.96	
CHLORINATOR	9270.00	
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TOTAL UNIT PROCESSES (INSTALLED)	\$ 404050.13	
CONTROL HOUSE	81877.54	
PLANT SITE	7579.74	
na an a		
PHYSICAL PLANT COST	\$ 493507.41	
ENGINEERING	43768.61	
a second state of the second st	國 義 勇 尊 章 章 章 章 章 章 章 章	
DIRECT PLANT COST	\$ 537276.02	
CONTRACTOR#S FEE (0.1 DPC)	53727.60	
CONTINGENCY (0.15 DPC)	2 Taylor (1988) 80591.40	
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FIXED CAPITAL COST	\$ ···	

Table 10.2. Capital Cost for Drury Lane Plant

are about average in the Chemical Process Industry. The derivation of the Fixed Capital Cost is summarized in Table 10.2, for the case of the Drury Lane, Burlington Water Pollution Control Plant. The actual plant cost in 1962 was \$676,033.78, while the estimated cost was \$671,595.03. The difference is less than 1%.

The operating and maintenance costs of the unit processes are normally reported in the literature together as one cost. Smith (46) and Eckenfelder and Barnard (2) have reported them in terms of annual costs (utilities excluded). This necessitates the use of a "Labour Index" to bring these costs up to date. However, we can avoid this by converting the annual costs to man-hours required per year, by dividing by the labour cost \$/man-hours for the year in which the correlations were obtained. This was calculated from the average wage of \$110/week or approximately \$3.00/man-hour in 1966 (26). The annual cost is then computed by multiplying the man-hours required by the current labour cost. Table 10.3 summarizes the operating and maintenance man-hours used in the program.

Again we have assumed that the duplication of a unit will double the number of man-hours required to operate and maintain the units.

The sum of the Operating and Maintenance Costs for the unit processes will give the total labour cost. This,

together with supervision, utilities, sludge haulage, chemical and plant supplies, will give the Direct Operating and Maintenance Cost. Supervision is assumed to be 10% of the Total Labour Cost. Chemical and plant supplies are assumed to be 6% and 5% of the Total Labour Cost, repectively. These estimates are obtained from the Ontario Water Resources Commission annual reports on their plant operations. The estimates of power consumption provided by Michel (26) can be correlated to give the following annual power consumption P in kwh:

> For Primary Plant $P = 101000 Q^{0.65}$ For Trickling Filter Plant $P = 162000 Q^{0.65}$

For Activated Sludge Plant P = $378000 Q^{0.65}$ where Q is the plant flow in migd.

Sludge haulage fees in Ontario vary and from the annual reports of the Ontario Water Resources Commission, an approximate correlation is obtained as:

Sludge haulage costs = $8000 (Q)^{0.5}$ \$/yr. where Q = sludge volume hauled, mig/yr.

Indirect costs are generally not significant, the most important of which is for laboratory analyses. The bigger plants usually conduct their own analyses, while the smaller ones tend to send them away to be done. The Net Operating and Maintenance cost will then be the sum of the Direct and Indirect costs.

To obtain the true annual cost, we have to take into account the cost of the Fixed Capital Investment. In the case of owned capital, we have to include depreciation and opportunity costs. The capital recovery factor to use is:

$$CRF = \frac{S (1 + S)^{n}}{(1 + S)^{n} - 1}$$
(10.2)

where CRF = Capital Recovery Factor

S = interest rate that owned capital can obtain
 elsewhere

and n = expected life of plant.

The cost of the Fixed Capital Investment to the municipality of company, annually, is then

 $R = CRF (L_0 - I_g)$

where R =amortization

I₀ = Fixed Capital Investment

and $I_s = scrap$ value of project at the end of n years.

In the case of borrowed capital, which is the usual case with most municipalities, we have to add debt retirement plus accrued interest to the Net Operating and Maintenance Cost to obtain the true Annual Cost. The factor can be calculated from the same equation, but the interest rate on the loan s and the debt reitrement period n may be different.

The derivation of Total Annual Cost is shown in Table 10.4, again, for the case of the Drury Lane, Burlington Water Pollution Control Plant. The actual operating cost in 1969 was \$103,520.52, while the projected cost was \$109,116.34, which is about 5% higher.

Unit	Capacity Variable	Annual Operation & Maintenance man/hr. Requirements		
Pretreatment	Plant flow, Q, migd	$133(Q) + 510(Q)^{0.37}$		
Primary Sedimentation	Surface area, SA, thousand	222 (SA) + 555 (SA) 0.5		
Activated Sludge including blower and final sedimentation	sq. ft. Aerator volume, V, mig and plant flow, Q, migd	720(Q) + 590(Q)/V ^{0.67}		
Anaerobic Digester	Digester volume, DV, thousand	$13(DV) + 144(DV)^{0.5}$		
Chlorinator	Plant flow, Q, migd	$30(Q) + 170(Q)^{0.37}$		
Vacuum Filter	Plant flow, Q, migd	$0.27(Q) + 154(Q)^{0.37}$		
Sludge Incinerator	Plant flow, Q, migd	$400(Q) + 1600(Q)^{0.37}$		
Sludge Drying Beds	Surface area, A, sq. ft.	$0.014(A) + 20(A)^{0.37}$		
Trickling Filter	Volume, FV, thousand cu. ft.	$10(FV) + 210(FV)^{0.5}$		

Table 10.3. Annual Operating and Maintenance Man-Hour Requirements

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DRURY LANE	SEWAGE TRE	ATMENT PLANT	- BURLINGTON.	TOTAL	ANNUAL COST IN	1969
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Table 10.4. Operating Costs for Drury Lane Plant

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CHAPTER 11

11. INTER-ACTIVE VERSION

An inter-active version of the simulation was written to enable the user to set up a dialogue with the computer. This can be used in the following ways:

- As a teaching tool to train operators and other students,
- (2) To enable plant personnel to predict plant performance due to changes in feed flows and/or operating conditions, and
- (3) As a quick and accurate aid to plant designers.

An important feature of the inter-active version is that it requires only a very minimum knowledge of the GEMCS system for its use. This is achieved through a series of questions and answers posed by the computer. Of course, a user with a good knowledge of GEMCS, can manipulate and use the inter-active version with greater flexibility.

The inter-active version consists of four subroutines and a slightly modified version of the main program of GEMCS. The four subroutines are FLWCHT, CASET, ICLOAD and ENSET. Their functions will be described below.

FLWCHT enables the user to set up his own plant layout, together with the physical dimensions of the various units. It has an enlarged process flow diagram covering most of the common unit processes encountered in wastewater treatment

plants. This is shown in Figure 11.1. The units are selected or by-passed by the use of the mixer-splitter module. The majority of the data set is read in by DLOAD1, to minimize the use of the teletype to input such information. Module numbers, module types, input and output stream numbers and s o on are predetermined for all the units in the process flow diagram and hence can be read in by DLOAD1. The calculation order and the physical dimensions of the various units are determined by a dialogue with the user.

CASET is used to generate a random case study. It generates a random feed by multiplying the average value of each component flow by a random factor ranging between 0.5 and 2.0. It also generates a failure in one of the plant units. The purpose of the subroutine is to teach operators what to do in the event that such a flow or plant failure occurs, as well as to locate the cause of such a failure.

ICLOAD is the analytical laboratory of the simulation. All information regarding stream flows and component concentrations are supplied through ICLOAD. A charge is levied for each analysis performed, the object of the game being to locate and correct a plant failure with the minimum number of analyses. A surcharge is also placed on any unsatisfactory effluent. Five cents is levied on each pound of suspended solids and Biochemical Oxygen Demand, above the regulatory level of 20 mg/1 for both. This provides the incentive to correct a plant failure with the minimum delay.

ENSET enables the user to change any operating conditions by changing the appropriate values of the EN vectors. The process flow diagram however, cannot be changed. This must be done through FLWCHT.

In addition to the above four subroutines, slight changes are made to the other modules, primarily to suppress excessive printing.

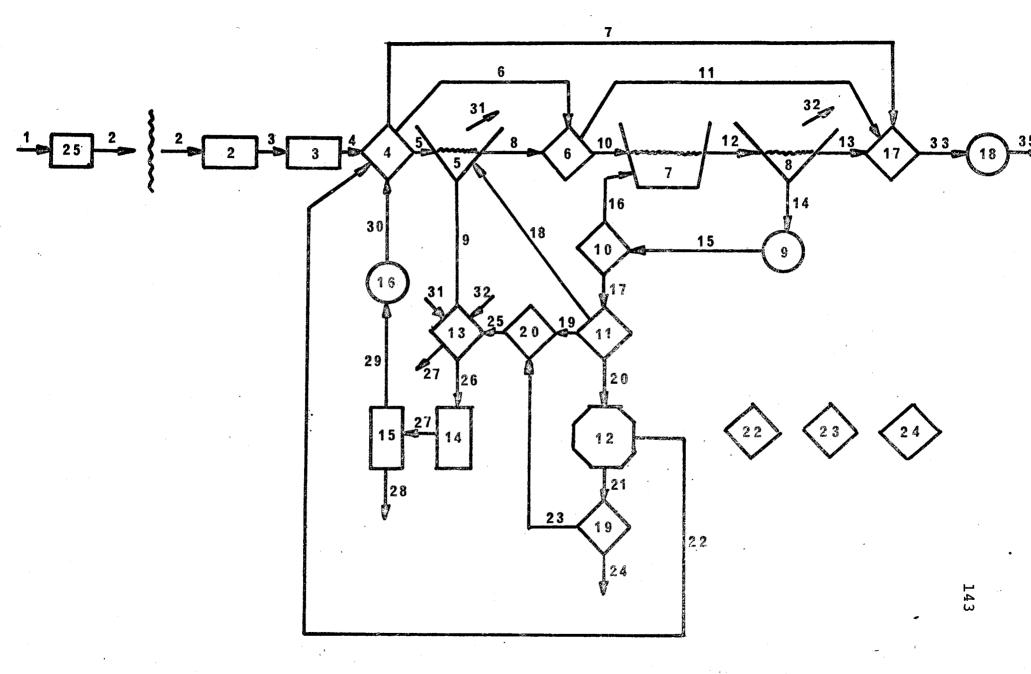


Figure 11.1. Process Flow Diagram for the Interactive Version

Key to Figure 11.1.

Module Number	Equipment
2	SCREEN1 - Screens
3	GRIT1 - Grit chambers
4	MIXER1
5	PRISET1 - Primary clarifiers
6	MIXER1
7	ACTSL1 - Activated sludge tanks
8	SECLARl - Secondary clarifiers
9	SETSP1
10	MIXER1 - Control sludge wasting
11	MIXERL
12	SEPAOl - (Thickener)
13	MIXERL
14	ANDIG1 - First stage digestion
15	ANDIG1 - Second stage digestion
16	SETSP1
17	MIXERL
18	CHLORl - Chlorinator
19	MIXERL
20	MIXERL
22	CØNTLI
23	CØNTLI
24	REPTOl - Report generator
	CASET - Case generator

CHAPTER 12

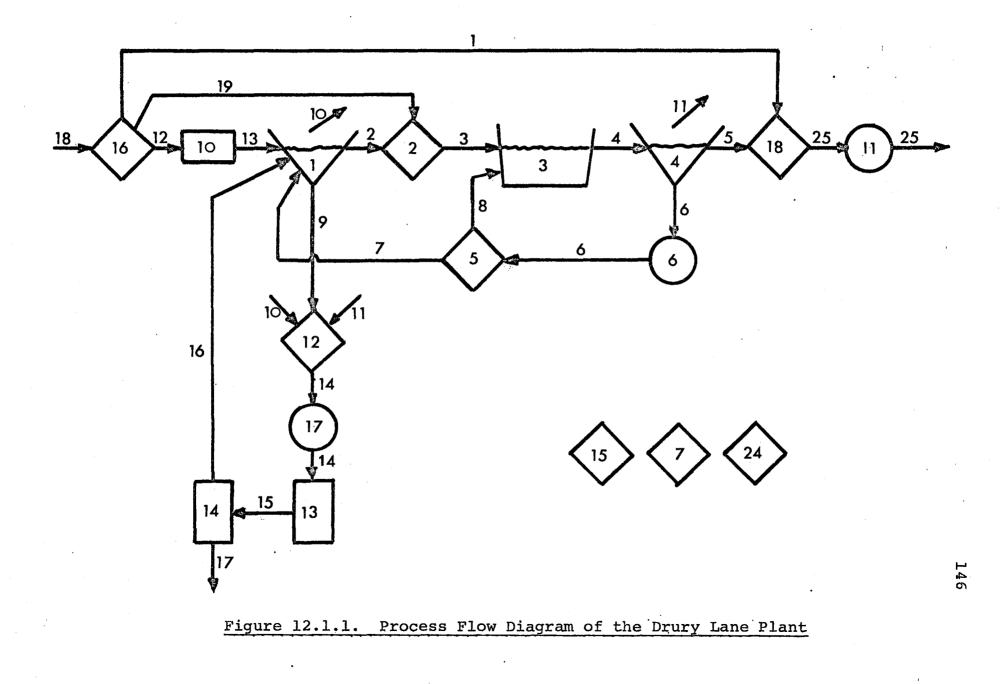
12. TEST RUNS

The two plants selected for simulation are the Drury Lane and the Skyway Sewage Treatment plants. The former is a conventional activated sludge process, whilst the latter employs the extended aeration modification. In addition to testing the various models developed, the two simulations should show that the same kinetics can be used for the activated sludge process, regardless of the process modification.

The base cases used will be the average flow and operating conditions for 1969, obtained from plant data and also from the Ontario Water Resources Commission's Annual Summaries. The 1969 data were chosen, as the operating and flow conditions were fairly constant throughout the year. Due to the diurnal variation in the feed flows, the steadystate models can, at best, predict the average performance of the various units. It is for this reason that no attempt was made to collect base case data on any particular day, as this would require at least a twenty four hour sampling schedule, for a week.

12.1 Simulation of the Drury Lane, Burlington Sewage Treatment Plant

The process flow diagram for the plant is shown in Figure 12.1.1, and the design data is summarized in Appendix I.



Key to Figure 12.1.1.

Module Number	Equipment
1	PRISET1 - Primary clarifiers
2	MIXER1
3	ACTSL1 - Activated sludge tanks
4	SECLAR1 - Secondary clarifiers
5	CØNTO2 - Controls sludge wasting
б	SETSP1
7	CØNTL1
10	GRIT1 - Grit chambers
11	CHLØR1 - Chlorinator
12	MIXER1
13	ANDIG1 - First stage digestion
14	ANDIG1 - Second stage digestion
15	CØNTL1
16	MIXER1
17	SETSP1
18	MIXERL
24	REPTOl - Report generator

It was originally designed for a flow of 2.5 migd. However, due to the expansion of the Skyway plant, part of the flow was diverted to the latter plant, and the actual flow received was 1.6 migd, in 1969. The average BOD and TSS for the raw sewage is 215 mg/l and 299 mg/l, respectively.

The pretreatment section consists of 1-inch bar screens and grit channels with a detention time of approximately one minutes. Primary treatment consists of two tanks with dimensions 29.3 ft. x 18 ft. x 12.25 ft. Secondary treatment is provided by two triple-pass activated sludge tanks, each with a total length of 321.5 ft., width of 18 ft. and depth of 10.7 ft. Total air blower capacity is 3000 scfm. Two circular secondary clarifiers are employed, each with a radius of 25 ft. and depth of 10.6 ft. A two-stage digestion system is used to handle the waste sludge generated. The first stage consists of two 40 ft. diameter tanks with a depth of 20 ft., and the second stage consists of one 40 ft. diameter tank with a depth of 18 ft. A summary of the operating conditions and the parameters used for the simulation is shown in Table 12.1.1.

The data set used for the simulation is shown in Appendix I. The computer printout for the case is also included in Appendix I. A summary of the results from the simulation, together with the plant data is shown in Table 12.1.2.

The removal of Total Suspended Solids by the primary clarifier was predicted to be 49% while the observed removal

was 48%. The BOD removal predicted was 30% compared to the actual removal of 33%. Both removals were predicted very well by the primary clarifier model.

The Total Suspended Solids in the final effluent is identical with the value observed, which was 20 mg/l. The BOD of the final effluent was predicted to be 22 mg/l, while the observed value was lower at 15 mg/l. The agreement is quite good.

The predicted Total Suspended Solids of the return activated sludge was 11,700 mg/1 with a volatile content of 57%. The observed values were 12,000 mg/1 and 60% respectively.

The simulation predicted that a greater volume of sludge should be sent to the digester than was actually observed. A raw sludge flow of 7600 igpd was calculated, while the reported flow was much lower at 5300 igpd. This is reasonable since a considerable amount of solids is lost during a rain, but which is not accounted for. The figures imply that about 30% of the solids which should go to the digester was lost over the wiers of the secondary clarifiers, either during a storm or due to a plant upset.

The simulation predicted a digested sludge flow of 3000 igpd at a Total Suspended content of 5.8% as compared to the volumes of 2300 igpd hauled away for disposal. The anaerobic digester model predicted a much higher volatile solids

destruction than was observed. This is reflected in the lower volatile solids concentration predicted for the digested sludge.

RAW SEWAGE:

average flow = 1.6 migd average TSS 299 mg/l = 195 mg/l average VSS = percent of voltile solids settleable = (77%) percent of inorganic solids settleable = (46%) average BOD 215 mg/1 = average DOC (61) mg/l Ξ

PRIMARY SETTLING TANKS: (using nomenclature of Chapter 4)

 $W_r = \emptyset \exp (-kt^{\alpha}/H^{\beta})$ where k = 0.23 $\alpha = 0.5$ $\beta = 0.25$ and $\emptyset = 0.85$

average underflow solids concentration = 60,000 mg/l
ACTIVATED SLUDGE TANKS: (using nomenclature of Chapter 5)

$$k_{1} = 0.0028 \text{ hr}^{-1}$$

$$k_{2} = 0.0014 \text{ hr}^{-1}$$

$$k_{3} = 0.000002 \text{ hr}^{-1}$$

$$k_{4} = 0.0010 \text{ hr}^{-1}$$

$$k_{5} = 2.2 \text{ hr}^{-1}/(\text{cu.ft.air/cu.ft.reactor})$$

y = 0.54

MLSS = 2200 mg/l

RTD model: 3 CSTR's in series.

SECONDARY SETTLING TANKS: (using nomenclature of Chapter 6) $XRSS = \frac{556 (GSS \ 0.494) \times 1.5}{(MLSS \ 1.82) (TM \ 1.5)}$

SVI = 56.1 + 113 (FM) (1.05^{T-20})

fraction of time return sludge pumps used = 0.3 ANAEROBIC DIGESTER: (using nomenclature of Chapter 7)

lst stage at high-rate, k = 0.082

2nd stage at conventional rate, $k = 0.026 \text{ day}^{-1}$

fraction of flow to supernate = 0.6

Figure 12.1.1. Operating Conditions and Parameters Used

In the Simulation of the Drury Lane Plant

· · · · · · · · · · · · · · · · · · ·	Flow	(migd)	BOD)(mg/l)	TSS (mg/1)	VSS	(mg/l)	· DOC	(mg/l)
Stream	Plant Data	Predicted								
Raw Feed	1.6	_	215		299	-	191	-	(61)	_
Primary Effluent	1.6	1.6	145	151	155	153	80	76		60.
Primary Clarifier Underflow	-	0.0069		-	-	63900	-	41500		60
Secondary Effluent	1.6	1,59	15	22	20	20	8	11	-	10
Return Activated Sludge	(0.32)	0,30	-		12000	11700	7200	6620		10
Raw Sludge to Digester	0.0053	0.0076		7	58000	65400	37000	45300		60
Digester Supernate	0.0028	0.0044	- ·	۳.	8000	4300	-	1600	۲.	82
Digested Sludge	0.0023	0.0030		-	57000	58600	22000	12100	_	82

Table 12.1.2. Simulation of the Drury Lane, Burlington, Water Pollution Control Plant 152

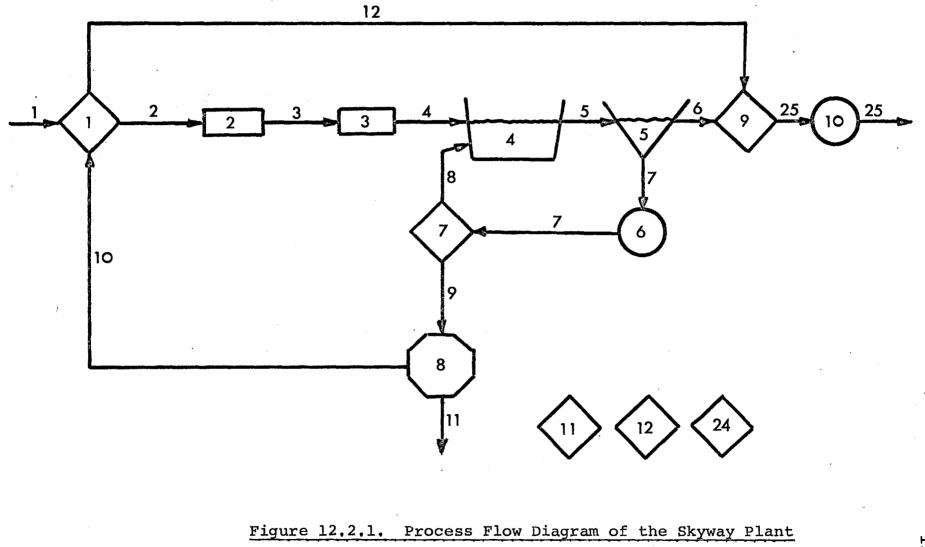
12.2 Simulation of the Skyway, Burlington, Sewage Treatment

'<u>Plant</u>

The process flow diagram for the plant is shown in Figure 12.2.1, and the design data is summarized in Appendix J. The average flow in 1969 was 3.3 migd, but due to an expansion completed in 1970, it is now receiving about 6.0 migd.

Pretreatment consists of a 2½ inch coarse screen followed by a 1 inch screen, both mechanically cleaned. Grit removal is achieved by an aerated grit chamber with a detention time of approximately 20 minutes. There are no primary tanks in the Skyway Plant. The activated sludge process employed is that of the extended aeration modification, and consists of six single-pass tanks with dimensions of 270 ft. x 27 ft. x 15 ft. The total blower capacity is 18000 scfm. Secondary clarification is provided by four settling tanks, 60 ft. square and 12 ft. deep. Waste sludge is thickened in a circular tank, 20 ft. in diameter and 9 ft. deep. A summary of the operating conditions and the parameters used in the simulation is shown in Table 12.2.1.

The data set for the simulation is shown in Appendix J, together with the computer printout for the case. A summary of the results from the simulation together with the plant data is shown in Table 12.2.2.



Key to Figure 12.2.1.

Module Number	Equipment
1	MIXER1
2	SCREEN1 - Screens
3	GRITI - Grit chambers
4	ACTSL1 - Activated sludge tanks
5	SECLARl - Secondary clarifiers
6	SETSP1
7	CØNTO2 - Controls sludge wasting
8	SEPAOl - (Thickener)
9.	MIXERL
1_0	CHLØRl - Chlorinator
11	CØNTLL
12	CØNTL1
24	REPTOl - Report generator

The simulation predicted a slightly better final effluent than was observed. The BOD and TSS predicted were 8 mg/l and 10 mg/l respectively, while the observed values were 12 mg/l and 12 mg/l respectively.

From plant data, the average suspended solids concentration in the return sludge was 10,000 mg/l, while the simulation predicted a value of 9,500 mg/l. The agreement is very good. However, the average volatile content of the sludge observed was 60% while the simulation gave a value of 42%.

A greater discrepancy occured here than was with the case of the Drury Lane Plant, in the volume of sludge hauled away. The simulation predicted that a volume of 22,000 igpd at a solids content of 2.1% should be hauled away, but actual plant figures indicated that only 4000 igpd at 2.0% solids were hauled away. Although upsets are frequent in a plant this size, the almost five-fold difference in the thickened sludge for disposal is unexpected. From a rough balance on the solids over the entire plant, the volume of sludge produced should be approximately

 $3.3 \times 10^6 \times (185 - 12)$

20,000

= 28,000 igpd, at 20% solids

This figure is in better agreement with the predicted value. Continuous sampling of the final effluent will probably show up the difference.

, 5 average flow = 3.3 migd average TSS = 185 mg/l average VSS = 85 mg/l average BOD = 155 mg/l average DOC = (62) mg/l

ACTIVATED SLUDGE Tanks: (using nomenclature of Chapter 5)

 $k_{1} = 0.0028 \text{ hr}^{-1}$ $k_{2} = 0.0014 \text{ hr}^{-1}$ $k_{3} = 0.000002 \text{ hr}^{-1}$ $k_{4} = 0.0010 \text{ hr}^{-1}$ $k_{5} = 2.2 \text{ hr}^{-1}$ Y = 0.54

MLSS = 4000 mg/l

RTD model: 3 CSTR's in series

SECONDARY SETTLING DATA: (using nomenclature of Chapter 6)

$$XRSS = \frac{556 (GSS \ 0.494) \times 1.5}{(MLSS \ 1.82) (TM \ 1.5)}$$

 $SVI = 56.1 + 113 (FM) (1.05 T^{-20})$

Fraction of time return sludge pumps used = 0.4 THICKENER: Fraction of liquid to overflow = 0.6

Fraction of solids to overflow = 0.1

Table 12.2.1. Operating Conditions and Parameters Used in the Simulation of the Skyway Plant

	Flow (migd)		BOD (mg/1)		TSS (mg/l)		VSS(mg/l)		DOC(mg/l)	
Stream	Plant Data	Predicted								
Raw Feed	3.3	_	155	-	185	-	84	-	(62)	
Secondary Effluent	3.3	3.3	12	8	12	10	3	4	-	4.
Return Activated Sludge	2.1	2,35	1	_	10000	9500	6000	4000	-	4
Thickened Sludge	0.004	0.022	-	-	20000	21000	12000	9000	-	-

Table 12.2.2. Simulation of the Skyway, Burlington, Water Pollution Control Plant

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CHAPTER 13

13. DISCUSSION OF RESULTS

13.1 Discussion of Experimental Data

13.1.1 Primary Settling Data

Our settling curve analysis is based on the settleable solids present in the raw waste. The same analysis when applied to the total suspended solids gave less consistent results between the runs, as can be seen from Table 13.1.1.

Run Number	Source	k	œ	β
A-1	Dundas (1971)	0.164	0.372	0.216
A-2	Drury Lane (1971)	0.202	0.403	0.285
A-3	Dundas (1970)	0.211	0.410	0.311
A-4	Dundas (1969)	0.150	0.397	0.252

Table 13.1.1. Correlations for the Batch Settling

Data, Based on Total Solids

An assumption made in our model is that the removal obtained is the same for all classes of settleable solids. The validity of the assumption, although not tested, can be checked by taking volatile suspended solids measurements, together with the usual suspended solids readings. A shortcoming of our data is the fact that our batch settling column is only 8 ft. tall. This means that an effective liquid depth of only 6 ft. may be used. Most primary settling tanks have liquid depths between 8 ft. and 12 ft., thereby necessitating the extrapolation of our data, to predict their performance.

13.1.2 Tracer Studies of Primary Sedimentation Tanks

In an attempt to evaluate the level of turbulence existing in the primary settling tanks, tracer studies were performed. The results were discouraging. Tracer recoveries were of the order of 80% and consequently the detention times estimated from the exit age distributions were much lower than the hydraulic detention times. The loss of the tail end of the exit age distribution also affects whatever model we try to fit to the flow pattern. Another tracer system should have been used where the recovery obtained is much higher.

The use of a pulse input is also a poor choice as it generates a tail end in the exit age distribution, which is very important but where the accuracy of measurement is poor. Time Series Analysis, using a series of random pulse inputs would largely remove this source of error. We had to revert to a simplified treatment to correct for the effects of turbulence by defining the efficiency factor. The fact that a fairly constant efficiency factor exists for the primary tanks in both the Dundas and Drury Lane plants seem to suggest that a correlation is possible, relating the efficiency factor with a parameter that represents the level of turbulence in the tanks.

13.1.3 Data From the Chemostat Experiments

Our Chemostat set-up is rather simple and lacks a good control over the operating conditions. A closer control should have been kept on the temperature in the reactor. With proper temperature control, the experiments could be repeated at other temperature levels. The flow rate from the constant head tank was found to vary and had to be adjusted about twice a day.

We have assumed in our calculations that the original solids in the feed are not significantly solubilized during the experiment, and that the increase in solids in the effluent is mainly due to synthesis of microbial cells. There will be a slight solubilization of the feed solids, and hence our calculated value for M, the synthesized microbial cell mass, will be low. This problems can be corrected by filtering the raw waste

through a filter press or by using a "synthetic" raw sewage, having only the soluble components.

Another assumption made was that the effluent soluble organic carbon was entirely degradable. This was not checked by taking the Biochemical Oxygen Demands of the filtered effluent. However, the least squares fit of the data points gave a correlation u = 1.04 $\times 10^{-3}$ (C + 0.55). The intercept on the horizontal axis is not significantly different from zero (but is in fact even negative). A significant positive intercept of say 5 mg/l or greater will indicate that not all of the effluent soluble carbon is degradable.

13.1.4 Data on the Aeration Studies of Activated Sludge

Since some of the biological rates in the kinetic scheme proposed in Chapter 5 are expected to be very slow, we had to aerate the return activated sludge for at least two weeks. No raw sewage was added to the return sludge. The reason is that the soluble degradable carbon in the raw sewage will have a very rapid rate of assimilation and hence creating a "stiff" condition in the solution of the differential equation describing its rate of reaction.

Several assumptions were made in the analysis of the results. The conversion factors obtained in Section 9.2,

were used in calculating the soluble degradable carbon and in calculating the carbon equivalent of the volatile solids. Equation 5.1.12, relating the oxygen uptake rate to the rate of carbon loss due to respiration, and hence to the concentration of microorganisms, was assumed to hold. The above assumptions can be tested by integrating the area under the oxygen uptake rate curve (which is the total oxygen consumption), and comparing with the amount of carbon loss. for Run No. D-1,

 $\frac{\Delta O_2}{\Delta C} = \frac{6200}{1700} = 3.65$

and for Run No. D-2,

$$\frac{\Delta O_2}{\Delta C} = \frac{5800}{1500} = 3.86$$

Both values are close to the factor of 3.73 obtained in Section 5.1.

A comparison of the results from the chemostat experiments and the extended aeration studies showed that the unit growth rate factor, k_4 , was much lower in the latter case. The value of k_4 obtained in the chemostat experiments was 0.0010 hr⁻¹ (mg/1)⁻¹ whilst that obtained from the extended aeration studies was 0.0006 hr⁻¹ (mg/1)⁻¹. The former experiments were run under conditions where the substrate concentrations were high, whilst the latter experiments were made under food-deficient conditions. This suggests that an extended food-deficient condition could lead to decreased microbial activity, in what is commonly known as a "lag-phase". This implies then that k₄ is not actually a constant but could depend on the length of substrate privation. However, in activated sludge processes, the former condition normally occurs. The food-deficient condition would occur if we use the aerobic biological reaction in a solids digestion process.

Only a few readings were taken of the soluble degradable organic carbon throughout the duration of the runs. This was because the BOD measurements required a fairly large volume of sample. A larger reactor should have been used.

13.2 Discussion of Plant Simulations

13.2.1 <u>Simulation of Drury Lane, Burlington, Sewage Treatment</u> Plant

The agreement between the predicted and observed values for the primary and final effluents were very good. The results show that primary sedimentation tanks can be modelled quite successfully if a sufficiently accurate and detailed breakdown of the feed solids are provided. But normally a complete solids analysis, as required by our stream list is never performed, and the simulation may require the use of a less detailed model.

The activated sludge process was also modelled very successfully. The kinetic rates, measured experimentally, predicted the final effluent very well. An examination of the converged values for the return sludge stream showed that 42% of the volatile solids is composed of microorganisms. This is in agreement with the value of approximately 50% obtained in Section 5.3, for the return sludge from the Skyway Plant,

Several other case studies were made under different operating and flow conditions. Table 13.2.1 shows the individual effects of increasing capacities in the primary clarifiers, aeration tanks, secondary clarifiers and of just increasing the mixed liquor suspended solids in the aeration tanks to 3000 mg/1.

