

DECLARATION OF CANDIDATE

I, Harold Arthur Nicholls, hereby declare
that this thesis is my own work and that
it has not been submitted for a degree at
another University.

Signed by candidate

January 1983

The copyright of this thesis vests in the author. No quotation from it or information derived from it is to be published without full acknowledgement of the source. The thesis is to be used for private study or non-commercial research purposes only.

Published by the University of Cape Town (UCT) in terms of the non-exclusive license granted to UCT by the author.

BIOLOGICAL NITROGEN AND PHOSPHORUS
REMOVAL ON LARGE PLANTS

BY

HAROLD ARTHUR NICHOLLS B Sc (IND CHEM)

A thesis submitted in partial fulfilment of
the requirements for the Degree of Master of
Science in the Faculty of Engineering,
University of Cape Town

The University of Cape Town has been given
the right to reproduce this thesis in whole
or in part. Copyright is held by the author.

S Y N O P S I S

Full scale implementation of biological excess phosphorus removal by means of the activated sludge process commenced in South Africa at the Johannesburg Alexandra Plant in 1974 . Originally the plant was designed to operate in the nitrifying extended aeration mode ; but by switching off some surface aerators around the common inlet point for the influent and underflow recycle, an anoxic zone was created to induce denitrification. This operational modification worked very successfully. Later additional aerators were switched off to create an anaerobic - anoxic zone at the head of the aeration basin, in this manner attempting to induce the prerequisite conditions for excess biological uptake of phosphorus as put forward by Barnard (i.e. by stimulating P release under anaerobic conditions) . This revised operational procedure was not successful due to difficulties experienced in controlling mixing in the anaerobic - anoxic zone. However the experience gained from this investigation together with research findings at laboratory and pilot scale level at the National Institute for Water Research, the University of Cape Town, and Johannesburg instilled sufficient confidence into City Council staff to proceed with the design of the 150 Ml/d Goudkoppie Plant, initially for nitrogen removal and to subsequently modify it to a 5 stage Phoredox process for both nitrogen and phosphorus removal. Before the plant was commissioned (1976) it was necessary to proceed with extensions to the Northern Works and these were also based on the 5 stage Phoredox process.

All the Johannesburg activated sludge plants were comprehensively monitored to determine their performance under cyclic flow and load conditions. The data collected from the Goudkoppie plant were particularly comprehensive and exhaustive.

Concomitant with the monitoring program extensive research into the single stage nitrification/denitrification/excess phosphorus removal process was being undertaken by the three agencies mentioned earlier. Of particular interest was the development by the University of Cape Town of a general kinetic model of the single sludge activated sludge process that described carbonaceous degradation, nitrification and denitrification behaviour under cyclic flow and load conditions. This dissertation critically evaluates the predictions of this model against the observed responses on the Goudkoppie and Northern Works.

As new information and hypotheses on the mechanism of excess biological removal became known from the three agencies checks were made on the above works to test compliance against observed performance. Particular attention was given to the University of Cape Town's hypothesis that a certain minimum concentration of readily biodegradable COD in the anaerobic zone was necessary to stimulate excess phosphorus uptake. Consideration was also given to the Johannesburg group's belief that any deficiency in the readily biodegradable COD in the influent sewage could be supplemented by permitting thickened raw sludge to ferment under acid conditions and returning the liquid phase directly to the anaerobic zone.

The principle findings from these investigations are briefly as follows:

1. The general model predicted the observed response of the large scale plants with remarkable accuracy. It can also be used for predicting the behaviour of a proposed plant under a range of assumed influent flow and load conditions and for the optimisation of existing works performance.
2. The readily biodegradable COD hypothesis for excess phosphorus removal appears to be supported by the observed phosphorus removal response of the Goudkoppie and Northern Works.

Important corollaries to this conclusion are :

- (a) Any action that increases this COD fraction in the influent improves excess phosphorus removal. In this regard industries discharging readily biodegradable COD in their effluents should not be encouraged to carry out pretreatment.
- (b) Any action that decreases this fraction in the influent should be avoided. At the Goudkoppie Works it was found that deposition and fermentation of solids took place in a flow balancing tank, which was situated upstream of the biological reactors and readily biodegradable COD was lost from the system as methane gas. Daily flushing of accumulated sludge virtually completely eliminated this effect.

3. Oxygen and nitrate entering the anaerobic reactor via the influent or the recycle from the final clarifiers enable facultative anaerobic bacteria or denitrifiers to metabolise some or all of the readily biodegradable COD available. This situation does not permit an adequate release of phosphorus and uptake in the subsequent aerobic process is detrimentally affected.

This problem was present at both the Goudkoppie and Northern Works and different procedures were adopted for its solution. At the Goudkoppie Works nitrates in the clarifier underflow recycle were reduced by limiting the oxygen input to the aerobic reactor, thus limiting the nitrification capability of the plant. This method of control however appeared to be a contributory factor to the development of very poor settling sludges. At the Northern Works the readily biodegradable component of the influent COD was supplemented by adding settled liquors from a digester operated under acid fermentation conditions.

4. From (3) above evidently the nitrification - denitrification behaviour of a plant can have a crucial effect on its propensity for excess phosphorus removal. In terms of the kinetics of denitrification as developed by the University of Cape Town, the denitrification contribution per unit volume of the secondary anoxic reactor is very minor compared to that of the primary anoxic reactor. This was verified on both the Goudkoppie and Northern Works. It was so apparent that the secondary anoxic zone has been omitted from the new works at Bushkoppies. The volume previously allocated to the secondary anoxic zone should preferably be allocated additionally to the primary anoxic reactor. However this reactor also has a denitrification capacity determined principally by its volume, the mass fraction of sludge it contains, and the magnitude and nature of the influent COD. The nitrification capacity of the process again is fixed largely by the TKN content of the influent. A qualitative measure for judging the nitrification - denitrification behaviour of a process is the TKN/COD ratio of the influent. Where this ratio is above a certain value it may be impossible to attain adequate denitrification - a situation that is present at the Northern Works. The phosphorus removal propensity is therefore dependent not only on the presence of an adequate level of readily biodegradable COD in the influent but also on the TKN/COD ratio which determines the denitrification capability of the plant.

5. The difficulties experienced in obtaining adequate denitrification at the Goudkoppie and Northern Works suggest that the Phoredox process, in which the nitrate concentration in the underflow recycle is the same as in the effluent, limits this process to TKN/COD ratios less than some upper limiting value. It also suggests that alternative process configurations, such as the UCT process, in which the nitrate discharge to the anaerobic reactor can be controlled independently of the concentration in the effluent, have advantages over the Phoredox system when the limiting TKN/COD ratio is exceeded.

ACKNOWLEDGEMENTS

This thesis was carried out under the supervision of Professor G.v.R. Marais, Professor of Water Resources and Public Health Engineering, and Mr. D.W. Osborn, Chief Scientific Officer, City of Johannesburg. Thanks are expressed for their excellent guidance and appreciation for all the hours spent in directing this thesis.

Thanks are also due to :

Mr A.R. Pitman and Mrs L. Melmed for fruitful discussions and advice.

The City Engineer and Officials in his department for making facilities available on all the Johannesburg Sewage Works

The Water Research Commission for their assistance

Friends and colleagues for assistance in some of the analyses

My wife, for without her help and encouragement I would not have completed this work.

LIST OF CONTENTS

	<u>Page</u>
<u>SYNOPSIS</u>	(i)
<u>ACKNOWLEDGEMENTS</u>	(vi)
<u>LIST OF CONTENTS</u>	(vii)
<u>LIST OF SYMBOLS</u>	(xii)
CHAPTER ONE : INTRODUCTION	1.1.
1. HISTORICAL BACKGROUND	1.1.
1.1. Johannesburg sewerage systems	1.1.
1.2. Nutrient removal from Johannesburg effluents	1.3.
CHAPTER TWO : REVIEW OF NUTRIENT REMOVAL	2.1.
1. INTRODUCTION	2.1.
2. NITROGEN REMOVAL	2.2.
2.1. Denitrification	2.3.
2.2. Denitrification processes configurations	2.4.
2.3. Nitrification in anoxic/aerobic systems	2.9.
2.4. Optimisation of amoxic - aerobic plants	2.10.
3. BIOLOGICAL PHOSPHORUS REMOVAL	2.13.
3.1. Introduction	2.13.
3.2. Chemical precipitation theories	2.14.
3.3. Biological removal theories	2.15.
3.3.1. Biological excess accumulation of phosphorus in microorganisms	2.15.
3.3.2. Non - metabolic accumulation of excess phosphorus in bacteria	2.17.
3.3.3. Excess phosphorus removal as applied to the activated sludge process	2.20.

3.3.3.1.	Acinetobacter	2.20.
3.3.3.2.	The anaerobic zone	2.20.
3.3.3.3.	Release of phosphorus in the anaerobic zone	2.21
3.3.4.	Phosphorus removal in single sludge systems	2.24.
4.	ACTIVATED SLUDGE MODELS	2.27. 2.27
CHAPTER THREE : NUTRIENT REMOVAL AT THE JOHANNESBURG ALEXANDRA PLANT		3.1.
1.	INTRODUCTION	3.1.
2.	DESIGN OF THE PLANT	3.2.
	2.1. Dissolved oxygen control	3.5.
	2.2. Sludge age control	3.6.
3.	EXPERIMENTAL INVESTIGATION INTO NITROGEN REMOVAL AND RESULTS	3.7.
	3.1. Sampling procedure	3.9.
	3.2. Results	3.10.
4.	DISCUSSION	3.10.
	4.1. MLSS concentration	3.10.
	4.2. Nitrification and denitrification	3.12.
	4.3. Settling properties of the sludge	3.14.
5.	OPERATION OF THE PLANT IN A NITROGEN AND PHOSPHORUS REMOVAL MODE	3.15.
	5.1. Switching out of aerators	3.16.
	5.2. Accumulation of sludge in the secondary clarifier	3.20.
	5.3. Storage of sludge on the floor of the aeration basin	3.22.
6.	GENERAL CONCLUSIONS	3.25.

	<u>Page</u>
CHAPTER FOUR : THE GOUDKOPPIE PLANT	
1. INTRODUCTION	4.1.
2. DESIGN FEATURES	4.3.
3. PLANT PERFORMANCE EVALUATION	4.9.
3.1. Objectives	4.9.
3.2. Conditions of test	4.9.
4. PERFORMANCE DATA	4.12.
4.1. Data validity	4.23.
5. MODEL PREDICTIONS	4.24.
5.1. Sewage characteristics	4.24.
5.1.1. Influent COD and TKN concentration and reactor temperature	4.26.
5.1.2. Soluble unbiodegradable COD fraction	4.26.
5.1.3. Unbiodegradable particulate COD fraction	4.27.
5.1.4. Easily biodegradable COD fraction	4.28.
5.1.5. Fraction of TKN in ammonia form	4.29.
5.2. Constants specific to a waste flow	4.29.
5.3. Constants invariant for different waste flows	4.30.
5.4. Simulated results	4.32.
6. DISCUSSION	4.36.
6.1. Model predictions	4.36.
6.1.1. Daily mass response	4.36.
6.1.2. Cyclic response	4.37.
7. NITROGEN REMOVAL	4.39.
8. PHOSPHORUS REMOVAL	4.42.
9. POSSIBLE MODIFICATION TO THE GOUDKOPPIE PLANT TO IMPROVE NUTRIENT REMOVAL UNDER THE EXISTING FLOW AND LOAD	4.44.
9.1. Changing to a Modified UCT Process configuration	4.45
9.1.1. Discussion	4.53.
9.2. Bypassing the settling tank	4.54.
9.3. Modification to operating the settling tank	4.64.

	<u>Page</u>
10. THEORETICAL EVALUATION OF THE EFFECT ON NUTRIENT REMOVAL WHEN THE BALANCING TANK CONTROL STRATEGY WAS APPLIED	4.65.
CHAPTER FIVE : NORTHERN WORKS	
1. INTRODUCTION	5.1.
2. DESIGN FEATURES	5.4.
3. PLANT PERFORMANCE EVALUATION	5.7.
3.1. Objectives	5.7.
3.2. Conditions of test	5.7.
3.3. Performance data	5.10.
4. EVALUATION OF EXPERIMENTAL PLANT RESPONSE	5.20.
5. MODEL PREDICTIONS AND COMPARISON WITH PERFORMANCE DATA	5.23.
5.1. Sewage characteristics	5.23.
5.2. Simulated results	5.29.
6. DISCUSSION	5.33.
6.1. Model predictions	5.33.
6.1.1. Average daily response	5.33.
6.1.2. Cyclic response	5.34.
7. OPERATIONAL CHANGES TO ACHIEVE HIGHER PHOSPHORUS AND NITROGEN REMOVAL	5.34.
CHAPTER SIX : CONCLUSIONS	6.1.
REFERENCES	R.1.
APPENDIX	
A1 List of volutin forming bacteria which accumulate polyphosphate	A.1.
A2 Alexandra Plant ; Raw data obtained when 0,1,2 and 3 aerators closest to the feed were off	A.2.

		<u>Page</u>
A3	Example calculation of mixed liquor suspended solids concentration using Marais Ekama equations	A.6.
A4	Measurement of oxygen consumption rate	A.8.
A5	Goudkoppie Plant : Raw data obtained during the experimental period	A.11.
A6	Goudkoppie Plant : Calculation of nitrogen and COD mass balance	A.19.
A7	Calculation of readily biodegradable influent COD fraction	A.25.
A8	Northern Works : Raw data obtained during the experimental period	A.28.
A9	Northern Works : Calculation of nitrogen and COD mass balance	A.37.
A10.	Northern Works : COD concentrations ; raw sewage, settled sewage, balancing tank effluent.	A.38.

LIST OF SYMBOLS

<u>Symbol</u>	<u>Description</u>	<u>Units</u>
ADP	Adenosine diphosphate	
ATP	Adenosine triphosphate	
b	Specific endogenous respiration rate Subscripts h or n refer to heterotrophic or nitrifiers respectively	
COD	Chemical oxygen demand	mg/l
f	Unbiodegradable fraction of the active organisms	mg VSS/mg VSS
f _{bs}	Rapidly biodegradable fraction of the influent COD	mg COD/mg COD
f _{ca}	Ratio easily : total biodegradable COD in the influent	mg COD/mg COD
f _n	Fraction of total nitrogen in the VSS	mg N/mg VSS
f _{na}	Fraction of influent TKN in the free and saline ammonia form	mg N/mg N
f _{nu}	Soluble unbiodegradable organic nitrogen fraction	mg N/mg N
f _{oe}	Fraction of nitrogen released through endogenous respiration in the organic nitrogen form	mg N/mg N
f _{os}	Fraction of nitrogen required for cell synthesis in the organic nitrogen form	mg N/mg N
f _{pe}	fraction of phosphorus as P in the endogenous residue	mg P/mg VSS
f _{pi}	fraction of phosphorus as P in the inert volatile solids	mg P/mg VSS

<u>Symbol</u>	<u>Description</u>	<u>Units</u>
f_{us}	Unbiodegradable soluble fraction of the influent COD	mg COD/mg COD
f_{up}	Unbiodegradable particulate fraction of the influent COD	mg VSS/mg COD
f_{xm}	Maximum allowable unaerated mass fraction to ensure nitrification	m^3/m^3
f_{xp}	Primary anoxic mass fraction	m^3/m^3
f_{xs}	Secondary anoxic mass fraction	m^3/m^3
k_a	Maximum specific nitrification rate constant	mg N/mg VASS/h
K_a	Substrate adsorption rate coefficient	1/mg VSS/d
K_{mp}	Maximum specific growth rate coefficient for heterotrophic organisms utilising stored particulate substrate	mg COD/mg VSS/d
K_n	half saturation value for nitrification	mg N/l
K_{sp}	Saturation coefficient for heterotrophic cells synthesised from particulate substrate	mg COD/l
K_{ss}	Saturation coefficient for heterotrophic cells synthesised from soluble substrate	mg COD/l
K_r	Conversion rate constant for organic nitrogen	1/mg VSS/d
K_{120}	First denitrification rate constant at 20°C	mg N/mg VASS/h
K_{220}	Second denitrification rate constant at 20°C	mg N/mg vass/h
K_{320}	Third denitrification rate constant at 20°C	mg N/mg VASS/h

<u>Symbol</u>	<u>Description</u>	<u>Units</u>
MLSS	Mixed Liquor Suspended Solids	mg/l
N_e	TKN concentration in the effluent	mg N/l
N_{ii}	TKN concentration in the influent	mg N/l
N_{nr}	Nitrate concentration in underflow	mg N/l influent
N_s	Nitrogen concentration required for sludge production	mg N/l
P	COD equivalent of volatile solids	mg COD/mg VSS
P_f	Phosphorus propensity factor	.
Q	Feed flow rate	l/d
Q_r	Returned sludge flow rate	l/d
R_s	Sludge age	d
s	Recycle ratio	
S	Unmetabolised biodegradable COD in the effluent	mg COD/l
SF	Safety factor	
S_{bi}	Biodegradable COD in the influent	mg COD/l
S_{bsi}	Readily biodegradable COD in the influent	mg COD/l
S_{bsa}	Readily biodegradable COD in the anaerobic reactor	mg COD/l
S_{pi}	Particulate unbiodegradable COD in the influent	mg COD/l
S_{ti}	Total COD in the influent	mg COD/l
S_{ui}	Soluble unbiodegradable COD in the influent	mg COD/l
S_{xii}	Unbiodegradable particulate COD in the influent	mg COD/l
T	Temperature	$^{\circ}$ C
TKN	Total Kjeldahl nitrogen concentration	mg N/l

<u>Symbol</u>	<u>Description</u>	<u>Units</u>
V	Volume of reactor	l
VASS	Volatile active solids concentration	mg/l
VSS	Volatile settleable solids	mg/l
X_a	Active volatile solids concentration in mixed liquor	mg VSS/l
X_e	Endogenous volatile solids concentration in mixed liquor	mg VSS/l
X_i	Inert volatile solids concentration in mixed liquor	mg VSS/l
X_{ii}	Unbiodegradable influent volatile solids concentration	mg/l
X_T	Total solids concentration in mixed liquor	mg/l
Y_h	Yield coefficient for heterotrophic organisms utilising stored COD	mg VSS/mg COD
Y_n	Yield coefficient for <u>Nitrosomonas</u>	mg VSS/mg N
θ	General parameter for pH dependency coefficients. Subscript h or n refers to heterotrophic organism or <u>Nitrosomonas</u> respectively	
U_h	Specific growth rate of heterotrophic organisms. Subscript m refers to maximum value	/d
U_n	Specific growth rate of <u>Nitrosomonas</u>	/d
ΔP	Phosphorus as P removed from the influent and incorporated into the waste sludge per day	mg/l/d
α	Fraction of phosphorus present in the activated sludge	mg P/mg VSS

CHAPTER ONE

INTRODUCTION

1. HISTORICAL BACKGROUND

Johannesburg was founded in 1866 when gold was discovered on the farm Langlaagte, and soon became a thriving mining camp. Today it is not only the largest city in South Africa but also the financial hub of the country. It is surrounded by satellite towns which form the Witwatersrand complex covering 17 000 square kilometres and accomodating a population of approximately 5 million people.

The Witwatersrand at an altitude of approximately 1 800 m above sea level runs in an east - west direction and forms a natural watershed, with drainage to the north to the Indian Ocean and to the south, the Atlantic Ocean. Before discharging to the oceans the rivers draining the Witwatersrand are impounded in a number of dams, two of which are in close proximity to Johannesburg, viz., the Vaal Dam in the southern and the Hartebeespoort Dam in the northern drainage basins respectively. In order to make use of gravity sewerage, advantage has been taken of the natural drainage by the siting of wastewater treatment plants near the banks of rivers feeding these dams, so that the treated effluents inevitably find their way to the storage dams.

1.1. Johannesburg sewerage systems

From the turn of the century until the early 1930's the only form of wastewater purification was land irrigation. From 1930 until the late

1960's most of the plants built in Johannesburg employed the biological filtration process with an anaerobic digestion of the solids.

In 1960 the financial climate resulted in a shortage of capital which led to a change in the type of treatment process ; activated sludge plants were predicted to be lower in overall costs than biological filter plants, and were adopted as the basis for future designs.

The first activated sludge plant constructed was a temporary works situated at Palmietfontein on the southeastern outfall sewer to the Olifantsvlei Works, designed as a modified Orbal extended aeration plant. (1971) . This plant produced a well nitrified effluent which discharged via maturation ponds into the Klip River, and hence to the Vaal Barrage. Although the effluent was of a high quality in terms of BOD and ammonia removal, it soon became evident that the presence of nitrates and phosphorus in the effluent caused eutrophication in the Barrage.

A second installation (1974), also temporary, was the Alexandra plant constructed to relieve the overloaded Bruma sewer flowing to the Northern Works. The plant was sited on a small tract of land between the existing sewer and the Jukskei River. It was of the extended aeration type but due to space limitations on the site, the waste sludge from the plant was returned to the sewer for final treatment in the digesters at the Northern Works, thus eliminating the need for sludge treatment facilities. The plant produced a well nitrified effluent which discharged into the Jukskei River and hence the Hartbeespoort Dam.

The first major permanent activated sludge installation, a nitrifying extended aeration activated sludge plant, was the Olifantsvlei Works (1974) . Its design was very similar in principle

to the Orbal process except that the four orbal rings were converted into four rectangular tanks in series. At the time, as far as is known, this was the largest plant of its kind in operation. The effluent discharged into the Klip River, flowing into the Vaal Barrage.

All the effluents from the plants described above found their way to the two storage dams, the Vaal Barrage and Hartebeespoort Dam. From a study of the historical development of eutrophication problems in these dams it became apparent that these effluents were the major cause, in particular in the Hartebeespoort Dam, of eutrophication. Eutrophication in this dam caused, at various times, prolific growth of algae and, at a later stage, an explosive growth of waterhyacinth (*E. CRASSIPES*), Scott and Steyn (1979).

Waterhyacinth was first noted in the tributaries of the Hartebeespoort Dam in 1959 and from then on was controlled by mechanical means (Anon. 1971). However, in 1971, when the level of the dam was increased the inundation of partially vegetated areas provided a favourable environment for rapid proliferation. By 1976, weed coverage was so large (\pm 70 percent of the surface area) that the recreational use of the dam was severely restricted. Chemical methods of control had to be resorted to (Scott and Steyn, 1979), but it was evident that the problem would be alleviated only by reducing the inflow of nitrates and phosphorus to the dam.

1.2. Nutrient removal from Johannesburg effluents

Johannesburg, being the main contributor of treated effluents to the two basins draining the Witwatersrand, quite early on (1973) recognised the importance of nutrient limitation in their discharges. Earlier attempts included the construction of large scale maturation ponds, still existant. Although these ponds induced a considerable

improvement in some pollution parameters, particularly the bacteriological and volatile solid parameters, they contributed little removal of phosphorus and nitrogen.

In 1974, Barnard reported that nitrate nitrogen could be removed in the activated sludge process by creating zones in the process where no oxygen is supplied (anoxic zones), the nitrate replacing the oxygen as the hydrogen ion (or electron) acceptor - the Bardenpho process. To Johannesburg this work presented the possibility of reducing the nitrogen content in some of the existing activated sludge plants, if these could be suitably modified. Modifications were made to the Alexandra and Olifantsvlei plants in 1974 and by trial and error operational procedures developed to optimise the nitrogen removal. The success attained in this endeavour verified the soundness of Barnard's proposals and the Johannesburg City Council decided as a policy that all future works should include nitrogen removal. Based on experience gained on the Alexandra and Olifantsvlei plants, and taking due account of theoretical developments on biological nitrogen removal elsewhere, the first plant designed specifically to incorporate nitrogen removal was the one to be located at Goudkoppie (1976)

While the Goudkoppie plant was under construction, Barnard announced further developments in nutrient removal. From research reported in the literature and experience on a pilot plant he hypothesised that if the organism mass in the activated sludge process is stressed under conditions of no oxygen and no nitrate (anaerobic condition) such that the release of phosphorus from the organism mass to the bulk liquid takes place, then, if the mass is subsequently aerated, the mass will take up the released phosphorus and some additional phosphorus from the bulk liquid, and thereby reduce the phosphorus content in the effluent. To achieve this he proposed a modification to the Bardenpho process called the Phoredox process (Fig. 2.8.)

Barnard's proposal again was of great interest to the Johannesburg City Council, but, before embarking on a policy of implementation in future designs, in the absence of full scale experience and adequate evaluation of the hypothesis it was considered essential to test Barnard's hypothesis at full scale.

A study of the Alexandra plant design indicated that by suitable modifications and appropriate changes in operation, it should be possible to evaluate the hypothesis. It was recognised that the modifications necessary to the Alexandra plant for both nitrogen and phosphorus removal were unlikely to induce optimal removals, because it was not possible to implement the modifications strictly as required by hypothesis. Furthermore, monitoring of the plant was expected to present some severe difficulties because the plant was not designed with these objectives in mind. For example it was not possible to measure readily, or to control, key process parameters governing nutrient removal such as sludge age, recycle ratio, oxygenation rate control, and others. Consequently it was expected that the results would be more of a qualitative nature lacking completeness and precision.

Despite the problems mentioned above evaluation of the process capability was crucial to Johannesburg for two reasons, it would ;

- (1) Establish whether nutrient removal on a large plant is possible by appropriate modification of the process or process operation, particularly the latter, thereby to set an example of successful application for other works.
- (2) Identify features in design and operation that require the attention of designers of future installations to operate plants optimally.

With objectives above in mind, and cognizant of the restrictions inherent in the existing plant, the Alexandra plant was modified and evaluated. The success achieved was considered by the City Council

to be sufficient to decide on a policy of implementation on all future works. This policy was put into effect immediately. At that time Goudkoppie Works was already under construction ; however, it was possible to make ad hoc changes to the design to incorporate the Phoredox proposals in so far as the physical limitations imposed by construction already completed did allow. At that time also the design of the Northern Works was in its initial stages ; it was possible therefore to incorporate the Phoredox proposals more completely, and this was accordingly done.

The plants above have been in operation now for a number of years and reflect the evolution of the biological nutrient removal concept, starting with modifications of existing plants to induce nutrient removal and culminating in the design of the plants at Goudkoppie and Northern Works.

Over this period research greatly increased understanding of the basic mechanisms of, and the factors controlling biological nitrogen and phosphorus removal. As the information became available so it was implemented by appropriate modifications to existing design (where this was possible) and/or modifications of the operating procedures. The experience thus gained, the problems encountered and procedures to resolve these all should be of interest to both designers and operators of such plants. These considerations formed the main incentive for this enquiry.