The effect of adding an extra primary clarifier was merely to improve the primary effluent to a BOD of 146 mg/l and a TSS of 141 mg/l. There was no significant improvement in the final effluent.

The effect of adding an extra aeration tank improved the final effluent to a BOD of 15 m/gl and a TSS of 17 mg/l. No improvements were expected to occur in the primary effluent.

Stream	BOD	TSS	DOC			
Raw Sewage	215	299	61			
CASE STUDY WITH NORMAL (OPERATING CONDI	TIONS				
Primary Effluent	151	153	61			
Secondary Effluent	22	20	10			
CASE STUDY WITH A THIRD	PRIMARY CLARII	TIER				
Primary Effluent	146	141	61 .			
Secondary Effluent	21	20	9			
CASE STUDY WITH A THIRD	AERATION TANK					
Primary Effluent	150	153	61			
Secondary Effluent	15	17	8			
CASE STUDY WITH A THIRD	SECONDARY CLAI	RIFIER				
Primary Effluent	150	153	61			
Secondary Effluent	21	17	9			
CASE STUDY WITH MLSS INCREASED TO 3000 mg/l						
Primary Effluent	151	153	61			
Secondary Effluent	16	16	7			

Table 13.2.1. Effect of Plant Alterations, at the Same Flow.

The extra secondary clarifier reduced the TSS in the final effluent to 17 mg/l, with not much improvement in the BOD.

The results of merely increasing the mixed liquor suspended solids in the aeration tanks to 3000 mg/l were quite surprising. The final effluent improved to a BOD of 16 mg/l and a TSS of 16 mg/l. This is just like having an extra aeration tank! This is quite logical since about 50% more solids will be carried in the aeration tanks at 3000 mg/l as compared to the operating value of 2200 mg/l.

Table 13.2.2 shows the effect of increasing the raw waste flow to 2.0 and 2.4 migd., with the same influent BOD and TSS, and with no change in equipment capacity. It should be noted that the plant was originally designed for a flow of 2.5 migd with an influent BOD of 200 and an influent TSS of 180 mg/l. However, both the influent BOD and TSS have increased over the years. The reported influent BOD and TSS In 1969 were, as mentioned before, 215 mg/l and 299 mg/l respectively.

At the operating conditions of 1969, with a feed flow of 1.6 migd, the final effluent BOD and TSS were predicted to be 22 mg/1 and 20 mg/1 respectively. This is just bordering on the limits set by the Ontario Water

Resources Commission. At a feed flow of 2.0 migd, the primary effluent deteriorated slightly to a BOD of 153 mg/l and a TSS of 160 mg/l, while the final effluent BOD and TSS rose to 28 mg/l and 25 mg/l respectively. At a feed flow of 2.4 migd, the primary effluent deteriorated further to a BOD of 156 mg/l and a TSS of 166 mg/l while the final effluent BOD and TSS predicted were 34 mg/l and 30 mg/l respectively, which is completely unsatisfactory.

The limiting equipment was found to be in the aeration tanks. If expansion is to be contemplated at the Drury Lane Plant to handle a higher flow, top priority should be given to expand the aeration tank capacity. The problem can be alleviated by maintaining a higher mixed liquor suspended solids in the activated sludge process.

Stream	BOD	TSS	DOC
Raw Sewage	215	299	61
CASE STUDY WITH 1.6 migd	FEED FLOW		
Primary Effluent	151	153	61
Secondary Effluent	22	20	10
CASE STUDY WITH 2.0 migd	FEED FLOW		
Primary Effluent	153	160	61
Secondary Effluent	28	25	12
CASE STUDY WITH 2.4 migd	FEED FLOW		
Primary Effluent	156	166	61
Secondary Effluent	34	30	15

Table 13.2.2. Effect of Increased Plant Flows, with no

Change in Present Equipment Capacities

13.2.2 Simulation of the Skyway, Burlington, Plant

In this case the activated sludge process model predicted a slightly better final effluent than was observed. However, the results show that the extended aeration process is not something special, but is merely the same activated sludge process, under a different name. The results also suggest that the 18 hour detention time required by the Ontario Water Resources Commission and other public agencies for a treatment plant without a primary clarifier is not necessary.

To show the effects of decreased detention time, or increased feed flow, on the final effluent, case studies were made at 6, 8 and 10 migd using the same feed composition. The two new final clarifiers, which are now completed, are added to the process flow diagram, together with the new (third) sludge return pump. The resulting detention times are 16, 12, and 10 hours respectively, all excluding sludge recycle.

The results are shown in Table 13.2.3. The final effluent at 10 migd feed flow, is predicted to have a BOD of 15 mg/l and a TSS of 20 mg/l, which is still acceptable. It should be pointed out that the feed to the plant (based on 1969 values) which has a BOD

of 155 mg/l and a TSS of 185 mg/l is approximately equivalent to the primary effluent of many conventional activated sludge plants, and hence a detention time of 8-12 hours, which is sufficient for a conventional activated sludge plant, should also be sufficient for the "extended aeration" modification. The only justification for requiring more aeration time in a "extended aeration" plant is when the influent BOD and TSS are both higher than say 200 mg/l.

Stream	BOD	TSS	DOC
Raw Sewage	155	185	62
CASE STUDY WITH 6.0	migd FEED FLOW		
Final Effluent	11	13	5
CASE STUDY WITH 8.0	migd FEED FLOW		
Final Effluent	13	17	6
CASE STUDY WITH 10.	0 migd FEED FLOW		
Final Effluent	15	20	7

Table 13.2.3. Effect of Increased Feed Flows, at the

Skyway Plant.

CHAPTER 14

14. CONCLUSION

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Simulation models for primary sedimentation, the activated sludge process, secondary sedimentation, anaerobic digestion and several other unit processes have been developed. The models were based on known and developed theories, supported by laboratory and plant data.

The primary clarifier was modelled on correlations obtained from batch column settling studies. It was found that if settleable suspended solids were used rather than the total suspended solids, better correlations resulted. This is to be expected as the total suspended solids include the non-settleable solids as well.

For the modelling of the activated sludge process, a new reaction scheme, based on the physiological activities of the bacterial and the bacterial cell mass, was proposed. Batch aeration studies of activated sludge were used to obtain estimates of the parameters used in the reaction scheme. This is an improvement over the use of mixed liquor suspended solids, as the fraction of "active mass" in the sludge is dependent on feed compositions and operating conditions in the plant.

For the secondary clarifier model, use was made of correlations developed by Villiers and the Rex Chainbelt

Company, for the fraction of solids escaping over the wiers and the Sludge Volume Index of the sludge produced, respectively. The anaerobic digester model assumed a first order rate for the liquefaction of organic solids, the rate constants being obtained from experimental data published in the literature.

Two plants were used in the simulation studies. The first is a conventional activated sludge plant, while the second is an extended aeration plant. The results from the simulation studies were in very good agreement with plant data for the liquid streams. The actual sludge volumes produced from both plants were lower than those predicted by the simulation. The loss of solids over the wiers of the secondary clarifiers, during plant upsets or storm flows, probably accounted for the lower production of sludge from both plants.

An inter-active version of the computer simulation was also developed and tested under varying conditions. This can be used as an effective teaching aid for operators and students in the wastewater treatment area.

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APPENDIX A

PRIMARY SETTLING DATA ON BATCH COLUMN

- RUN #A.1 Raw Sewage from Dundas W.P.C.P.
- RUN #A.2 Raw Sewage from Drury Lane, Burlington W.P.C.P.
- RUN #A.3 Raw Sewage from Dundas W.P.C.P., 1970 Data.
- RUN #A.4 Raw Sewage from Dundas W.P.C.P., 1969 Data.

<u>RUN #A.1</u>.

Raw Sewagė from Dundas W.P.C.P. Settling Data: Original suspended solids -

average of duplicates.

Time (min) Depth (ft)	0	15	30	60	90	120	24 Hrs.
1.5	252	106	156	132	124	116	82
2.5	247	210	164	146	129	122	82
3.5	253	210	179	152	137	129	83
4.5	248	209	180	156	147	151	82
5.5	248	220	188	159	152	151	84
Depth Correction (ft)	0	0.1	0.2	0.3	0.4	0.5	

Average of initial solids = 250 mg/l (rounded up)
Average of final non-settleable solids = 82 mg/l (rounded up)
Hence total settleable solids = 168 mg/l
% of settleable solids = 100 (SS - 82)/168

Weight % settleable solids as a function of time and depth:

<u>Time (min)</u> Depth (ft)	. 0	15	30	60	90	120	24 Hrs.
1.5	100	62	44	30	25	20	0
2.5	100	77	49	38	28	24	0
3.5	100	77	58	42	33	28	0
4.5	100	77	58	44	39	35	0
5.5	100	82	63	46	42	35	0
Depth Correction (ft)	0	0.1	0.2	0.3	0.4	0.5	

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Fraction of settleable solids removed, as a function of time and depth, using Equation 4.1.6:

Depth (ft)	Time (min)	Fraction Settleable Solids Removed
2.0	16	0.47
	21	0.57
	28	0.61
	35	0.65
	48	0.69
	65	0.73
3.0	20	0.46
	28	0.57
	35	0.61
	45	0.65
	62	0.69
	- 84	0.74
4.0	23	0.46
	31	0.57
	41	0.61
	53	0.65
	74	0.69
	101	0.74

(Con't.)

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Depth (ft)	Time (min)	Fraction Settleable Solids Removed
5.0	25	0.45
-	34	0.56
	46	0.60
	60	0.64
	85	0.69
	116	0.73
6.0	26	0.44
	37	0.56
	51	0.60
	66	0.64
	95	0.70
	130	0.74

RUN #A.2.

Raw Sewage from Drury Lane, Burlington W.P.C.P. Settling Data: Original suspended solids average of duplicates

Time(min) Depth(ft)	0	10	20	30	45	60	75	90	120	24 Hrs.
2.0	282	208	178	157	142	135	131	128	120	88
3.0	272	225	181	167	161	154	144	128	124	88
4.0	276	244	204	176	169	152	137	138	133	86
5.0	278	251	203	176	168	154	140	135	132	89
6.0	273	238	208	189	175	169	156	146	132	88
Depth Correctior (ft)	0	0.1	0.2	0.3	0.3	0.4	0.5	0.6	0.7	

Average total initial suspended solids = 275 mg/l Average final suspended solids = 88 mg/l Total settleable solids = 187 mg/l Weight % settleable solids = 100 (SS - 88)/187 Temperature = 20[°]C

Weight % settleable solids as a function of time and depth:

<u>Time(min)</u> Depth(ft)	0	10	20	30	45	60	75	90	120	24 Hrs.
2.0	100	66	48	37	29	25	23	21	17	0
3.0	100	73	50	42	39	35	30	21	19	0
4.0	100	83	62	47	43	34	26	26	23	0
5.0	100	88	62	47	42	35	28	25	23	0
6.0	100	81	64	54	47	43	37	31	-23	0
Depth Correction (ft)	0	0.1	0.2	0.3	0.3	0.4	0.5	0.6	0.7	

Weight fraction of settleable solids removed, as a function fo time and depth, using Equation 4.1.6:

Depth (ft)	Time (min)	Weight Fraction Settleable Solids Removed
2.0	8	0.50
	13	0.59
	17	0.64
	22	0.67
	30	0.71
	38	0.75
	50	0.78
3.0	12	0.49
	18	0.58
	24	0.63
	31	0.67
	42	0.72
	54	0.75
	68	0.78
4.0	15	0.49
	23	0.57
	· 30	0.64
	39	0.67
	53	0.72

(Con't.)

Depth (ft)	Time (min)	Weight Fraction Settleable Solids Removed
4.0	68	0.75
	85	0.78
5.0	17	0.49
	26	0.57
	35	0.63
	46	0.67
	62	0.71
	80	0.75
	100	0.78
6.0	19	0.47
	29	0.57
	40	0.63
	53	0.67
	70	0.72
	92	0.75
	114	0.78

RUN #A.3.

Raw Sewage from Dundas W.P.C.P., data

of 1970.

Settling Data: Original suspended solids - average of duplicates.

<u>Time (min)</u> Depth (ft)	0	15	30	45	60	75	90	105	120	24 Hrs.
2.1	234	164	138	130	123	115	111	110	106	66
3.1	243	186	151	140	132	126	120	116	112	74
4.1	246	193	162	147	139	133	127	122	118	80
5.1	262	198	172	153	143	138	133	129	123	72
6.1	264	200	175	158	143	141	136	132	128	68
Depth Correction (ft)	0	0.1	0.2	0.3	0.5	0.6	0.7	0.8	0.9	

Average total initial suspended solids = 250 mg/l Average final suspended solids = 73 mg/l Total settleable solids = 177 mg/l Weight % settleable solids = 100 (SS - 72)/177 Weight % settleable solids as a function of time and depth:

<u>Time (min)</u> Depth(ft)	0	15	30	45	60	75	90	105	120	24 Hrs.
2.1	100	50	37	32	28	24	21	21	18	0
3.1	100	64	44	38	33	30	27	24	21	0
4.1	100	68	50	42	37	34	30	28	25	0
5.1	100	71	56	45	40	37	34	32	28	0
6.1	100	72	58	48	40	38	35	33	30	0
Depth Correction (ft)	0	0.1	0.2	0.3	0.5	0.6	0.7	0.8	0.9	

Weight fraction of settleable solids removed, as a function of time and depth, using Equation 4.1.6:

Depth (ft)	Time (min)	Weight Fraction Settleable Solids Removed						
2.0	9	0.50						
	13	0.58						
	18	0.63						
	23	0.66						
	31	0.70						
	44	0.74						
3.0	13	0.50						
	18	0.57						
•	24	0.62						
	32	0.66						
	43	0.70						
	60	0.74						
4.0	16	0.49						
	22	0.57						
	30	0.62						
	39	0.66						
	53	0.70						
	75	0.75						

Depth (ft)	Time (min)	Weight Fraction settleable Solids Removed					
5.0	18	0.49					
	26	0.52					
	34	0.62					
	46	0.66					
	63	0.70					
	88	0.75					
6.0	21	0.48					
	29	0.56					
	39	0.61					
	53	0.66					
	73	0.70					
	100	0.75					
	<u>.</u>	1					

RUN #A.4.

Raw Sewage from Dundas W.P.C.P.

Settling Data: Original suspended solids average of duplicates

- data of Hudspith et al, 1969

<u>Time (min)</u> Depth (ft)	0	15	30	45	60	75	90	105	120	24 Hrs.
1.0	(218)*	192	158	145	133	152	160	144	140	88
2.0	(232)*	187	166	165	167	160	146	150	145	93
3.0	248	204	178	163	156	158	157	188	145	95
4.0	244	208	176	174	162	162	138	168	157	93
5.0	246	202	176	172	164	164	144	156	158	(158)

Average of initial solids = 246 mg/l Average of final non-settleable solids = 92 mg/l Hence total settleable solids = 154 mg/l % settleable solids = 100 (SS - 92)/154

*Insufficient mixing?

Weight % settleable solids as a function

of time and depth:

Time (min)Depth (ft)	0	15	30	45	60	75	90	105	120	24 Hrs.
2.2	100	66	43	34	27	39	44	34	31	0
3.2	100	62	48	47	49	44	37	38	34	0
4.2	100	73	56	47	42	43	42	62	34	0
5.2	100	72	55	53	45	45	30	49	42	0
6.2	100	71	55	52	47	47	34	42	43	0
Depth Correction (ft)	0	0.1	0.2	0.3	0.4	0.5	0.6	0.7	0.8	

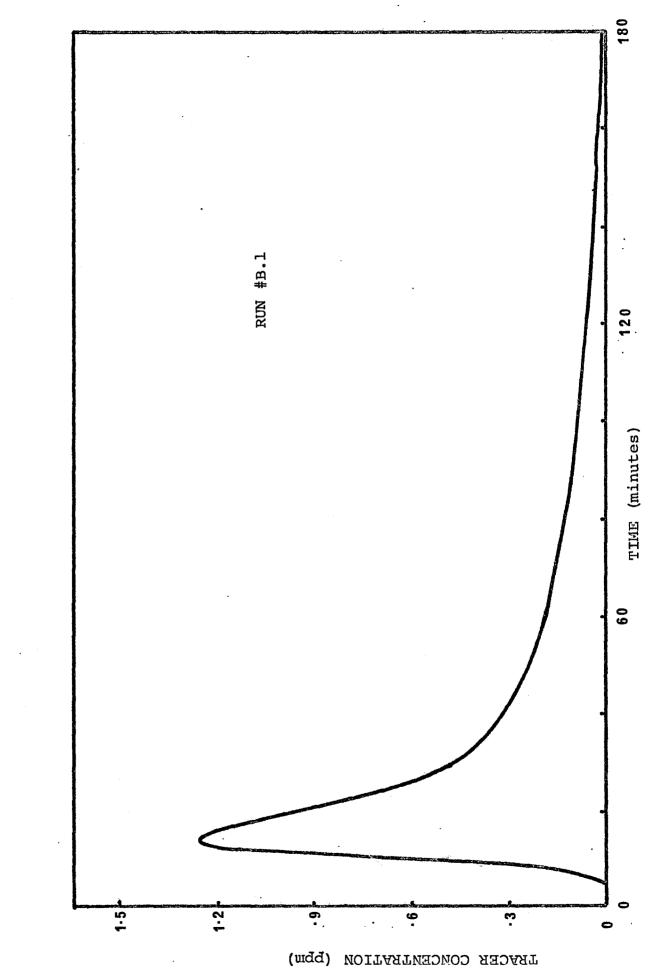
Weight fraction of settleable solids removed as a function of time and depth, using Equation 4.1.6:

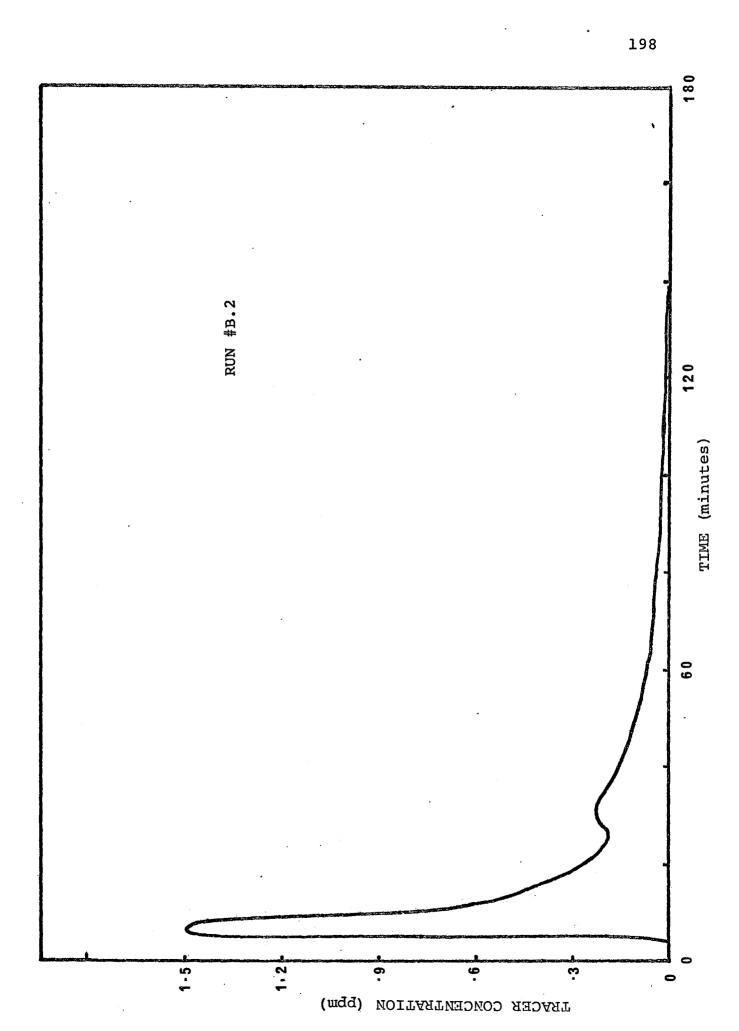
Depth (ft)	Time (min)	Weight Fraction settleable Solids Removed
2.0	10	0.46
	15	0.52
	20	0.58
	27	0.60
	38	0.67
	60	0.74
3.0	13	0.45
	20	0.52
Ň	27	0.56
	37	0.60
	53	0.67
	80	0.73
4.0	16	0.44
	24	0.52
	34	0.56
	46	0.59
	65	0.66
	97	0.72

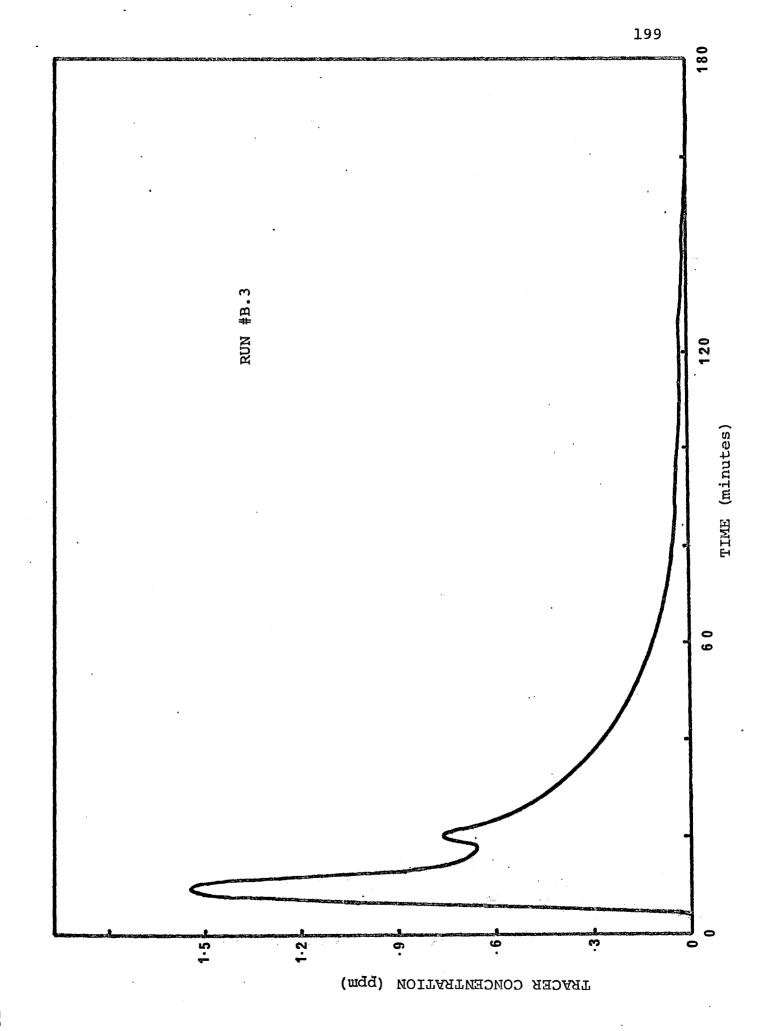
Depth (ft)	Time (min)	Weight Fraction Settleable Solids Removed
5.0	18	0.44
•	27	0.53
	39	0.57
	53	0.61
	76	0.66
	111	0.74
6.0	20	0.44
	29	0.50
	43	0.55
	60	0.60
	85	0.66
	123	0.75
		[·

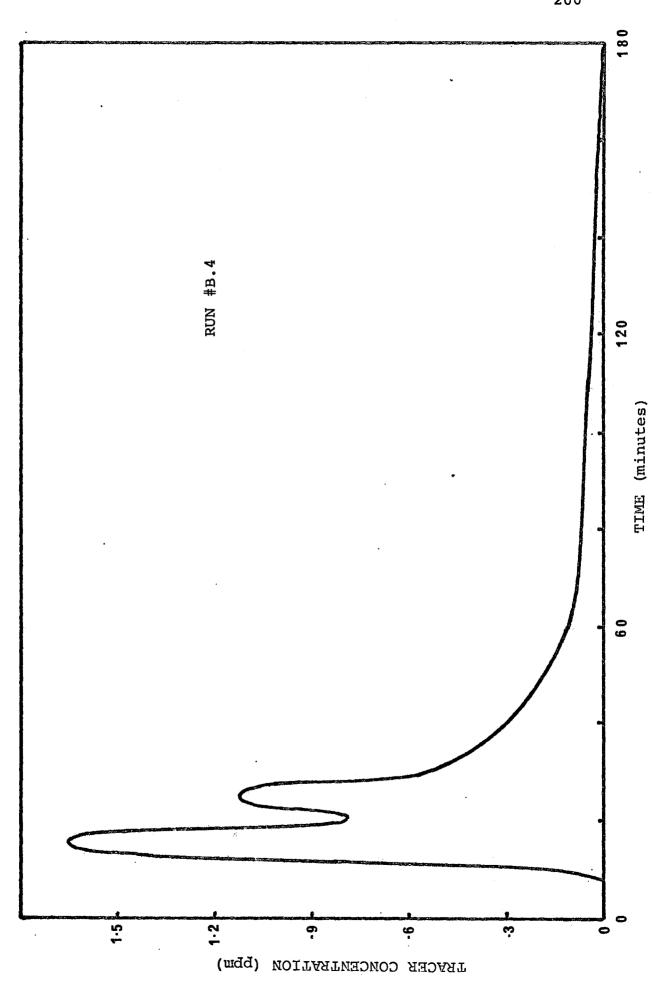
APPENDIX B

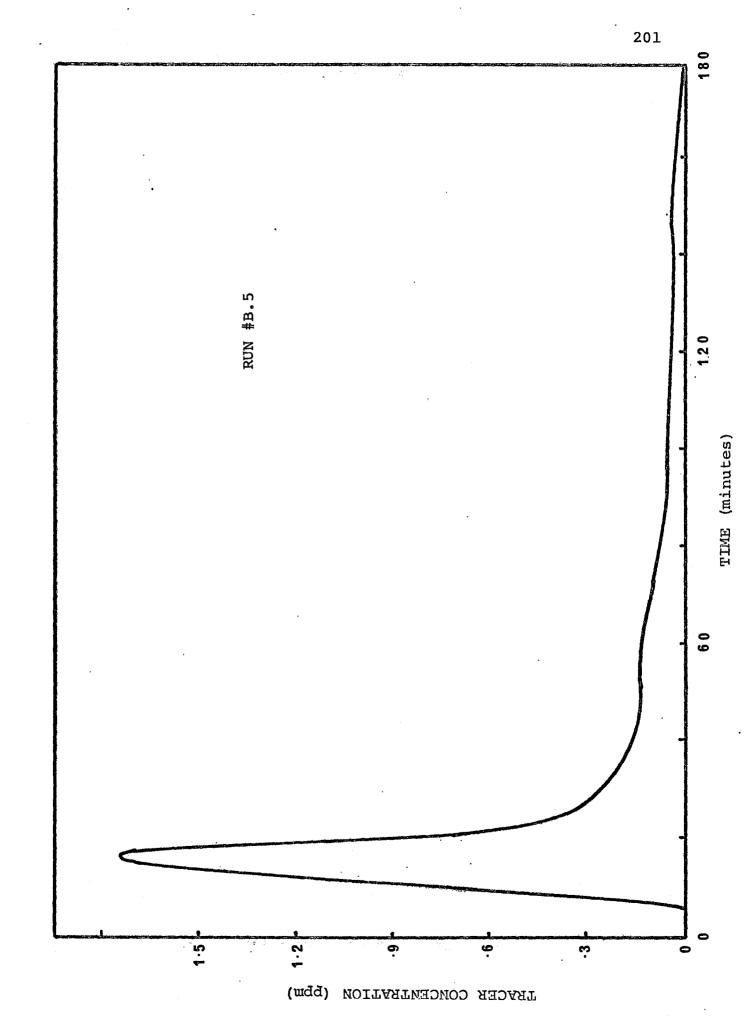
Exit Age Distributions of Primary Tanks RUN #B.1-3 Dundas Plant RUN #B.4-6 Drury Lane Plant SETTLEABLE SOLIDS REMOVAL DATA

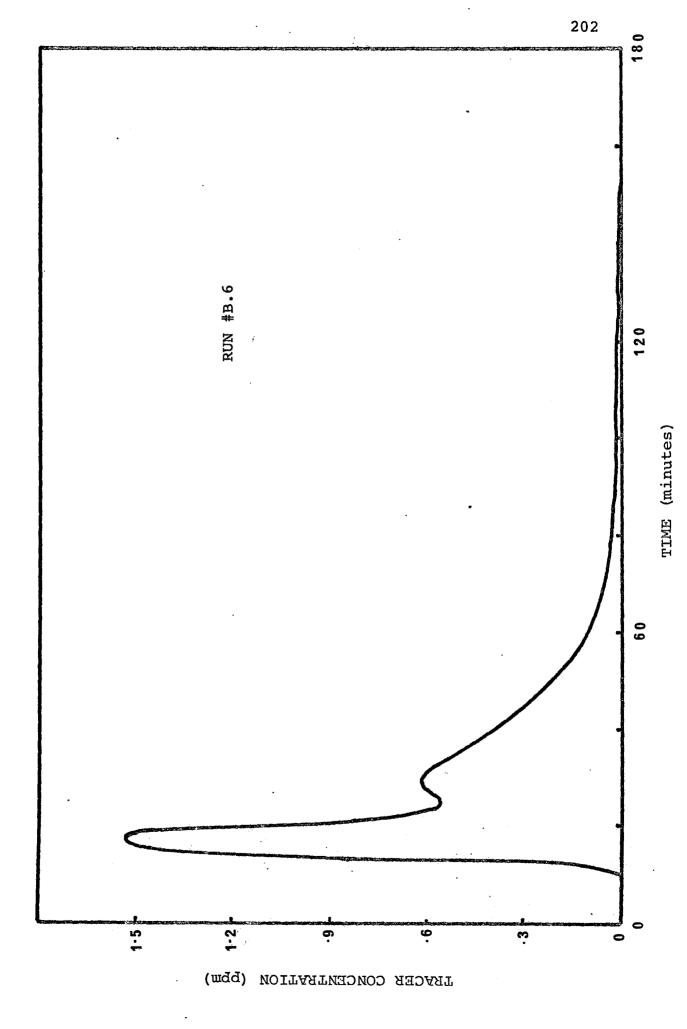












Run Number	Raw Flow (migd)	Settleable Suspended Solids Influent (Effluent		% Removal
B.1 B.2	2.1	285 240	171 155	0.40
в.3	1.8	252	159	0.37
B.4 B.5	2.2 2.0	244 207	100 75	0.59
в.6	1.6	210	71	0.66

Table B.1. Settleable Solids Removal Data

APPENDIX C

DATA FROM THE CHEMOSTAT EXPERIMENTS

RUN	#C.1.	Detention	Time	n ,	40 hours
RUN	#C.2.	Detention	Time	II ,	45 hours
RUN	#C.3.	Detention	Time	=	60 hours
RUN	#C.4.	Detention	Time	=	30 hours
RUN	#C.5.	Detention	Time	, III	45 hours
RUN	#C.6.	Detention	Time	=	70 hours
RUN	#C.7.	Detention	Time	=	50 hours

							· · ·	
Run	Detention			EFFLUENT		$M = \frac{\Delta SS}{2.1}$	$U = k_4 C_1$	
Number	Time (Hrs.)	C _o (mg/1)	SS _o (mg/1)	C(mg/l)	SS(mg/l)	(mg/1)	(mg/1) ⁻¹ (hr) ⁻¹	Y
1	40	69	26	45	54	14	4.9×10^{-2}	0,58
2	45	70	22	43	56	17	3.6×10^{-2}	0.62
3	60	65	25	35	59	16	3.2×10^{-2}	0.52
4	30	65	28	50	47	ş (5.5×10^{-2}	
5	45	62	30	35	64	16	3.8x10 ⁻²	0.59
6	70	60	24	25	. 64	19	2.6×10^{-2}	0.54
7	50	72	25	30	73	23	3.6×10^{-2}	0.53
1				1				

Average value of Y = 0.57

Standard deviation = 0.037

Temperature = $25^{\circ}C + 3^{\circ}C$

Source: Filtered raw sewage from Skyway, Burlington

(<u>NOTE</u>: M is obtained from the increase in suspended solids by the conversion factor of 2.1. Both C and M are expressed in terms of carbon).

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APPENDIX D

EXTENDED AERATION STUDIES ON ACTIVATED SLUDGE

- (a) Two runs using return sludge from the Skyway, Burlington W.P.C.P.
- (b) Replicates analyses using five simulaneously drawn samples.