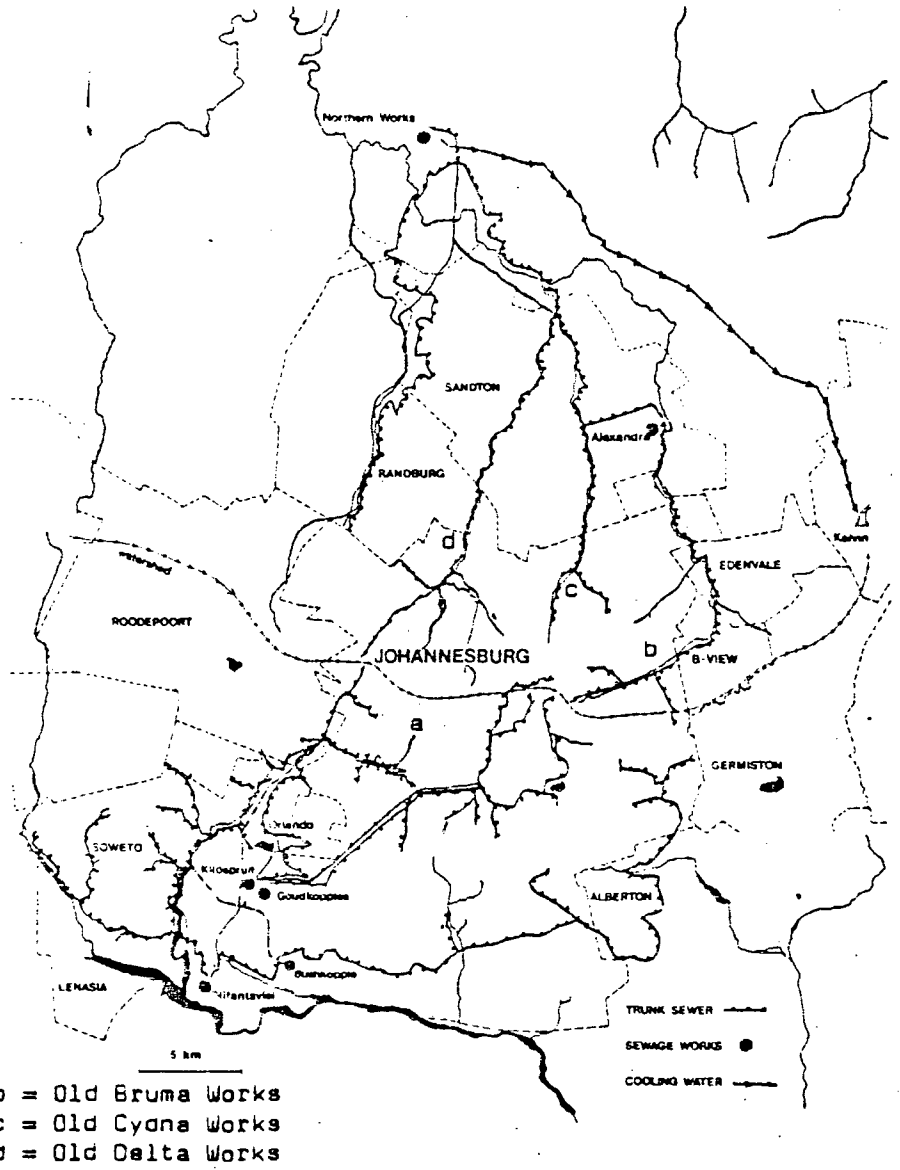


Fig. 1.1. Johannesburg Metropolitan Drainage Area

C H A P T E R T W O

REVIEW OF NUTRIENT REMOVAL

1. INTRODUCTION

The primary objective of this investigation was to enquire into the behaviour of large scale plants that either have been modified to induce nutrient removal (i.e. removal of nitrogen and phosphorus) or have been designed specifically towards this end. It is not possible to describe the work done in this regard if there is no understanding of the basic principle on which these processes are presumed to operate. Accordingly a review of these principles is essential. It is not the intention here to survey, in depth, the literature on nutrient removal associated with the activated sludge process; this has been done quite competently by a number of authors, Christensen and Harremoës (1972, 1974), Stern and Marais (1974), Menar and Jenkins (1970), Harold (1966), Dawes and Senior (1973), Kulaev (1975) and van Haandel, Ekama and Marais (1981). Rather the intention is to provide only such background as is necessary to appreciate the basis for the modification in the design and operation of existing plants to induce nutrient removal, and for the design of new plants with nutrient removal as an integral part.

The relevant literature review on nutrient removal is best set out by subdividing the discussion into three parts:

- (i) Nitrogen removal
- (ii) Phosphorus removal
- (iii) Modelling of the activated sludge process including the removal of nutrients

2. NITROGEN REMOVAL

Nitrogen removal by biological denitrification in the single sludge activated sludge process will be considered as this is the method by which nitrogen is removed in the activated sludge plants under investigation.

In the single stage activated sludge process, to remove nitrogen and the free and saline ammonia in the raw wastewater, ammonia first must be oxidised to nitrate. This is accomplished under aerobic conditions by the Nitrosomonas sp, which oxidise the ammonia to nitrite, and by the Nitrobacter sp, which oxidise the nitrite to nitrate.

Considering the kinetics of nitrification in the activated process, Downing, Painter and Knowles (1964) showed that the conversion of ammonia to nitrite is the rate limiting step because of the slow rate of reproduction of Nitrosomonas. In order that this species be retained in the process, the sludge age must exceed the replication time of this organism. From a mass balance of the nitrifiers over a completely aerobic process the minimum sludge age necessary to just maintain this organism in the system is given by :

$$R_m = \frac{1}{(\mu_{nmT} - b_n)} \quad (1)$$

where R_m = minimum sludge age in days

μ_{nmT} = maximum specific growth rate of Nitrosomonas at T C (d^{-1})

b_n = specific endogenous respiration of Nitrosomonas (d^{-1})

The magnitude of b_n is small relative to μ_{nmT} at temperatures greater than 12°C and consequently can be neglected for all practical purposes, in which event, $R_m = 1 / \mu_{nmT}$ (2)

The specific growth rate μ_{nmT} is temperature dependent and can be expressed in terms of the simplified Arrhenius equation

$$\mu_{nmT} = \mu_{nm20} (1,123)^{T - 20} \quad (3)$$

where μ_{nm20} can have values ranging from 0,2 to 0,7 depending on the source of waste water.

Once nitrification has occurred it becomes possible to remove nitrogen via denitrification.

2.1. Denitrification

Denitrification consists of a series of energy yielding biochemical reactions in which the organic material serves as the electron donor, and nitrate serves as the final electron acceptor. The pathways involved in this reaction series have been reported extensively in the literature and are summarised by Christensen and Harremoës (1972, 1974) .

If free oxygen is present, the enzymatic system for the utilisation of nitrate or nitrite as an electron acceptor is inhibited; consequently, in the activated sludge process, if denitrification is to be successful, it is essential that oxygen free zones are created. This may be achieved in a number of ways:

- (i) Inclusion in the process configuration of a reactor that is stirred but not aerated whereby the dissolved oxygen concentration is maintained at zero.
- (ii) Aeration at a number of points in the reactor chosen such that the dissolved oxygen concentration falls to zero between any two.
- (iii) Maintenance of the oxygen concentration at approximately 0,5 mg/l at the surface of the reactor causing regions of even lower dissolved oxygen concentration towards the

bottom of the aeration basin. At these low oxygen tensions it is possible that the activated sludge floc can not maintain an aerobic condition throughout and an oxygen free zone is created within the floc.

Quantitative expression of denitrification in (ii) and (iii) above is very difficult because the magnitude of the unaerated zones or the anoxic state in the particles is difficult to express in a reliable manner, and hence, consistent results can not be guaranteed.

2.2. Denitrification processes configurations

The potential of biological denitrification using endogenous respiration as the source of electron donors was first recognised by Wuhrmann (1957) when he incorporated a stirred reactor following the aeration basin (see Fig. 2.1.)

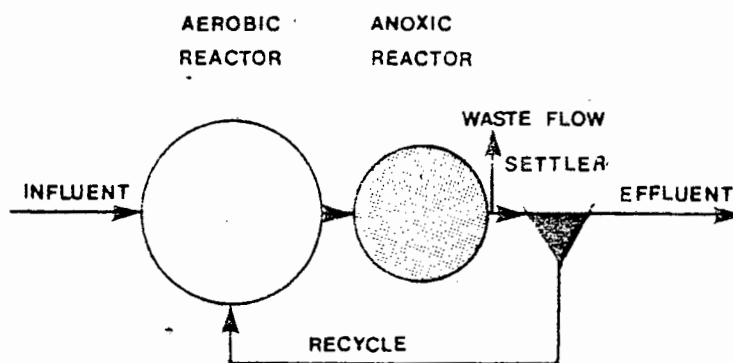


Fig. 2.1. The Wuhrmann configuration for biological nitrogen removal

Wuhrmann (1957) initially claimed that the denitrification rate in the stirred reactor was $0,0017 \text{ mg N/mg MLVSS/h}$, however Christensen, Hårremoes and Roed-Jensen (1975), on repeating Wuhrmann's experiments, found that the rate of nitrate utilisation was dependent on the COD adsorbed onto the floc, and obtained rates ranging from $0,0001 - 0,0002 \text{ mg N/mg MLVSS/h}$.

Ludzack and Ettinger (1962) proposed an alternative method in which raw sewage is utilised as the electron source in the denitrification reaction. A stirred unaerated reactor is placed ahead of the main aeration basin, the two reactors being in partial communication, the nitrified sludge from the main aeration basin circulating through this denitrification zone by the stirring action of the mixers and aerators (see Fig. 2.2.) Because of the ready availability of both electron donors and electron acceptors enhanced denitrification was achieved; Ludzack and Ettinger reported a denitrification rate of 0,0053 mg N/mg VSS/h .

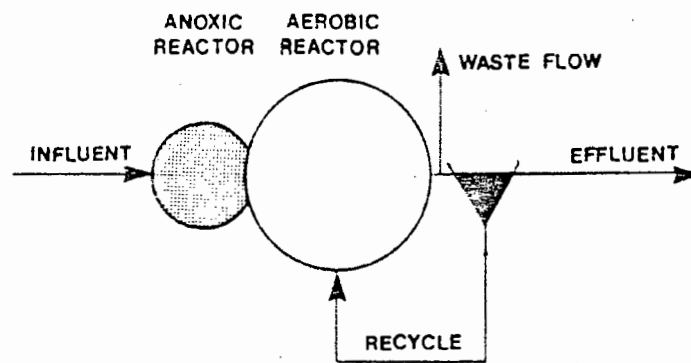


Fig. 2.2. The Ludzack Ettinger denitrification system

Based on the ideas of Ludzack and Ettinger and Wuhrmann, Barnard (1973) designed a process which included a primary denitrification reactor before - and a secondary denitrification reactor after the main basin plus a further reaeration basin in series (Fig. 2.3.) The primary denitrification reactor was separated completely from the aeration basin and a positive recycle ("a" recycle) instituted from the aerobic reactor to the predenitrification reactor; the underflow recycle discharged to the predenitrification reactor; the underflow recycle discharged to the predenitrification reactor ("s" recycle). This process was patented and became known as the Bardenpho process (Fig. 2.3.)

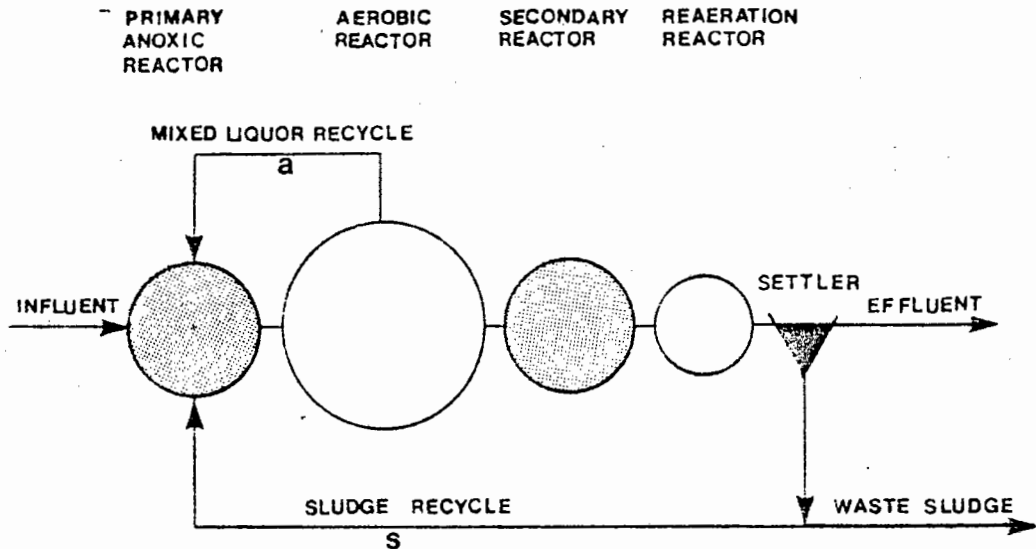


Fig. 2.3. Layout of the Bardenpho process

Barnard reported denitrification rates in the primary and secondary reactors of approximately 0,0036 and 0,0013 mg N/mg MLVSS/h respectively.

Investigations into the Bardenpho process at laboratory scale by Stern and Marais (1974) indicated two denitrification rates in the primary reactor: a rapid rate and a slower rate, both zero order reactions with respect to nitrate and proportional to the active mass X_a . The removal rates can be expressed as follows:

$$dN_n/dt = K_a X_a \quad (4)$$

K_a = denitrification rate constant (mg N/mg VASS/h)

N_n = nitrate concentration (mg N/l)

X_a = active mass concentration (mg/l)

Investigating the temperature dependency of equation (4) Marsden and Marais (1977) included a temperature correction factor into the rate equation expressed in the primary reactor:

* First phase

$$K_{1T} = K_{120} (1,2)^{T-20} \quad (5)$$

where

$$\begin{aligned} K_{120} &= 0,030 \quad (\text{mg N/mg VASS/h}) \\ &= \text{first rate constant } K_1 \text{ at } 20 \text{ }^\circ\text{C} \end{aligned}$$

* Second phase

$$K_{2T} = K_{220} (1,1)^{T-20} \quad (6)$$

where

$$\begin{aligned} K_{220} &= 0,0042 \quad (\text{mg N/mg VASS/h}) \\ &= \text{second rate constant } K_2 \text{ at } 20 \text{ }^\circ\text{C} \end{aligned}$$

Considering the second anoxic reactor in the Bardenpho process, Stern and Marais (1974) showed that there was only one denitrification rate which also could be expressed by equation (6) with K having the following temperature dependence:

$$K_{3T} = K_{320} (1,03)^{T-20} \quad (7)$$

where

$$\begin{aligned} K_{320} &= 0,003 \quad (\text{mg N/mg VASS/h}) \\ &= \text{third rate constant } K_3 \text{ at } 20 \text{ }^\circ\text{C} \end{aligned}$$

Two theories have been proposed to explain the different rates observed:

- (i) Barnard (1975) suggested that the redox potential of the activated sludge in the immediate vicinity of the added raw sewage was lower than that in the remainder of the denitrification reactor. This lower potential resulted in rapid denitrification which continued until the raw sewage feed and the activated sludge were evenly dispersed, at which stage the second denitrification stage became operative.

(ii) Dold, Ekama and Marais (1980) proposed a bi-substrate theory in which it was postulated that raw sewage contained two fractions of different biodegradability:

- * a readily biodegradable soluble fraction which is absorbed and metabolised at a very high rate resulting in a rapid utilisation of electron acceptors, in this case nitrate.

- * a particulate fraction which must first be adsorbed onto the organism and solubilised by extracellular enzymes prior to metabolism by the organism.

The solubilization stage is slow and consequently the rate of nitrate metabolism is slow.

Depending on the mass of slowly biodegradable fraction available in the pre and post denitrification reactors, the rates K_2 and K_3 differ correspondingly. Integrating equation (4),

$$N_n = KX_a R \quad (8)$$

where R is the nominal retention time over which the reaction takes place. Van Haandel, Dold and Marais (1982) showed that the easily biodegradable material is rapidly utilised, in a matter of minutes i.e. R is very small; furthermore the mass concentration of nitrate removed in this reaction is proportional to the mass concentration of easily biodegradable material. Consequently provided the retention time in the reactor is longer than a few minutes the removal of nitrate due to easily biodegradable material is constant for a particular waste water. The removal rate of nitrate due to the K_2 and K_3 reactions remains constant per unit of retention time so that the mass concentration removed increases as the retention time increases.

As a consequence of this behavioural pattern, the equation for nitrate removal by the plant can be written:

$$Ns' = Y S_{bi} + K_2 X_a R_1 + K_3 X_a R_3 \quad (9)$$

where

Ns' = mass of nitrate removed (mg N/l)

Y = constant depending on the fraction of readily biodegradable COD available in the incoming COD

S_{bi} = biodegradable COD concentration (mg/l)

K_2 = second denitrification rate constant in the primary anoxic reactor (mg N/mg X_a /h)

K_3 = denitrification rate constant in the secondary anoxic reactor (mg N/mg X_a /h)

X_a = active mass concentration (mg/l)

R_1 = nominal retention time in the primary anoxic reactor (h)

R_3 = nominal retention time in the secondary anoxic reactor (h)

Research by Stern and Marais (1974) and van Haandel et al (1982)

showed:

- (i) Equation (9) applies only if nitrate is still present in the effluent from the reactor.
- (ii) Over a day the mean concentration of nitrate removed is approximately the same irrespective of whether the flow and load is cyclic or constant.

2.3. Nitrification in anoxic/aerobic systems

Denitrification can not occur if nitrification does not take place. Van Haandel, Dold and Marais (1982) analysed the conditions which must be satisfied to ensure a high efficiency of nitrification in

an anoxic - aerobic system. They found that when an anoxic zone is present the minimum sludge age for nitrification is given by:

$$R'_{sm} = 1 / [(1 - f_x) U_{nmT} - b_{nT}] \quad (10)$$

where f_x is the fraction of the mass of sludge in an anoxic state.

To ensure high efficiency of nitrification the sludge age R_s must be greater than R'_{sm} and they found that provided $R_s > 1,25 R'_{sm}$ nitrification efficiency will always be high, greater than 95 per cent, and stable under both cyclic and steady flow conditions. As R'_{sm} is affected by the value of U_{nmT} , for design the lowest U_{nmT} , at the lowest temperature must be inserted.

2.4. Optimization of anoxic - aerobic plants

In design, if the sludge age R_s and the specific growth rate of the nitrifiers U_{nm} are specified, then the maximum anoxic mass fraction allowable can be calculated from equation (10). The critical anoxic mass fraction is given by the smallest value of U_{nm} , hence the calculation must be done at the lowest temperature expected. By choosing a number of sludge ages it is possible to develop a curve relating the anoxic mass fraction f_x to the sludge age R_s .

At about 30 days sludge age the anoxic mass fraction is about 0,5 if $U_{nm20} = 0,35$, and Siebritz, Ekama and Marais (1980) found that there is a limit to the magnitude of f_x ; for a f_x greater than 0,5 poor settling sludges were observed. The empirical rule therefore, is that under no condition should f_x exceed 0,5 - see Fig. 2.4.

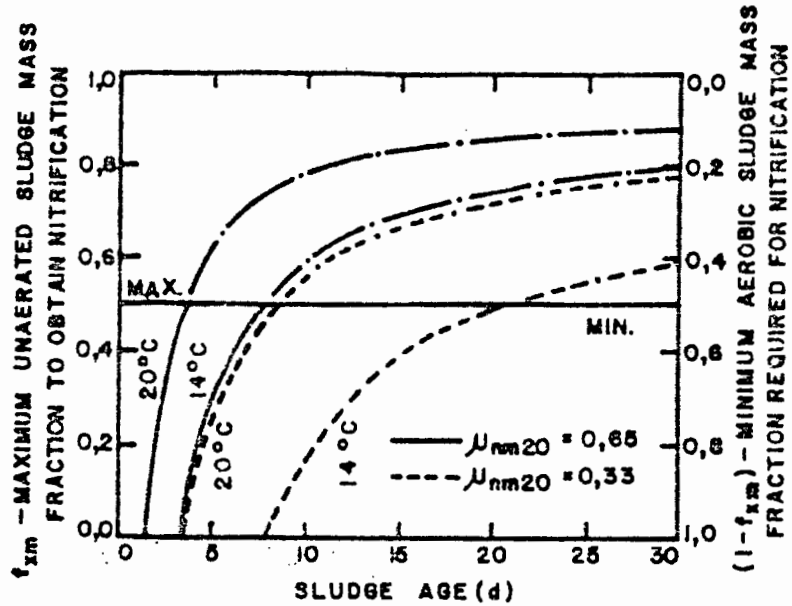


Fig. 2.4. Maximum permissible anoxic volume fraction or minimum aerobic volume fraction required to sustain nitrification versus sludge age for different maximum specific growth rate constants of the nitrifiers at 20 C and 14 C

Having obtained the relationship between f_x and R_s , van Haandel, Dold and Marais (1982) then derived an equation whereby, specifying the "a" and "s" recycles (Fig. 2.3.) the anoxic mass fraction can be subdivided into pre- and post- denitrification sub-fractions. This needs to be done at the maximum expected temperatures. The reason for this is that at low temperatures below 14 C K_2 and K_3 become identical. At this lower temperature because $K_2 = K_3$ the division of f_x into f_{x1} and f_{x2} can be as one pleases, but at the higher temperatures $K_2 > K_3$, and removals greater than at lower temperatures are achieved if f_{x1} and f_{x2} are appropriately selected. Having divided f_x into f_{x1} and f_{x2} , then by equating the denitrification potential (van Haandel, Ekama and Marais, 1981) plus the nitrogen required for synthesis to the influent TKN, the TKN/COD ratio the plant can handle for complete denitrification can be determined.

In this manner a maximum TKN/COD ratio versus sludge age can be plotted - see Fig. 2.5.

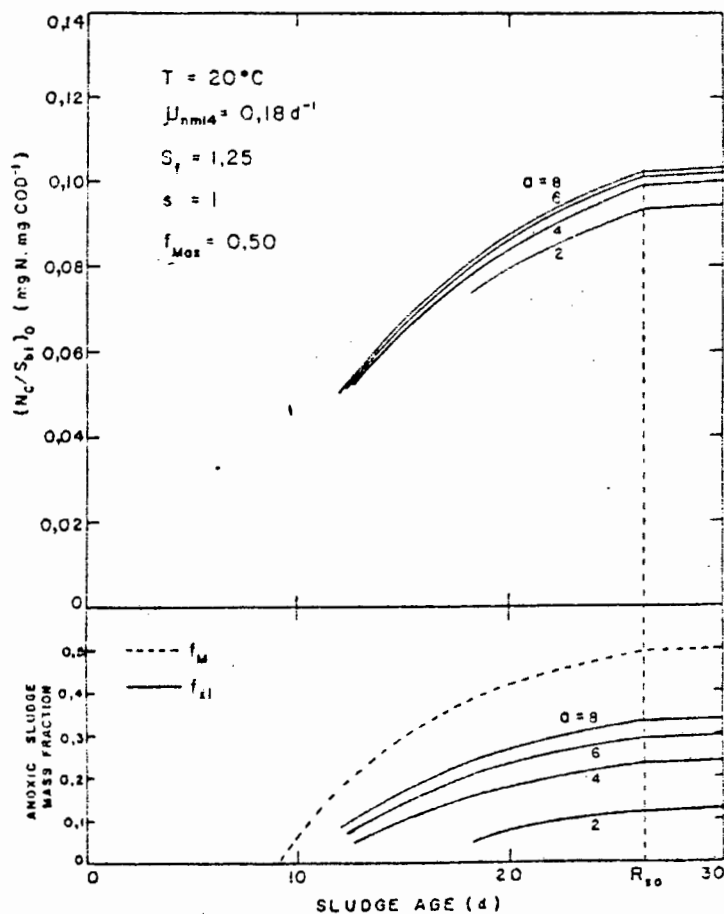


Fig. 2.5. Design chart for selection of optimal nitrogen removal configuration

N_c = Nitrification capacity (mg N/l influent)

S_{bi} = Biodegradable COD concentration of the influent (mg/l)

N_{ti} = Total nitrogen concentration of the influent (mg N/l)

S_{ti} = Total COD concentration of the influent (mg/l)

μ_{nm} = Specific growth rate for Nitrosomonas (d^{-1})

S_f = Safety factor

a = MLSS recycle ratio

s = Sludge return ratio

$(N_c/S_{bi})_0$ = Maximum ratio for which completa nitrification is possible

$(N_c/S_{bi})_i$ = Maximum ratio for Budzack Ettinger type process

It is now a simple matter, knowing the TKN/COD ratio and the minimum temperatures, to select the appropriate sludge age and the anoxic mass fraction f_x . Knowing the maximum temperatures and recycle ratios, f_x is subdivided into f_{x1} and f_{x2} .

If the TKN/COD ratio to be treated is so high that it does not intercept the TKN/COD vs R_g curve (Fig. 2.5.), one would select the maximum allowable $f_x = 0,50$, subdivide it into $f_{x1} = 0,5$, $f_{x2} = 0$ i.e. a Ludzack - Ettinger process. The plant now no longer can remove all the nitrate generated but it will remove the maximum possible.

The description above essentially contains the basic ideas for optimising nitrate removal from a single sludge nitrification - denitrification system; it allows a rapid assessment of an existing plant as to whether the plant is optimally designed.

3. BIOLOGICAL PHOSPHORUS REMOVAL

3.1. Introduction

Significant but unexplained removals of phosphorus in the activated sludge process were first noted at the Rilling Road plant at San Antonio Texas by Vacker, Connell and Wells (1967). This appeared to be contrary to the findings of earlier workers such as Sawyer (1944), Wuhrmann (1957), Levin and Shapiro (1965) who, investigating the removal of phosphorus from wastewater by biological means, concluded that normal bacterial metabolism could not bring about substantial phosphorus removal.

The phosphorus content of the activated sludge at the Rilling Road plant, which operated at a sludge age of 3,2 days, was found to be 5 - 7 mg P/mg of dry sludge, which is 2 - 3 times that normally found in similar plants. Endeavouring to explain this high

phosphorus content, operational parameters at Rilling Road were compared with other activated sludge plants and one major difference was noted: at the Rilling Road plant oxygen concentration at the head of the plant was zero. Similar observations on high phosphorus removal were made on the Baltimore plant in Maryland USA by Scalf, Pfeffer, Lively and Witherow (1969), where again the dissolved oxygen concentration was virtually zero at the head of the aeration basin. Milbury, McCauley and Hawthorne (1971) in a detailed study of the phosphorus removal behaviour in a conventional (i.e. semi plug flow) activated sludge plant showing excess phosphorus removal, found that the phosphorus profile across the length of the aeration basin showed a high concentration at the influent end tailing off to very low concentrations at the effluent point.

In order to explain excess phosphorus removal two main theories have been developed, both of which are still being debated by many research investigators today.

3.2. Chemical precipitation theories

Menar and Jenkins (1970) and Ferguson, Jenkins and Eastman (1973), from an examination of solubility characteristics of various phosphate compounds, postulated that phosphorus could be precipitated at pH values of 7,5 - 8,5 as calcium hydroxy apatite $\text{Ca}_5(\text{OH})(\text{PO}_4)_3$. They further suggested that where the aeration intensity was low, large quantities of carbon dioxide were produced which remained in solution, resulting in a depression of pH and resolution of precipitated calcium hydroxy apatite. Considering the precipitation of phosphorus further, De Boice and Thomas (1975) developed a steady state computer model to predict the conditions required for precipitation of a number of calcium and

phosphorus containing salts. Applying this data to the Baltimore plant they showed that removal of phosphorus as an insoluble calcium salt was theoretically feasible.

Working along somewhat different lines Arvin and Christensen (1979) showed experimentally that denitrification could initiate the precipitation of calcium phosphate inside biofilms. This was said to be a consequence of high pH values which can be created by a combination of the production of alkalinity and the diffusional resistance to its export from the bacterial cell.

3.3. Biological removal theories

Biological removal of phosphorus in the activated sludge process implies the incorporation of phosphorus into microorganisms, and its subsequent removal from the process with the waste sludge.

In general terms this may be achieved in two ways:

- * phosphorus may be taken into the cell to meet normal metabolic requirements
- * phosphorus in excess of that required for immediate metabolism could accumulate within specific areas usually as an osmotically inert polyphosphate

3.3.1. Biological excess accumulation of phosphorus in microorganisms

Phosphorus, an essential element in the life cycle of all living organisms, can enter or leave bacteria in only one form: ortho-phosphate.