Aeration of return activated sludge from the

Skyway, Burlington W.P.C.P.

Original Data:

Aeration Time (days)	Suspended Solids (mg/l)	Oxygen Uptake Rate (mgO ₂ /1/hr)	Organic Carbon	BOD ₅ (filtrate) (mg/l)
0	6970	30.0	24	16
1	6710	21.6	25	
2	6260	18.6	29	10
3	6040	16.2	35	
7	5550	15.0	39	
8	5200	10.2	45	
9	4710	12.0	52	12
10	4880	10.8	46	
11	4560	9.6	50	
14	4670	9.0	53	
15	3960	6.6	55	
16	4250	6.0	53	8
17	3940	6.6	60	
23	3840	3.6	67	
24	3520	3.0	65	
25	3660	2.4	67	14

Temperature = $23 \pm 2^{\circ}C$

 $pH = 7.5 \pm 0.3$

Air Rate = 1.0 litre/hr/litre reactor volume Ash content of sludge:

initial = 1650 mg/l
final = 1480 mg/l
average = 1520 mg/l

Converted Data:

· · · · · · · · · · · · · · · · · · ·		-	-	
Aeration Time (days)	VSS (mgC/l)	OUR (mgO ₂ /l/hr)	SNDOC (mgC/l)	SDOC (mgC/l)
0	2600	20.0	16	8
1	2470	21.6	19	
2	2260	18,6	23	5
3	2150	16.2	30	
7	1920	15.0	34	
8	1750	10.2	40	
9	1520	12.0	46	6
10	1600	10.8	41	
11	1450	9.6	45	
14	1500	9.0	48	
15	1160	6.6	50	
16	1300	6.0	49	4
17	1150	6.6	55	
23	1120	3.6	62	
24	950	3.0	60	
25	1020	2.4	60	7

VSS = Volatile suspended solids (mg/ 1) (see Section 9.1) = (Total Suspended Solids - 1520)/2.1

SDOC	Ħ	Soluble degradable organic carbon (mgC/l)
	=	BOD ₅ (filtrate)/1.9

SNDOC = Soluble non-degradable organic carbon (mhC/l)
= soluble organic carbon - SDOC

Aeration of return activated sludge from the

Skyway, Burlington W.P.C.P.

Original Data:

Aeration Time (days)	Suspended Solids (mg/1)	Oxygen Uptake Rate (mgO ₂ /1/hr)	Soluble Organic Carbon (mgC/l)	BOD ₅ (filtrate) (mg/l)
0	7900	30.6	21	16
1	7370	25.2	23	
2	7260	21.6	29	
3	7100	21.6	33	10
4	6530	18.6	37	
7	5800	15.0	43	
8	5900	15.6	48	8
9	5290	13.8	50	
10	5380	11.4	55	
11	5290	10.8	57	
14	4700	7.8	59	12
15	4950	6.6	64	
16	4850	6.6	59	
17	4910	5.4	61	12
(

Temperature = $23 \pm 2^{\circ}C$ pH = 7.4 ± 0.4 Air Rate = 1.0 litre air/hr/litre reactor volume <u>Ash content of sludge</u>: initial = 1700 mg/l final = 1890 mg/l average = 1800 mg/l

Converted Data:

[
Aeration Time (days)	VSS (mgC/l)	OUR (mgO ₂ /l/hr)	SNDOC (mgC/l)	SDOC (mgC/l)
0	2900	30.6	13	8
1	2650	25.2	18	
2	2600	21.6	24	
3	2520	21.6	28	5
4	2250	18.6	32	
7	1900	15.0	38	
8	1950	15.6	44	4
9	1660	13.8	45	
10	1700	11.4	50	
11	1660	10.8	52	
14,	1380	7.8	53	6
15	1500	6.6	59	
.6	1450	6.6	54	
17	1480	5.4	55	6
x				

VSS = volatile suspended solids (mgC/l) (see Section 9.1)

- SDOC = soluble degradable organic carbon (mgC/l)
 = BOD₅ (filtrate)/l.9
- SNDOC = soluble non-degrddable organic carbon
 (mgC/l)

= soluble organic carbon - SDOC

(b) Replicate Analyses on Five Simultaneously-drawn

Samples:

Original Data:

	Volatile Suspended Solids (mg/l)	Oxygen Uptake Rate (mgO ₂ /1/hr)	Soluble Organic Carbon (mgC/l)	BOD ₅ (mg/1)
1	3820	12.6	47	10.4
2	3550	9.6	51	7.6
3	3890	9.6	45	8.4
4	3920	10.2	44	11.6
5	3610	12.0	51	6.4

 <u>NOTE</u>: (a) 10 ml samples were used for VSS
 (b) dilution for BOD₅ is 25%. Filtrate is close to saturation. Hence no need for O₂ correction.

The above data is converted to the similar values used in the four responses for the objective function:

Sample Number	P+M (mgC/l)	OUR (mgO ₂ /1/hr)	SNDOC (mgC/l)	SDOC (mgC/1)		
1	1820	12.6	41	5,5	•	
2	1690	9.6	47	4.0		
3	1850	9.6	41	4.4		
4	1870	10.2	38	6.0		
5	1720	12.0	48	3.3		
NOTE: (a) P+M = volatile suspended solids/2 1						
	(b) SD00	$C = BOD_5/1.9$	9			

(c) SNDOC = soluble organic carbon - SDOC

Variance - Covariance Matrix (18)

Component 1	=	volatile suspended solids (mgC/l)
Component 2	=	oxygen uptake rate (mgO ₂ /hr/l)
Component 3	=	soluble non-degradable carbon (mgC/l)
Component 4	=	soluble degradable carbon (mgC/l)

6450.0	7.5	-332.5	69,5
-7.5	1.98	0.75	0.045
-332.5	0.75	18.5	-4.325
69.5	0.045	-4.325	1.213

Inverse of Variance - Covariance Matrix

0.139	-1.22	4.19	7.00
-1.22	11.2	-36.8	-61.9
4.19	-36.8	126.3	212.0
7.00	-61.9	212.0	358.0

APPENDIX E

DATA FOR THE STRIPPING RATE OF PURGEABLE SOLUBLE ORGANIC CARBON BY AERATION WITH AIR

RUN	#E.1	Air	Rate =	2.1	cu.ft.air/hr/cu.ft.	reactor
RUN	#E.2	Air	Rate =	1.0	cu.ft.air/hr/cu.ft.	reactor
RUN	#E.3	Air	Rate =	0.5	cu.ft.air/hr/cu.ft.	reactor
RUN	#E.4	Air	Rate =	1.5	cu.ft.air/hr/cu.ft.	reactor

RUN #E.1.

Air Rate = 2.1 cu.ft.air/hr/cu.ft. reactor

Time	(minutes)	SOC (mg/l)	SPOC (mg/l)	
	0	123	24	
	1	118	19	
	3	116	17	
	5	115	16	
	7	110	11	
	10	113	14	
	13	106	7	
	17	106	7.	
	20	103	4	
	25	102	3	
	30	101	2	
	40	99	0	
	50	99	0	

Soluble non-purgeable organic carbon assumed to be = 99 mg/l SOC = total soluble organic carbon SPOC = soluble purgeable organic carbon SPOC = SOC - 99

<u>RUN #E.2</u>.

Air Rate = 1.0 cu.ft.air/hr/cu.ft. reactor

Time	(minutes)	SOC	(mg/l)	SPOC	(mg/1)
	0	1	.26	2	20
	2		.24		.8
	5		.21		.5
	10	1	24	. 1	.8
	15	1	.18	1	.2
	20	1	.16	ני	.0
	25	1	.18	1	.2
	30	1	.14		8
	40	1	.12		6
	50	נ	.10		4
	60	1	.09		3
	70	្រា	.08		2
	80	106			0
	90	1	_06		0

Soluble non-purgeable organic carbon assumed to be = 106 mg/l SOC = total soluble organic carbon SPOC = soluble purgeable organic carbon SPOC = SOC - 106

RUN #E.3.

Air Rate = 0.5 cu.ft.air/hr/cu.ft. reactor

Time	(minutes)	SOC	(mg/l)	SPOC	(mg/l)	
	0	1	31	2	20	
	5	1	29]	.8	
	10	1	28 .]	.7	
	15		27	16		
	20		126		15	
	30	124		13		
	40	122		11		
	50	120			9	
	60	1	19		8	
	80	1	17		6	
	100		115		4	
	120		111		0.	
140		1	11		0	

Soluble non-purgeable organic carbon assumed to be = lll mg/l SOC = total soluble organic carbon SPOC = soluble purgeable organic carbon SPOC = SOC - lll RUN #E.4.

Air Rate = 1.5 cu.ft.air/hr/cu.ft. reactor

Time	(minutes)	soc	(mg/l)	SPOC	(mg/1)
			i i		
	0	1	30	3	32
	2	1	26	2	28
	5	1	21	. 2	23
	10	116		18	
	15		113		.5
	20	109		11	
	25	106			8
	30	105			7
	40	102			4
	50	l	00		2
	60		98		0
70		98			0

Soluble non-purgeable carbon assumed to be = 98 mg/l SOC = total soluble organic carbon SPOC = soluble purgeable organic carbon SPOC = SOC - 98

APPENDIX F

DATA OF ACTIVATED SLUDGE

SETTLING FROM P. LEUNG

Aeration Time (Hrs.)	MLSS (mg/l)	Effluent SS (mg/l)	SVI	Prediction Effluent SS (mg/l)
			· · · · · · · · · · · · · · · · · · ·	
0.5	2100	[*] 35	101.3	38
1.0	600	74	83.0	68 [.]
1.0	3800	20	156	23
3.0	220	93	91.3	96
3.0	2100	34	97.9	26
3.0	4000	9.5	171	18
5.0	600	50	82.7	49
5.0	3800	14.5	153	17
5.5	2100	30	99.4	23

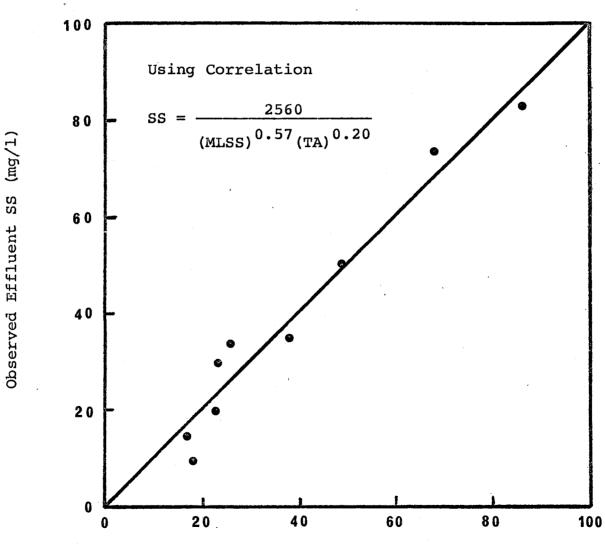
Data of Peter Leung

The predicted effluent SS_e was found to be

 $SS_{e} = \frac{2560}{(MLSS)^{0.57} (TA)^{0.20}}$ or XRSS = $\frac{SS_{e}}{(MLSS)} = \frac{2560}{(MLSS)^{1.57} (TA)^{0.20}}$

The SVI was not found to correlate well.

Figure F.l. Comparison of Observed Effluent Solids with Correlation



(Data of P. Leung)

Predicted Effluent SS (mg/l)

APPENDIX G

ANAEROBIC DIGESTION DATA

- (a) Conventional Digester -Data of Rankine (32).
- (b) High Rate Digester Data of Torpey (50), Roy and
 Sawyer (34) and Estrada (12).

(a) <u>Conventional Digester</u>

Data of Rankine (32).

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Detention Time (days)	Fraction Reduction of Volatile Solids =(1-R)	(l-R)/R
17	0.42	0.71
26	0.39	0.64
27	0.50	1.00
29	0.37	0.59
35	0.45	0.82
35	0.63	1.70
37	0.44	0.78
39	0.52	1.09
50	0.52	1.09
53	0.55	1.22
55	0.56	1.27
65	0.64	1.77
67	0.70	2.33
75	0.57	1.33
62	0.53	1.13

(b) High Rate Digester

Source	Detention Time (days)	Fraction Reduction of Volatile	(1-R)/R
	(uuys)	Solids = (1-R)/R	
Torpey (50)	14.0	0.56	1.27
	10.3	0.45	0.82
	8.3	0.47	0.70
	6.4	0.45	0.82
	3.7	0.37	0.59
Roy (34)	20.0	0.58	1.38
	15.0	0.57	1.33
	10.0	0.56	1.27
	8.0	0.54	1.17
	6.0	0.52	1.09
Estrada(12)	22.5	0.61	1.58
	22.0	0.54	1.17
	19.8	0.64	1.77
	18.8	0.61	1.58
	18.0	0.63	1.70
	15.9	0.68	2.12
	15,7	0.57	1.33
	14.8	0.48	0.92
	13.9	0.40	0.67
	13.3	0.52	1.09
	12.0	0.66	[.] 1.94
	10.1	0.42	0.71

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APPENDIX H

INTER-RELATIONSHIPS BETWEEN ORGANIC CARBON,

BIOCHEMICAL OXYGEN DEMAND AND VOLATILE SUSPENDED SOLIDS

Sample	Raw S TBOD	Gewage TOC	VSS	Filt: DBOD		DBOD SOC	TBOD-DBOD TOC-SOC	VSS TOC-SOC
l	220	160	230	114	60	1.90	1.06	2.30
2	300	226	320	110	58	1.96	1.13	1.90
3	260	177	224	110	68	1.62	1.37	2.05
4	289	192	236	145	72	2.02	1.20	1.98
5	315	218	286	149	85	1.75	1.23	2.14
6	310	187	244	149	74	2.01	1.42	2.15
7	308	196	272	151	70	2.15	1.25	2.15
Average =				1.91	1.24	2.09		
Standard Deviation =					0.17	0.12	0.11	

TBOD = total BOD_5 (mg/l)

DBOD = BOD_5 of soluble organics (mg/l)

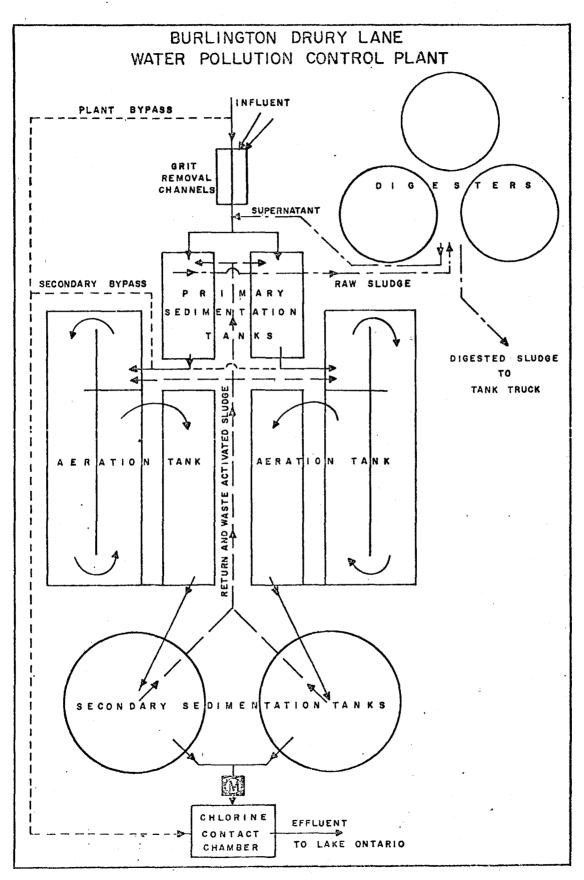
TOC = total organic carbon (mg/1)

SOC = soluble organic carbon (mg/1)

APPENDIX I

SIMULATION DATA AND RESULTS FOR THE DRURY LANE,

BURLINGTON, SEWAGE TREATMENT PLANT



 $\mathbf{2}$

DESIGN DATA

PROJECT NO.	2-0051-60	TREATMENT Acti	vated Sludge
DESIGN FLOW	2.5 mgd	DESIGN POPULATION	30,000
BOD – Raw Sewage – Removal	200 mg/l 90%	SS – Raw Sewage – Removal	180 mg/l 90%

PRIMARY TREATMENT

Screening

1" bar screens

Grit Removal

Type: Grit channels Retention: 0.8 min

Primary Sedimentation

Type: Walker Process Size: Two 49.3' x 18' x 12.25' (135,700 gal) Retention: 1.3 hr Loading: Surface, 1400 gal/ft²/day Weir, 17,100 gal/ft/day

SECONDARY TREATMENT

Aeration Tanks

Type: Diffused air; triple-pass Size: Two tanks, each with 2 passes 118' x 18' x 10.7' 1 pass 85.5' x 18' x 10.7' (833,000 gal. total) Retention: 8.0 hours

Air Supply

One Sutorbilt - 1500 cfm Two Roots-Connerville - 750 cfm

Diffusers - (each tank)

1) 132 Schumacher Brandel tubes in first two passes 2) 41 Spargers on 2' centres in third pass

Secondary Sedimentation

Type: Rex Unitabe Tow-Bro Size: Two 50' dia x 10.6' swd (260,000 gal) Retention: 2.5 hr Loading: Surface, 1000 gal/ft²/day Weir, 8500 gal/ft/day

CHLORINATION

Type: Kent

Chlorine Contact Chamber

- in outfall

OUTFALL

- to Lake Ontario

SLUDGE HANDLING

Digestion System

Type: Two-stage

Primary --Size: Two 40' dia tanks (313,000 gal total), depth = 20' Loading: 2.7 lb/ft³/mo

Secondary --Size: One 40' dia tank (143,000 gal) Loading Total: 1.9 lb/ft³/mo depth = 18' ,

DAI	A SET FOR	THE SIMULATION	OF THE DRU	URY LANE PL
RY L	ANE, BURLI 1.0 0.0 25.0 10.0 4.0 13.0 11.0	NGTON, STP 1.0 0.0 1.0 1.0 6.0 14.0 24.0	$1 \cdot 0$ $0 \cdot 0$ $2 \cdot 0$ $5 \cdot 0$ $15 \cdot 0$	0.0 0.0 5.0 12.0 7.0
	1.0 1.0 1.0 1.0 1.0	1.0 1.0 1.0 1.0 1.0 1.0	1.0 1.0 1.0 1.0 1.0	1 • 0 1 • 0 1 • 0 1 • 0 1 • 0
0.0	7.0 4.0 0.0 0.0 23.0 9.0	66000.0 0.0 48.0 18.0 100.0 0.5	25 • 0 0 • 0 56 • 0 3 • 0 0 • 0 6 • 0	5.0 0.0 10.0 6.0 0.0 0.0
97	9 • 0 0 • 0 0 • 0 0 • 0 4 4 9 9 • 4 8 • 7 4 0 • 0	0.5 186.7 0.0 2751.5 131.2 15796.0 0.0 15060.6 0.0 5086.3 0.0 144.2	$\begin{array}{c} 0 & 0 \\ 56 & 0 \\ 3 & 0 \\ 0 & 0 \\ 6 & 0 \\ 40 & 0 \\ 14 & 5 \\ 0 & 0 \\ 14 & 5 \\ 0 & 0 \\ 979 & 1 \\ 25 & 04 \\ 0 & 0 \\ 0 & 0 \end{array}$	0.00 6.5 0.0 163.9 28.4 0.0 0.0 4.0
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5	8 • 74 0 • 0 0 • 0	144.2 31.71 911.98 0.0 5086.3	25.04 0.0 0.0	4.5 0.0 1306.4

DRURY LAN	E, BURLINGTO	N, STP 1.0	1.0
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16.0 3.0 17.0	4•0 13•0	1•0 6•0 14•0 24•0	2•0 5•0 15•0
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4.0 18.0 660000.0 0.0 55.0 0.0 16.0 1863.97 0.0 451.5 54.1 0.0 6.0 149860.6 0.0 149860.6 0.0 1845.8 9.44 0.0 7.0 9074.6 0.0 1845.8 9.44 0.0 1845.8 9.44 0.0 1845.8 9.44 0.0 1845.8 9.44 0.0 1845.8 9.44 0.0 1845.8 9.44 0.0 1845.8 9.44 0.0 1845.8 9.44 0.0 1845.8 9.44 0.0 1845.8 9.44 0.0 1845.8 9.44 0.0 1845.8 9.44 0.0 1845.8 9.44 0.0 1845.8 9.44 0.0 1845.8 9.44 0.0 1845.8 9.0 0.0 1.0 0.0 0.0 0.0 0.0 0.0 0	$7 \cdot 0$ $4 \cdot 0$ $0 \cdot 0$ $0 \cdot 0$ $23 \cdot 0$ $9 \cdot 0$ $0 \cdot 0$ $0 \cdot 0$ $0 \cdot 0$ $0 \cdot 0$ $0 \cdot 0$ $4499 \cdot 4$ $8 \cdot 74$ $0 \cdot 0$ $0 \cdot 0$ $4499 \cdot 4$ $8 \cdot 74$ $0 \cdot 0$ $0 \cdot 0$ $0 \cdot 0$ $0 \cdot 0$ $0 \cdot 0$ $12 \cdot 75$ $8 \cdot 74$ $0 \cdot 0$ $0 \cdot 0$	$\begin{array}{c} 66000 \bullet 0 \\ 0 \bullet 0 \\ 48 \bullet 0 \\ 18 \bullet 0 \\ 100 \bullet 0 \\ 0 \bullet 5 \\ 186 \bullet 7 \\ 0 \bullet 0 \\ 2751 \bullet 5 \\ 131 \bullet 2 \\ 15796 \bullet 0 \\ 0 \\ 15060 \bullet 6 \\ 0 \bullet 0 \\ 15060 \bullet 6 \\ 0 \bullet 0 \\ 15060 \bullet 6 \\ 0 \bullet 0 \\ 15086 \bullet 3 \\ 0 \bullet 0 \\ 144 \bullet 2 \\ 31 \bullet 71 \\ 911 \bullet 98 \\ 0 \bullet 0 \\ 5086 \bullet 3 \\ 0 \bullet 0 \\ 144 \bullet 2 \\ 31 \bullet 71 \\ \end{array}$	$25 \cdot 0$ $0 \cdot 0$ $56 \cdot 0$ $3 \cdot 0$ $0 \cdot 0$ $6 \cdot 0$ $40 \cdot 0$ $0 \cdot 0$ $14 \cdot 5$ $0 \cdot 0$ $14 \cdot 5$ $0 \cdot 0$ $0 \cdot 0$ $979 \cdot 1$ $25 \cdot 04$ $0 \cdot 0$ $0 \cdot 0$ $0 \cdot 0$ $0 \cdot 0$ $8 \cdot 0$ $25 \cdot 04$ $0 \cdot 0$ $0 \cdot 0$ $8 \cdot 0$ $25 \cdot 04$ $0 \cdot 0$ $0 \cdot 0$ $8 \cdot 0$ $25 \cdot 04$ $0 \cdot 0$ $0 \cdot 0$ $8 \cdot 0$ $25 \cdot 04$ $0 \cdot 0$ $0 \cdot 0$ $8 \cdot 0$ $25 \cdot 04$ $0 \cdot 0$ 0
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1 • 0 3 • 0 1 • 0 2 5 0 0 • 0 5 • 0 1 • 0 2 • 0 6 • 0 1 • 0 1 • 0 7 • 0 0 • 0 0 • 0 0 • 0	4 • 0 5 • 0 0 • 20 0 • 3 17 • 0 6 • 0 8 • 0 4 • 0 6 • 0 8 • 0 4 • 0 6 • 0 2 • 0 0 • 0	0 0 0 9 8 2 0 0 9 8 2 0 1 5 0 0 0 0 0 0 0 0 0 0 0 0 0	2.0 80000.0 2.0 0.0 0.0 0.0 0.0 0.0 101.0 0.0 0.0

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$\begin{array}{c} 1 & 0 & 0 \\ 1 & 3 & 0 \\ 1 & 4 & 0 \\ 1 & 4 & 0 \\ 1 & 4 & 0 \\ 1 & 4 & 0 \\ 1 & 4 & 0 \\ 1 & 4 & 0 \\ 1 & 4 & 0 \\ 1 & 4 & 0 \\ 1 & 4 & 0 \\ 1 & 1 \\ 1 & 0 \\ 1 &$	12.0 14.0 15.0 20.0 1.0 12.0 15.0 16.0 18.0 0.08 2.0 0.0 0.0 0.0 0.0 0.0 0.0 0.0	24.0 0.0 2.0 0.0 24.0 0.0 17.0 1.0 0.0 50.0 0.0 0.0 0.0 0.0 0.0	2.0 0.0 40.0 1.0 1.0 0.0 0.0 0.0 0.0 0.0		

RUN WITH AVERAGE FLOW CONDITIONS -1969

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EN D ***(OF CON	L OOP Verged	6		
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END ***	OF CON	L OOP VERGED	2		
END ###	O F C O N'	L OOP VERGEI	102		
\$\$\$	\$\$	FINAL	. OUTPU	T STREAMS	FOR MODULE
6	587	25.000 57.355 13.547 2.127 6.194 0.001	7	8.746 0.000 7.653 0.000 .211 8.401	65876.308 10.050 5.894 0.000 95.515 20.892

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25.000

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MIXER1

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65876.308

FINAL OUTPUT STREAMS FOR MODULE

8.746

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25.042 6.406 0.000

0.000

0.000 5.269

25.042

CFLOR1

\$\$\$\$\$

\$\$ \$ \$ \$

1.50014.3241.517

0.002

0.000

1.500

658757.355 13.547 2.127 6.194 0.000	0.000 7.653 0.000 .211 8.401	10.050 5.894 0.000 95.515 20.892		6.406 0.000 0.000 0.000 5.269	14.324 1.517 002 0.000 0.000	•	· · ·	
\$\$\$\$\$ FINAL	OUTPUT STREAMS	FOR MODULE	16	- MIXER1	\$\$ \$ \$ \$			
$\begin{array}{r} 12.000\\660000.000\\197.538\\40.920\\36.300\\0.000\end{array}$	7.000 2.640 128.898 0.000 0.000 15.180	66008.582 101.640 31.680 11.880 66.000 .330		25.000 40.260 36.960 1.980 0.000 3.960	5.000 142.230 6.600 3.960 0.000 0.000			
$\begin{array}{c} 1 & 0 & 0 & 0 \\ 0 & 0 & 0 & 0 \\ 0 & 0 & 0$	7 • 0 0 0 0 • 0 0 0			25.000 0.000 0.000 0.000 0.000 0.000 0.000	5 • 0 0 0 0 • 0 0 0			
19.000 0.000 0.000 0.000 0.000 0.000 0.000	7 • 0 0 0 0 • 0 0 0			25.000 0.000 0.000 0.000 0.000 0.000 0.000	5.000 0.000 0.000 0.000 0.000 0.000 0.000			· · · · · · ·
\$\$\$\$\$ FINAL	OUTPUT STREAMS	FOR MODULE	10	- GRIT1	£\$ £ £ £		•	
$\begin{array}{r} 13 \cdot 000 \\ 660 \cdot 000 \cdot 000 \\ 194 \cdot 176 \\ 40 \cdot 920 \\ 36 \cdot 300 \\ 0 \cdot 000 \end{array}$	7.000 2.640 128.898 0.000 0.000 15.180	66008.582 101.640 28.318 11.880 66.000 .330		25.000 40.260 36.960 1.980 0.000 3.960	4.500 142.230 6.600 3.960 0.000 0.000			
\$\$\$\$ FINAL	OUTPUT STREAMS	FOR MODULE	1	- PRISET1	25222			•
2.000 667861.462 102.242 16.872 36.319 0.000	8.992 0.000 49.809 0.000 23.570	66790.415 64.035 15.615 1.212 96.833 .617		25.042 40.316 36.818 1.971 0.000 5.830	3.500 100.662 3.663 3.995 0.000 0.000			
9.000 2800.000 183.813 41.644 .152 0.000	8.992 0.000 119.376 0.000 .000 .099	287.893 57.015 64.283 0.000 .406 .003		25.042 .169 .154 .008 0.000 .024	3.500 50.303 15.194 .017 0.000 0.000		 	232
10.000 276.080 22.957 .007 .015 0.000	8.992 0.000 22.935 0.000 .000 .010	28.745 10.938 006 10.912 040 000		25.042 017 015 001 0.000 .002	3.500 13.136 002 002 0.000 0.000			.