Entry into the cell is generally by one or other of the following:

(Potgieter, 1980)

- (i) Diffusion which requires that a concentration gradient of phosphorus exists across the cell membrane; high on the outside, low on the inside of the cell;

phosphorus will transfer from the higher to the lower concentration across the cell wall.

- (ii) Active transport in which phosphorus is forced into the cell against a concentration gradient using adenosine triphosphate (ATP) as an energy source.

Normal domestic sewage contains organic phosphate, polyphosphate and orthophosphate. The first two must be converted into the ortho form by extracellular enzymes e.g. alkaline phosphatase before they can be utilised by the cell. Having entered the cell orthophosphate is involved in a large number of biochemical reactions, one of the most important being the formation of adenosine triphosphate (ATP) which serves as a source of highly utilisable energy to the organisms. When ATP is converted in the metabolic process to adenosine diphosphate (ADP) a large amount of energy, 16Kcal/mole is liberated.

In order to estimate this mass of phosphorus which has entered the cell to satisfy metabolic requirements Marais and Ekama (1976) derived the following equation:

$$\Delta P = \frac{Y_{S_{bi}}}{(1 + bR_s)} (\alpha + f_{pe} f_b R_s) + f_{pi} X_{ii} \quad (11)$$

ΔP = concentration of phosphorus removed from the influent and incorporated into the sludge wasted / day

$$P = (P_{inf} - P_{eff}) \quad (\text{mg/l})$$

α = fraction of phosphorus as P in the active sludge mass, normally about 0,03 but much higher if excess removal takes place (mg P/mg VASS)

f_{pe} = fraction of phosphorus as P in the endogenous residue
= 0,015 (mg P/mg VSS)

f_{pi} = fraction of phosphorus as P in the volatile inert fraction in the influent
= 0,015 (mg P/mg VSS)

Y = growth yield coefficient
 = 0,45 (mg VASS/mg COD)

S_{bi} = biodegradable influent COD
 = $S_{ti} (1 - f_{us} - 1,48 f_{up})$ (mg/l)

where

f_{us} = fraction of soluble unbiodegradable COD
 in influent (mg COD/mg COD)

f_{up} = fraction of particulate unbiodegradable volatile solids
 in influent (mg VSS/mg COD)

S_{ti} = total influent COD concentration (mg/l)

b = specific endogenous respiration rate
 = 0,24 (mg VASS/mg VASS/d)

R_s = sludge age (d)

X_{ii} = unbiodegradable influent volatile solids
 = $f_{up} S_{ti} / 1,46$ (mg/l)

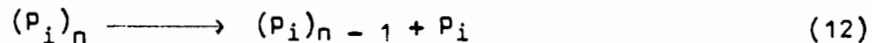
From equation (11) the concentration of phosphorus removed is a function of influent COD concentration and sludge age. In the case of normal domestic sewage, the ΔP /COD ratio can range from 0,01 to 0,03, the higher value being where excess phosphorus removal takes place.

3.3.2. Non - metabolic accumulation of excess phosphorus in bacteria

The most likely form in which phosphorus may accumulate within the bacterial cell is as high energy, osmotically inert polyphosphate (Dawes and Senior, 1973). This combines with ribonucleic acid, protein lipids and Mg^{2+} to form volutin granules. (A list of some of the organisms known to accumulate polyphosphates is given in Appendix A1). These volutins are spherical in shape and vary in size from 40 to 110 nm in diameter, the larger particles possibly being aggregates of the smaller. (Goodman, Sauer, Sauer and Rush, 1969; as quoted by Dawes and Senior, 1973, Buchan, 1981)

From the literature (Harold, 1966; Dawes and Senior, 1973) it appears that the main function of polyphosphate in volutin granules is twofold:

- (i) To act as a reserve "phosphate pool" . Under conditions where the phosphorus requirements of the cell cannot be satisfied by the exogenous supply, the polyphosphates in the "phosphate pool" hydrolyse, releasing orthophosphate for use by the cell.
- (ii) To serve as a reserve "energy pool" . This can be used to supply the cell with additional energy if required; Dawes and Senior (1973) have shown that the energy associated with the reaction given below releases approximately 9 Kcal/mole which is slightly lower than the energy associated with ATP .



P_i = orthophosphate concentration (mole/l)

n = number of orthophosphate molecules in the polymer

In order to form these pools phosphorus must enter the cell and be synthesized into polyphosphate.

Harold (1966) and Dawes and Senior (1973) reporting on transportation of phosphorus within a bacterial cell suggest two principles by which polyphosphate may accumulate in volutin granules:

- (i) In growing cells, incoming phosphorus is used in the synthesis of new cells which requires the formation and utilisation of ATP . If growth is stopped because of exhaustion of a nutrient other than phosphorus, competition for ATP is relieved, and polyphosphate

accumulates via a reaction catalysed by the enzyme polyphosphate kinase, indicated in Fig. 2.6. by route 2. This phenomenon is generally referred to as luxury uptake.

- (ii) In cells that have been subjected to prior phosphorus starvation, very rapid synthesis of polyphosphate results when phosphorus is added, again using polyphosphate kinase and route 2 in Fig. 2.6.

This is sometimes referred to as the overplus mechanism.

The polyphosphates, having accumulated within the volutins, may degrade via two routes:

- * using polyphosphate kinase (route 3 in Fig. 2.6.)
- * using polyphosphatase (route 1 in Fig. 2.6.)

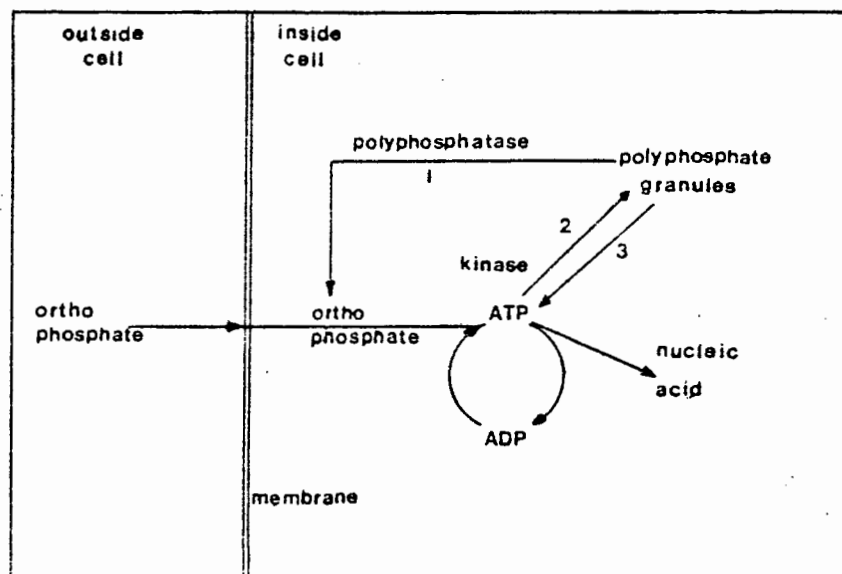


Fig. 2.6. Simplified biochemical pathways depicting phosphorus accumulation

3.3.3. Excess phosphorus removal as applied to the activated sludge process

The theories offered above are not easily applied to the activated sludge process because of the complex nature of sewage and the wide variety of living organisms generated by the process. Yall, Boughton, Knudsen and Sinclair (1970) and Garber (1972) have applied theories to the activated sludge process but to date the precise mechanism of phosphorus accumulation has not been fully elucidated. Nevertheless research and development in this field is continuing and highlights from this work are presented.

3.3.3.1. Acinetobacter

Fuhs and Chen (1975) suggested that the bacterium Acinetobacter, a strict aerobe, and one of the most abundant organisms in sewage (Water Pollution Research, 1968), is the main micro-organism responsible for biological phosphorus removal. Using activated sludge taken from a pilot plant removing phosphorus, Lawson (1980) was able to identify 20 different strains of Acinetobacter, some of which exhibited a marked increase in size and change in shape (pleomorphism) concomitant with phosphorus removal. Using electron microscopy to examine the sludge from a number of activated sludge plants which were removing phosphorus Buchan (1981) also noted pleomorphism in an organism strongly resembling Acinetobacter which accumulated significant quantities of phosphorus in volutin granules.

3.3.3.2. The anaerobic zone

A critical evaluation of the above data suggests that a necessary prerequisite for phosphorus removal in the activated sludge process is the presence of a non-aerated, well mixed zone

which should desirably be free of nitrates. Davelaar, Davies and Wiechers (1978) indicated in pilot plant experiments that the organisms responsible for phosphorus removal are always present in activated sludge and only need some form of activation to enhance excess phosphorus accumulation. Fuhs and Chen (1975) suggest that the function of this zone is to ferment substrate into a form which can be readily assimilated by Acinetobacter. Furthermore Siebritz, Ekama and Marais (1980) are of the opinion that 25 mg/l of readily biodegradable COD must be present in the anaerobic zone in order to trigger the phosphorus removal mechanism. This COD could well be in the form of volatile acids which are the end product of fermentation.

The degree of anaerobiosis required to achieve fermentation or to trigger the mechanism to accumulate phosphorus was considered by van Haandel et al (1979) who expressed it in terms of an anaerobic potential which is an alternate method for expressing redox potentials.

3.3.3.3. Release of phosphorus in the anaerobic zone

Good phosphorus uptake in the aerobic zone has been found to be associated with an adequate release in the anaerobic zone. This release was always associated with uptake of phosphorus in the aerobic zone (Barnard (1974, 1976); McClaren and Wood (1976); Simkins and McClaren (1978). The last named suggested that the release by itself might not be involved directly in the phosphorus removal mechanism, but indicated rather that appropriate conditions have been created in the anaerobic zone which promoted phosphorus uptake in the aerobic zone.

Van Haandel, Ekama and Marais (1979) and Siebritz, Ekama and Marais (1980) endeavoured to define the anaerobic state that resulted in release. They accepted that release is associated with uptake but attempted to establish in terms of measurable parameters when release would occur.

Siebritz, Ekama and Marais (1980) examining the hypothesis that the anaerobic capacity is the critical parameter, observed that phosphorus release could occur in the presence of nitrate if a sufficiently large mass concentration of COD is added to the mixed liquor; this tended to negate the hypothesis that a redox effect was responsible for phosphorus release. From further investigations they concluded that the conditions for phosphorus release are:

- (i) Nitrate must be absent
- (ii) The concentration of easily biodegradable COD, S_{bsa} in the anaerobic reactor must be greater than 25 mg/l. The S_{bsa} concentration in the anaerobic reactor may be estimated using the following equation (Siebritz et al, 1980)

$$S_{bsa} = \frac{S_{bsi} Q - 8,6 N_{nr} Q_r}{(Q + Q_r)} \quad (13)$$

where

S_{bsa} = readily biodegradable COD remaining
in the anaerobic reactor (mg COD/l)

S_{bsi} = readily biodegradable COD in the feed (mg COD/l)

N_{nr} = nitrate concentration in the recycle (mg N/l)

Q = feed flow rate (l/d)

Q_r = returned sludge flow rate (l/d)

The mass concentration of phosphorus released will depend on $(S_{bsa} - 25)$ and the actual time of residence in the anaerobic reactor. Apparently the higher the release the greater the uptake that occurs. This led to the establishment of a phosphorus release factor P_f defined as

$$P_f = (S_{bsa} - 20) f_{xa} \quad (14)$$

where f_{xa} is the fraction of sludge in the anaerobic reactor. Siebritz et al (1980) found that a relationship existed between P_f and the percentage concentration of phosphorus with respect to the VASS, and plotted observed against P_f as shown in fig. 2.7.

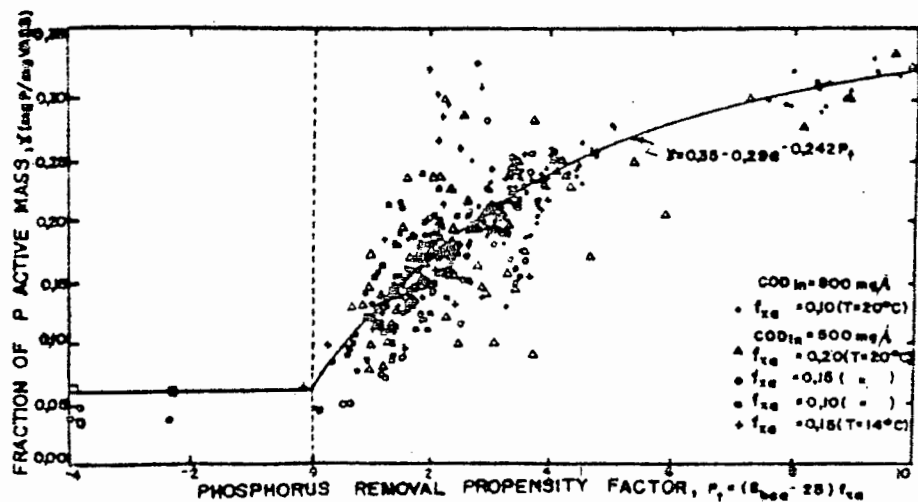


Fig. 2.7. Coefficient of excess phosphorus removal versus phosphorus removal propensity factor observed in the UCT (University of Cape Town) Process

3.3.4.- Phosphorus removal in single sludge systems

A number of laboratory and pilot plant investigations have been conducted by various workers to develop a single sludge process to remove phosphorus. Barnard (1974) (Fig. 2.3.) in pilot plant studies using the Bardenpho process, noted excess phosphorus removal when the nitrate concentration in the second anoxic zone fail to zero. A release of phosphorus in this zone was observed with subsequent uptake in the reaeration zone. To obtain release in a most expeditious manner he modified the Bardenpho process by the incorporation of a nitrogen- and oxygen- free anaerobic zone at the head of the primary anoxic zone as shown in Fig. 2.8. and called the system the Phoredox process. He assumed that nitrates returned to this zone in the underflow from the final clarifiers would be negligible.