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.\$\$\$\$\$ FINAL OUTPUT STREAMS FOR MODULE 2 - MIXER1	\$\$ \$ \$ \$
3.000 8.988 66790.415 25.042 667861.462 0.000 64.035 40.316 102.242 49.809 15.615 36.8 <u>1</u> 8	3.500 100.662 3.663 3.995
16.872 0.000 1.212 1.971 36.319 .003 96.833 0.000	3.995 0.000
0.000 23.570 .617 5.830	0.000
\$\$\$\$\$ FINAL OUTPUT STREAMS FOR MODULE 5 - CCNT02	\$\$ \$ \$ \$
$\begin{array}{cccccccccccccccccccccccccccccccccccc$	4.500 314.603
	185.441
1.324 0.000 0.000 1.796 4.465 1.126	0000.0
7.000 8.746 911.871 25.042 9073.592 0.000 28.801 .088	4.500
106.741 60.297 46.443 0.000 16.761 0.000 0.000 0.000	4.500 20.276 11.952 .000
085 003 1.316 0.000 0.000 .116 .288 .073	0.000
\$\$\$\$\$ FINAL OUTPUT STREAMS FOR MODULE 3 - ACTSL1	\$\$ \$ \$ \$
4.000 8.746 80940.066 25.042 808648.472 3.790 486.895 7.861	2.500
808648.472 3.790 486.895 7.861 1779.025 1005.970 773.055 0.000 278.858 0.000 1.207 0.000 7.600 .259 117.246 0.000	2.500 350.522 198.969
7.600 .259 117.246 0.000 0.000 10.313 25.646 6.468	\$00. 000.0 000.0
\$\$\$\$\$ FINAL OUTPUT STREAMS FOR MODULE 4 - SECLAR1	\$\$ \$ \$ \$ \$
5.000 8.746 65876.306 25.042 658757.336 0.000 10.048 6.404	1.500
13.547 (.652 5.895 0.000	1.500 14.319 1.517 .002
2.126 0.000 0.000 0.000 6.191 .211 95.513 0.000 0.000 8.402 20.892 5.269	
149860,604 0,000 475,640 1,457	1.500 334.754 197.452
	000.00
0.000 1.911 4.753 1.199	0.000
11.000 8.746 3.179 25.042 30.533 0.000 1.207 .000 2.535 2.535 .000 0.000	1.500
	•000
•000 •000 •004 0•000 •000 •000 •001 •000	0 • 0 0 0 0 • 0 0 0
\$\$\$\$ FINAL OUTPUT STREAMS FOR MODULE 6 - SETSP1	\$\$ \$ \$ \$
6.000 8.746 15060.581 25.042 149860.604 0.000 475.640 1.457	4.500 334.754

1762.044 276.731 1.408 0.00(995.784 0.000 048 1.911	767.160 0.000 21.728 4.753	0.000 U.000 0.000 1.199	197.452 .000 0.000 0.000
\$\$\$\$\$ FINAL	OUTPUT STREAMS	FOR MODULE	5 - CCNT02	\$\$ \$ 9 9
8 • 000 140787 • 128 1656 • 204 259 • 976 1 • 323 0 • 000	8.746 0.000 935.493 0.000 .045 1.796	14148.721 446.841 720.711 0.000 20.413 4.465	25.042 1.369 0.000 0.000 0.000 1.126	4.500 314.486 185.497 .000 0.000 0.000
7.000 9073.475 106.739 16.755 .085 0.000	$\begin{array}{r} 8.746 \\ 0.000 \\ 60.291 \\ 0.000 \\ .003 \\ .116 \end{array}$	911.859 28.798 46.449 0.000 1.316 .288	25.042 .088 0.000 0.000 0.000 .073	4.500 20.268 11.955 .000 0.000 0.000
\$\$\$\$\$ FINAL	OUTPUT STREAMS	FOR MODULE	12 - MIXER1	25299
14.000 3106.613 209.305 41.651 .168 0.000	8.945 U.U00 144.846 0.000 .000 .109	319.816 69.160 64.290 12.119 .450 .004	25.042 186 170 009 0.000 .027	1.500 64.888 15.196 .018 0.000 0.000
\$\$\$\$ FINAL	OUTPUT STREAMS	FOR MODULE	17 - SETSP1	\$\$ \$ \$ \$
$\begin{array}{r} 14.000\\ 3106.613\\ 209.305\\ 41.651\\ .168\\ 0.000 \end{array}$	8.945 0.000 144.846 0.000 .000 .109	319.816 69.160 64.290 12.119 .450 .004	25.042 186 170 009 0.000 .027	10.500 64.888 15.196 018 0.000 0.000
\$\$\$\$ FINAL	OUTPUT STREAMS	FOR MODULE	13 - ANDIG1	\$\$ \$ \$ \$
15.000 3106.613 93.557 8.367 .168 0.000	9.000 0.000 29.098 0.000 12.786	314.448 14.065 64.290 2.435 45.717 .004	40.000 205 170 0022 0.000 2.783	8.500 13.362 3.053 .042 0.000 0.000
\$\$\$\$ FINAL	OUTPUT STREAMS	FOR MODULE	14 - ANDIG1	\$\$ \$ \$ \$ \$
16.000 1863.968 8.095 .841 .101 0.000	9.000 0.000 2.925 0.000 .000 8.383	186.728 1.546 5.143 29.962 .002	40.000 154 027 000 0.000 1.824	6.500 1.595 .053 0.000 0.000
17.000 1242.645 74.645 4.416	9.000 0.000 15.356 0.000	127.218 7.415 59.146 1.285	40.000 102 142 001	6.500 7.036 1.611 .035

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	\$\$\$\$ FINAL	OUTPUT STREAM	S FOR MODULE	18	- MIXER1	\$\$ \$ \$ \$		t	
	25.000 658757.336 13.547 2.126 6.191 0.000	8.746 0.010 7.652 0.000 .211 8.402	65876.306 10.048 5.895 0.000 95.513 20.892		25.042 6.404 0.000 0.000 0.000 5.269	1.500 14.319 1.517 0002 0.000 0.000			
	\$\$\$\$ FINAL	OUTPUT STREAM	S FOR MODULE	11	- CHLOR1	£\$ £ £ ₹ ₹	•		
··· ·	25.000 658757.336 13.547 2.126 6.191 0.000	8.746 0.000 7.652 0.000 .211 8.402	65876.306 10.048 5.895 0.000 95.513 20.892	٠	25.042 6.404 0.000 0.000 0.000 5.269	1.500 14.519 1.517 0.000 0.000			

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SUMMARY REPORT ON PLANT OPERATIONS - CASE

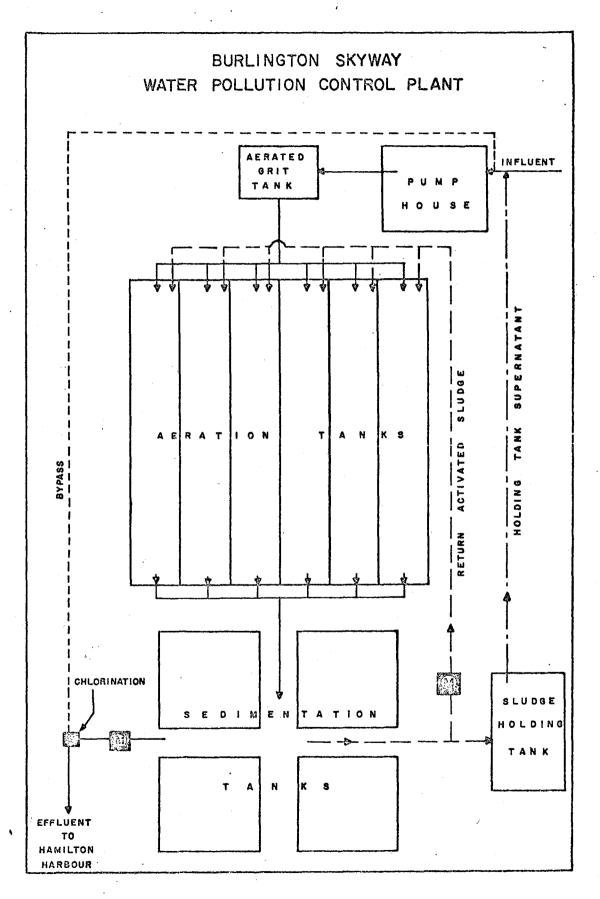
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STREAM	NUMBER	FLOW	DO	TOC	DOC	EOD	SS	VSS	
RAW FEED	18	66008.58	4.00	153.98	61.00	215.47	299.26	195.27	
INPUT TO PRI. CLAR.	13	66008.58	4.00	153.98	61.00	215.47	294.17	195.27	
EFF. OF PRI. CLAR.	2	66790.41	0.00	95.87	60.37	150.71	153.08	74.58	
UNDERFLOW PRI CLAR	9	287.89	0.00	19804.14	60.37	17472.90	63847.75	41465.40	
OUTPUT OF AER TANKS	4	80940.07	4.69	601.55	9.72	433.06	2197.95	1242.86	
EFF. OF SEC. CLAR.	5	65876.31	0.00	15.25	9.72	21.74	20.56	11.62	
UNCERFLOW SEC CLAR	6	15060.58	0.00	3158.18	9.72	2222.72	11705.68	6611.85	
RETURN ACT. SLUDGE	8	14148.72	0.00	3158.18	9.72	2222.72	11705.68	6611.85	
WASTE ACT. SLUDGE	7	911.86	0.00	3158.18	9.72	2222.72	11705.68	6611.85	
INPUT TO DIGESTER	14	319.82	0.00	21624.99	59,87	20289.09	65445.39	45290.35	
DIGESTER SUPERNATE	16	186.73	0.00	828.20	82.43	854.23	4335.31	1566.42	
DIGESTED SLUDGE	17	127.22	0.00	5828.41	82.43	5530.88	58674.73	12070.57	

APPENDIX J

SIMULATION DATA AND RESULTS FOR THE SKYWAY,

BURLINGTON, SEWAGE TREATMENT PLANT



DESIGN DATA

PROJECT NO.	2-0105-62	DESIGN FLOW	6 mgd
BOD - Raw Sewage	200 mg/l	SS - Raw Sewage	200 mg/l

PRETREATMENT

Coarse Screening

Type: Link Belt, mechanically-cleaned Size: $2\frac{1}{2}$ " space

Lift Station Pumps (@ 30' tdh)

2 @ 6 mgd each; 2 @ 5 mgd each

Screening

Type: Link Belt, mechanically-cleaned Size: 1" space

Grit Removal

Type: Air degritter Size: One 20.3' x 25' x 14.2' Retention: 20.8 min

SECONDARY TREATMENT

Aeration Tanks

Type: Diffused air, single pass Size: Six 270' x 27' x 15' (660, 300 ft. or 4.12 mil gal) Retention: 16.5 hr Loading: 9.5 lb BOD/1000 ft³

Air Supply

Type: Hoffman multi-stage

centrifugal blowers Size: 3 - 4000 cfm (max); 1-6000 cfm (max) Diffusers

Type: Saran-covered flexofusers Spacing: 250 tubes per tank

Secondary Sedimentation

Type: Eimco Size: Four 60' x 60' x 12' swd (538,000 gal) (only two completed) Retention: 4.1 hours Loading: Surface, 434 gal/ft²/day Weir, 7,100 gal/ft/day

CHLORINATION

Type: Fischer & Porter Size: Two 2000 lb/day

Chlorine Contact Chamber

– nil

- chlorination in outfall

OUTFALL

- to Hamilton Harbour

SLUDGF HANDLING

Type: Thickening tank, decanted Size: One 20' dia x 9' depth (2830 ft³ or 1760 gal) i - 3

0.0 14.0 1.0 5.0 12.0	ON SKYWAY ST 1.0 0.0 25.0 2.0 6.0 9.0	P 1.0 0.0 1.0 3.0 7.0 10.0	1.0 0.0 7.0 8.0 24.0	0.0 0.0 4.0 11.0
-25.0 1.0 1.0 1.0 1.0 1.0 1.0	1•0 1•0 1•0 1•0 1•0	1 • 0 1 • 0 1 • 0 1 • 0 1 • 0	1 • 0 1 • 0 4 • 0 1 • 0 1 • 0	1 • 0 1 • 0 1 • 0 1 • 0 1 • 0
3.0 1.0 1400000.0 0.0 18.0 55.0 0.0 7.0 998359.7 0.0 1214.9 3.47 0.0 1214.9 3.47 0.0 13.0 0.0 13.0 0.0 13.0	7 • 4 4 • 0 0 • 0 0 • 0 20 • 0 8 • 5 8 5 0 • 0 0 •	$140000 \cdot 0$ $40 \cdot 0$ $5 \cdot 0$ $100 \cdot 0$ $1 \cdot 0$ $100227 \cdot 0$ $0 \cdot 0$ $5602 \cdot 0$ $99 \cdot 84$ $34 \cdot 54$ $1371 \cdot 4$ $0 \cdot 0$ $900 \cdot 0$ $0 \cdot 0$ $11 \cdot 0$ $60 \cdot 0$	25 • 0 0 • 0 60 • 0 8 • 0 0 • 0 6 • 0 25 • 0 0 • 0 0 • 0 0 • 0 6 • 15 25 • 0 0 • 0 6 • 15 25 • 0 0 • 0 0 • 0 0 • 0 1 • 0 1 • 0	5 0 0 0 0 0 0 0 0 0 0 0 0 0
1.0 2.0 1.0 2.0 1.0 1.0 1.0 1.0 1.0 1.0 1.0 1.0 2.0 1.0 2.0 1.0 2.0 1.0 2.0 0.0 2.0 0.0 2.0 0.0 0.0 0.0 0.0 0	$ \begin{array}{c} 1 \cdot 0 \\ 1 \cdot 0 \\ 2 \cdot 0 \\ 0 \cdot 0 \\ 15 \cdot 0 \\ 2 \cdot 0 \\ 3 \cdot 0 \\ 11 \cdot 0 \\ 3 \cdot 0 \\ 11 \cdot 0 \\ 3 \cdot 0 \\ 4 \cdot 0 \\ 7 \cdot 0 \\ 8 \cdot 0 \\ 5 \cdot 0 \\ 27 \cdot 0 \\ 0 \cdot 0 \\ 0 \cdot 0 \\ 0 \cdot 0 \end{array} $	$17 \cdot 0$ $10 \cdot 0$ $12 \cdot 0$ $15 \cdot 0$ $0 \cdot 0$ $15 \cdot 0$ $0 \cdot 0$ $46 \cdot 0$ $46 \cdot 0$ $46 \cdot 0$ $46 \cdot 0$ $15 \cdot 2$ $2 \cdot 2$ $0 \cdot 0 \cdot 0$ $15 \cdot 2$ $2 \cdot 2$ $0 \cdot 0 \cdot 0$ $0 \cdot 0$	0.0 0.0 0.0 0.0 0.0 0.0 0.0 0.0 0.0 0.0	0.0 0.0 0.0 0.5 0.0 0.5 0.0 0.5 0.0 0.5 0.0 0.0
240000.0 5.0 1.0 3.0 1.25000.0 6.0 1.0 1.0 1.0 2.0 8.0 1.0 2.0 8.0 1.0 2.0 9.0 2.0 1.0 1.0 1.0 1.0 1.0 1.0 1.0 1	$9 \cdot 0$ $5 \cdot 0$ $6 \cdot 0$ $0 \cdot 4$ $4 \cdot 0$ $7 \cdot 0$ $17 \cdot 0$ $7 \cdot 0$ $8 \cdot 0$ $5 \cdot 0$ $10 \cdot 0$ $1 \cdot 0$ $5 \cdot 0$ $10 \cdot 0$ $1 \cdot 0$ $2 \cdot 0$ $1 \cdot 0$ $1 \cdot 0$ $2 \cdot 0$ $1 \cdot 0$	$22 \cdot 0$ $0 \cdot 0$ $7 \cdot 0$ $36 \cdot 0 \cdot 0$ $15 \cdot 0$ $0 \cdot 0$ $15 \cdot 0$ $0 \cdot 0$ $9 \cdot 0$ $15 \cdot 0$ $0 \cdot 0$ $15 \cdot 0$ $0 \cdot 0$ $11 \cdot 0$ $16 \cdot 0$ $12 \cdot 0$ $0 \cdot 0$ $15 \cdot 0$ $0 \cdot 0$	2.0 0.0 13.0 80000.0 0.0 0.0 0.0 0.0 0.0 0.0 0.0 0	1.0 0.0 2.0 4.0 0.0 0.0 0.0 0.0 0.0 0.0 0.0 0.0 0

$1 \cdot 0 \\ 11 \cdot 0 \\ 0 \cdot 0 \\ 9 \cdot 0 \\ 1 \cdot 0 \\ 0 \cdot 0 \\ 1 \cdot 0 \\ 0 \cdot 0 \\ 1 \cdot 0 \\ 0 \cdot 0 \\ 0 \cdot 0 \\ 1 \cdot 0 \\ 0 \\$	25.0 2.0 0.0 10.0 0.0 0.0 0.0 0.0 0.0	0.0 50.0 0.0 5.0 0.0 0.0 0.0 0.0	0.0 101.0 0.0 0.001 0.0 0.0 0.0 0	24 10.0 0.0 0.0 1.0 0.0 0.0 0.0 0.	41
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RUN WITH AVERAGE FLOW CONDITIONS IN 1969

END OF	LCOP	1				
END OF	L 00P	2				
END OF	LCOP	3				
END OF	LCOP	4				
'END OF ***CON	L O O P V E R G E D	5			ŀ	•
END OF	L 00P	101				
END OF	LCOP	1				
END OF ***CON	L COP Verged	2				
END OF	LOOP	102				
END OF	LCOP	1				
END OF ***Con	L COP Ve Rge D	2				
END OF	L 0 0 P	103				
END OF	LCOP	1				••
END OF ***CON	L COP VERGED	2			1.16	
END OF	L C O P V E R G E D	104				
1\$\$\$\$	FINAL	OUTPUT	STREAMS	FOR MODULE	9	- MIXER1
13907	25.000 59.785 14.364 1.822 4.845 0.000	·	8.582 0.000 5.890 0.000 1.271 .075	139076.568 8.922 8.474 0.000 139.076 48.069		25.000 6.117 8.000 0.000 0.000 8.550
1\$\$\$\$	FINAL	OUTPUT	STREAMS	FOR MODULE	10	- CHLOR1
13907	25.000 59.785 14.364 1.822 4.845 0.000		8.582 0.000 5.890 0.000 1.271 .075	139076.568 8.922 8.474 0.000 139.076 48.069		25.000 6.117 0.000 0.000 0.000 8.550

\$\$\$\$\$ FINAL OUTPUT STREAMS FOR MODULE

\$\$ \$ \$ \$ 1.500 11.394 .983 .001

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1.500 11.394 .983

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2 • 000 1413731 • 237 - 279 • 356 27 • 960 77 • 048 0 • 000	7.703 5.600 126.522 0.000 .013 28.001	141384.741 147.109 68.835 7.000 141.373 1.875		25.000 86.860 84.000 11.200 0.000 8.484	5.000 220.403 14.089 5.800 0.000 0.000
12.000 0.000 0.000 0.000 0.000 0.000 0.000 0.000	7 • 7 0 3 0 • 0 0 0 0 • 0 0 0 0 • 0 0 0 0 • 0 0 0 0 • 0 0 0	$\begin{array}{c} 0 & 0 & 0 & 0 \\ 0 & 0 & 0 & 0 \\ 0 & 0 &$		$\begin{array}{c} 25 & 0 & 0 & 0 \\ 0 & 0 & 0 & 0 \\ 0 & 0 & 0$	5 • 0 0 0 0 • 0 0 0
\$\$\$\$\$ FINAL	OUTPUT STREAMS	FOR MODULE	2	- SCREEN1	\$\$ \$ \$ \$
3.000 1413731.237 266.632 21.900 77.048 0.000	7.703 5.600 113.797 0.000 .013 28.001	141384.159 141.050 68.835 7.000 141.373 1.875		25.000 86.860 84.000 11.200 0.000 8.484	4.500 213.131 14.089 9.800 0.000 0.000
\$\$\$\$\$ FINAL	OUTPUT STREAMS	FOR MODULE	3	- GRIT1	\$\$ \$ \$ \$
4.000 1413731.237 251.669 21.900 77.048 0.000	7.703 5.600 113.797 0.000 .013 28.001	141384.159 141.050 53.872 7.000 141.373 1.875		25.000 86.860 84.000 11.200 0.000 8.484	$\begin{array}{r} 4.000\\ 213.131\\ 14.089\\ 5.800\\ 0.000\\ 0.000\\ 0.000\end{array}$
\$\$\$\$ FINAL	OUTPUT STREAMS	FOR MODULE	7	- CCNT02	2223
8 000 975565 103 9310 721 1181 065 3 398 0 000	8.585 0.000 3818.087 0.000 .892 .053	97938.673 1822.428 5492.634 0.000 97.556 33.719		25.000 4.291 0.000 0.000 0.000 5.997	5.500 1423.737 637.072 .001 0.000 0.000
9 000 22795 252 217 556 27 597 079 0 000	8.585 0.000 89.214 0.000 .021 .001	2288.455 42.583 128.342 0.000 2.280 .788		25.000 100 0.000 0.000 0.000 .140	5.500 33.267 14.886 000 0.000 0.000
\$\$\$\$ FINAL	OUTPUT STREAMS	FOR MODULE	4	- ACTSL1	22222
5.000 2389296.340 9557.206 1210.113 8.324 0.000	8.585 17.105 3926.700 0.000 2.183 .129	239322.029 1880.367 5630.506 6.934 238.930 82.578		25.000 10.510 0.000 0.000 0.000 14.687	2.500 1476.276 652.810 .002 0.000 0.000
\$\$\$\$\$ FINAL	OUT PUT STREAMS	FOR MODULE	5	- SECLAR1	2222
6.000 1390760.532	8.585 0.000	139076.643 8.922	و میگرین میں از میں ا	25.000 6.117	1.500 11.395

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	14.364 1.822 4.845 0.000	5.889 0.000 1.271 .075	8.475 0.000 139.076 48.067		0.000 0.000 0.000 8.549	- 983 - 001 0-000 0-000		
g	7.000 360.390 528.279 208.291 3.478 0.000	8.585 0.000 3906.250 0.000 .912 .054	$\begin{array}{c} 100227.123\\ 1864.510\\ 5622.030\\ 0.000\\ 99.836\\ 34.505 \end{array}$		25.000 4.391 0.000 0.000 0.000 6.137	1.500 1456.560 651.828 .001 0.000 0.000		a an ann an Anna Anna Anna Anna Anna An
• •	13.000 175.418 14.562 .000 .001 0.000	8.585 0.090 14.561 0.000 .000 .000	18.263 6.935 001 6.934 018 006	•	25.000 001 0.000 0.000 0.000 0.000	1.500 8.322 000 0.000 0.000 0.000	•	
\$\$\$ \$\$	FINAL OU	JTPUT STREAMS	FOR MODULE	6	- SETSP1	35 2 5 2		
9	7.000 360.390 528.279 208.291 3.478 0.00	8.585 0.000 3906.250 0.000 .912 .054	100227.123 1864.510 5622.030 0.000 99.836 34.505		25.000 4.391 0.000 0.000 0.000 6.137	5.500 1456.560 651.828 001 0.000 0.000		n n san sa
\$\$\$ \$\$	FINAL OU	JTPUT STREAMS	FOR MODULE	7	- CCNT02	25212		
91	8.000 561.112 310.685 180.698 3.399 0.000	8.585 0.000 3817.044 0.000 .891 .053	97938.264 1821.931 5493.641 0.000 97.556 33.717		25.000 4.291 0.000 0.000 0.000 5.997	5.500 1423.297 636.942 .001 0.000 0.000	•	, par - e - e - e - e - e - e - e - e - e -
22	9.000 799.278 217.595 27.593 .079 0.000	8.585 0.000 89.206 0.000 .021 .001	2288.859 42.579 128.389 0.000 2.280 .788		25.000 100 0.000 0.000 0.000 .000 .140	5.500 33.263 14.886 000 0.000 0.000		, e ang an
22323	FINAL OU	TPUT STREAMS	FOR MODULE	8	- SEPA01	\$\$ 1 1 1		
13	10.000 733.153 21.759 2.759 .048 0.000	8.585 0.000 8.921 0.000 .013 .001	1374.208 4.308 12.639 0.000 1.373 .475		25.000 060 0.000 0.000 0.000 .000	5.500 3.402 1.489 000 0.000 0.000		24 4
9	11.000 066.125 195.835 24.834	8.585 0.000 80.285 0.000	914.650 38.271 115.550 0.000		25.000 040 0.000 0.000	5.500 29.861 13.397	•	a too anar oo a
	•032 0•000	•008 •000	• 907 • 313		0.000 .056	0.00000000000000000000000000000000000		

	1.500	25.000	139076.643	8.582	25.000
	11.395	6.117	8.922	0.000	1390760.532
	.983	0.000	8.475	5.889	14.364
	.001	0.000	0.000	0.000	1.822
	0.000	0.000	139.076	1.271	4.845
	0.000	8.549	48.067	.075	0.000
	\$\$ \$ \$?	LO - CHLOR1	FOR MODULE	OUTPUT STREAMS	\$\$\$\$\$ FINAL
· · · ·	1.500	25.000	139076.643	8.582	25.000
	11.395	6.117	8.922	0.000	1390760.532
	.983	0.000	8.475	5.885	14.364
	.001	0.000	0.000	0.000	1.822
	0.000	0.000	139.076	1.271	4.845
	0.000	8.549	48.067	.075	0.00

 $\mathbf{p}(x) = \mathbf{p}(x) + \mathbf{p}$

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SUMMARY REPORT ON PLANT OPERATIONS - CASE

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STREAM	NUMBER	FLOW	DO	TOC	DOC	EOC	SS	VSS	
RAW FEED	4 44	0010.72	4.00	101.99	62.00	154.99	183,99	83.99	
OUTPUT OF AER TANKS		9322.03	4.00 7.1 6	785.71	4.40	616.86	3993.45	1640.76	
EFF. OF SEC. CLAR.	6 13	9076.64	0.00	6.41	4.40	8.19	10.33	4.23	and a
UNDERFLOW SEC CLAR	7 10	0227.12	0.00	1860.29	4.40	1453.26	9506.69	3897.40	nan - Northe
RETURN ACT. SLUDGE	89	7938.26	0.00	1860.29	4.40	1453.26	9506.69	3897.40	
WASTE ACT. SLUDGE	9	2288.86	0.06	1860.29	4.40	1453.26	9506.69	3897.40	on the March and The
THICKENED SLUDGE	. 11	914.65	0.00	4184.22	4.40	3264.73	21410.93	8777.71	
PLANT BYFASS	12	0.00	0.00	0.00	0.00	0.00	0.00	0.00	وروب والمحمول المحمول

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APPENDIX K

Computer Listings

LISTING OF THE GEMCS EXECUTIVE

PROGRAM TST (INPUT, OUTPUT, TAPE5=INPUT, TAPE6=OUTPUT) GEMCS/6400 **** PRINTING CONTROL CHARACTERS ***** KPRNT(1)=1 CAUSES PRINTING OF NCONT,LLST,NS KPRNT(1)=0 SUPPRESSES ABOVE PRINTING KPRNT(2)=1 CAUSES PRINTING OF INITIAL STREAMS KPRNT(2)=0 SUPPRESSES ABOVE PRINTING KPRNT(3)=1 CAUSES PRINTING OF MODULES SETS KPRNT(3)=0 SUPPRESSES ABOVE PRINTING KPRNT(4)=1 CAUSES PRINTING OF NPOINT KPRNT(4)=0 SUPPRESSES ABOVE PRINTING KPRNT(5)=1 CAUSES PRINTING OF SN TABLE ON ENTERING LOOP KPRNT(5)=0 SUPPRESSES ABOVE PRINTING ISP =1 PRINTS BOTH THE INPUT STREAMS AND OUTPUT STREAMS A MODULE - CAN BE SET BY HAVINGMSN NEGATIVE IN DATA SET ISP=0 SUPPRESSES ABOVE PRINTING - DONE AUTOMATICALLY BY AFTER TWO LOOPS STREAMS LEAVING GEMCS TWO LOOPS AFTER COMMON DECLARATIONS - JUNE 8 ***** **** LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10) COMMON IS, NE, JJ, LOOP, NIN, NOUT, MŚN, ISP, NC, III, NCALC, NOCOMP, NSR MODE, NPLNT, LOOPC SN (25, 30), EEN (600), NPOINT (25, 2) COMMON COMMON COMMON DIMENSION NAME(20) NAME(1)=8H MIXER1 NAME(2)=8H CONTL1 NAME(3)=8H SETST1 NAME(4)=8H SETSP1 NAME(4)=8H SETSP1 NAME(5)=8H SEPA01 NAME(6)=8H PRISET1 NAME(6)=8H PRISET1 NAME(7)=8H ACTSL1 NAME(7)=8H ACTSL1 NAME(9)=8H SECLAR1 NAME(10)=8H TRICK1 NAME(10)=8H TRICK1 NAME(11)=8H GRIT1 NAME(12)=8H ANDIG1 NAME(13)=8H COST1 NAME(13)=8H COST1 NAME(16)=8H REPT01 NAME(16)=8H REPT01 NAME(16)=8H CONT02 NAME(18)=8H CONV01 NAME(19)=8H CASET SCREEN1 REPT01 CONVUI CASET NAME(19)=8H NAME (20) = 8H THICK1 III=25CALL DLOAD1 LOOPC=1 WRITE(6,196)_LOOPC READ(5,198) TITLE WRITE(6,200) TITL TITLE CALCULATING EQUIPMENT IN CALCULATION ORDER LIST 5 USED AS AN INDICATOR FOR A STRAIGHT THROUGH CALCULATION FOR A RECYCLE OPERATION IK IS USED İK=1 IK=0 IK=0 IF(LOOP.EQ.999) IK=1 CONTINUE NC=1MODE=0 KTEST=0 LOOP=1IF(IK.EQ.0) GO TO 98

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MODE=1 KTEST=1 LOOP=999 98 C C 102 CONTINÚÉ READ EN VECTOR IF (KPRNT(5)) 108,108,104 CONTINUE 104 (6,182) LOOP 6 IKE=1,III WRITE DO 106 WRITE (6,194) (SN(IKE,J),J=1,JJ) CONTINUE 106 C 108 NE=LLST(NC) MM=NE CALL DISKIO (1,MM) NN=EN(3)+.001 NIN=EN(6)+•001 NOUT=EN(11)+•001 NTYPE=ABS(EN(2))+•001 KSW=0 IF (EN(2).LE.0.) KSW=1 IF (LOOP.GT.2) ISP=0 IF(LOOP.LE. 2) ISP=1 CCC FINDING INPUT STREAMS IF (NIN.EQ.U) GO TO 128 DO 122 I=1.NIN S=EN(I+6) CALL STREAM (S) IF (IS-III) 114,114,110 IF (15-111) 1 CONTINUE SI(1,1)=S DO 112 J=2,JJ SI(1,J)=0 CONTINUE 110 112 GO TO 122 DO 116 J=1,JJ SI(I,J)=SN(IS,J) CONTINUE IF (INT(S+.001)) 114 116 (INT(S+.001)) 122,122,118 CONTINUE M=S+.001 IF (NS(M)-6) 122,120,120 118 120 CONTINUE SN(IS,1)=0. CONTINUE 122 IF (ISP) 128,128,124 WRITE (6,184) NE ,NAME(NTYPE) 124 DO 126 IKE=1,NIN WRITE(6,194) WRITE (6 CONTINUE (6,194) (SI(IKE,J),J=1,JJ) 126 C 128 C C - CONTINUE CALL MODULE (NTYPE) STORING OUTPUT STREAMS AND PRINTING IF(NOUT) 166,166,136 CONTINUE DO 150 I=1,NOUT S=EN(I+11) SO(I,1)=S M=S+.001 IF(NS(M)-6) 140 136 ٩ 140,150,140 CALL STREAM IF (IS-III) CALL STREAM IF (IS-III) 140 (-S)146,146,142 (O.) 142 146,146,144

WRITE (6,190) GO TO 150 DO 148 J=1,JJ SN(IS,J)=SO(I,J) CONTINUE 250 144 146 148 150 C CONTINUE IF (ISP) 158,158,154 WRITE (6,186) NE , NAME(NTYPE) DO 156 IKE=1,NOUT ·IF 154 WRITE (6,194) WRITE (6,194) (SO(IKE,J),J=1,JJ) CONTINUE 156 Č 158 IF (LOOP-999) 166,160,160 IF (KSW.NE.U) GO TO 166 WRITE(6,188) NE, NAME(NTYPE) DO 164 IKE=1,NOUT WRITE(6,194) WRITE (6,194) (SO(IKE,J),J=1,JJ) 160 164 CONTINUE 166 NC=NC+1IF (NC-NCALC) 102,102,168 C 168 CONTINUE IF (LOOP-999) 170,172,172 CONTINUE 170 NC=0 TO 166 (KTEST) 174,174,176 GO TO IF (KTE NC=0 KTEST=1 172 174 MODE=1 GO TO 166 LOOPC=LOOPC+1 IF (LOOPC-NPLNT) 178,178,180 WRITE(6,196) LOOPC CALL DLOAD3 GO TO 100 176 178 с 180 WRITE (6,192) STOP С (1H0,32H\$\$\$\$\$ (1H0,31H\$\$\$\$\$ SN TABLE ON ENTERING LOOP ,15,7H INPUT STREAMS FOR MODULE ,15,4H -182 FORMAT \$\$\$\$\$ } 184 FORMAT \$\$\$\$\$) 1A8,7H FORMAT (1H0,32H\$\$\$\$\$ 27H \$\$\$\$\$) OUTPUT STREAMS FOR MODULE , 15,4H - , A8, 186 7H \$\$\$\$\$) FORMAT (1H0, A8,7H \$\$\$\$ 2 (1H0,38H\$\$\$\$ \$\$\$\$\$) FINAL OUTPUT STREAMS FOR MODULE , 15,4H 188 3 (1H0,25H***** 190 FORMAT ERROR IN SN **** FORMAT (1H0,38H*********** END OF FORMAT (1H ,10F12.3) FORMAT(1H1,28H\$ 192 194 END OF EXECUTION ****** 1 5555555555555 CASE 196 ,14,24H 1\$\$\$\$\$\$\$ FORMAT(10A8) FORMAT(1H0,10A8) 198 ŹÓÖ END

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SUBROUTINE STREAM (S)

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COMMON LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10) COMMON IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR COMMON MODE,NPLNT,LOOPC COMMON SN(25,30),EEN(600),NPOINT(25,2) DO 2 IS=1,III IF (ABS(SN(IS,1)-ABS(S))-.001) 4,2,2 CONTINUE IS=IS+1 RETURN END

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SUBROUTINE DISKIO (IPNT, MM)
COMMON LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10)
COMMON IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR
COMMON MODE,NPLNT,LOOPC
COMMON SN(25,30),EEN(600),NPOINT(25,2)
THIS SUBROUTINE SIMULATES DISKIO ON360/30
IF IPNT=1. READS FROM MODULE SETS TABLE
IF IPN=2. WRITES ONTO MODULE SETS TABLE
IF IPNT=1. REA
IF IPN=2. WRITE
MQ=NPOINT(MM,1)
ML=NPOINT(MM,2)
IF (IPNT-1) 6,2,6
DO 4 I = 1.0 ML
EN(I) = EEN(MQ+I)
CONTINUE
GO TO 14
    (IPNT-2) 12,8,12
IF
DO 10 I=1,ML
EEN(MQ+I)=EN(I)
CONTINUE
ĞŌ
GO TO 14
WRITE (6,16) IPNT,MM
RETURN
FORMAT (1HU,22H DISKIO ERROR-IPNT,MM=,15,1X,15)
END
```

SUBROUTINE DLOAD1

COMMON.LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10) COMMON IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR COMMON MODE,NPLNT,LOOPC COMMON SN(25,30),EEN(600),NPOINT(25,2) DIMENSION TITLE (10) DIMENSION AKPRNT(10), ALLST(10) DIMENSION ANS(100) DO 2 I=1,100 EN(I)=0. CONTINUE DO 6 I=1,30 DO 4 J=1,4 SI(J,I)=0. SO(J,I)=0. CONTINUE DO 6 IK=1,III SN(IK,I)=0. CONTINUE DO 6 IK=1,1II SN(IK,I)=0. CONTINUE DO 7 I=1,25 NPOINT(I,1)=0.0 NPOINT(I,2)=0.0 NCOUNI=0 WRITE(6,79)

8 C C C	WRITE(6,81) READ (5,80) TITLE WRITE(6,82) TITLE WRITE(6,81) READ (5,100) AKPRNT DO 8 I=1,10 KPRNT(I)=AKPRNT(I) CONTINUE REPORT ON PRINTING EXPECTED
C 102 146 180 C C C C C	IF (KPRNT(1)-1) 12,10,12 WRITE (6,84) IF (KPRNT(2)-1) 16,14,16 WRITE (6,86) IF (KPRNT(3)-1) 20,18,20 WRITE (6,88) CONTINUE
	READ NO. OF MODULES IN CALCULATION ORDER, AND NO. OF CONPOMENTS
22 24 26	READ (5,100) ANCALC,ANOCOM,ANPLNT NCALC=ANCALC NOCOMP=ANOCOM NPLNT=ANPLNT IF (NPLNT.GT.C) WRITE (6,108) NPLNT IF (KPRNT(1)-1) 24,22,24 WRITE(6,95) NOCOMP WRITE(6,95) NOCOMP WRITE(6,92) NCALC CONTINUE IF (-NCALC) 28,26,26 NCALC=-NCALC LOOP=999 GO TO 30 LOOP=1
28 C C 30	READ CALCULATION ORDER
32 34	READ (5,100) (ALLST(I),I=1,NCALC) DO 32 I=1,NCALC LLST(I)=ALLST(I) CONTINUE IF (KPRNT(1)-1) 36,34,36 WRITE(6,93) WRITE (6,94) (LLST(I),I=1,NCALC) CONTINUE
36 C C C C	READ STREAM CODES,MSN IS THE MAXIMUM STREAM NUMBER, IF NEGATIVE OUTPUT STREAMS WILL BE PRINTED DURING EXECUTION
	READ (5,100) AMSN MSN=AMSN IF (-MSN) 40,38,38
38 40 42 44	MSN=-MSN ISP=1 IF (KPRNT(1)-1) 44,42,44 WRITE (6,96) MSN READ (5,100) (ANS(I),I=1,MSN) DO 46 I=1,MSN
46	NŠ(I)=ANŠ(I) CONTINUE
48 50	IF (KPRNT(1)-1) 50,48,50 WRITE (6,94) (NS(I),I=1,MSN) JJ=NOCOMP+5 READ (5,100) ANSR
52 54	NSR=ANSR IF (KPRNT(2)-1) 54,52,54 WRITE (6,98) NSR DO 60 I=1,NSR READ (5,100) (SN(I,J),J=1,JJ) CALL CONVERT(2,1,I,1,I) CALL BALNCE(1,I)
56	IF (KPRNT(2)-1) 58,56,58 WRITE (6,102) (SN(I,J),J=1,JJ) WRITE(6,102)
58	CONTINUÉ

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CONTINUE READING MODULE SETS, NOE IN NUMBER READ (5,100) ANOE NOE=ANOE IF (KPRNT(3)-1) 64,62,64 WRITE (6,104) NOE DO 72 I=1,NOE 72 I=1,NOE 66 IZ=1,100 DŌ $EN(IZ)=\overline{0}$ CONTINUE READ (5,100) (EN(N),N=1,5) NN = EN(3)NN=EN(3) NCOUNT=NCOUNT+NN READ (5,100) (EN(N),N=6,NN) IF (KPRNT(3)-1) 70,68,70 WRITE (6,102) (EN(N),N=1,NN) WRITE(6,102) MM=EN(1)+.001 NPOINT(MM,1)=NCOUNT-NN NPOINT(MM,2)=NN CALL DISKIO (2,MM) CALL DISKIO (2,MM) CONTINUE (KPRNT(4).GT.0) GO TO 74 ĨĘ GO (6,106) I=1,25 (6,90) WRITE WRITE (6,90) I,NPOINT(I,1), CONTINUE CONTINUE FORMAT STATEMENTS FOR DLOAD I,NPOINT(I,1),NPOINT(I,2) RFTURN 2\$\$\$) FORMAT(1H0,1H\$,19X,10A8,19X,1H\$) FORMAT(1H0,22HPRINTING NCONT,LLST,NS) FORMAT(1H0,22HPRINTING INITIAL STREAMS) FORMAT(1H0,20HPRINTING MODULE SETS) FORMAT(1H0,20HPRINTING MODULE SETS) FORMAT(1X,3I6) 8888999999999911110 02468 02468 FORMAT (1X,316) FORMAT (1HU,41HNUMBER OF MC FORMAT (1HU,27HCALCULATION FORMAT (1HU,27HCALCULATION FORMAT (1HU,23HNUMBER OF CC FORMAT (1HU,33HNUMBER OF S FORMAT (1HU,33HNUMBER OF S FORMAT (1HU,33HNUMBER OF S FORMAT (1HU,33HNUMBER OF MC FORMAT (1HU,29HNUMBER OF MC FORMAT (1HU,7/,20H MODULE FORMAT (1HU,7/,20H MODULE FORMAT (1HU,25HNUMBER OF S IN CALCULATION ORDER = LIST IS- ,/) MODULES N ORDER ,15) COMPONENTS = ,15) STREAM CODES READ = INITIAL STREAMS READ ,15,/) = , I5,/) SETS R LENGTH CASES MODULE READ = ,15,/) ,/) PLANT ,151 END

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SUBROUTINE MODULE(NT)

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С	GO TO (1,2,3,4,5,6,7,8,9,10,11,12,13,14,15,16,17,18,19,20),NT
C 1	CALL MIXER1
2	GO TO 50 CALL CONTL1
3	GO TO 50 CALL SETST1 GO TO 50
4	CALL SETSP1 GO TO 50
5	CALL SEPA01 GO TO 50
6	CALL PRISET1
7	GO TO 50 CALL ACTSL1 GO TO 50
8	CALL DPTR1 GO TO 50
9	CALL SECLARI GO TO 50
10	CALL TRICK1 GO TO 50
11	ČĂLL GRITI GO TO 50
12	CALL ANDIGI GO TO 50
13	CALL CHLOR1 GO TO 50
14	CALL COSTI GO TO 50
15	CALL SCREENI GO TO 50
16	CALL REPTOI GO TO 50
17	CALL CONTO2 GO TO 50
18	CALL CONVOI GO TO 50
19	CALL CASET GO TO 50
20 C	CALL THICK1
50	RETURN END

SUBROUTINE DLOAD3

000000 WRITTEN BY P. TAN 1971 MODULE TO RUN NEW CASES READS IN NEW STREAMS AND MODULE SETS COMMON LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10) COMMON IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR COMMON MODE,NPLNT,LOOPC COMMON SN(25,30),EEN(600),NPOINT(25,2) DIMENSION TITLE(10) C READ(5,100) TITLE FORMAT(10A8) WRITE(6,101) TITLE 100 101 C C FORMAT(1H0,10A8) READ NUMBER OF STREAM CHANGES READ(5,112) ANSR FORMAT(5F12,4) 112 NSR=ANSR+0.001 IF(NSR.LE. 0) GO TO 300 WRITE(6,111) NSR 111 FORMAT(1HC,15,25HINITIAL STREAMS ARE READ) C C READ STREAMS DO 530 I=1,NSR BORROWS VECTOR FROM SO(4,J), J=1,JJ READ(5,112) (SO(4,J),J=1,JJ) CALL CONVERT(2,2,4,2,4) C CALL BALNCE(2,4) WRITE(6,113) WRITE(6,113) (SO(4,J),J=1,JJ) 113 FORMAT(1X,10F12.4) S=SO(4,1) CALL STREAM(S) IF(III-IS) 400,401,401 400 CALL STREAM(0.0) IF(III-IS) 403,401,401 401 DO 402 J=1,JJ 402 SN(IS,J)=SO(4,J) CONTINUE 53Ū C READ NUMBER OF MODULE CHANGES READ(5,112) ANOE NOE=ANOE+0.001 300 IF(NOE.LE. U) GO TO 301 WRITE(6,114) NOE 114 FORMAT(1HU,15,21HMODULE SETS ARE READ) C READ MODULES DO 540 I=1.NOE DO 541 IZ=1.100 EN(IZ)=0.0 541 READ(5,112) (EN(N),N=1,5) NN=EN(3)+0.001 NM=EN(3)+0.001 MM=EN(1)+0.001 READ(5,112) (EN(N),N=6,NN) WRITE(6,113) WRITE(6,113)(EN(N),N=1,NN) CALL DISKIO(2,MM) CALL DIS CONTINUE 540 301 CONTINUE RETURN С 403 WRITE(6,405) FORMAT(1H0,* ERROR IN SN - VECTOR SPACE INSUFFICIENT *,//) 405 С RETURN END

SUBROUTINE SETST1

COMMON LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10) COMMON IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR COMMON MODE,NPLNT,LOOPC COMMON SN(25,30),EEN(600),NPOINT(25,2)

ALTERS THE TEMPERATURE IN ANY STREAM EN(4) = CHANGE IN TEMP (+VE FOR INCREASE AND -VE FOR DROP) WRITTEN BY P. TAN FOR WWTP SIMULATION

DO 1 I=2,JJ SO(1,I)=SI(1,I) SO(1,4)=SI(1,4)+EN(4)

RETURN END

SUBROUTINE SETSP1

COMMON LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10) COMMON IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR COMMON MODE,NPLNT,LOOPC COMMON SN(25,30),EEN(600),NPOINT(25,2)

WRITTEN BY P. TAN FOR WWTP SIMULATION ALTERS THE HEAD IN ANY STREAM EN(5)= CHANGE IN HEAD (+VE FOR GAIN, -VE FOR LOSS)

DO 1 I=2,JJ SO(1,I)=SI(1,I) SO(1,5)=SI(1,5)+EN(5) RETURN END

SUBROUTINE CONTL1

COMMON LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10) COMMON IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR COMMON MODE,NPLNT,LOOPC COMMON SN(25,30), EEN(600), NPOINT(25,2)

WRITTEN BY A.I.JOHNSON AND T.TOONG MAY 1968

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- 1. 2. 3.
- 4.
- LIST EQUIPMENT NUMBER EQUIPMENT TYPE -2 LENGTH OF EN LIST-21 LARGEST LOOP IDENTIFICATION NUMBER IN PROCESS FLOW DIAGRAM PRINTING CONTROL-NUMBER OF LOOPS BETWEEN PRINTING OF THE STREAM UNDER TEST JONLY NON ZERO COMPONENTS* AND THEIR FRACTIONAL CHANGES -15. TEN ZEROES NO. OF MODULES IN CALC ORDER LIST CONTROLLED BY CONTL1 MAXIMUM NO. OF LOOPS STREAM NUMBER OF STREAM TESTED FRACTIONAL TOLERANCE LOOP IDENTIFICATION NO. 5.

6.-15.

- 16. 17.
- 18.
- 19.
- <u>2</u>0. LOOP IDENTIFICATION NO.
- 21. LOOP IDENTIFI 22.-(19+JJ) = 0.0

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13300	IF(999-LOOP)13030,13030,13300 NN=EN(3)+.001
19900	LOOP=EN(21)+.001 ML=EN(20)+.001
	$\begin{array}{c} \text{ME} = \text{E}[0(2 -)\text{I} + 1] \\ \text{WRITE}(6 + 13) = 1 \\ \text{LOOP} \end{array}$
13001	FORMAT(1HC)11HEND OF LOOP + 15)
	D=FLOAT(LL)/EN(5) C=FLOAT(INT(D))-D
	E=EN(18) IF(INT(E+•001))13011•13015•13011
	CALL STREAM (E) IF(LL-2)13015,13150,13150
13150	DO 13-13 J=3,JJ
	S=SN(IS,J) IF(INT(S+.001)) 13016,13013,13016
13016	TEST=(S-EN(J+19))/S IF(ABS(TEST)-EN(19)) 13017,13017,13170
13170 13017	L=L+1 IF(J-3)13024,13019,13024
13019	WRITE(6,13004) E,TEST IF(C)13013,13130,13130
13130 13077	WRITE(6,13077) Format(1X,30HJValueFrac)
13024 13135	IF(ABS(C)001)13135,13135,13013 WRITE(6,13005) J,S,TEST
13004	FORMAT(1X,13,F12,2,F12,6) FORMAT(1H0,36HFRACTIONAL CHANGE IN TOTAL OF STREAM,F5.0,2HIS,F10.8
13013	
13020	IF(L) 13022,13020,13022 WRITE(6,13301)
13301	FORMAT(1H +12H***CONVERGED) EN(21)=EN(20)
	IF(FLOAT(LOOP).GE.EN(4)) LOOP=999 GO TO 13023
13022 13025	IF(LL-INT(EN(17)+.001)) 13015,13025,13015 WRITE(6,13302)
13302	FORMAT(1H +16H***NOT CONVERGED) EN(21)=EN(20)
	IF(FLCAT(LOUP).GE.EN(4)) LOUP=999 GO_TO_13023
13015	EN(21)=LOOP+1 LOOP=LOOP+1
	NC=NC-INT(EN(16))-1 IF(INT(E+.UU1)) 13023,13030,13023 D013021 J=3,JJ
13023 13021	DO13021 J=3,JJ EN(J+19)=SN(IS,J)
	MM=NE CALL DISKIO(2,MM)
С	CONTINUE
13030	RETURN END

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SUBROUTINE CONV01

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COMMON LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10) COMMON IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR COMMON MODE,NPLNT,LOOPC COMMON SN(25,30), EEN(600), NPOINT(25,2) WRITTEN BY P. TAN FOR WWTP SIMULATION 25/6/71 CONVERGENCE PROMOTION ON A STREAM EN VECTOR ****** NLOOP = EN(4)+0.01 IF(LOOP.LT.NLOOP) RETURN KOUNT=EN(16)+0.01 KOUNT=KOUNT+1 S=EN(5) CALL STREAM(S) GO TO (1,2,3), KOUNT C 1 600 DO 600 J=3,JJ EN(14+J)=SN(IS,J) EN(16)=KOUNT RETURN C 2 610 DO 610 J=3,JJ EN(12+JJ+J)=SN(IS,J) EN(16)=KOUNT RETURN DO 630 J=13,JJ DENOM=2.0*EN(12+JJ+J)-EN(14+J)-SN(IS,J) IF(DENOM.EQ. 0.0) GO TO 630 SN(IS,J)=(EN(12+JJ+J)*EN(12+JJ+J)-SN(IS,J)*EN(14+J))/DENOM EN(14+J)=SN(IS,J) CONTINUE EN(16)=1.0 I=S+0.07 630 $I = S + 0 \cdot 01$ CALL BAENCE(1,I) RETURN = END

LISTING OF THE PRIMARY SETTLING TANK MODEL

SUBROUTINE PRISET1 LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10) IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR MODE,NPLNT,LOOPC SN(25,30),EEN(600),NPOINT(25,2) COMMON COMMON COMMON COMMON SIMULATES A PRIMARY SETTLING TANK WRITTEN BY P. TAN FOR WWTP SIMULATION FOR REVISED STREAM LIST 2/9/71 EN VECTOR STANDARD FORMAT NUMBER OF TANKS IN PARALLEL HEAD LOSS THROUGH TANKS (FT OF WATER) FIRST INPUT STREAM (RAW WASTE STREAM) SECOND INPUT STREAM (SUPERNATE FROM THE DIGESTERS -THIRD INPUT STREAM (WASTE ACTIVATED SLUDGE - IF ANY FIRST OUTPUT STREAM (UNDERFLOW) THIRD OUTPUT STREAM (UNDERFLOW) THIRD OUTPUT STREAM (SCUM TROUGH FLOW) LENGTH OF TANK (FEET) = 0.0 IF TANK (FEET) = 0.0 IF TANK (FEET) = RADIUS OF TANK (FEET) DEPTH OF TANK (FEET) NOT CURRENTLY USED SOLIDS CONCENTRATION OF SCUM FLOW (MG/L) EFFICIENCY FACTOR FOR TANKS FRACTION OF FLOCCULATED SOLIDS FROM WASTE ACTIVATED SLUDGE AND DIGESTER SUPERNATE THAT IS RESUSPENDED NUMBER OF UNDERFLOW SLUDGE PUMPS CAPACITY OF PUMPS IGPH EACH FRACTION OF CAPACITY AT WHICH PUMPS ARE RUN ******* 1.-15. 4.5.7 8 9 12 13 14 IF ANY)) 16. 17. 18. 19 20 21 22 23• 24• 25. CALL ALKMIX(PH) DO 700 J=2,JJ SO(1,J)=0.0 SO(2,J)=0.0 SO(3, J)=0.0 IF(NIN.EQ.1) GO TO 40 INPUT STREAMS 2 AND 3 AR FASTER THAN SOLIDS IN RA FROM INPUT STREAMS 2 AND STREAMS - HENCE SETTLE MUCH DISP IS THE FRACTION OF SOLIDS ARE RESUSPENDED. ARE SLUDGE RAW WASTE. AND 3 WHICH DISP=EN(22) DISP1=1.0-DISP D0 620 J=13,19 D0 620 I=2,NIN S0(2,J)=S0(2,J)+SI(I,J)*DISP1 S0(1,J)=S0(1,J)+SI(I,J)*DISP DO 610 J=20,JJ DO 610 I=2,NIN SI(1,J)=SI(1,J)+SI(I,J) SI(1,4)=SI(1,4)*SI(1,3) DO 630 I=2,NIN SI(1,3)=SI(1,3)+SI(I,3) SI(1,4)=SI(1,4)+SI(I,4)*SI(I,3) SI(1,6)=SI(1,6)+SI(I,6) IF(SI(I,5)+LT+SI(1,5)) SI(1,5)= SI(1,4)=SI(1,4)/SI(1,3) SI(1,5) = SI(1,5)

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CONTINUE CALCULATE EXIT HEAD SO(1,5)=SI(1,5)-EN(5) SO(2,5)=SI(1,5)-EN(5) SO(3,5) = SI(1,5) - EN(5)SO(1,2)=PH SO(2,2)=PH SO(3,2)=PH SO(1,4)=SI(1,4) SO(2,4)=SI(1,4) SO(3,4)=SI(1,4) CALCULATE SURFACE AREA IF(EN(16).LE. U.U) GO TO 730 AREA=EN(16)*EN(17)*EN(4) GO TO 720 AREA=3•142*EN(17)*EN(17)*EN(4)/4•0 CONTINUE 730 720 C C CALCULATING RESIDENCE TIME TBAR (MINUTES) TBAR=AREA*EN(18)/SI(1,3)*374.4 CALCULATING OVERFLOW RATE OR (IGPD/SQ.FT.) OR=SI(1,3)*24.0/AREA CALCULATE THE THEORETICAL REMOVAL FROM BATCH TEST DATA REMOVAL=1.0-EXP(-0.23*TBAR**0.5/EN(18)**0.25) ACCOUNT FOR THE DECREASE IN EFFICIENCY DUE TO TURBULENCE REMOVAL=REMOVAL*EN(21) MATERIAL BALANCE SO(2,13)=SO(2,13)+SI(1,13)*REMOVAL SO(2,15)=SO(2,15)+SI(1,15)*REMOVAL SO(2,16)=SO(2,16)+SI(1,16)*REMOVAL SO(2,17)=SO(2,17)+SI(1,17)*REMOVAL REM1=1.0-REMOVAL SO(1,13)=SO(1,13)+SI(1,13)*REM1 SO(1,15)=SO(1,15)+SI(1,15)*REM1 SO(1,16)=SO(1,16)+SI(1,16)*REM1 SO(1,17)=SO(1,17)+SI(1,17)*REM1 CALCULATE UNDERFLOW FLOW SO(2,6)=EN(25)*EN(23)*EN(24)*10.0 SSF=(SO(2,13)+SO(2,15)+SO(2,16)+SO(2,17))/SO(2,6) ASSUME THAT THE MAXIMUM UNDERFLOW CONCENTRATION IS 60000 MG/L IF(SSE_LE.0.06) GO TO 19 INSUFFICIENT WITHDRAWAL OF SLUDGE LEADS TO CARRY-OVER OF SOLIDS SO(1,13)=SO(1,13)+SO(2,13)*(1.0-0.06/SSF) SO(1,15)=SO(1,15)+SO(2,15)*(1.0-0.06/SSF) SO(1,16)=SO(1,16)+SO(2,16)*(1.0-0.06/SSF) SO(1,17)=SO(1,17)+SO(2,17)*(1.0-0.06/SSF) SO(2,13)=SO(2,13)*0.06/SSF SO(2,15)=SO(2,15)*0.06/SSF SO(2,16)=SO(2,16)*0.06/SSF SO(2,17)=SO(2,17)*0.06/SSF SO(2,17)=SO(2,17)*0.06/SSF 19 C C SET SCUM FLOW SO(3,18)=SI(1,18)*0.9 SO(1,18)=SO(1,18)+SI(1,18)-SO(3,18) SO(2,18)=0.0 SO(3,6)=SO(3,18)*(1.0-EN(20)*1.0E-6)/EN(20)*2.2E6 EFFLUENT FLOW SO(1,6)=SI(1,6)-SO(2,6)-SO(3,6) SET SCUM VARIABLES (OTHER THAN SCUM) SIMILAR TO EFFLUENT F=SO(3,6)/(SO(1,6)+SO(3,6)) So(3,13)=So(1,13)*F So(1,13)=So(1,13)-So(3,13) So(3,15)=So(1,15)*F So(1,15)=So(1,15)-So(3,15)

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SO(3,16)=SO(1,16)*F SO(1,16)=SO(1,16)-SO(3,16) SO(3,17)=SO(1,17)*F SO(1,17)=SO(1,17)-SO(3,17) COMPONENTS IN SOLUTION SPLIT ACCORDING TO FLOW COMPONENTS IN SOLUTION SPLIT ACCORDING TO DO 600 I=20,JJ SO(1,I)=SI(1,I)*SO(1,6)/SI(1,6) SO(2,I)=SI(1,I)*SO(2,6)/SI(1,6) SO(3,I)=SI(1,I)-SO(1,I)-SO(2,I) CONTINUE SO(1,14)=SI(1,14)*SO(1,6)/SI(1,6)+SO(1,14) SO(2,14)=SI(1,14)*SO(2,6)/SI(1,6)+SO(2,14) SO(3,14)=SI(1,14)*SO(3,6)/SI(1,6)+SO(2,14) SO(1,19)=SI(1,19)*SO(1,6)/SI(1,6)+SO(2,19) SO(2,19)=SI(1,19)*SO(2,6)/SI(1,6)+SO(2,19) SO(3,19)=SI(1,19)*SO(3,6)/SI(1,6)+SO(3,19) 600 SET EXIT OXYGEN CONCENTRATIONS TO ZERO ALL SO(1,7) = 0.0SO(2,7) = 0.050(3,7)=0.0CALL BALNCE(2,1) CALL BALNCE(2,2) CALL BALNCE(2,3) IF(ISP) 1,1,2 CONTINUE WRITE(6,300) REMOVAL FORMAT(1H0,17HSOLIDS REMOVAL = ,F10.3) WRITE(6,400) OR FORMAT(1X,17HOVERFLOW RATE = ,F10.3, WRITE(6,410) TBAR FORMAT(1X,77HDETENTION TIME = ,F10.3, 300 400 ,F10.3,12H IGPD/SQFT) 410 FORMAT(1X, 17HDETENTION TIME = ,F10.3,10H MINUTES) CONTINUÊ

RETURN END

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SUBROUTINE ACTSL1

LLST(50), NS(100), FN(100), SI(4,30), SO(4,30), KPRNT(10) IS, NE, JJ, LOOP, NIN, NOUT, MSN, ISP, NC, III, NCALC, NOCOMP, NSR COMMON COMMON COMMON MODE, NPLNT, LOOPC COMMON SN(25,30), EEN(600), NPOINT(25,2) COMMON /A/ TAER, TM, FM, DELS SIMULATES THE ACTIVATED SLUDGE PROCESS CAN ALSO BE USED TO SIMULATE THE STEP AERATION PROCESS CALLS THE BASIC CSTR PROGRAM FOR AEROBIC BIOLOGICAL REACTIONS EN VECTOR ****** NUMBER OF REACTORS IN PARALLEL TOTAL HEAD LOSS THROUGH ACT. SL. TANKS (FT OF WATER) NUMBER OF INPUT STREAMS (= 2.0) FIRST INPUT STREAM (= RETURN SLUDGE FLOW) SECOND INPUT STREAM (= FEED STREAM) NUMBER OF OUTPUT STREAMS (= 1.0) OUTPUT STREAM NUMBER STANDARD FORMAT NUMBER OF REACTORS TOTAL HEAD LOSS TH 1.-15. 4. 5. 6. 7. 8 11 12 STREAM NUMBER OUTPUT NOTE EN(16.-32.) LENGTH OF REACTOR WIDTH OF REACTOR ARE USED BY CSTR2 R (FEET) 16.17.(FEET) (FEET) 18. DEPTH OF REACTOR (FEFT) = 1.0 IF NOT USING A RESIDENCE TIME DISTRIBUTI = EN(34) IF USING A RTD MODEL LYSIS RATE (K2 /HR) BASAL METAB. RATE (K1 /HR) RATE OF SOLUBILIZATION OF DEGRADABLE SOLIDS (STRIPPING RATE OF VOLATILE ORGANICS (K5 /HR) MICROORGANISMS YIELD FACTOR (Y DIMENSIONLESS) FRACTION OF LYSIS PRODUCTS TO SOLUBLE DEGRADAB FRACTION OF LYSIS PRODUCTS TO SOLUBLE NONDEGRA RATE OF AIR ADDITION TO EACH REACTOR (SCF/HR) UNIT GROWTH RATE CONSTANT (K4 /HR) FRACTION OF MLSS THAT IS NITRIFIERS N/C RATIO IN RAW FEED DEPTH OF REACTOR A RESIDENCE TIME DISTRIBUTION MODEL 19. 20. 21. (K3 /HR) DEGRADABLE CARBON NONDEGRADABLE CARBON 29. N/C RATIO IN RAW FEED P/C RATIO IN RAW FEED ABSORPTION NUMBER (FOR OXYGEN TRNSFER) 30. 31. 32. NOT PRESENTLY USED NUMBER OF CSTR'S IN SERIES - EITHER AN RTD MODEL OR THE ACTUAL NUMBER OF TANKS IN A STEP AERATION PROCESS 33. 34. MAXIMUM OF TEN TANKS IN SERIES FRACTION OF FEED STREAM TO FIR FRACTION OF FEED STREAM TO SEC 35. 36. FIRST TANK SECOND TANK 37. ETC FRACTION OF FEED STREAM TO TENTH TANK (IF ANY) NUMBER OF BLOWERS USED 44. 45. . 46. - CAPACITY OF EACH BLOWER SCF/HR EN(27)=EN(45)*EN(46)/EN(4)/EN(34) SIN=SI(1,11)+SI(2,11)FEED STREAM IN SI(4,J) STORE DO 600 J=2,JJ SI(4,J)=SI(2,J) CALCULATING THE VARIABLES FOR USE BY THE FOOD TO MICOORGANISM RATIO AND SECONDARY CLARIFIER TAER IS, THE AERATION TIME TANKS=FN(34) IF(EN(19).GT.1.0) TANKS=1.0 TAER=FN(16)*EN(17)*FN(18)*TANKS *6.24/(SI(1.3)+SI(2.3))*EN(4) TM=TAFR/EN(34) FM=SI(2,10)/SI(1,12)/TAER*24.0

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600 C C C C C NCSTR=EN(34)+0.01 С DO 610 IJ=1,NCSTR IF(EN(34+IJ).LE. 0.0) GO TO 20 С SI(2,3)=SI(4,3)*EN(34+IJ) DO 900 KB=6,JJ SI(2,KB)=SI(4,KB)*EN(34+IJ) 900 CALL MIXERI EN(29)=EN(29)*SI(1,11)/SO(1,11) DO 910 KB=2,JJ SI(1,KB)=SO(1,KB) 910 Č 20 C CONTINUE CALL CSTR2 C D0 620 J=2,JJ SI(1,J)=SO(1,J) 620 C 610 C C CONTINUE

SO(1,5)=SI(2,5)-EN(5) DELS=SO(1,11)-SIN

SUBROUTINE CSTR2

RETURN END

COMMON LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10) COMMON IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR COMMON MODE,NPLNT,LOOPC COMMON SN(25,30),EEN(600),NPOINT(25,2) COMMON /A/ TAER,TM,FM,DELS THIS IS THE BASIC CSTR ROUTINE TO DESCRIBE THE AEROBIC BIOLOGICAL PROCESSES AND IS CALLED BY ACTSL WRITTEN BY P. TAN EN VECTOR ********* NOTE-CSTR2 USES ONLY EN(16,-32.) EN(1-15.) AND EN(33.-) ARE USED BY ACTSL NOTE-CSTR2 HAS ONLY ONE INPUT STREAM,THE MIXED LIQUOR FLOW 16. LENGTH OF REACTOR (FEET) 17. WIDTH OF REACTOR (FEET) 18. DEPTH OF REACTOR (FEET) 19. = 1.0 IF NOT USING A RESIDENCE TIME DISTRIBUTION MODEL = EN(34) IF USING A RESIDENCE TIME DISTRIBUTION MODEL 20. LYSIS RATE (K2 /HR) 21. BASAL METAB. RATE (K1 /HR) 22. RATE OF SOLUBILIZATION OF DEGRADABLE SOLIDS (K3 /HR) 23. STRIPPING RATE OF VOLATILE ORGANICS (K5 /HR) 24. MICRODRGANISMS YIELD FACTOR (Y DIMENSIONLESS) 25. FRACTION OF LYSIS PRODUCTS TO SOLUBLE DEGRADABLE CARBON 26. FRACTION OF LYSIS PRODUCTS TO SOLUBLE DEGRADABLE CARBON 27. RATE OF AIR ADDITION TO EACH REACTOR (SCF/HR) 28. UNIT RATE GROWTH CONSTANT (K4 /HR) 29. FRACTION OF LYSIS THAT IS NITRIFIERS 30. N/C, RATIO IN RAW FEED 32. ABSORPTION NUMBER (FOR OXYGEN TRANSFER)

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	CALL CONVERT(1,3,1,3,1)	264
C 600	$DO \ 600 \ J=2,JJ$ SO(1,J)=SI(1,J)	
600 C	SI15=SI(1,15) SI1619=SI(1,16)+SI(1,17)+SI(1,18)+SI(1,19)	
C C	CALCULATING OXYGEN SATURATION AT INLET TEMPERATURE CS=14.16-SI(1,4)*(0.3943-SI(1,4)*(0.007714-0.0000646*S	I(1,4))) [.]
C C	CALCULATING VOLUME OF REACTOR V=EN(16)*EN(17)*EN(18)/EN(19)	
C C	CALCULATING OXYGEN TRANSFER COEFFICIENT KLA AKL=EN(32)*((EN(27)/60.0)**0.85)*(EN(18)**0.7)/V	
	BACTIN IS THE CONCENTRATION OF NITROSOMONAS BACT=EN(29)*SI(1,11) BACTIN=BACT	
c c	KOUNT=1	
	INITIAL GUESS OF DEGRADABLE CARBON IS HALF FEED VALUE INITIAL GUESS OF OTHER VARIABLES ARE FEED VALUES	