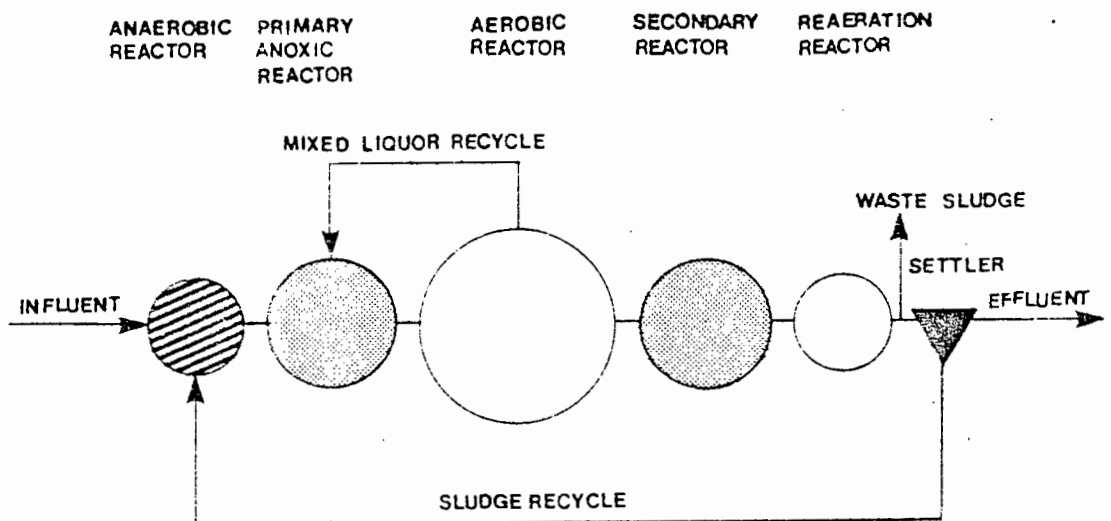


Fig. 2.8. The Phoredox activated sludge process configuration

Investigations into the behaviour of the Phoredox process have indicated that : (Simkins and McClaren, 1978; Marsden and Marais, 1977)

- (i) when nitrate was not zero in the effluent excess removal of phosphorus declined sharply
- (ii) the second anoxic zone removed virtually no phosphorus

In attempting to quantify this removal with the Phoredox process, Marsden and Marais (1977) and Hoffman and Marais (1977) found that the combined effect of metabolic uptake and luxury uptake is limited and related to the influent COD concentrations and the sludge age.

$$P = S_{ti} \frac{(1 - f_{us} - f_{up}) Y_h (\alpha + f_p f_{bHT} R_s)}{(1 + b_{HT} R_s)} + \frac{f_p f_{up}}{P} \quad (15)$$

where

S_{bi} = biodegradable COD (mg/l)

f_p = fraction of phosphorus present in the active mass (mg P/mg VASS)

VASS = volatile active suspended solids (mg/l)

f_{up} = fraction of phosphorus present in the inert and endogenous mass (mg P/mg VSS)

P = phosphorus concentration per unit influent flow discharged with the waste sludge (mg P/l)

Siebritz, Ekama and Marais (1980) using the Modified Phoredox i.e. a Phoredox system without a secondary anoxic zone, found that as the TKN/COD ratio of the plant increased above about 0,07 the nitrate in the effluent increased proportionally with the increase. As the effluent nitrate concentration is also the concentration recycled to the anaerobic reactor, if equation (13) is applied to determine S_{bsa} in the anaerobic reactor, the S_{bsa} concentration in the anaerobic reactor will be reduced to less than 25 mg/l if the nitrate

concentration in the recycle attains a few mg/l in value, phosphorus release will cease and excess uptake will terminate.

In order to overcome the situation where TKN/COD ratio is so high that the Modified Phoredox process can not establish the minimum S_{bsa} for phosphorus removal, Siebritz, Ekama and Marais (1980) developed a further modification of the Phoredox process (Fig. 2.9.) called the UCT process.

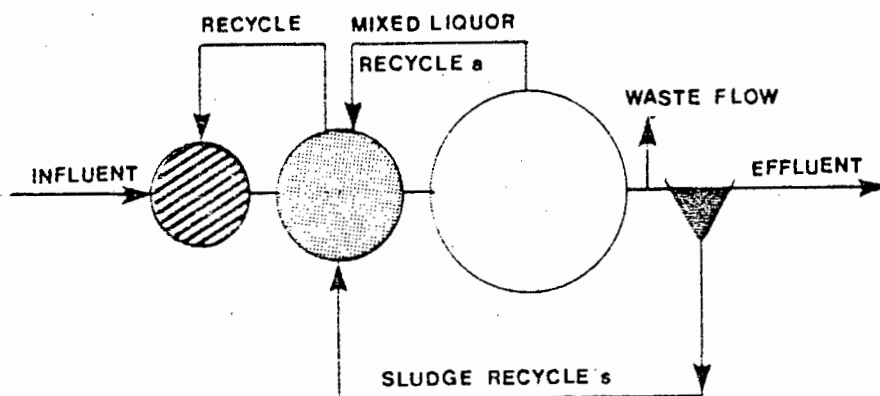


Fig. 2.9. The process configuration of the University of Cape Town Modification of the activated sludge process

By returning the underflow from the final clarifiers to the primary anoxic zone the nitrate can be reduced to zero by controlling the "a" recycle. An additional recycle is introduced from the anoxic zone to the anaerobic zone and, because no nitrate is present in the recycle, the S_{bsa} is not depleted, and conditions favourable for phosphorus release are established (see Fig. 2.9.) i.e. in terms of Siebritz's hypothesis the highest value of S_{bsa} will be maintained in the anaerobic reactor. This modification lends itself to a situation where the feed wastewater has a TKN/COD ratio higher than 0,07 to 0,08.

From the discussion above it would appear that in terms of the findings of the Cape Town group, the Phoredox five stage process is appropriate only if the TKN/COD ratio is less than about 0,07. At higher TKN/COD ratios the appropriate system would appear to be the UCT process.

The prerequisite conditions for phosphorus release and uptake that have emerged are of great utility both for design of new works, and assessment of the phosphorus removal property in existing works and the effect of modification of such works.

4. ACTIVATED SLUDGE MODELS

The models described in the previous section for nitrification and denitrification, are derived from a general model, and, are strictly valid only for the constrained conditions of constant flow and load. The nitrification and denitrification behaviour can be analysed also by means of the general model under constant flow and load conditions. The response, however, of the general model is so similar to that using the approximate model that there is little point in using the general model when analysing constant flow and load conditions, particularly when the general model requires the use of a large computer.

When cyclic load and flow conditions are imposed on the plant then the theoretical response can be obtained only by the use of the general activated sludge model. A general model, which will be used in analysis of the response of large scale plants is that developed by Dold, Ekama and Marais (1980) for aerobic processes, and extended to incorporate denitrification by van Haandel, Ekama and Marais (1981). It is not the intention here to go into a detailed description of the model. This is extensively discussed in two papers of Dold et al (1980)

and van Haandel et al (1981) and in a report by van Haandel and Marais (1981) . Use will be made of the model to simulate two large scale plants, that of Goudkoppies and the Northern Works respectively, these being the only plants where it was possible to monitor all the important variables such as COD, flow, TKN, recycles (a and s recycles), and process variables (MLVSS, oxygen utilisation rate, nitrate, ammonia, temperature, etc.). If the predicted reponses correspond closely with the observed, not only will it contribute to the reliability of the model but will allow both assessment of the phosphorus removal property and the modifications that would promote optimal removal of nitrate and phosphorus.

C H A P T E R T H R E E

NUTRIENT REMOVAL AT THE JOHANNESBURG ALEXANDRA PLANT

1. INTRODUCTION

The main outfall sewers to the Northern Works generally follow the river valleys as can be seen in Fig. 1.1. By 1970 it was evident that the outfall connecting the now defunct Bruma Works to the Northern Works was rapidly becoming overloaded and that urgent remedial measures would have to be instituted. Relief was provided by the construction of a temporary 27 Ml/d facility at Alexandra Township, after which the Works was named. The only available site was a very narrow strip of land sandwiched between the outfall sewer and the Jukskei River, necessitating special design features:

- * a small head loss through the plant
- * compact design
- * low capital cost as the plant was expected to operate for only six years
- * limitation of area and close proximity to a residential area precluded the establishment of sludge handling facilities. Sludge could however be returned to the sewer for digestion at the Northern Works.

When all these points were considered there was no economic alternative to the extended aeration activated sludge process.

2. DESIGN OF THE PLANT

The feed to the plant, which included nightsoil from the Alexandra Township, was drawn directly from the Bruma sewer, rough control being exercised by hand operation of a penstock valve, and the rate of flow being monitored by a flume type flow meter. Typical hourly flow and daily flow characteristics are presented in Figs. 3.1. and 3.1. respectively.

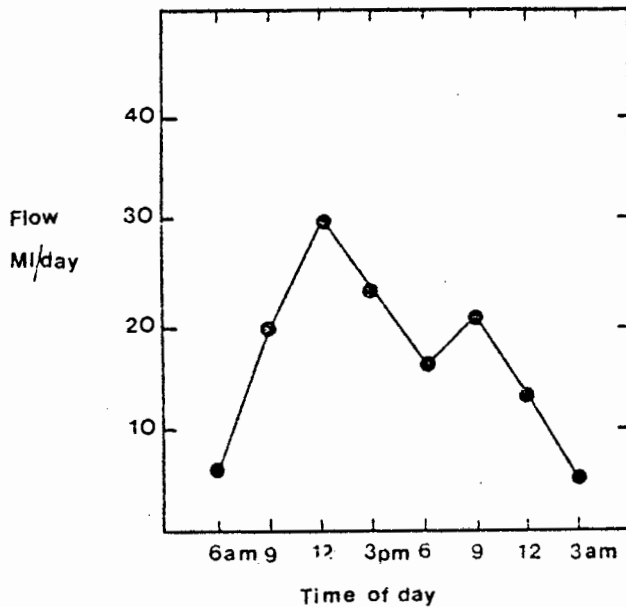


Fig. 3.1.

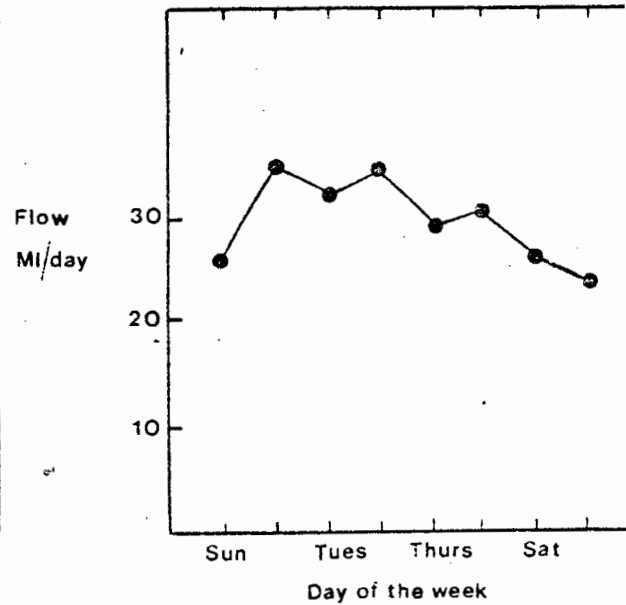


Fig. 3.2.

Johannesburg Alexandra Plant

Hourly feed flow variation
to plant over a day

Daily feed flow variation
to plant over a week

COD and TKN concentrations were monitored at 8 day intervals by taking 0,5 l grab samples every hour and compounding these over 24 hours. The plant, designed to nitrify completely, consisted of two mechanically aerated basins in series, the first comprising 60% of the total volume. Plant layout is shown in Fig. 3.3., design data in Table 3.1.

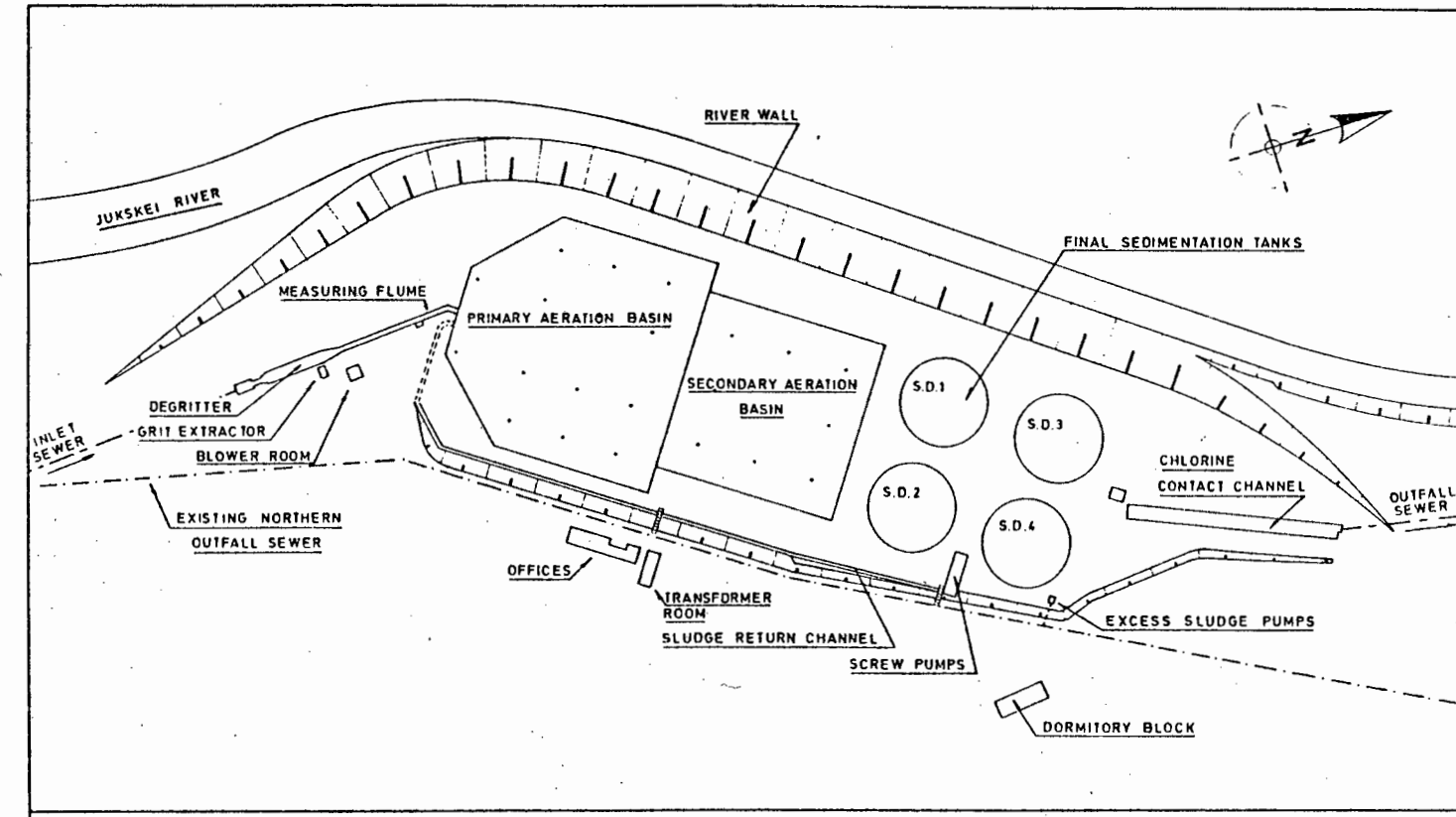


Fig. 3.3. Johannesburg Alexandra Plant design layout

Table 3.1.