c	SO(1,20)=0.5*SO(1,20) SO(1,21)=0.5*SO(1,21)	
С 10	BEGIN ITERATION FOR EFFLUENT CONCENTRATIONS OF CSTR CONTINUE	
c	CALCULATING RATE OF SOLUBLE CARBON REMOVAL CC=SO(1,20)+SO(1,21) RFC=EN(28)*CC	
c	SO(1,20)=SI(1,20)/(1.0+TM*(RFC*SO(1,15)/CC+EN(23)*EN(2 SO(1,21)=(SI(1,21)+TM*SO(1,15)*(EN(20)*EN(25)+EN(22)*(1 1 SO(1,18)+SO(1,19)))/(1.0+TM*RFC*SO(1,15)/CC)	7)/V)) 50(1,16)+
c c c	SO(1,22)=SI(1,22)+TM*EN(20)*EN(26)*SO(1,15)	
	CALCULATING EXIT SOLIDS SO(1,15)=SI(1,15)/(1.0-TM*(EN(24)*RFC-EN(20)-EN(21))) SO(1,16)=SI(1,16)/(1.0+TM*SO(1,15)*(EN(22)-EN(20)*(1.0) 1EN(26))/SO(1,16))) SO(1,18)=SI(1,18)/(1.0+TM*EN(22)*SO(1,15)) SO(1,19)=SI(1,19)/(1.0+TM*EN(22)*SO(1,15))	-EN(25)-
CCC	BALANCE ON NITROGEN AVNIT IS THE AVAILABLE AMMONIACAL NITROGEN AVNIT=SI(1,27)+(SI(1,15)-SO(1,15))*0.23 +(SI(1,8)-SI(1 ISO(1,16)-SO(1,17)-SO(1,18)-SO(1,19)-SO(1,20)-SO(1,21)-	•15)- SO(1•22))*
.c	2EN(30) CALCULATE CONVERSION OF AMMONIA TO NITRATES EFF=SO(1,7)	
	IF(SO(1,7).GE.1.0) EFF=1.0 SO(1,27)=AVNIT/(1.0+0.275*BACT*TM*EFF/(1.0+SO(1,27))) IF(SO(1,27).LT. 0.0) GO TO 40	
40	GO TO 41 CONTINUE	
410 41	SO(1,27)=0.0 WRITE(6,410) FORMAT(1H0,28HDEFICIENCY IN FEED NITROGEN) CONTINUE	
	SO(1,28)=SI(1,28)+(AVNIT-SO(1,27)) BACT=BACTIN+(AVNIT-SO(1,27))*0.05	
C C	CALCULATING EXIT OXYGEN CONCENTRATION SO(1,7)=(SI(1,7)-TM*((RFC*(1.0-EN(24))+EN(21))*SO(1,15) 1+(AVNIT-SO(1,27))*4.57-AKL*CS))/(1.0+TM*AKL) IF(SO(1,7).LT. U.O) SO(1,7)=0.0)*2•66
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	KOUNT=KOUNT+1
C 2 0 4 0 0 2 0 2 0 2 0 C 2 0 C 2 0 C 2 0 C 2 0 C 2 0 C 2 0 0 0 C 2 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0	ÎF(KOUNT.GT. 20) GO TO 20 IF(ABS((CC-SO(1,20)-SO(1,21))/CC).GT. 0.005) GO TO 10 GO TO 30
	CONTINUE WRITE(6,400) FORMAT(1H0,20HCSTR2 NOT CONVERGED) CONTINUE
	PHOSPHORUS BALANCE
	SO(1,29)=SI(1,29)+(SI15-SO(1,15))*0.05 + (SI1619-SO(1,16)-SO(1,17) 1-SO(1,18)-SO(1,19))*EN(31) IF(SO(1,29).LT.0.0) GO TO 50 GO TO 51
_50	CONTINUE 50(1,29)=0.0
420 51 C	ŴŔĬŤĖ(6,420) Format(1h0,3uhdeficiency in Feed phosphorus) Continue
	DUE TO THE HIGH DEGREE OF FLOCCULATION IN THE AERATION TANKS COLLOIDAL COMPONENTS ARE CONVERTED TO SETTLEABLE SO(1,13)=SO(1,13)+SO(1,14) SO(1,14)=U.U SO(1,16)=SO(1,16)+SO(1,19) SO(1,16)=SO(1,16)+SO(1,19) SO(1,19)=U.U
	CALCULATING THE FRACTION OF NITRIFIERS IN THE EXIT SOLIDS EN(29)=BACT/SO(1,1)
	CALCULATING EXIT PH ASSUME CARBON DIOXIDE CONCENTRATION IS IN EQUILIBRIUM WITH AIR SO(1,2)=-ALOG1U(2.6E-7/SI(1,23))
	CALL CONVERT(2,2,1,2,1) CALL BALNCE(2,1)
	RETURN END

LISTING OF THE SECONDARY SETTLING TANK MODEL

SUBROUTINE SECLAR1

COMMON COMMON LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10) IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR MODE, NPLNT, LOOPC SN(25,30), EEN(600), NPOINT(25,2) /A/ TAER, TM, FM, DELS COMMON COMMON COMMON WRITTEN BY P. TAN FOR WWTP SIMULATION FOR REVISED STREAM LIST 2/9/71 SIMULATES A SECONDARY CLARIFIER EN VECTOR **** *** STANDARD FORMAT NUMBER OF CLARIFIERS IN PARALLEL HEAD LOSS THROUGH CLARIFIER (FEET OF WATER) FIRST OUTPUT STREAM (EFFLUENT) SECOND OUTPUT STREAM (UNDERFLOW) THIRD OUTPUT STREAM (SCUM TROUGH FLOW) MODE OF OPERATION = 1.0 FOR FIXED FRACTION OF RECYCLE = 2.0 FOR MAXIMUM CONCENTRATION OF UNDERFLOW FRACTION OF RECYCLE IF MODE=1 SURFACE AREA OF CLARIFIER (PER TANK) - SQ.I SCUM FLOW CONCENTRATION (MG/L) NUMBER OF UNDERFLOW SLUDGE PUMPS 1.-15. 4 5 12 13 14 16. 17. 18. 19. - SQ.FT. 20. 21. 22. NUMBER OF CAPACITY FRACTION F UNDERFLOW SLUDGE PUMPS OF PUMPS IGPH EACH OF CAPACITY USED EN(17)=EN(20)*EN(21)*EN(22)/SI(1,3) SO(1,2)=SI(1,2) SO(2,2)=SI(1,2) SO(3,2)=SI(1,2) SO(1,4)=SI(1,4) SO(2,4)=SI(1,4) SO(3,4)=SI(1,4) CALCULATE EXIT HEADS SO(1,5)=SI(1,5)-EN(5) SO(2,5)=SI(1,5)-EN(5) SO(3,5)=SI(1,5)-EN(5) CALCULATING THE SLUDGE VOLUME INDEX SVI=56.1+113.J*FM*(1.05**(SI(1,4)-20.0)) CALCULATING THE MAXIMUM UNDERFLOW CONCENTRATION CU CU=1.0E+06/SVI CALCULATING THE MIXED LIQUOR SUSPENDED SOLIDS XMLSS XMLSS=SI(1,11)/SI(1,6)*1.0E6 METHOD=EN(16)+0.001 GO TO (1.2).METHOD CONTINUE FOR FIXED FRACTION OF RECYCLE SO(2,6)=SI(1,6)*EN(17) SO(1,6)=SI(1,6)-SO(2,6) CALCULATING THE OVERFLOW RATE GSS (USGPD/SQFT) GSS=SO(1,6)*2.879/EN(18)/EN(4) APPLY CORRELATION THE FACTOR 1.5 IS SUGGESTED BY RICH TO ACCOUNT FOR THE FACTOR 1.5 IS SUGGESTED BY RICH TO ACCOUNT FOR THE EXTRAPOLATION OF BATCH SETTLING TESTS TO ACTUAL TANK XRSS=556.U*(GSS**0.494)/(XMLSS**1.82)/(TAER**0.439)*1.5 MATERIAL BALANCE SO(1,11)=SI(1,11)*XRSS CONTINUE TANKS

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SO(2,11)=SI(1,11)-SO(1,11) UNDERFLOW CONCENTRATION CANNOT BE GREATER THAN CU IF((SO(2,11)/SO(2,6)*1.0E6) .LT. CU) GO TO 10 SO(2,11)=CU*1.0E-6*SO(2,6) SO(1,11)=SI(1,11)-SO(2,11) GO TO 10 C CONTINUE FOR MAXIMUM CONCENTRATION OF UNDERFLOW ROUGH ESTIMATE OF UNDERFLOW FLOW SO(2+6)=SI(1+11)*1.0E6/CU SO(1+6)=SI(1+11)*1.0E6/CU SO(1+6)=SI(1+10)-SO(2+6) CALCULATING THE OVERFLOW RATE GSS (USGPD/SQFT) GSS=SO(1+6)*2.879/EN(18)/EN(4) APPLY CORRELATION XRSS=556.U*(GSS**0.494)/(XMLSS**1.82)/(TAER**0.439)*1.5 SO(1+11)=SI(1+11)*XRSS SO(2+11)=SI(1+11)-SO(1+11) CALCULATING CORRECTED UNDERFLOW FLOW С C CALCULATING CORRECTED UNDERFLOW FLOW SO(2,6)=SO(2,11)*1.0E6/CU SO(1,6)=SI(1,6)-SO(2,6) С CONTINUE ASSUME THAT THE FLOCS ARE UNIFORM IN COMPOSITION ALSO ASSUME THAT THE OVERFLOW AND UNDERFLOW SOLIDS HAVE THE SAME COMPOSITION AS THE INCOMING FLOCS P1=SO(1,11)/SI(1,11) DO 600 I=13,19 SO(1,I)=P1*S1(1,I) SO(2,I)=SI(1,I)-SO(1,I) CONTINUE 600 C C SPLIT OVERFLOW INTO EFFLUENT AND SCUM FLOW SO(3,18)=SI(1,18) SO(1,18)=U.U SO(2,18)=0.U SO(2,18)=0.U SO(3,6)=SO(3,18)*(1.0-EN(19)*1.0E-6)/EN(19)*2.2E6 SO(1,6)=SO(1,6)-SO(3,6) P1=SO(3,6)/(SO(1,6)+SO(3,6)) D0 61U 1=13,17 SO(3,1)=SO(1,1)*P1 SO(1,1)=SO(1,1)-SO(3,1) CONTINUE CONTINUE SO(3,19)=SO(1,19)*P1 SO(1,19)=SO(1,19)-SO(3,19) 610 C C SPLIT SOLUBLE COMPONENTS ACCORDING TO FLOW P1=SO(1,6)/SI(1,6) P2=SO(2,6)/SI(1,6) DO 620 I=20,JJ SO(1,I)=SI(1,I)*P1 SO(2,I)=SI(1,I)*P2 SO(3,I)=SI(1,I)-SO(1,I)-SO(2,I) .620 C C C CONTINUE SET ALL EXIT OXYGEN CONCENTRATIONS AT ZERO SO(1,7)=0.0 SO(2,7)=0.0 SO(3,7)=0.0 C BALNCE(2,1) BALNCE(2,2) BALNCE(2,3) CALL CALL CALL C IF(ISP) 30,30,31 WRITE(6,300) TAER FORMAT(1HU,22HAERATION TIME WRITE(6,301) FM FORMAT(1X,22HFOOD TO MICRO RATIO = WRITE(6,302) GSS 4 31 300 = •F10.3,7H HOURS) 301 ,F10.3)

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CONTINUE

LISTING OF THE ANAEROBIC DIGESTER MODEL

FORMAT(1X,22HOVERFLOW RATE = WRITE(6,303) SVI FORMAT(1X,22HSLUDGE VOLUME INDEX =

SUBROUTINE ANDIG1

C C

COMMON LLST(50),NS(100),EN(100);SI(4,30),SO(4,30),KPRNT(10) COMMON IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR COMMON MODE,NPLNT,LOOPC COMMON SN(25,30),EEN(600),NPOINT(25,2)

•F10.3•12H

,F10.3)

SIMULATES AN ANAEROBIC DIGESTER WRITTEN BY P. TAN AUG/71

EN VECTOR *******

STANDARD FORMAT NUMBER OF DIGESTERS IN PARALLEL HEAD LOSS IN DIGESTERS (FT OF WATER) DIAMETER OF DIGESTER (FEET) DEPTH OF DIGESTER (FEET) 1.-15. 4. 5. 16. 18.

DEPTH OF DIGESTER (FEET) = 1.0 FOR CONVENTIONAL DIGESTER OPERATION = 2.0 FOR HIGH RATE DIGESTER OPERATION TEMPERATURE OF OPERATION FRACTION OF WATER TO FIRST OUTPUT STREAM (SUPERNATANT) = 1.0 IF FIRST STAGE OF A TWO-STAGE SYSTEM FRACTION OF SOLIDS TO FIRST OUTPUT STREAM (SUPERNATANT) (EXCLUDES SETTLEABLE INORGANIC SOLIDS) = 1.0 IF FIRST STAGE OF A TWO-STAGE SYSTEM FRACTION OF SETTLEABLE INORGANIC SOLIDS TO SUPERNATE = 1.0 IF FIRST STAGE OF A TWO-STAGE SYSTEM FRACTION OF SETTLEABLE INORGANIC SOLIDS TO SUPERNATE = 1.0 IF FIRST STAGE OF A TWO-STAGE SYSTEM STORAGE SPACE FOR GAS PRODUCED FRACTION OF VOLUME THAT IS EFFICIENTLY USED IN DIGESTION 19. 20. 21.

22.

- 23.
- 24.
- USING SO(4, JJ) AS TEMPORARY STORAGE OF INTERMEDIATE RESULTS

DO 600 IK=2,JJ SO(4,IK)=SI(1,IK) CONTINUE

SO(1,4) = EN(19)SO(2,4) = EN(19)SO(2,4)=EN(19) SO(1,5)=SI(1,5)-EN(5) SO(2,5)=SI(1,5)-EN(5) SO(4,5)=SI(1,5)-EN(5)

CALCULATING VOLUME OF DIGESTER (CU FT) VOL=3.142*EN(16)*EN(16)*EN(17)/4.0*EN(4) VOL=VOL*EN(24)

CALCULATING DIGESTER RESIDENCE TIME (DAYS) RESTIME=VOL*6.24/SI(1,3)/24.0

CALCULATING VOLATILE SOLIDS LOADING ON DIGESTER LB/CUFT/DAY SSLOAD=SI(1,12)*24.0/VOL

ALKALINITY CALCULATIONS ALK=20.0*SI(1,23)/SI(1,6)

USGPD/SQFT)

C C

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600 C

CN=71.43*SI(1,27)/SI(1,6) PH=SI(1,2) CALL ALKPH(1,ALK,ACT,CN,PH) CT=ACT 269 C SET REACTION CONSTANT FOR HIGH RATE OR CONVENTIONAL OPERATION NCOR=EN(18)+0.01 GO TO (1,2),NCOR ĀK1=0.026*1.04**(EN(19)-35.0) 1 GO TO 10 AK1=0.080*1.04**(EN(19)-35.0) 2 10 C CONTINUE CALCULATE FRACTION OF VOLATILE SOLIDS DESTROYED FRAC=1.0-1.0/(1.0+AK1*RESTIME) C C STORE RESULTING LIQUOR IN SO(4,J) - PRIOR TO SEPARATION DO 610 IK=15,19 SO(4,IK)=SI(1,IK)*(1.0-FRAC) 610 CALCULATE VOLUME OF GAS LIBERATED GAS=17.0*SI(1.12)*FRAC EN(23)=GAS С CALCULATE VOLATILE ACID CARBON IN EFFLUENT AKMAX=0.28*EXP(-0.036*(35.0-EN(19))) VAC=200.0*EXP(0.12*(35.0-EN(19)))/(1.0+AKMAX*RESTIME) SO(4,20)=SI(1,20)+SI(1,6)*VAC*1.0E-6 С C NUTRIENT BALANCE SO(4,27)=SI(1,27)+SI(1,12)/2.1*FRAC*0.23 SO(4,29)=SI(1,29)+SI(1,12)/2.1*FRAC*0.05 CALL BALNCE(2,4) CALCULATE EFFLUENT PH ACID=(VAC-SI(1,20)/SI(1,6)*1.0E6)/24.0/1000.0 CNNEW=71.43*SO(4,27)/SO(4,6) С ALK=ALK+(CNNEW-CN)-ACID CALL ALKPH(2,ALK,CT,CNNEW,APH) SO(4,2)=APH SO(1,2)=APH SO(2,2)=APH SO(2,2)=APH SO(4,23)=ALK*SO(4,6)/20.0 C C CALCULATING EFFLUENT AND UNDERFLOW FLOWS IF(EN(20) •GE• 1•0) GO TO 31 C SETTING THE SUPERNATANT FLOW SO(1,6)=SO(4,6)*EN(20) DO 620 IK=20,JJ DO 620 IK=20,00 SO(1,1K)=SO(4,1K)*EN(20) CONTINUE SO(1,13)=SO(4,13)*EN(22) DO 640 IK=14,19 SO(1,1K)=SO(4,1K)*EN(21) 620 640 CONTINUE CALL BALNCE(2,1) . C C CALCULATE DIGESTED SLUDGE FLOW BY MATERIAL BALANCE D0 650 IK=6,JJ S0(2,IK)=S0(4,IK)-S0(1,IK) CONTINUE 650 SO(2,3)=SO(4,3)-SO(1,3) GO TO 32 с 31 CONTINUE ONLY ONE STREAM DRAWN OFF - THE FIRST OUTPUT STREAM DO 660 IK=2,JJ SO(1,IK)=SO(4,IK) SO(2,IK)=0.0 С 660 CONTINUE C 32 CONTINUE

	х				
С	SET ALL EXIT OXYGEN CONCENTRATIONS SO(1,7)=0.0 SO(2,7)=0.0	το Ζ	ZERO	270	
C C					
21	IF(ISP) 20,20,21 CONTINUE WRITE(6,400) RESTIME				
400	FORMAT(1H0,25HDETENTION TIME	=	,F12.3,6H	DAYS)	
410	WRITE(6,410) SSLOAD FORMAT(1X,25HSOLIDS LOADING GAS=GAS*24.0 WRITE(6,430) GAS	=	,F12.3,15H	LBS/CUFT/DAY)
430	FORMAT(1X,25HGAS PRODUCED WRITE(6,420) FRAC	=	•F12•3•11H	CUFT/DAY)
420 20 C	FORMAT(1X,25HFRACTION VSS DESTROYED CONTINUE	=	•F12•3)		
-	RETURN END				

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LISTING OF THE TRICKLING FILTER MODEL

SUBROUTINE TRICK1

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COMMON LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10)
COMMON IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR
COMMON MODE,NPLNT,LOOPC
COMMON SN(25,30),EEN(600),NPOINT(25,2)
  SIMULATES A TRICKLING FILTER
UTILIZES THE EQUATION PROPOSED BY ECKENFELDER
WRITTEN BY P. TAN SEPT/71
  EN VECTOR
 DO 600 J=2,JJ
SO(1,J)=SI(1,J)
SO(1,5)=SI(1,5)-EN(5)
  CALCULATE SURFACE AREA
AREA=3.142*EN(17)*EN(17)/4.0*EN(4)
 CALCULATE THE LIQUID LOADING RATE (USGPD/SQFT)
Q=SI(1,3)*28.79/AREA
 CALCULATE REMOVAL OF BOD ACCORDING TO EQUATION
REM = FRACTION OF DEGRADABLE CARBON REMAINING
REM=EXP(-EN(18)*EN(16)**EN(19)/Q**EN(20))
 ASSUME THAT THE REMOV/
SO(1,16)=SI(1,16)*REM
DO 610 J=18,22
SO(1,J)=SI(1,J)*REM
                       THE REMOVAL APLLIES TO ALL DEGRADABLE COMPONENTS
ASSUME MICROORGANISM YIELD FACTOR IS 0.54
SO(1,15)=0.54*(SI(1,16)+SI(1,18)+SI(1,19)+SI(1,20)+SI(1,21)
1 +SI(1,22))*(1.0-REM)
 CALL BALNCE(2,1)
 RETURN .
  END
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LISTING OF THE SCREEN MODEL

SUBROUTINE SCREEN1

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COMMON LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10) COMMON IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR COMMON MODE,NPLNT,LOOPC COMMON SN(25,30),EEN(600),NPOINT(25,2) COMMON COMMON COMMON EN VECTOR \$***
STANDARD FORMAT
= SCREENINGS, IF KNOWN FROM PLANT OPERATIONS (CUFT/MIG)
= 0.0 IF USING BARMINUTER
= ANY NEGATIVE NUMBER IF WANT TO USE INTERNAL SCREENINGS
ESTIMATE
HEAD LOSS ACROSS SCREEN (FT OF WATER)
SCREEN SIZE, IF USING INTERNAL ESTIMATE OF SCREENINGS (IN) 1.-15. 4. 5.16. DO 600 J=2,JJ SO(1,J)=SI(1,J) SO(1,5)=SI(1,5)-EN(5) CALCULATE VOLUME OF SCREENINGS PER DAY IF(EN(4))1,2,3 SCR=19.0*10.0**(-0.8*EN(16)) GO TO 4 RETURN SCR=EN(4) VOL=SI(1,3)*SCR*24.0*1.0E-6 ESTIMATED DRY WEIGHT OF SCREENINGS IS 30 LBS/CUFT CALCULATING MASS OF SCREENINGS PER HOUR WRAGS=30.0*VOL/24.0 CONVERT TO EQUIV CARBON WRAGS=WRAGS/2.1 SO(1,16)=SI(1,16)-WRAGS CALL BALNCE (2,1) IF(ISP) 10,10,11 WRITE(6,400) VOL FORMAT(1H0,13HSCREENINGS = CUFT/DAY) ,F8.2,10H CONTINUÉ RETURN END

LISTING OF THE GRIT REMOVAL MODEL

SUBROUTINE GRIT1

COMMON LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10) COMMON IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR COMMON MODE,NPLNT,LOOPC COMMON SN(25,30),EEN(600),NPOINT(25,2)

WRITTEN BY P. TAN 27/8/71 DEGRITTER FOR SEWAGE TREATMENT SIMPLE MODEL USING THE AVERAGE AMOUNT OF GRIT COLLECTED /MIG

EN VECTOR

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1.-15. STANDARD FORMAT 4. CU.FT. OF GRIT PER MILLION GALLONS OF RAW FEED 5. HEAD LOSS THROUGH CHAMBER (FT OF WATER)

DO 600 J=2,JJ SO(1,J)=SI(1,J) SO(1,5)=SI(1,5)-EN(5) 600 CUFT=SI(1,3)*EN(4)*1.0E-6 GRIT=CUFT*0.40*62.4*2.65 SO(1,11)=SI(1,11)-GRIT SO(1,13)=SI(1,13)-GRIT IF (ISP) 1,1,2 CUFT=CUFT*24.0 WRITE(6,400) CUFT FORMAT(1H0,25HVOLUME OF GRIT REMOVED = CONTINUE 400 1 C CU FT/DAY •F8.2•11H)

RETURN END

LISTING OF THE CHLORINATOR MODEL

SUBROUTINE CHLOR1

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COMMON LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10)
COMMON IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR
COMMON MODE,NPLNT,LOOPC
 COMMON SN(25,30), EEN(600), NPOINT(25,2)
DIMENSION CL(6)
 EN VECTOR
 *****
                 STANDARD FORMAT
DISPOSITION OF WASTE
= 1.0 FOR RAW SEWAGE
= 2.0 FOR SEPTIC RAW SEWAGE
= 3.0 FOR SETTLED RAW SEWAGE
= 4.0 FOR SEPTIC SETTLED RAW SEWAGE
= 5.0 FOR BIOLOGICAL TREATMENT EFFLUENT
= 6.0 FOR SAND FILTER EFFLUENT
USED FOR STORING AMOUNT OF CHLORINE USED PER DAY
 1.-15.
 4.
5.
CL(1)=10.0
CL(2)=20.0
CL(3)=8.0
CL(4)=16.0
CL(5)=6.0
CL(5)=6.0
CL(6) = 3.0
DO 600 J=2,JJ
SO(1,J)=SI(1,J)
I=EN(4)+0.01
CL2=SI(1.3)*24.0*10.0*CL(I)*1.0E-6
EN(5)=CL2
IF(ISP) 10,10,11
WRITE(6,400) CL2
FORMAT(1H0,17HCHLORINE USAGE = CONTINUE
                                                                             ,F8.2,10H
                                                                                                        LBS/DAY
                                                                                                                                 )
RETURN
END
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LISTING OF THE MIXER-SPLITTER MODEL

SUBROUTINE MIXER1

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COMMON LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10) COMMON IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR COMMON MODE,NPLNT,LOOPC COMMON SN(25,30),EEN(600),NPOINT(25,2) EN VECTOR 1.-15. STANDARD FORMAT FRACTION OF FLOW TO FIRST OUTPUT STREAM 16. 17: IF(NIN.GT.1) GO TO 30 DO 650 J=2,JJ SO(NOUT,J)=SI(1,J) 650 GO TO 40 CONTINUE CALL ALKMIX(PH) SUM OF INPUT STREAMS IS TEMPORARILY STORED IN SO(NOUT,J) DO 600 J=6,JJ SO(NOUT,J)=0. SO(NOUI,J)=0. DO 600 I=1,NIN SO(NOUT,J)=SO(NOUT,J)+SI(I,J) SO(NOUT,2)=PH SO(NOUT,3)=0.0 SO(NOUT,4)=0.0 SO(NOUT,5)=SI(1,5) DO 610 I=1.NIN 600 SO(NOUT,5)=SI(1,5) DO 61C I=1,NIN SO(NOUT,3)=SO(NOUT,3)+SI(I,3) SO(NOUT,4)=SO(NOUT,4)+SI(I,3)*SI(I,4) IF(SI(I,5),LT,SO(NOUT,5)) SO(NOUT,5)=SI(I,5) CONTINUE SO(NOUT,4)=SO(NOUT,4)/SO(NOUT,3) 610 CONTINUE OUTPUT IF(NOUT.EQ.1) RETURN DO 620 I=1:NOUT SO(I,2)=SO(NOUT.2) SO(I,3)=SO(NOUT,3)*EN(I+15) SO(I,4)=SO(NOUT,4) SO(I,5)=SO(NOUT,5) D0 620 J=6,JJ S0(I,J)=S0(NOUT,J)*EN(I+15) CONTINUE 620 C C RETURN END

LISTING OF THE SIMPLE PHASE SEPARATOR MODEL

SUBROUTINE SEPA01

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COMMON LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10)
COMMON IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR
COMMON MODE,NPLNT,LOOPC
COMMON SN(25,30),EEN(600),NPOINT(25,2)
                EN VECTOR

    1.-15. STANDARD FORMAT
    4. FRACTION OF FLOW TO FIRST OUTPUT STREAM (SUPERNATE)
    5. FRACTION OF SOLIDS TO FIRST OUTPUT STREAM (SUPERNATE)

                SO(1,2)=SI(1,2)
SO(2,2)=SI(1,2)
SO(1,4)=SI(1,4)
SO(2,4)=SI(1,4)
SO(1,5)=SI(1,5)
SO(2,5)=SI(1,5)
               IF(SI(1,6).GT. 0.0) GO TO 10
DO 650 J=6,JJ
SO(1,J)=0.0
SO(2,J)=0.0
CONTINUE
 650
                SO(1,3)=0.0
SO(2,3)=0.0
RETURN
                CONTINUE
                SO(1,6)=EN(4)*SI(1,3)*10.0
SO(2,6)=SI(1,6)-SO(1,6)
               SEPARATE SOLIDS
D0 600 J=13,19
SO(1,J)=SI(1,J)*EN(5)
                SO(2,J)=SI(1,J)-SO(1,J)
CONTINUE
600
C
C
               SOLUBLE COMPONENTS ARE SPLIT ACCORDING TO FLOW

P1=SO(1,6)/SI(1,6)

DO 610 J=20,JJ

SO(1,J)=SI(1,J)*P1

SO(2,J)=SI(1,J)-SO(1,J)
610
Č
                CONTINUE
                CALL BALNCE(2,1)
CALL BALNCE(2,2)
               RETURN
END
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LISTING OF THE PROGRAM TO CONTROL THE MLSS IN THE AERATION TANKS

SUBROUTINE CONTO2

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COMMON LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10)
COMMON IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR
COMMON MODE;NPLNT,LOOPC
COMMON SN(25,30),EEN(600),NPOINT(25,2)
COMMON /A/ TAER,TM,FM,DELS
DATA DELS/0.0/
   WRITTEN BY P. TAN
FOR WWTP SIMULATION
   CONTROLS THE MLSS IN THE AERATION TANKS
  EN VECTOR
                     STANDARD FORMAT
FEED STREAM TO AERATION TANKS (EXCLUDING RECYCLE FLOW)
MLSS REQUIRED IN AERATION TANKS
FIRST OUTPUT STREAM (RECYCLE STREAM)
SECOND RECYCLE STREAM (WASTE SLUDGE FLOW)
   1.-15.
  4•
  5
12
13
  S=EN(4)
CALL STREAM(S)
CALCULATE FRACTION OF INCOMING FLOW TO RECYCLE - A
A=(EN(5)*1.0E-6*SN(IS,6)-SN(IS,11)-DELS)/(SI(1,11)-EN(5)*SI(1,6)*
1 1.0E-6)
  IF(A.LT.0.0) A=0.0
IF(A.GT.1.0) A=1.0
DO 600 I=3,JJ
SO(1,I)=SI(1,I)*A
SO(2,I)=SI(1,I)-SO(1,I)
  SO(1,2)=SI(1,2)
SO(2,2)=SI(1,2)
SO(1,4)=SI(1,4)
SO(2,4)=SI(1,4)
SO(1,5)=SI(1,5)
SO(2,5)=SI(1,5)
SO(2,5)=SI(1,5)
SO(1,7)=0.0
SO(2,7)=0.0
  RETURN
  END
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LISTING OF SUPPORTING MODELS TO HANDLE STREAM VARIABLES 278

SUBROUTINE BALNCE(NVECT,I)
COMMON LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10) COMMON IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR COMMON MODE,NPLNT,LOOPC COMMON SN(25,30),EEN(600),NPOINT(25,2)
WRITTEN BY P. TAN FOR WASTE-WATER SIMULATION BALANCES THE STREAM VARIABLES IN A STREAM (ELEMENTS 8-12)
NVECT=1 FOR SN VECTORS NVECT=2 FOR SO VECTOR C1 = RATIO OF BOD5 TO SOLUBLE DEGRAD ORGANICS (EQUIV CARBON) C2 = RATIO OF BOD5 TO PARTICULATE DEGRAD ORGANICS (EQUIV CARBON) C3 = RATIO OF VSS TO PARTICULATE ORGANICS (EQUIV CARBON)
C1=1•9 C2=1•2 C3=2•1
GO TO (1,2),NVECT
CONTINUE PC=SN(I,15)+SN(I,16)+SN(I,17)+SN(I,18)+SN(I,19) SN(I,12)=C3*PC SN(I,11)=SN(I,12)+SN(I,13)+SN(I,14) SN(I,9)=SN(I,20)+SN(I,21)+SN(I,22) SN(I,8)=SN(I,9)+PC SN(I,10)=C1*(SN(I,20)+SN(I,21))+C2*(SN(I,16)+SN(I,18)+SN(I,19)) SN(I,3)=0.0378*(SN(I,13)+SN(I,14)) + 0.096*(SN(I,15)+SN(I,16))
1 + SN(I, 17) + SN(I, 19)) + 0.104 + SN(I, 18) + 0.1 + SN(I, 6) RETURN
CONTINUE PC=SO(I,15)+SO(I,16)+SO(I,17)+SO(I,18)+SO(I,19) SO(I,12)=C3*PC SO(I,11)=SO(I,22)+SO(I,13)+SO(I,14) SO(I,9)=SO(I,2C)+SO(I,21)+SO(I,22) SO(I,8)=SO(I,9)+PC SO(I,8)=SO(I,9)+PC SO(I,10)=C1*(SO(I,20)+SO(I,21))+C2*(SO(I,16)+SO(I,18)+SO(I,19)) SO(I,3)=0.0378*(SO(I,13)+SO(I,14)) + 0.096*(SO(I,15)+SO(I,16)) 1 +SO(I,17)+SO(I,19)) + 0.104*SO(I,18) + 0.1*SO(I,6)
RETURN

RETUR

C C

. *

C C 2

ç

C C 1

SUBROUTINE CONVERT(M,KIN,JIN,KOUT,JOUT)

C

C 1 600

2 610

3, 620 C 10 C

700

C40 CCCC

C 4 C C

630 C C

640

C5CC

```
LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10)
IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR
MODE,NPLNT,LOOPC
SN(25,30),EEN(600),NPOINT(25,2)
ION V(30)
COMMON
COMMON
COMMON
COMMON
DIMENSION
CONVERTS STREAM VARIABLES FROM POUNDS PER HOUR TO PPM
OR VICE VERSA
INPUT VARIABLES
        1
2
                 FOR CONVERTING LBS/HR TO PPM
M =
            FOR CONVERTING PPM TO LBS/HR
VECTOR TO BE READ FROM
М
    Ξ
KIN
        =
KIN = VECTOR TO BE READ FROM

= 1 FOR SN VECTOR

= 2 FOR SO VECTOR

= 3 FOR SI VECTOR

JIN = ROW NUMBER OF INPUT VECTOR

KOUT = VECTOR TO BE WRITTEN ON TO

= 1 FOR SN VECTOR

= 2 FOR SO VECTOR

= 3 FOR SI VECTOR

JOUT = ROW NUMBER OF OUTPUT VECTOR
VARIABLES 2. TO 6. ARE NOT CHANGED
READ FROM INPUT VECTOR
GO TO (1,2,3),KIN
DO 600 J=2,JJ
V(J)=SN(JIN,J)
GO TO 10
DO 610 J=2,JJ
V(J)=SO(JIN,J)
GO TO 10
DO 620 J=2,JJ
V(J)=SI(JIN,J)
CONTINUE
IF(V(6).GT. 0.0 ) GO TO 40
V(3)=0.0
D0 700 J=6,JJ
V(J)=0.0
GO TO 11
CONTINUE
CONVERSION
GO TO (4,5),M
CONTINUE
CONVERTING FROM LBS/HR TO PPM
SOLUBLE COMPONENTS
V(7)=V(7)/V(6)*1.0E6
V(9)=V(9)/V(6)*1.0E6
DO 630 J=20,JJ
V(J)=V(J)/V(6)*1.0E6
PARTICULATE COMPONENTS
V(8)=V(8)/V(3)*1.0E5
D0 640 J=10,19
V(J)=V(J)/V(3)*1.0E5
G0 TO 11
CONVERTING FROM PPM TO LBS/HR
SOLUBLE COMPONENTS
V(7)=V(7)*V(6)*1.0E-6
V(9)=V(9)*V(6)*1.0E-6
DO 650 J=20, JJ
```

650 C C	$V(J) = V(J) * V(6) * 1 \cdot 0E - 6$
	PARTICULATE COMPONENTS V(8)=V(8)*V(3)*1.0E-5 D0 660 ==10.19
660	DO 660 J=10,19 V(J)=V(J)*V(3)*1.0E-5
11	CONTINUE
660 C 11 C C C	WRITING ON TO OUTPUT VECTOR
c	GO TO (6,7,8),KOUT
C 6 670	DO 670 J=2,JJ Sn(JOUT,J)=V(J) GO TO 12
680	GO TO 12 DO 680 J=2,JJ SO(JOUT,J)=V(J) GO TO 12
8 690 12 C	DO 690 J=2,JJ SI(JOUT,J)=V(J)
12	CONTINUE
L.	RETURN END

2 80 .

SUBROUTINE ALKMIX(PH) COMMON LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10) COMMON IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR COMMON MODE,NPLNT,LOOPC COMMON SN(25,30),EEN(600),NPOINT(25,2) SUBROUTINE TO CALCULATE THE RESULTING PH WHEN STREAMS ARE MIXED SUMALK=0.0 SUMCN=0.0 SUMCT=0.0 Some T=0.00 DO 600 I=1,NIN IF(SI(I,6).LE.0.0) GO TO 600 IF(SI(I,2).LE.0.0) SI(I,2)=7.0 ALK=20.0*SI(I,23)/SI(I,6) CN=71.43*SI(I,27)/SI(I,6) PH=SI(I,2) CALL ALKPH(1,ALK,CT,CN,PH) SUMALK=SUMALK+ALK SUMCT=SUMCT+CT SUMCN=SUMCN+CN 600 Č CONTINUE CALL ALKPH(2, SUMALK, SUMCT, SUMCN, APH) PH=APH RETURN END SUBROUTINE ALKPH(NZ,ALK,CT,CN,PH) DIMENSION XCOF(4) SUBROUTINE TO CALCULATE CT OR PH GIVEN THE OTHER THREE VARIABLES PK1=6.3 PK2=10.3 PKN=9.3 GO TO (1,2), NZ CALCULATE CT CONTINUE CT=(ALK-CN/(1.0+10.0**(PKN-PH)))*(10.0**(PK1-PH)+1.0+10.0** 1(PH-PK2))/(1.0+10.0**(PH-PK2)*2.0) RETURN CALCULATE PH CONTINUE CONTINUE A1=ALK-CN A2=ALK-CT A3=ALK-CN-CT A4=ALK-CN-2.0*CT A5=ALK-CN-2.0*CT XCOF(1)=10.0**(-PK1-PK2-PKN)*A5 XCOF(2)=(A3*10.0**(-PK1-PKN) + A4*10.0**(-PK1-PK2)) XCOF(3)=(A1*10.0**(-PKN) + A2*10.0**(-PK1)) XCOF(4)=ALK XCOF(4) = ALKCALL BISECT(XCOF,H)

PH=-ALOG10(H)

RETURN END

C

CCC

С

C

C C 1

С

C C 2

C

С

С

SUBROUTINE BISECT(XCOF,H)

DIMENSION XCOF(4)

F(XCOF,H)= XCOF(1)+H*(XCOF(2)+H*(XCOF(3)+H*XCOF(4)))

HL=1.0E-5 HR=1.0E-9

10

IF(F(XCOF,HL)*F(XCOF,HR)) 10,20,30
H=SQRT(HL*HR)
IF(F(XCOF,HL)*F(XCOF,H).LT. 0.0) GO TO 2
HL=H
GO TO 4
HR=H
IF(ABS((HL,HD))(HL).CT.0.01) CO TO 200

IF(ABS((HL-HR)/HL).GT.0.01) GO TO 10

H=HL RETURN