ALEXANDRA PLANT DESIGN DATA

Flow D W F	(Ml/d)	27	
Raw sewage composition	BOD (mg/l)	350	
	TKN (mg N/l)	35	
<u>Aeration Basin</u>			
Volume (both basins)	(m ³)	33 500	
Retention period at design flow	(h)	28	
Oxygenation capacity	(kg O ₂ /h)	1562	
Mixed liquor suspended solids	(mg/l)	3 500 - 5 000	
Water depth in tank	(m)	4,57	
Basin number		1	2
Volume	(m ³)	20 450	13 050
Number of aerators		14	9
Power per aerator at design flow	(kW)	37	30
Energy density at design flow	(W/m ³)	27	21
<u>Secondary Sedimentation Tanks</u>			
Number		4	
Diameter	(m)	26,7	
Side water depth	(m)	3,73	
Surface loading at D W F	(m/h)	0,5	
Volume	(m ³)	2 350	
Retention time at D W F	(h)	8,3	
<u>Returned Sludge Pumps</u>			
Number		2	
Power per pump	(kW)	11	
Maximum pumping capacity	(m ³ /sec/pump)	0,24	
Head	(m)	2,1	

The rotational direction of the aerators is shown in Fig. 3.4.

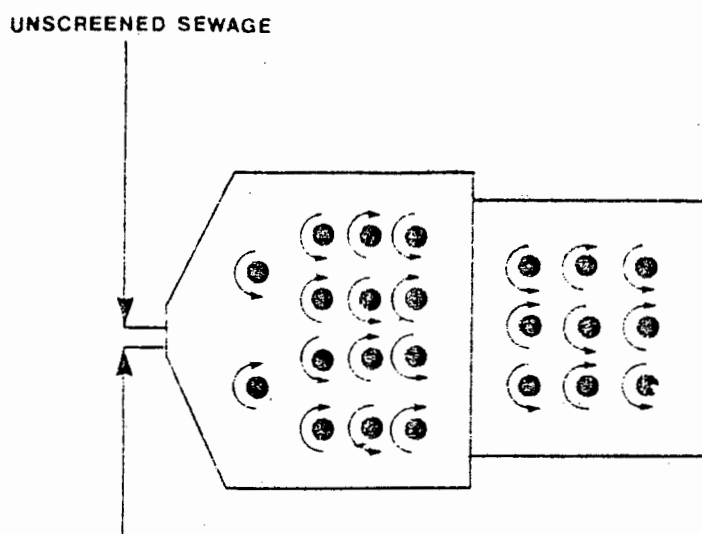


Fig. 3.4. Johannesburg Alexandra Plant

Direction of rotation of surface aerators.

Specific control features were provided as follows:

2.1. Dissolved oxygen control

Two methods were provided:

- (i) A manual system accomplished by changing the depth of immersion of the mechanical aerators by raising or lowering a level control weir at the discharge end of the second aeration basin.
- (ii) A semi-automatic system involving the connection of 12 aerators in the first aeration basin to a time switch which allowed the appropriate number of aerators to be switched in or out in accordance with a pre-determined load pattern.

2.2. Sludge age control

The utilitarian design made no provision for measuring the mass of sludge wasted per day. Consequently in the initial stages of plant operation it was not possible to determine accurately the sludge age. Limited control of sludge age was imposed by endeavouring to keep the mixed liquor suspended solids (MLSS) at a constant value. Initially the best concentration was established by changing the MLSS until a sludge with optimal settling properties was obtained. However, because the actual sludge age was indeterminate, it was virtually impossible to operate the plant in a scientific manner and data obtained during this period is of little value.

In 1976 improved control of the sludge age was made possible by the installation of a pump to discharge a preselected volume of mixed liquor from the aeration basin. With this method the sludge age could be maintained at any selected value by simply adjusting the flow to be abstracted for wastage each day, the sludge age, R_s , being defined by

$$R_s = \frac{\text{plant reactor volume}}{\text{volume wasted per day}}$$

The minimum sludge age for nitrification, R_{sm} was calculated from the formula derived by Downing et al (1964):

$$1 / R_{sm} = U_{nm20}^{(T - 20)} \quad (2)$$

where

U_{nmT} = specific growth rate of the nitrifiers at T°C

In the absence of any information on U_{nmT} it was decided to accept a very conservative, low value, $U_{nm20} = 0,3$. The lowest mean water temperature expected during winter was approximately 12°C.

Substituting $\mu_{nm20} = 0,3$ and $T = 12$ in equation (2), R_{sm} was estimated at 7,8 days. For efficient nitrification, i.e. 95% conversion, the sludge age must be greater than R_{sm} ; Ekama and Marais (1978) found that if a safety factor of 1,25 is applied to R_{sm} the plant will always nitrify efficiently even under cyclic flow conditions. Applying this factor of safety, the sludge age was specified at $1,25 \times 7,8 = 10$ days.

At the time the plant was put into operation it was already decided to utilise it as an experimental facility for testing denitrification behaviour. In order to allow for the effect of an anoxic zone of up to $1/3$ total volume, the sludge age was increased to $10/(1 - 1/3) = 15$ days.

3. EXPERIMENTAL INVESTIGATION INTO NITROGEN REMOVAL AND RESULTS

The objective of the experimental investigation was to determine the effect of anoxic conditions on the performance of the plant. Anoxic conditions were to be imposed by switching off one, two or more aerators in the first or primary basin adjacent to the discharge point of the raw sewage and underflow return. The investigation consisted of four experiments, with zero, one, two and three aerators switched off respectively.

When one, two or more aerators were switched off in the primary basin, correspondingly an anoxic zone of increasing size was created. Nitrates produced in the remaining aerated section of the primary basin cycled through the artificially created anoxic zone by the strong pumping action of the remaining aerators (as illustrated in Fig. 3.5. (a) and (b)) and were denitrified in the anoxic zone.

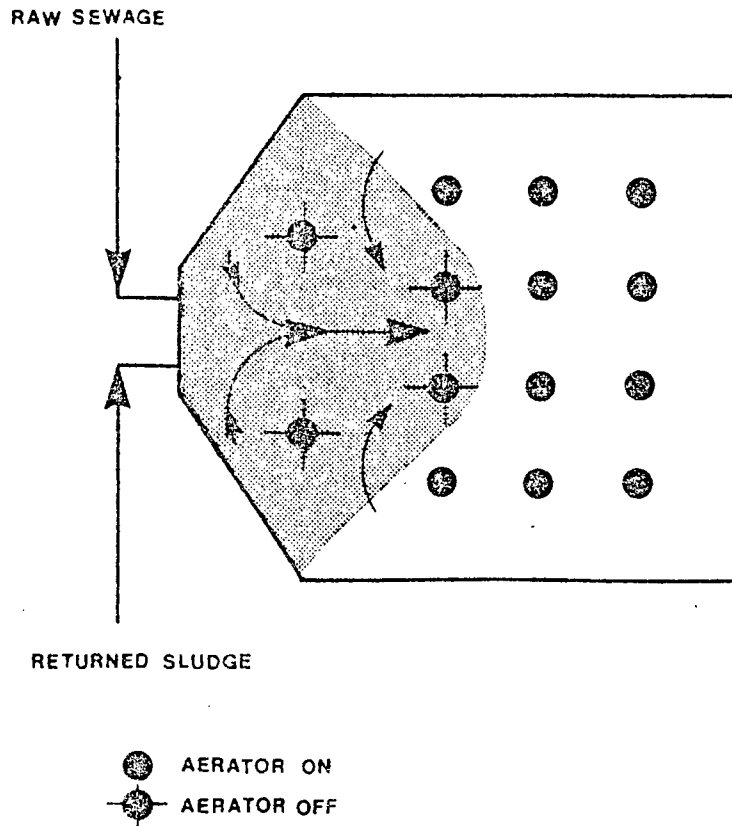


Fig. 3.5.(a) First aeration basin depicting the surface wave pattern and the current patterns obtained with aerators off

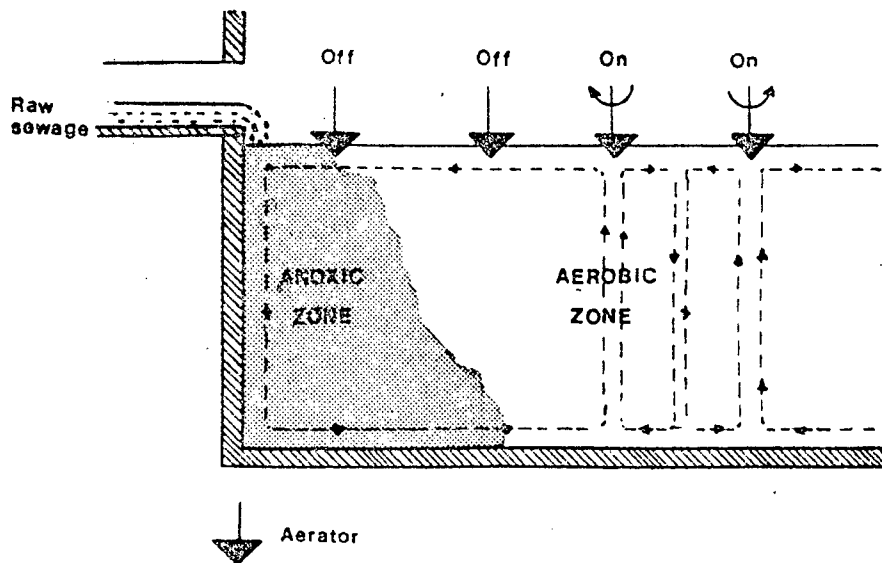


Fig. 3.5.(b) Flow pattern with four aerators off

Johannesburg Alexandra Plant

Preliminary tests, when operating the plant near the design load conditions and at a sludge age of 15 days, showed excessively high oxygen concentrations in the mixed liquor, particularly in the second aeration basin. Partial disruption of the activated sludge flocs was observed, resulting in fine colloidal particles in the effluent, higher than allowed by regulations promulgated under the Water Act No 54 of 1956. Since the plant effluent must comply with these regulations, corrective measures were taken by reducing the oxygenation input to the second basin by switching off three aerators; the dissolved oxygen concentration in the second aeration zone reduced to 1 - 2 mg/l, flocculation improved and the effluent quality fell below the minimum required. However, during the test in which three aerators were to be switched off in the primary basin the oxygen level reduced to zero in the second basin and it became necessary to switch in two of the three aerators i.e. only one remained off in the second basin.

3.1. Sampling procedure

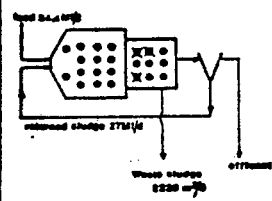
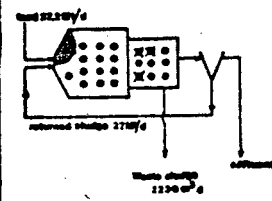
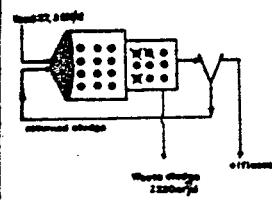
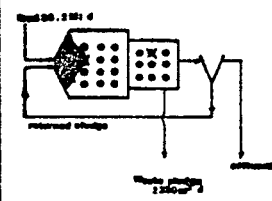
Sampling in each of the four experiments commenced only after the plant had been operated in the particular mode for a period of one month i.e. two sludge ages. Snap samples of influent just before the discharge into the plant, and effluent just before discharge into the river were taken hourly and compounded in equal proportions at the end of a 24 hour sampling period. Simultaneously hourly samples of mixed liquor were taken from a site adjacent to one of the aerators near the centre of the first basin and subsequently compounded to determine an average MLSS value. When the plant was sampled over weekends the samples were preserved by the addition of 1 ml of 0,85 per cent mass/mass mercuric chloride.

3.2. Results

A summary of the aerators which were switched in or out for each test run is given in Table 3.2. together with the summarised mean data for the feed and effluent during the test period. The raw data obtained during the various experiments are to be found in Appendix A2.

Table 3.2.