```
H=HR
IF(F(XCOF,HL).EQ. 0.0) H=HL
RETURN
```

C 30

C C

С

С

С

24.

C 2 0

С

H=HR IF(ABS(F(XCOF,HL)).LT. ABS(F(XCOF,HR))) H=HL

7

RETURN END

LISTING OF THE REPORT GENERATOR

SUBROUTINE REPTU1 LLST(50), NS(100), EN(100), SI(4,30), SO(4,30), KPRNT(10) IS, NE, JJ, LOOP, NIN, NOUT, MSN, ISP, NC, III, NCALC, NOCOMP, NSR MODE, NPLNT, LOOPC SN(25,30), EEN(600), NPOINT(25,2) COMMON COMMON COMMON COMMON WRITTEN BY P. TAN FOR WWTP SIMULATION REPORTS ON THE VARIOUS STREAMS IN THE PLANT FORMAT LENGTH 35 Ξ =0.0 RAW FEED TO PLANT (STREAM NUMBER) INPUT TO PRIMARY CLARIFIER EFFLUENT OF PRIMARY CLARIFIER UNDERFLOW OF PRIMARY CLARIFIER INPUT TO AERATION TANKS OUTPUT OF AERATION TANKS INPUT TO SECONDARY CLARIFIER INPUT TO SECONDARY CLARIFIER EFFLUENT OF SECONDARY CLARIFIER UNDERFLOW OF SECONDARY CLARIFIER RETURN ACTIVATED SLUDGE WASTE ACTIVATED SLUDGE INPUT TO DIGESTERS DIGESTER SUPERNATANT DIGESTED SLUDGE INPUT TO THICKENER THICKENED SLUDGE INPUT TO TRICKLING FILTER OUTPUT OF TRICKLING FILTER PLANT BYPASS ĒN(35) = SET STREAM NUMBER TO 0.0 IF DO NOT WISH TO REPORT ON STREAM IF (MODE.EQ. U) RETURN WRITE(6,79) WRITE(6,80) WRITE(6,82) WRITE(6,80) WRITE(6,85) LOOPC DO 50 I=1,20 S=EN(I+15) IF(S.LE.0.0) GO TO 50 CALL STREAM(S) IT=IS CALL CONVERT(1,1,IT,3,1) SI(1,1)=S IW=29+I WRITE(6,IW) SI(1,1),SI(1,3),(SI(1,J),J=7,12) CONTINUE WRITE(6,79) MODE=0 ,F5.0,7F10.2) ,F5.0,7F10.2) ,F5.0,7F10.2) ,F5.0,7F10.2) ,F5.0,7F10.2) ,F5.0,7F10.2) ,F5.0,7F10.2) FORMAT(1H0,20HRAW FEED FORMAT(1H0,20HINPUT TO PRI. FORMAT(1H0,20HEFF. OF PRI. FORMAT(1H0,20HUNDERFLOW PRI FORMAT(1H0,20HUNDERFLOW PRI FORMAT(1H0,20HINPUT TO AER FORMAT(1H0,20HOUTPUT OF AER CLAR. CLAR I CLAR TANKS TANKS

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С

C C 50

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CC331233333333333

FORMAT(1HU,20HINPUT TO SEC. CLAR.	,F5.0,7F10.2)
FORMAT(1HU,20HEFF. OF SEC. CLAR.	•F5•0•7F10•2)
FORMAT(1H0,20HUNDERFLOW SEC CLAR	•F5•0•7F10•2)
FORMAT(1H0,20HRETURN ACT. SLUDGE	•F5•0•7F10•2)
FORMAI(1HU,20HWASIE ACT. SLUDGE	•F5•Q•7F10•2)
EORMAI(1H0,20HINPUT TO DIGESTER	•F5•Q•7F10•2)
FORMAI(1HU,2UHDIGESTER SUPERNATE	•F5•0•7F10•2)
FORMAI(1HU,20HDIGESTED SLUDGE	•E5•0•7E10•2)
FORMAI(1H0,20HINPUT TO THICKENER	•F5•0•7F10•2)
FORMAT(1HU,20HTHICKENED SLUDGE	•F5•U•7F10•2)
FORMAT(1HU,20HTRICK. FILT. INPUT	•F5•0•7F10•2)
EORMAI(1H0,20HTRICK, FILT, OUTPUT	•F5•V•7F10•2)
FORMAT(1H0,20HPLANT BYPASS	• E5 • C • 7 E 10 • 2)
FORMAT(1HU,2UH	•F5•0,7F10•2)
FORMAT(1H1)	
FORMAT(1H0,120H\$\$\$\$\$\$\$\$\$\$\$\$\$\$\$	
155555555555555555555555555555555555555	\$
2 5 5 1 1 1 1 1 1 1 1 1 1	
FORMAT(1Hu,1H\$,2UX,42HSUMMARY REPO	RI ON PLANT OPERATIONS - CASE ,
1_15,51X,1H\$)	
FORMAT(/7/,7X,6HSTREAM, 8X,6HNUMBE 17X,3HDOC,7X,3HBOD,8X,2HSS,7X,3HVSS	R,5X,4HFLOW,6X,3H DO,7X,3HIOC,
T (V) 3HDOC (V) 3HROD 8X 95H22 (X) 3HA22	9//)

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RETURN END

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LISTING OF THE COST ESTIMATION PROGRAM

SUBROUTINE COST1

LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10) IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR MODE,NPLNT,LOOPC SN(25,30),EEN(600),NPOINT(25,2) ION.TITLE (10),ELECT(3) COMMON COMMON COMMON COMMON DIMENSION.

CALCULATES THE FIXED CAPITAL INVESTMENT, THE OPERATING AND MAINTENANCE COSTS OF A WASTEWATER TREATMENT PLANT, PLACES A SURTAX ON EFFLUENT IF THE EFFLUENT QUALITY IS POU COSTS TAKEN FROM ECKENFELDER AND BARNHARDT, AND FROM SMITH IS POUR. SMITH.

EN VECTOR

C

- 1. MODULE NUMBER 2. MODULE TYPE 3. MODULE LENGTH 4.-15. = TWELVE ZEROES 2.3 UNIT PROCESS PRETREATMENT NO. OF PRETREATMENT UNITS PRIMARY SEDIMENTATION NO. OF PRI. SETTLING TANKS ACTIVATED SLUDGE TANKS NO. OF AERATION TANKS AIR BLOWERS NO. OF AIR BLOWERS NO. OF AIR BLOWERS SLUDGE RETURN PUMPS NO. OF FINAL TANKS SLUDGE RETURN PUMPS NO. OF SLUDGE RETURN PUMPS ANAEROBIC DIGESTION NO. OF ANAEROBIC DIGESTERS CHLORINATION NO. OF CHLORINATION UNITS
- CHLORINATION UNITS FILTRATION VACUUM FILTERS INCINERATION SLUDGE INCINERATERS DRYING BEDS DRYING BEDS 1234567890 12333333333344443 NO. OF VACUUM NO. OF SLUDGE NO. OF SLUDGE NO. OF THICKENING THICKENERS HAULAGE SLUDGE NO. OF SLUDGE
- SLUDGE HALL 0.0 TRICKLING FILTERS NO. OF TRICKLING FILTERS AERATED LAGOONS NO. OF LAGOONS TYPE OF PLANT (FOR UTILITIES COST) 44 • 45 • 46.

Ξ 47. NOT NOT PRESENTLY USED 48. ŬŠĒĎ DESIGNED FLOW MIGPD LABOUR COST \$/MAN-HOUR ENR INDEX FOR YEAR LABORATORY COSTS \$/YR STREAM NUMBER OF EFFLU ACTUAL PLANT FLOW MIGP OF EFFLUENT FLOW MIGPD

PARAMETER DESIGNED FLOW MIGPD SURFACE AREA PER UNIT /1000 SQFT VOLUME PER UNIT MIGPD CAPACITY PER UNIT CPM/1000 SURFACE AREA/1000 SQFT PER UNIT CAPACITY MIGPD VOLUME PER UNIT /1000 CUET DESIGNED FLOW MIGPD SURFACE AREA PER UNIT/100 SQFT LBS SOLIDS/DAY PER UNIT SURFACE AREA SQFT SURFACE AREA PER UNIT SQFT SLUDGE VOLUME MIG/YR VOLUME PER UNIT/1000 CUFT SURFACE AREA PER UNIT/1000 SQFT PRIMARY PLANT TRICKLING FILTER PLANT ACTIVATED SLUDGE PLANT 1.0 2.0 3.0 FOR FOR = = FOR

(FOR SURCHARGE PURPOSES)

A=1H* B=1H-ELECT(1)=101000.0 ELECT(2)=162000.0 ELECT(3)=378000.0

C C	CAPITAL COST ESTIMATE . 286	
C	WRITE(6,400) WRITE(6,401) (A,I=1,22) READ (5,200) TITLE WRITE(6,550) TITLE	
C C	CALCULATE THE CAPITAL COSTS OF THE VARIOUS UNIT PROCESSES Q=EN(50) ENRFAC=EN(52)/1120.0 UPC=0.0	
C	DO 600 I=1,15 IF(EN(14+2*I).LE. 0.0) GO TO 600 GO TO (1,2,3,4,5,6,7,8,9,10,11,12,13,14,15),I	
C 1	COST=21800.0*(Q/EN(17))**0.63*ENRFAC*EN(17) WRITE(6.451) COST GO TO 610	
2	COST=(17300.0*EN(18)+6700.0*EN(18)**0.1)*ENRFAC*EN(19) WRITE(6,452) COST	
3	GO_TO_610 COST=(27000.0*EN(20)+67000.0)*ENRFAC*EN(21) WRITE(6,453) COST GO_TO_610	
4	COST=(13600.0+7600.0*EN(22))*ENRFAC*EN(23) WRITE(6,454) COST GO TO 610	
5	COST=(16200.0*EN(24)+6900.0/EN(24)**0.13)*ENRFAC*EN(25) WRITE(6,455) COST GO TO 610	
6	COST=(4700.0+1740.0*Q/EN(27))*ENRFAC*EN(27) WRITE(6,456) COST GO TO 610	
7	COST=(1340.0*EN(28)+13800.0*EN(28)**0.13)*ENRFAC*EN(29) WRITE(6,457) COST GO TO 610	
8	GOST=(12600.0*(Q/EN(30))**0.47)*ENRFAC*EN(31) WRITE(6,458) COST GO TO 610	
9	COST=(16500.0+48.0*EN(32))*ENRFAC*EN(33) WRITE(6,459) COST GO TO 610	
10	COST=(7.1*EN(34)/EN(35)+0.3*(EN(34)/EN(35))**1.61)*ENRFAC*EN(35 WRITE(6,460) COST GO TO 610	i)
11	COST=2.23*EN(36)*EN(37) WRITE(6,461) COST GO TO 610	
12	COST=EN(38)*(24200.0+11700.0/EXP(EN(38)/13.3))*ENRFAC*EN(39) WRITE(6,462) COST GO TO 610	
13 14	GO TO 600 COST=66000.0*EN(42)**0.6*ENRFAC*EN(43) WRITE(6.464) COST	
15	GO TO 610 COST=2.5*EN(44)*EN(45)*ENRFAC WRITE(6,465) COST	
C 610 600	UPC=UPC+COST CONTINUE	
c	WRITE(6,402) (B,I=1,14) WRITE(6,403) UPC	
C C	CALCULATE THE CAPITAL COST OF THE CONTROL HOUSE COST=58600.0*0*0*7*ENRFAC	
с'	WRITE(6,466) COST PPC=UPC+COST CALCULATING THE CAPITAL COST OF THE PLANT SITE + IMPROVEMENTS COST=4600.0*0**0.88*ENRFAC WRITE(6,467) COST WRITE(6,467)	
С	PPC=PPC+COST WRITE(6,402) (B,I=1,14)	

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287
           WRITE(6,404) PPC
           CALCULATE THE ENGINEERING COST FACTOR
CENG=0.08*(1.0E6/PPC)**0.146
CALCULATE THE ENGINEERING COSTS
COST=CENG*PPC
WRITE(6,4.5) COST
DPC=PPC+COST
           WRITE(6,402) (B,I=1,14)
WRITE(6,406) DPC
           CALCULATING THE CONTRACTOR'S FEE COST=DPC*U.1
           WRITE(6,407)
                                    COST
           FCI=DPC+COST
CALCULATING THE CONTINGENCY COSTS
COST=DPC*0•15
WRITE(6,408) COST
FCI=FCI+COST
           WRITE(6,402) (B,I=1,14)
WRITE(6,409) FCI
WRITE(6,402) (B,I=1,14)
           ******************
           OPERATING AND MAINTENANCE COST ESTIMATE
           WRITE(6,410)
WRITE(6,401)
                                    (A, I=1, 21)
           READ(5,200) TITLE
           WRITE(6,550)
WRITE(6,411)
                                    TITLE
           SUMHRS=0.0
           UPOM=0.0
Q=EN(55)
           DO 620 I=1,15
           IF(EN(14+2*I).LE.0.0) GO TO 620
GO TO (21,22,23,24,25,26,27,28,29,30,31,32,33,34,35),I
C
21
           HOUR=133.0*Q+510.0*Q**0.37
COST=HOUR*EN(51)
           WRITE(6,501) HOUR, COST
           GO TO 630
HOUR=(222.0*EN(18)+555.0*EN(18)**0.5)*EN(19)
COST=HOUR*EN(51)
WRITE(6,502) HOUR.COST
GO TO 630
HOUR=(720.0*Q+590.0*Q/EN(20)**0.67)
22
23
           COST=HOUR*EN(51)
WRITE(6,503) HOUR,COST
GO TO 630
GO TO 620
24
           GO TO 620
GO TO 620
HOUR=(13.0*EN(28)+144.0*EN(28)**0.5)
COST=HOUR*EN(51)
25
26
27
           WRITE(6,507) HOUR,COST
GO TO 630
HOUR=(30.0*Q+170.0*Q**C.37)
COST=HOUR*EN(51)
28
           WRITE(6,508) HOUR,COST
GO TO 630
HOUR=(0.27*Q+154.0*Q**0.37)
COST=HOUR*EN(51)
29
           WRITE(6,509) HOUR,COST
GO TO 630
HOUR=(400.0*Q+1600.0*Q**0.37)
30
           COST=HOUR*EN(51)
WRITE(6,510) HOUR,COST
GO TO 630
HOUR=(0.014*EN(36)+20.0*EN(36)**0.37)
31
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288 COST=HOUR*EN(51) WRITE(6,511) HOUR,COST GO TO 630 GO TO 650 GO TO 620 GO TO 620 HOUR=(10.0*EN(42)+120.0*EN(42)**0.5) COST=HOUR*EN(51) 32 33 34 WRITE(6,514) HOUR,COST GO TO 630 HOUR=(50.0*EN(44)+100.0*EN(44)**0.5) COST=HOUR*EN(51) WRITE(6,515) HOUR,COST 35 C 630 CONTINUE SUMHRS=SUMHRS+HOUR UPOM=UPOM+COST 620 C CONTINUE WRITE(6,415) (B,I=1,28) WRITE(6,413) SUMHRS, UPOM C CALCULATE THE SUPERVISION COST COST=0.1*UPOM WRITE(6.414) COST DOM=UPOM+COST CALCULATE THE UTILITIES COST C CALCULATE THE UTILITIES COST KI=EN(46)+0.01 COST=ELECT(KI)*Q**0.65*EN(47) WRITE(6,416) COST DOM=DOM+COST CALCULATE SLUDGE HAULAGE COSTS COST=8000.0*EN(40)**0.5 WRITE(6,513) COST DOM=DOM+COST CALCULATE THE CHEMICALS COST COST=0.06*UPOM WRITE(6,417) COST C C WRITE(6,417) COST DOM=DOM+COST CALCULATE PLANT SUPPLIES COST=0.05*UPOM WRITE(6:418) COST DOM=DOM+COST C WRITE(6,419) (B, WRITE(6,420) DOM (B,I=1,14) CALCULATING LABORATORY COSTS COST=EN(53) С WRITE(6,421) COST OMNET=DOM+COST CALCULATING SURCHARGE ON EFFLUENT (IF EFFLUENT QUALITY IS POOR) IF(EN(54).LE. 0.0) GO TO 50 . • C S= EN(54) CALL STREAM(S) LBS=SN(IS,3)*3.0E-4 IF(SN(IS,10).LT.LBS) GO TO SURBOD=(SN(IS,10)-LBS)*0.05 GO TO 52 SURBOD=0.0 LE(SN(IS,11).LT.LBS) CO TO 51 51 SURBOD=0.0 IF(SN(IS,11).LT.LBS) GO TO SURSS=(SN(IS,11)-LBS)*0.05 GO TO 54 SURSS=0.0 TOTSUR=SURBOD+SURSS WRITE(6,422) TOTSUR OMNET=OMNET+TOTSUR CONTINUE 52 53 53 54 50 C WRITE(6,419) WRITE(6,423) (B,I=1,14) OMNET CCCCC THE COST OF THE FIXED CAPITAL INVESTMENT. INTEREST RATE OF 8.0 PERCENT AND A PLANT LIFE OF 25 THAT THE ANNULTY PAYMENTS REPRESENT DEET RETIREMENT CALCULATING BASED ON AN YEARS. NOTE INTEREST AND CHARGED CASE OF BORROWED CAPITAL IN THE OR

CCC	DEPRECIATION AND OPPORTUNITY COST IN THE CASE OF OWNED CAPITAL. CAPITAL RECOVERY FACTOR = S*(1+S)**N/((1+S)**N-1) WHERE S IS THE INTEREST RATE AND N THE LIFE OF THE PLANT CF=(1.0+0.08)**25.0 COST=FCI*0.08*CF/(CF-1.0) WRITE(6,424) COST
с	WRITE(6,419) (B,I=1,14) TAC=OMNET+COST WRITE(6,428) TAC WRITE(6,419) (B,I=1,14) WRITE(6,430)
CC24440006789013	<pre>WRITE(6,430) FORMAT(10A8) FORMAT(1H1,30X,22HFIXED_CAPITAL_ESTIMATE_) FORMAT(1H0,30X,22A1) FORMAT(1H0,50X,14A1) FORMAT(1H0,10X,40HDTAL_UNIT_PROCESSES (INSTALLED) \$,F14.2) FORMAT(1H0,10X,40HDHYSICAL_PLANT_COST \$,F14.2) FORMAT(1H0,10X,40HDIRECT_PLANT_COST \$,F14.2) FORMAT(1H0,10X,40HDIRECT_PLANT_COST \$,F14.2) FORMAT(1H0,10X,40HCONTRACTOR'S FEE (0.1 DPC) \$,F14.2] FORMAT(1H0,10X,40HCONTINGENCY (0.15 DPC) \$,F14.2] FORMAT(1H0,10X,40HFIXED_CAPITAL_COST \$,F14.2] FORMAT(1H0,10X,40HFIXED_CAPITAL_COST \$,F14.2] FORMAT(1H0,10X,40HFIXED_CAPITAL_COST \$,F14.2] FORMAT(1H0,15X,15HUNIT_PROCESSES ,15X,9HMAN-HOURS,10X,9HCOST \$/YR) FORMAT(1H0,10X,30HTOTAL_UNIT_PROCESSES 0_AND_M,F14.2,4X,1H\$,</pre>
415 415 417 412 412 422 422 422 422 422 422 422 422	<pre>1 F14.2) FORMAT(1HU,1UX,3UHSUPERVISION (0.1 L) FORMAT(1HU,1UX,3UHSUPERVISION (0.1 L) FORMAT(1HU,1UX,3UHUTILITIES FORMAT(1HU,1UX,3UHUTILITIES FORMAT(1HU,1UX,3UHUTICALS (0.06 L) FORMAT(1HU,1UX,3UHUTICALS (0.06 L) FORMAT(1HU,1UX,3UHUTICALS (0.05 L) FORMAT(1HU,1UX,3UHUTICAL ANNUAL COST)18X,1H\$,F14.2) FORMAT(1HU,1UX,3UHITICAL ANNUAL COST)18X,1H\$,F14.2)</pre>
4 4 4 4 4 4 4 4 4 4 4 4 4 4 4 4 4 5	FORMAT(1H1)\$,F14.2)FORMAT(1H0,10X,40HPRETREATMENT\$,F14.2)FORMAT(1H0,10X,40HPRIMARY SETTLING TANKS;F14.2)FORMAT(1H0,10X,40HACTIVATED SLUDGE TANKS;F14.2)FORMAT(1H0,10X,40HAIR BLOWERS;F14.2)FORMAT(1H0,10X,40HAIR BLOWERS;F14.2)FORMAT(1H0,10X,40HAIR BLOWERS;F14.2)FORMAT(1H0,10X,40HANAEROBIC DIGESTERS;F14.2)FORMAT(1H0,10X,40HANAEROBIC DIGESTERS;F14.2)FORMAT(1H0,10X,40HCRINATOR;F14.2)FORMAT(1H0,10X,40HCRINATOR;F14.2)FORMAT(1H0,10X,40HSLUDGE INCINERATORS;F14.2)FORMAT(1H0,10X,40HSLUDGE THICKENERS;F14.2)FORMAT(1H0,10X,40HTRICKLING FILTERS;F14.2)FORMAT(1H0,10X,40HTRICKLING FILTERS;F14.2)FORMAT(1H0,10X,40HTRICKLING FILTERS;F14.2)FORMAT(1H0,10X,40HCNTROL HOUSE;F14.2)FORMAT(1H0,10X,40HRETREATMENT;F14.2;5X;F14.2)FORMAT(1H0,10X,30HPRETREATMENT;F14.2;5X;F14.2)FORMAT(1H0,10X,30HACTIVATED SLUDGE;F14.2;5X;F14.2)FORMAT(1H0,10X,30HACTIVATED SLUDGE;F14.2;5X;F14.2)FORMAT(1H0,10X,30HACTIVATED SLUDGE;F14.2;5X;F14.2)FORMAT(1H0,10X,30HACTIVATED SLUDGE;F14.2;5X;F14.2)FORMAT(1H0,10X,30HACTIVATED SLUDGE;F14.2;5X;F14.2)FORMAT(1H0,10X,30HACTIVATED SLUDGE;F14.2;5X;F14.2)FORMAT(1H0,10X,30HACTIVATED SLUDGE;F14.2;5X;F14.2)FORMAT(1H0,10X,30HACTIVATED SLUDGE;F14.2;5X;F14.2)FORMAT(1H0,10X,30HACTIVATED SLUDGE;F14.2;5X;F14.2)FORMAT(1H0,10X,30HACTIVATED SLUDGE;F14.2;5X;F14.2)
550 C	FORMAT(1H0,5X,10A8,///)

RETURN. END

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PROGRAM WWTP(INPUT,OUTPUT,WWRUN,TAPE2=INPUT,TAPE6=OUTPUT, ITAPE5=WWRUN) LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10) IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR MODE,NPLNT,LOOPC SN(25,3u),EEN(600),NPOINT(25,2) ION TITLE(10) ION NAME(20) COMMON COMMON COMMON DIMENSION DIMENSION GEMCS/INTERCOM ***** PRINTING CONTROL CHARACTERS ***** ***** PRINTING CONTROL CHARACTERS *****
KPRNT(1)=1 CAUSES PRINTING OF LLST:NS
KPRNT(2)=1 CAUSES PRINTING OF INITIAL STREAMS
KPRNT(2)=0 SUPPRESSES ABOVE PRINTING
KPRNT(3)=1 CAUSES PRINTING OF MODULES SETS
KPRNT(3)=0 SUPPRESSES ABOVE PRINTING
KPRNT(4)=1 CAUSES PRINTING OF NPOINT
KPRNT(4)=1 CAUSES PRINTING OF SN TABLE ON ENTERING LOOP
KPRNT(5)=1 CAUSES PRINTING OF FINAL OUTPUT STREAMS
KPRNT(5)=0 SUPPRESSES ABOVE PRINTING
KPRNT(6)=1 CAUSES PRINTING OF FINAL OUTPUT STREAMS
PRINTING OF FINAL OUTPUT STREAMS OF UNIMPORTANT
MODULES MAY BE SUPPRESSED BY PUTTING A NEGATIVE
SIGN ON THE MODULE TYPE FOR THE EQUIPMENT.
KPRNT(6)=0 SUPPRESSES ABOVE PRINTING
KPRNT(7)=1 PRINTS THE FRACTIONAL CHANGES IN STREAMS TESTED-CONTL1
KPRNT(7)=0 SUPPRESSES ABOVE PRINTING
KPRNT(8)=1 SUPPRESSES ABOVE PRINTING
ALL PRINTING OF INPUT AND OUTPUT STREAMS LEAVING A MODULE
ARE SUPPRESSED BY SETTING ISP=0. REWIND5 NAME(1)=8H MIXER1 NAME(2)=8H CONTL1 CONTLI NAME(3) = 8HSETST1 NAME(3)=0 NAME(4)=8H NAME(5)=8H NAME(6)=8H NAME(7)=8H NAME(8)=8H NAME(9)=8H SETSPI SEPÃ01 PRISET1 ACTSL1 DPTRI SECLARI TRICKI NAME(10)=8H NAME(11)=8H NAME(11)=8H NAME(12)=8H NAME(13)=8H NAME(14)=8H NAME(16)=8H GRIT1 ĂNDIG1 CHLORI COSTI SCREEN1 REPI01 NAME(17) = 8HCONT02 NAME(18)=8H CONV01 NAME(19)=8H NAME(20)=8H CASET THICK1 WRITE(6,234) FORMAT(1HU,5UHTO ANSWER QUESTIONS BELOW TYPE 1.0/YES OR 0.0/NO) III=25CALL DLOAD1 IK IS AN INDICATOR AS TO WHETHER A STRAIGHT THROUGH CALCULATION IS USED (IK=1) OR WHETHER A RECYCLE IS PRESENT (IK=0) IK=0 IF(LOOP.EQ.999) IK=1

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WRITE(6,240) FORMAT(1H0,30HWANT TO SET UP A NEW PLANT -READ(2,221) FLAG IF(FLAG.LE.U.0) GO TO 92 240) CONTINUE CALL FLWCHT 90 CALL FLW CONTINUE 92 CONTINUE WRITE(6,238) FORMAT(1X,45HWANT TO SET A RANDOM FLOW AND CASE STUDY -READ(2,221) FLAG IF(FLAG.LE.U.U) GO TO 93 96 238) Ċ 236 Ċ Ċ Ċ CONTINUE SETS A CASE RUN AROUND INPUT FEED CASET MUST HAVE EQUIPMENT NUMBER 25 AND SHOULD NOT BE INCLUDED IN THE CALCULATION ORDER. CALL DISKIO(1,25) CALL CASET GO TO 1000 C 93 CONTINUE CALL STREAM(1.0) IN=IS CALL STREAM(2.0) IF(IS.LE.III) GO CALL STREAM(0.0) SN(IS.1)=2.0 TO 94 CONTINUE 94 DO 600 J=2,JJ SN(IS,J)=SN(IN,J) 600 1000 CONTINUE LOOPC=1 WRITE (6,196) LOOPC C 100 C C CONTINUE CALCULATING EQUIPMENT IN CALCULATION ORDER LIST NC = IMODE=U KTEST=0 LOOP=1 IF(IK•EQ•U) GO TO 98 MODE = 1KTEST=1 L00P=999 CONTINÚÉ 98 C C 102 READ EN VECTOR IF (KPRNT(5)) 108,108,104 CONTINUE 104 WRITE (6,182) LOOP DO 106 IKE=1,III IF (SN(IKE,1).LE.C.O) GO TO 106 WRITE (6,194) (SN(IKE,J),J=1,JJ) CONTINUE 106 Ĉ 108. NE=LLST(NC) MM=NE CALL DISKIO (1,MM) NN=EN(3)+.001 NIN=EN(6)+.001 NOUT=EN(11)+.001 NTYPE=ABS(EN(2))+.001 KSW=0 IF (EN(2).LE.U.) KSW=1 ISP=0 SUPPRESSES ALL PRINTING OF INPUT AND OUTPUT STREAMS C ĪSP=0 CCC FINDING INPUT STREAMS

IF(NIN.EQ.0) GO TO 128 DO 122 I=1.NIN S=EN(I+6) S=EN(1+6) CALL STREAM (S) IF (IS-III) 114,114,110 CONTINUE SI(I,1)=S DO 112 J=2,JJ SI(I,J)=0. CONTINUE CONTINUE 110 112 GO TO 122 DO 116 J=1,JJ SI(I,J)=SN(IS,J) CONTINUE IF (INT(S+.U01)) 122,122,118 114 116 118 CONTINUE M=S+.001 IF (NS(M)-6) 122,120,120 IF (NS(M)-6) 122,120,120 CONTINUE SN(IS,1)=0. CONTINUE IF (ISP) 128,128,124 WRITE (6,184) NE ,NAME(NTYPE) DO 126 IKE=1,NIN WRITE (6,194) WRITE (6,194) (SI(IKE,J),J=1, 120 122 124 (SI(IKE,J),J=1,JJ)126 C 128 C CONTINUE CONTINUE CALL MODULE (NTYPE) STORING OUTPUT STREAMS AND PRINTING IF(NOUT) 166,166,136 CONTINUE DO 150 I=1,NOUT 136 IF(\$0(1,3).LE.U.0) GO TO 150 S=EN(1+11) S=EN(I+11) SO(I,1)=S M=S+.001 IF(NS(M)-6) 140,150,140 CALL STRFAM(S) IF (IS-III) 146,146,142 CALL STREAM (0.) IF (IS-III) 146,146,144 WRITE (6,190) GO TO 150 DO 148 J=1,JJ SN(IS,J)=SO(I,J) CONTINUE CONTINUE 140 142 144 146 148 150 C ·IF (ISP) 158,158,154 WRITE (6,186) NE , NAME(NTYPE) DO 156 IKE=1,NOUT 154 WRITE(6,194) WRITE (6,194) (SO(IKE,J),J=1,JJ) CONTINUE 156 C IF (LOOP-999) 166,160,160 IF (KSW.NE.U) GO TO 166 IF(KPRNT(6).LT.1) GO TO 166 WRITE(6,188) NE, NAME(NTYPE) 158 160 DO 164 IKE=1.NOUT WRITE(6,194) WRITE(6,195) (SO(IKE,J),J=1,JJ) FORMAT(1X,5F12.3) CONTINUE 195 164 C 166 NC=NC+1 IF (NC-NCALC) 102,102,168

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C 168 IF (LOOP-999) 170,172,172 CONTINUE NC=0 170 ĜÕ TO 166 IF (KTEST) 174,174,176 IF (KTĒ NC=0 KTEST=1 172 174 MODE=1 GO TO 166 176 CONTINUE CONTINUE CALL ICLOAD WRITE(6,220) FORMAT(1X,35HWANT TO MAKE ANY PLANT CHANGES -READ(2,221) FLAG IF(FLAG.LE.0.0) GO TO 230 CALL ENSET LOOPC=LOOPC+1 WRITE(6,196) LOOPC GO TO 100 CONTINUE WRITE(6,238) 220) 230 WRITE(6,238) FORMAT(1X,45HWANT TO GENERATE A NEW RANDOM CASE STUDY -READ(2,221) FLAG IF(FLAG.LE.U.U) GO TO 95 GO TO 236 232) CONTINUE WRITE(6,240) READ(2,221) FLAG IF(FLAG.GE.1.0) GO TO 90 95 STOP С 182 (1HU,32H\$\$\$\$\$ (1HU,31H\$\$\$\$\$ SN TABLE ON ENTERING LOOP ,15,7 INPUT STREAMS FOR MODULE ,15,4H FORMAT ,I5,7H \$\$\$\$\$) FORMAT 184 1A8,7H FORMAT \$\$\$\$\$ (1HU,32H\$\$\$\$\$ OUTPUT STREAMS FOR MODULE , 15,4H - ,A8, 186 188 FINAL OUTPUT STREAMS FOR MODULE , 15,4H ERROR IN SN *****) *** END OF EXECUTION 190 192 194 *****) 196 **** RUN, 14,26H FORMAT(10A8) 198 200 221 C FORMAT(1H0,10A8) FORMAT(F12.3)

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END

c c	SUBROUTINE FLWCHT	294
CC	COMMON LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(1 COMMON IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP COMMON MODE,NPLNT,LOOPC COMMON SN(25,30),EEN(600),NPOINT(25,2)	0) ,nsr
νονου	SETS THE CALCULATION ORDER OF A GIVEN PLANT AND SETS THE VARIOUS EQUIPMENT PARAMETERS	
-	LP1=0 LP2=0 LLST(1)=2 LLST(2)=3 LLST(3)=4 NCALC=3	·
C 400 200 C	WRITE(6,400) FORMAT(1H0,22HANY PRIMARY TANKS –) READ(2,200) F FORMAT(F12.3)	
	SETTING SPLITTER BEFORE PRIMARY SETTLER MQ=NPOINT(4,1) EEN(MQ+16)=F EEN(MQ+17)=1.0-F EEN(MQ+18)=0.0	
C C	IF(F.LT.1.0) GO TO 10	
C C	ASKING FOR DETAILS ON PRIMARY SETTLER NCALC=NCALC+1 LLST(NCALC)=5 MQ=NPOINT(5,1)	
500	WRITE(6,500) FORMAT(1X,20HPRIMARY CLARIFIER) WRITE(6,510)	
510	FORMAT(5X,30HNUMBER OF TANKS IN PARALLEL =) READ(2,200) F EEN(MQ+4)=F WRITE(6,520)	
52Ŭ	FORMAT(5X,26HLENGTH OF TANK (IN FT) =) READ(2,200) F EEN(MQ+16)=F	·
530	WRITE(6,530) FORMAT(5X,25HWIDTH OF TANK (IN FT) =) READ(2,200) F EEN(MQ+17)=F	
540	WRITE(6,540) FORMAT(5X,25HDEPTH OF TANK (IN FT) =) READ(2,200) F EEN(MQ+18)=F	
550	WRITE(6,550) FCRMAT(5X,26HNUMBER OF SLUDGE PUMPS =) READ(2,200) F EEN(MQ+23)=F	
560	WRITE(6,560) FORMAT(5X,35HCAPACITY OF SLUDGE PUMPS (IGPH) =) READ(2,200) F EEN(MQ+24)=F	
570	WRITE(6,570) FORMAT(5X,40HFRACTION OF CAPACITY PUMPS ARE USED =) READ(2,200) F EEN(MQ+25)=F	
C 10	NCALC=NCALC+1 LLST(NCALC)=6	
C 410	WRITE(6,410) Format(1H0,30Hany activated sludge tanks -) Read(2,200) F	
C C	SETTING THE SPLITTER BEFORE THE ACTIVATED SLUDGE TANKS MQ=NPOINT(6,1)	

	EEN(MQ+16)=F EEN(MQ+17)=1.0-F	Z
c	IF(F•LT•1•0) GO TO 40	
C	ASKING FOR DETAILS ON THE ACTIVATED SLUDGE TANKS LP1=1 NGALC-NGALC-1	
	NCALC=NCALC+1 LLST(NCALC)=7 MQ=NPOINT(7,1)	
580	WRITE(6,580) FORMAT(1X,25HACTIVATED SLUDGE TANKS)	
	WRITE(6,510) READ(2,200) F	
	EEN(MQ+4)=F WRITE(6,520)	
	READ(2,200) F EEN(MQ+16)=F	
	WRITE(6,530) READ(2,200) F EEN(MQ+17)=F	
-	WRITE(6,540) READ(2,200) F	
	EEN(MQ+18)=F WRITE(6,545)	
545	FORMAT(5X,37HNUMBER OF STIRRED TANKS IN SERIES = READ(2,200) F)
	EEN(MQ+34)=F WRITE(6,555)	
555	FORMAT(5X,30HARE YOU USING STEP AERATION) READ(2,200) F EEN(MQ+19)=1.0	
	$IF(F \cdot LE \cdot 0 \cdot 0) = EEN(MQ + 19) = EEN(MQ + 34)$ IF(F \ LE \ 0 \ 0) GO TO 35	
	SUM=0.0 DO 600 KM=1.10	
565	WRITE(6,565) KM FORMAI(1X,25HFRACTION OF FEED TO TANK ,13,4H =)	
	READ(2,200) F EEN(MQ+34+KM)=F	
600	SUM=SUM+F IF(SUM•GE•1•0) GO TO 35 CONTINUE	
35 575	WRITE(6,575) FORMAT(5X,25HNUMBER OF AIR BLOWERS =)	
212	READ(2,200) F EEN(MQ+45)=F	
585	WRITE(6,585) FORMAT(5X,32HCAPACITY OF BLOWERS (SCF/HR) =)	
~	READ(2,200) F EEN(MQ+46)=F	
С	NCALC=NCALC+1 LLST(NCALC)=8	
C C	ASKING FOR DETAILS OF THE SECONDARY CLARIFIERS	
	MQ=NPOINT(8,1) WRITE(6,590)	
590	FORMAT(1HU,2UHSECONDARY CLARIFIER) WRITE(6,510)	
	READ(2,200) F EEN(MQ+4)=F WRITE(6,505)	
505	FORMAT(5X,25HSURFACE AREA (SQ FT) =) READ(2,200) F	
	EEN(MQ+18)=F WRITE(6,550)	
	READ(2,200) F EEN(MQ+20)=F	
١	EEN(MQ+20)=F WRITE(6,560) READ(2,200) F EEN(MQ+21)=F	
	WRITE(6,570) READ(2,200) F	
	ÊÊN(MQ+22)=F	

C NCALC=NCALC+1 LLST(NCALC)=9 NCALC=NCALC+1 LLST(NCALC)=10 c ASKING FOR DETAILS OF THE WASTE SLUDGE POLICY MQ=NPOINT(10,1)WRITE(6,535) FORMAT(1X,30HFRACTION OF SLUDGE WASTED = 535) READ(2,200) F EEN(MQ+17)=FEEN(MQ+16)=1.0-F NCALC=NCALC+1 LLST(NCALC) = 11С WRITE(6,420) FORMAT(1X,20HIF WASTE ACT SL TO 420 ,/,5X,20HPRI CLAR -TYPE 1.0 1,/,5X,31HDIRECTLY TO DIGESTER -TYPE 2.0 ,/,5X,21HTHICKENER -TYPE 23.0 ;/) READ(2.200) F JF=F+0.01 MQ=NPOINT(11,1) EEN(MQ+16)=0.0 EEN(MQ+17)=0.0 EEN(MQ+18)=0.0 EEN(MQ+15+JF)=1.0 IF(F.LT.3.0) GO TO 40 NCALC=NCALC+1 LLST(NCALC)=12 NCALC=NCALC+1 LLST(NCALC)=19 WRITE(6,430) FORMAT(1X,40HIS THICKENED SLUDGE SENT TO DIGESTER -} 430 READ(2,200) F MQ=NPOINT(19,1) EEN(MQ+17)=F EEN(MQ+16)=1.0-F **C** 40 NCALC=NCALC+1 LLST(NCALC)=20 С WRITE(6,440) FORMAT(1H0,18HANY DIGESTERS -440) READ(2,200) F IF(F.LT.1.0) GO TO 50 LP2=1 NCALC=NCALC+1 LLST(NCALC)=13 CC ASKING FOR DETAILS OF THE DIGESTERS WRITE(6,450) FORMAT(1X,30HONE OR TWO STAGE DIGESTION -READ(2,200) F MQ=NPOINT(13,1) EEN(MQ+16)=F-1.0 EEN(MQ+16)=F-1.0 EEN(MQ+17)=1.0-EEN(MQ+16) IF(F.LT.2.0) GO TO 60 NCALC=NCALC+1 LLST(NCALC)=14 MQ=NPOINT(14,1) 450) MQ=NPOINT(14,1) WRITE(6,460) FORMAT(1X,22HFIRST STAGE DIGESTER WRITE(6,510) READ(2,200) F 460) EEN(MQ+4) = FWRITE(6,515) FORMAT(5X,3UHDIAMETER OF TANK (IN FT) = READ(2,200) F EEN(MQ+16)=F WRITE(6,540) 515) READ(2,200) F EEN(MQ+17)=FWRITE(6,525)

525	FORMAT(5X)35HTEMPERATURE OF DIGESTION DEG C =)
.C	READ(2,200) F EEN(MQ+19)=F
	NCALC=NCALC+1 LLST(NCALC)=15
470	WRITE(6,470) FORMAT(1X,23HSECOND STAGE DIGESTER) GO TO 65
с 60	NCALC=NCALC+1 LLSI(NCALC)=15
480 65	WRITE (6,480) FORMAT(1X,20HANAEROBIC DIGESTER) CONTINUE
20	MQ=NPOINT(15,1) WRITE(6,510)
	READ(2,200) F EEN(MQ+4)=F WRITE(6,515)
	READ(2,200) F EEN(MQ+16)=F
	WRITE(6,540) READ(2,200) F EEN(MQ+17)=F
	WRITE(6,525) READ(2,200) F EEN(MQ+19)=F
C	NCALC=NCALC+1 LLST(NCALC)=16
С 50	IF(LP1.NE.1) GO TO 70 NCALC=NCALC+1 LLST(NCALC)=22
	MQ=NPOINT(22•1) EEN(MQ+4)=(FLOAT(LP1+LP2-1))*100•0 +1•0
	EEN(MQ+20)=1.0 EEN(MQ+21)=1.0 EEN(MQ+17)=10.0
70	IF(LP2•NE•1) EEN(MQ+17)=20•0 IF(LP2•NE•1) GO TO 80 NCALC=NCALC+1
	LLST(NCALC)=23 MQ=NPOINT(23,1)
	EEN(MQ+4)=(FLOAT(LP1+LP2-1))*100.0 +1.0 EEN(MQ+20)=EEN(MQ+4) EEN(MQ+21)=EEN(MQ+4)
C	EEN(MQ+17)=5.0 IF(LP1.NE.1) EEN(MQ+17)=20.0
C 80	CONTINUE NCALC=NCALC+1 LLST(NCALC)=17
	NCALC=NCALC+1 LLST(NCALC)=18
C C C	ASKING FOR AVERAGE FEED CONDITIONS
750.	WRITE(6,750) FORMAT(1H0,30HWANT TO SET UP FEED STREAM) READ(2,200) F
	IF(F.LE.0.0) RETURN
700	CALL STREAM(1.0) WRITE(6,700) FORMAT(1H0,40HAVERAGE FLOW TO YOUR PLANT (MIGD) =)
	READ(2,200) F FLOW=F*1.0E6/24.0 FRAC=FLOW/SN(IS,3)
610	DO 610 J=6,JJ SN(IS,J)=SN(IS,J)*FRAC SN(IS,3)=FLOW
760	WRITE (6,760) FORMAT (1X,50HNOTE THAT BOD APPROX GREATER THAN TWICE DOC

)

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WRITE(6,710)
FORMAT(1X,31HAVERAGE BOD5 TO YOUR PLANT = )
READ(2,200) F
SN(IS,10)=FLOW*10.0*F*1.0E-6
WRITE(6,720)
FORMAT(1X,43HAVERAGE SUSPENDED SOLIDS TO YOUR PLANT =
READ(2,200) F
SN(IS,11)=FLOW*10.C*F*1.0E-6
WRITE(6,730)
 710
 720
                                                                                                                                                                                      .)
                 WRITE(6,730)
FORMAT(1X,36HAVERAGE DISSOLVED ORGANIC CARBON =
READ(2,200) F
730
                                                                                                                                                                       )
                 SN(IS,9)=FLOW*10.0*F*1.0E-6
                 SN(IS,9)=FLOW*I0.0*F*I.0E=6

SN(IS,22)=0.0

SN(IS,20)=0.2*SN(IS,9)

SN(IS,21)=SN(IS,9)-SN(IS,20)-SN(IS,22)

BODSOL=1.9*(SN(IS,20)+SN(IS,21))

BODP=SN(IS,10)-BODSOL

VSS=BODP/1.2

SN(IS,15)=5LOW#1().0#E_0E=6
                 VSS=BODP/1.2

SN(IS,15)=FLOW*10.0*5.0E-6

SN(IS,16)=0.5*VSS

SN(IS,17)=0.0

SN(IS,18)=0.2*VSS

SN(IS,19)=0.3*VSS

SN(IS,12)=2.1*(VSS+SN(IS,15))

SN(IS,13)=0.6*(SN(IS,11)-SN(IS,12))

SN(IS,14)=0.4*(SN(IS,11)-SN(IS,12))

SN(IS,8)=SN(IS,9)+VSS+SN(IS,15)

IV=IS
                 IV = \overline{I}S
                 CALL STREAM(2.0)
DO 620 J=2,JJ
SN(IS,I)=SN(IV,I)
620
C
                 CALL STREAM(16.0)
DO 640 J=6,JJ
                 SN(IS,J)=SN(IS,J)*FRAC
                 CONTINUE
SN(IS,3)=SN(IS,3)*FRAC
640
                 PURGE SN FILE
DO 630 I=1.III
IS=SN(I.1)+0.01
C
                  IF((IS.EQ.1).OR.(IS.EQ.16).OR.(IS.EQ.2)) GO TO 630
                 SN(1,1)=0.0
CONTINUE
630
C
                 RETURN
                 END
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SUBROUTINE CASET

COMMON LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10) COMMON IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR COMMON MODE, NPLNT, LOOPC SN(25,30), EEN(600), NPOINT(25,2) COMMON COMMON /C/ IND GENERATES A CASE STUDY AROUND THE AVERAGE RAW FLOW EN VECTOR **** 1.-15. STANDARD FORMAT STREAM NUMBER OF USUAL RAW WASTE FLOW (=1.0) 6. 7. 11. 1.0 = STREAM NUMBER EQUIPMENT NUM EQUIPMENT NUM 12. 16. GENERATED INPUT STREAM (=2.0) OF PRIMARY SETTLER (=5.0) OF THE ACTIVATED SLUDGE TANKS ABER OF NUMBER 17• 18• NUMBER (=7.0)NUMBER THE SECONDARY EQUIPMENT OF CLARIFIERS (=8.0)18. 20 EQUIPMENT N ANY NUMBER THE 1.0E6 DIGESTERS TO 1.0E8 NUMBER O BETWEEN (=15.0) USED AS OF THE STARTER IN THE RANDOM NUMBER GENERATOR IF(EN(20).GE.1.0E6) GO TO 9 WRITE(6,470) FORMAT(1X,50HTYPE ANY NUMBER BETWEEN 1 MILLION TO 100 MILLION READ(2,200) F FORMAT(F12.3) MQ=NPOINT(25,1) EEN(MQ+20)=F FON(AQ+20)=F 470 200 EN(20) = FIND=EN(20) IND=IND*2+1 CONTINUE WRITE(6,500) FORMAT(1X,20HREADY TO PLAY GAMES 500 C ,/,20H....CLUES.... S=EN(7) CALL STREAM(S) IIN=IS S=EN(12) CALL STREAM(S) IF(IS.LE.III) GO TO 22 CALL STREAM(0.0) CONTINUE IO=IS SN(IO,1) = EN(12)GENERATING A RAW WASTE FLOW SET WATER FLOW CONTINUE CALL RANDOM(Z) IF(Z+LT+0+25) GO TO 10 SN(IO+6)=SN(IIN+6)*Z*2+0 SET COMPONENT FLOWS DO 600 J=13,JJ CONTINUE CALL RANDOM(Z) IF(Z.LT.0.25) GO TO 20 SN(I0,J)=SN(IIN,J)*Z*2.0 CONTINUE 600 C C SET TOXIC COMPONENT FLOW CALL RANDOM(Z) SN(I0,30)=Z SN(IO,2) = SN(IIN,2)SN(10,4)=SN(11N,4) SN(10,5)=SN(11N,5) SN(10,7)=SN(11N,7) CALL BALNCE(1, IO)

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C C 24 GENERATING EQUIPMENT FAILURES CALL RANDOM(Z) KO=Z*11.0+1.0 GO TO (1,1,2,2,3,3,4,4,4,4,4,5,5),KO MM=EN(16)+0.01 1 DO 610 I=1,NCALC IF(LLST(I).EQ.MM) GO TO 620 610 GO TO 24 CONTINUE MQ=NPOINT(MM,1) 620 IF(K0.EQ.2) GO TO 11 EEN(MQ+4)=EEN(MQ+4)-1.0 WRITE(6,400) FORMAT(1X,25HONE PRIMARY TANK IS DOWN 400 ١ GO TO 5 EEN(MQ+23)=EEN(MQ+23)-1.0 WRITE(6,410) FORMAT(1X,50HONE OF THE PRIMARY SLUDGE PUMPS IS NOT WORKING 11 • 410 GO TO 5 C 2 MM=EN(17)+0.01 DO 630 I=1,NCALC IF(LLST(I).EQ.MM) GO TO 640 DO 630 630 ĞΟ ŦŌ 24 640 CONTINUE MQ=NPOINT(MM,1) IF(K0.EQ.4) GO TO 12 EEN(MQ+4)=EEN(MQ+4)-1.0 WRITE(6,420) FORMAT(1X,35HONE ACTIVATED SLUDGE TANK IS DOWN 420) EEN(MQ+45)=EEN(MQ+45)-1.0 WRITE(6,430) FORMAT(1X,40HONE OF THE AIR BLOWERS IS DOWN GO TO 5 12 430 ١ C 3 MM = EN(18) + 0.01DO 650 I=1.NCALC IF(LLST(I).EQ.MM) GO TO 660 650 GO TO 24 CONTINUE MQ=NPOINT(MM,1) IF(K0.EQ.6) G0 T0 13 EEN(MQ+4)=EEN(MQ+4)-1.0 660 WRITE(6,440) FORMAT(1X,35HONE OF THE SECONDARY CLARIFIERS IS DOWN) 440 GO TO 5 EEN(MQ+20)=EEN(MQ+20)-1.0 13 WRITE(6,450) FORMAT(1X,45HONE OF THE SECONDARY SLUDGE PUMPS IS DOWN GO TO 5 450) С MM=EN(19)+0.01 4 GO 670 1=1,NCALC IF(LLST(I).EQ.MM) GO TO 680 GO TO 24 DO 670 GO CONTINUE 680 MQ=NPOINT(MM,1) MQ=NPOINT(MM,1) KO=KO-6 GO TO (14,15,16,17),KO IF(EEN(MQ+4).LT.2.0) GO TO 24 EEN(MQ+4)=EEN(MQ+4)-1.0 WRITE(6,460) FORMAT(1X,25HONE DIGESTER IS DOWN GO TO 5 CONTINUE GO TO 5 CALL RANDOM(Z) 14 460 1 15 CALL RANDOM(Z) 16 EEN(MQ+24)=ZGO TO 5 17 5 EEN(MQ+19)=25.0 CONTINUE

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RETURN END

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SUBROUTINE RANDOM(Z)

GENERATES RANDOM NUMBERS HAVING A UNIFORM DISTRIBUTION BETWEEN 0-1 USES THE MIXED MULTIPLICATIVE CONGRUENTIAL METHOD

COMMON /C/ IND DATA IL/1/

IF(IL•EQ•0) GO TO 1 IL=0 M=2**30 FM=M N=2**15+3 IX=IND CONTINUE IX=MOD(N*IX,M) FX=IX Z=FX/FM

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RETURN

C	SUBROUTINE ICLOAD 302
	COMMON LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10) COMMON IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR COMMON MODE,NPLNT,LOOPC COMMON SN(25,30),EEN(600),NPOINT(25,2) DIMENSION COST(15)
c	COST(1)=0.0 COST(2)=8.0 COST(3)=20.0 COST(4)=2.0 COST(5)=0.0 COST(6)=20.0 COST(6)=20.0 COST(7)=8.0 COST(7)=8.0 COST(10)=10.0 COST(10)=10.0 COST(11)=8.0 COST(11)=8.0 COST(12)=15.0 COST(12)=15.0 COST(13)=20.0 COST(14)=20.0 COST(15)=20.0
νουνου	ALLOWS PLAYER TO MAKE DECISIONS AS TO PLANT OPERATIONS AND TO MAKE ANALYSES OF ANY STREAM COMPONENT COSTS ARE CALCULATED FOR ANY ANALYSIS AND A SURCHARGE IS APPLIED TO ANY PLANT DISCHARGING AN UNSATISFACTOY EFFLUENT THE FINAL EFFLUENT MUST BE STREAM 35.0
400 410 420	CALL STREAM(35.0) FLOW=SN(IS,3)*24.0*1.0E-6 WRITE(6,400) FLOW FORMAT(1X.15HEFFLUENT FLOW = .F12.6.6H MIGD) BOD=SN(IS.10)*1.0E5/SN(IS.3) WRITE(6.410) BOD FORMAT(1X.15HEFFLUENT BOD = .F8.1) TCOST=TCOST+COST(10) SS=SN(IS.11)*1.0E5/SN(IS.3) WRITE(6.420) SS FORMAT(1X.15HEFFLUENT SS = .F8.1) TCOST=TCOST+COST(11)
C C 480	APPLY SURCHARGE ON EFFLUENT BODCH=SN(IS,10)-30.0*SN(IS,3)*1.0E-5 IF(BODCH.LE.0.0) BODCH=0.0 SSCH=SN(IS,11)-30.0*SN(IS,3)*1.0E-5 IF(SSCH.LE.0.0) SSCH=0.0 SURCH=1.2*(BODCH+SSCH) WRITE(6,480) SURCH FORMAT(1H0,35HSURCHARGE ON EFFLUENT = \$,F8.2,6H /DAY)
480 C 430 i10	ASKING FOR ANALYSES WRITE(6,430) FORMAT(1H0,25HWANT ANY ANALYSES DONE –) READ(2,200) AN IF(AN.LE.0.0) GO TO 100 CONTINUE WRITE(6,440)
440 200. 210	FORMAT(1X,20HTYPE STREAM NUMBER) READ(2,200) S FORMAT(F12.3) FORMAT(2F12.4) IF(S.LE.0.0) GO TO 100 CALL STREAM(S) CALL CONVERT(1,1,1,15,3,1)
120 450	WRITE(6,450) FORMAT(1X,20HTYPE ELEMENT NUMBER) READ(2,200) SJ IF(SJ.LE.0.0) GO TO 110 IJ=SJ+0.01 WRITE(6,460) SI(1,1J)
460	FORMAT(1X,20HANALYSIS GIVES ,F12.1) IF(IJ.EQ.30) IJ=15

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303 TCOST=TCOST+COST(IJ) GO TO 120 C 100 CONTINUE WRITE(6,470) TCOST FORMAT(1HU,30H....TOTAL ANALYTICAL COST = \$ 470 C >F8.2) WRITE(6,520) FORMAT(1HU,45HWANT TO SEE THE SUMMARY REPORT ON STREAMS -READ(2,200) AN IF(AN.LT.1.0) GO TO 190 CALL DISKIO(1,24) CALL REPTO1 CONTINUE 520) 190 C RETURN END SUBROUTINE ENSET C COMMON LLST(50),NS(100),EN(100),SI(4,30),SO(4,30),KPRNT(10) COMMON IS,NE,JJ,LOOP,NIN,NOUT,MSN,ISP,NC,III,NCALC,NOCOMP,NSR COMMON MODE,NPLNT,LOOPC COMMON SN(25,30),EEN(600),NPOINT(25,2) C C C 140 ALLOWS PLAYER TO MAKE PLANT ADJUSTMENTS CONTINUE WRITE(6,500) 500 FORMAT(1X,22HTYPE EQUIPMENT NUMBER) READ(2,200) S FORMAT(F12.3) IF(S.LE.0.0) GO TO 130 MM=S+0.01 200 MM=5+0.01 MQ=NPOINT(MM,1) CONTINUE WRITE(6,510) FORMAT(1X,40HTYPE ELEM. NUMBER AND VALUE CHANGED TO READ(2,210) SJ,EL FORMAT(2F12.3) IF(SJ.LE.U.0) GO TO 140 VI=5 1+0.01 150 510 ١ 210 IJ=SJ+0.01 EEN(MQ+IJ)=EL GO TO 150 C 130 C CONTINUE RETURN END

D	ata set for th	e interactive	version	•	304
	WASTEWATER TR		SIMULATION 0.0 0.0	- INTERACTIVE 0.0 0.0	VERSION
19.0 2.0 7.0 20.0 22.0 -35.0 1.0	25.0 3.0 8.0 13.0 23.0	10.0 4.0 9.0 14.0 17.0	5•0 10•0 15•0 18•0	6.0 11.0 16.0	
1 • 0 7 • 0 7 • 0 1 • 0 1 • 0 1 • 0 7 • 0 2 • 0	1 • 0 1 • 0 1 • 0 1 • 0 7 • 0 7 • 0 7 • 0 7 • 0	7 • 0 1 • 0 1 • 0 7 • 0 7 • 0 1 • 0 7 • 0	1 • 0 1 • 0 7 • 0 7 • 0 6 • 0 7 • 0 7 • 0 7 • 0	1 • 0 1 • 0 1 • 0 7 • 0 7 • 0 1 • 0 1 • 0 1 • 0	
1 • 0 6 6 0 0 0 • 0 3 7 • 0 5 0 • 0 1 6 • 0 1 6 • 0 1 6 • 0 1 0 0 • 0 1 0 0 • 0 1 0 0 • 0 1 0 0 • 0 2 3 • 0	7 • 0 4 • 0 0 • 0 1 0 • 0 5 • 0 2 3 • 0 7 • 0 0 • 0 0 • 0 1 0 • 0 1 0 • 0	$ \begin{array}{c} 66000 \cdot 0 \\ 0 \cdot 0 \\ 78 \cdot 0 \\ 5 \cdot 0 \\ 100 \cdot 0 \\ 0 \cdot 5 \\ 16000 \cdot 0 \\ 0 \cdot 0 \\ 2000 \cdot 0 \\ 90 \cdot 0 \\ 5 \cdot 0 \end{array} $	$25 \cdot 0$ $0 \cdot 0$ $30 \cdot 0$ $30 \cdot 0$ $0 \cdot 0$ $6 \cdot 0$ $25 \cdot 0$ $0 \cdot 0$ $1500 \cdot 0$ $1000 \cdot 0$ $0 \cdot 0$ $4 \cdot 0$	0 • 0 0 • 0 5 • 0 1 0 • 0 0 • 0 0 • 0 0 • 0 1 0 0 0 0 • 0 0 • 0 1 0 0 0 0 • 0 0 • 0 1 0 0 0 •	• •
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1 • 0 3 • 0 1 • 0 1 • 0 4 • 0 3 • 0 3 • 0 1 • 0	11.0 3.0 4.0 1.0 4.0 5.0 0.0	$ \begin{array}{c} 15.0\\ 0.0\\ 18.0\\ 22.0\\ 6.0\\ 0.0\\ 0.0\\ 0 \end{array} $	0 • 7 0 • 0 0 • 0 0 • 0 3 0 • 0 7 • 0	0 • 0 0 • 0 0 • 0 0 • 0 0 • 0 0 • 0	
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