Performance data for the Alexandra Plant with 0,1,2 and 3 aerators switched out in the first basin

	Total Nitrogen mg/l	Ammonia as N mg/l	Nitrate as N mg/l	Alkalinity as CaCO ₃ mg/l	Total COD mg/l	Soluble COD mg/l	Suspended solids mg/l	pH	
	Feed	39,9	20,3	nil	240	590	230	290	7,6
	Effluent	32,3	4,9	23,5	150	74	52	25	7,5
No aerators off in first basin					MLSS 2 600 mg/l				
	Feed	38,4	22,4	nil	240	520	270	230	7,4
	Effluent	24,4	4,4	16,5	120	82	58	31	7,2
One aerator off in first basin					MLSS 2 560 mg/l				
	Feed	39,2	18,9	nil	240	480	220	290	7,4
	Effluent	16,7	3,9	8,8	150	66	53	30	7,5
Two aerators off in first basin					MLSS 2 260 mg/l				
	Feed	38,0	20,1	nil	220	590	280	310	7,2
	Effluent	12,4	3,0	6,6	130	56	43	21	7,3
Three aerators off in first basin					MLSS 3 100 mg/l				

4. DISCUSSION

4.1. MLSS concentration

The MLSS provided an opportunity to check the theoretical predictions of the model of Marais and Ekama (1976) . A sample calculation of the concentration of the various sludge fractions of the MLSS is shown in Appendix A3 . A summary of the calculated and observed mixed liquor concentrations is listed in Table 3.3.

Table 3.3.

Average experimental and theoretical MLSS concentrations during nitrogen removal experiments

	Experiment no.				Ave
	1	2	3	4	
Aerators off first basin	0	1	2	3	
Sludge age (d)	15	15	15	15	15
Theoretical MLSS (mg/l)	2695	1878	1740	2759	2268
Experimental MLSS (mg/l)	2620	2560	2260	3100	2635
% Difference	+3	-27	-23	-11	-14

The calculated and observed concentrations consistently show that the calculated concentrations were less than the observed. It is not possible to check the validity of the results because in 1974 when these tests were done the importance of the oxygen utilisation rate had not been realised and this measurement was not undertaken. Consequently it is not possible to do an energy or electron balance on the carbonaceous material i.e. check if the daily (mass of COD in the effluent) + (mass of COD in the sludge wastage) + (mass of oxygen utilised) equals the daily (mass of COD in the influent). Without a check on the COD balance it is not possible to make a comparison between the theoretical prediction and the observed MLSS. In addition the mean daily influent sample was obtained from grab samples, not compounded according to the flow at the time the sample was taken. This can have a serious effect when calculating the mass of COD that passes through the plant each day. Ekama and Marais estimated that, depending on the cyclicality of both flow and load, the mass of COD determined in the manner employed here, can result in underestimation of the load by up to 20%. On this basis the apparent underestimation of 14% is not unexpected.

4.2. Nitrification and denitrification

The introduction of an anoxic zone into the process configuration did not affect final effluent quality detrimentally. Both organically bound nitrogen and ammonia contained in the influent sewage will pass through the anoxic zone unaffected, except for dilution, to enter the downstream aerobic area where the organic nitrogen is converted to ammonia. At the sludge age provided (15 days) the major proportion of the bound and free saline nitrogen should be oxidised biologically to nitrate.

In particular the ammonia should be reduced to below 1 mg/l .
 However the ammonia concentrations in the last basin and effluent were in the range 3 - 5 mg N/l; no acceptable explanation for these high values can be advanced.

The mass of nitrate removed per day for each experiment is listed in Table 3.4.

Table 3.4.

Johannesburg Alexandra Plant

Mass of nitrogen removed per mass of incoming COD

	Experiment number			
	1	2	3	4
Number of aerators off in first basin	0	1	2	3
Mass N removed (kg/d)	261	380	514	901
Mass N removed / Mass COD (% removed)	1,4	3,0	5,3	4,6
Total N (mg N/l)				
Influent	39,9	38,4	39,2	38,0
Effluent	32,3	24,4	16,7	12,4
Drop in alkalinity (mg/l CaCO ₃)	90	120	90	90

It is clear that as an increasing number of aerators were switched off, so the mass of nitrate removed increased. The degree of nitrate removal from the system is dependent upon how much of the nitrate enters the anoxic zone and to what degree it is then denitrified. The nitrate entering the anoxic zone depends on the direction of rotation of the aerators, the recycle flow rate, the number and position of rotation of decommissioned aerators; whereas the degree of denitrification depends on the fraction of nitrate entering the anoxic zone and the concentration of COD in the influent. No reliable estimation of the mixing into the anoxic zone was possible so that no quantitative prediction of the concentration of nitrate entering the zone could be made. The only positive statement that appears to be valid is that as the anoxic zone increased, the concentration of nitrate in the effluent decreased. Consequently it would appear that the factor that limited denitrification was the size of the anoxic zone created. This was too small to denitrify all the nitrates entering the zone - as the zone increased so the nitrate removal increased correspondingly, as indicated in Table 3.4.

Surprisingly, from Table 3.4. , with one exception, there was no significant change in the loss in alkalinity through the process. It is to be expected that with nitrification and denitrification there will be a loss of alkalinity and this loss should decrease as the denitrification effect increases.

4.3. Settling properties of the sludge

The inclusion of a non-aerated zone into the original process configuration did not appear to produce any observable effects on

on the settling properties of the sludge. At that time comparative measurements on sludge settling rates were not appreciated so that little can be said quantitatively with regard to the effect of the anoxic zone on settling. Other effects were visible; over periods in which anoxic zones were present a dark brown scum formed during the cold weather on the surface of the aeration tank. Initially it was believed to be due to the presence of oil or fat but later was found to be due to floating accumulations of an organism Nocardia Spp. The reason why this particular organism should flourish in anoxic/aerobic systems, particularly in winter, has not yet been established, but without doubt the possibility of generation of this organism is a factor to be considered in these plants.

5. OPERATION OF THE PLANT IN A NITROGEN AND PHOSPHORUS REMOVAL MODE

Barnard (1974) , in testing the Bardenpho system, noted that on occasions when an anoxic reactor contained no nitrate phosphorus release was observed in that reactor and excess phosphorus removal was obtained in the aerobic reactor. From these observations it can be suggested that anaerobic conditions could be established by placing such a reactor at the head of the system to receive the influent flow and the underflow recycle - Fig. 2.8. This reactor, because of the low nitrate concentration in the underflow conceivably would utilise the mass of nitrate entering the reactor and establish an anaerobic condition sufficiently intense to cause release of phosphorus. This hypothesis was tested at the Alexandra plant by making due modifications to the operation of the plant.

Before modifying the Alexandra plant so that it would reduce phosphorus removal, it was necessary to examine a number of options whereby it may be possible to create 'anaerobic conditions' such that phosphorus will be released. Three options were investigated:

- (i) Switch out more than three aerators around the influent point of the main basin
- (ii) Store the sludge in the secondary clarifier
- (iii) Store the sludge on the floor of the main aeration basin.

It was not possible to monitor the system as extensively as deemed necessary; provision was not available on site for research investigations, only for essential monitoring required by law. Because of these difficulties only the bare minimum testing, to determine if denitrification and phosphorus removal took place, could be undertaken.

5.1. Switching out of aerators

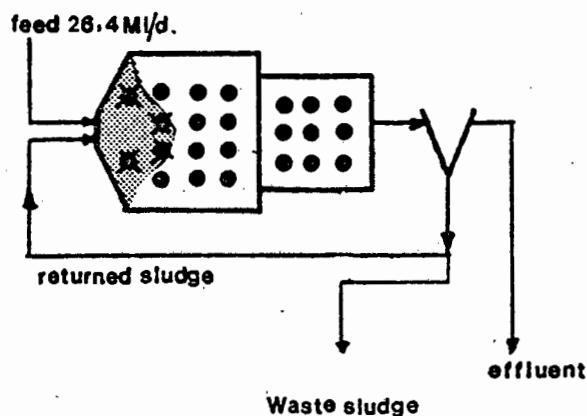
The plant was operated in the same manner as for the denitrification experiments except that four and later six aerators were switched out around the influent point.

Consider the experiment in which four aerators were switched out; the configuration of aerators switched in and out are shown in Table 3.5. , also the effluent response to the plant.

Table 3.5.

Johannesburg Alexandra Plant

Summary of operating parameters and chemical analysis under conditions where 4 aerators at the start of the aeration basin were off



		Concentration	
		Feed	Effluent
Total nitrogen	(mg N/l)	42,5	6,5
Ammonia as N	(mg N/l)	25,0	2,5
Nitrate as N	(mg N/l)	nil	1,9
Alkalinity as CaCO ₃	(mg/l)	220	140
Total COD	(mg/l)	690	85
Suspended solids	(mg/l)	-	7,0
pH		7,1	7,4
O - phosphate	(mg P/l)	5,5	5,0

Nitrate removal was excellent, the effluent concentrations ranged around 1,9 mg N/l. Phosphorus removal was relatively minor, indicating that the required anaerobic conditions for phosphorus release had not been attained. Nitrification was unaffected by the anoxic zone. This was indicated by the low ammonia nitrogen value of 2,5 mg N/l .

Notwithstanding the fairly large unaerated volume there was no significant deterioration of the settling properties. A typical settling velocity curve over this period is depicted in Fig. 3.6.

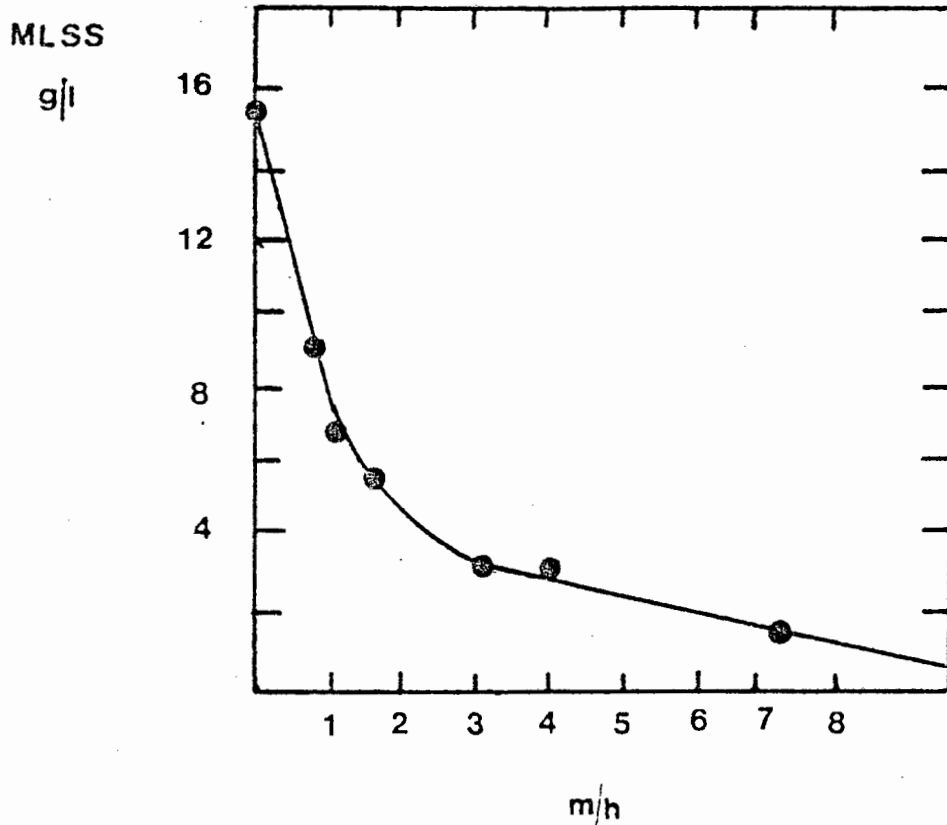


Fig. 3.6. Johannesburg Alexandra Plant

Relation between settling velocities of sludge and sludge concentration.

In order to increase the probability of anaerobic conditions six aerators were switched off around the inlet point - see Fig 3.7. This operational modification gave excess phosphorus removal sporadically. The plant was operated in this mode for four months; the influent and effluent phosphorus over this period is shown in Fig. 3.8.

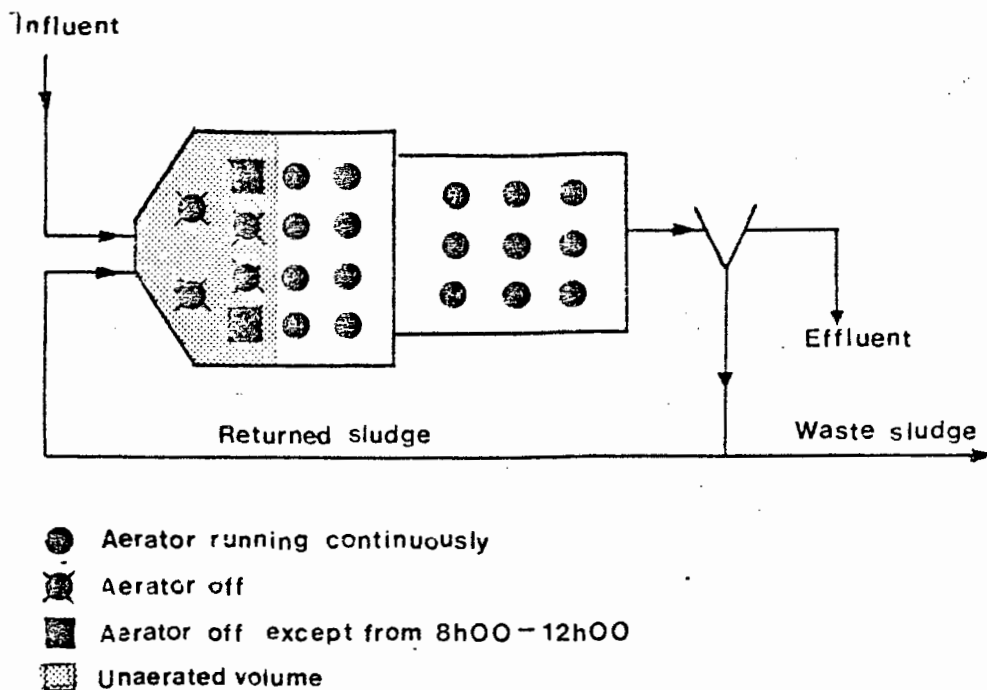


Fig. 3.7. Creation of an enlarged anaerobic zone at the head of the first aeration basin

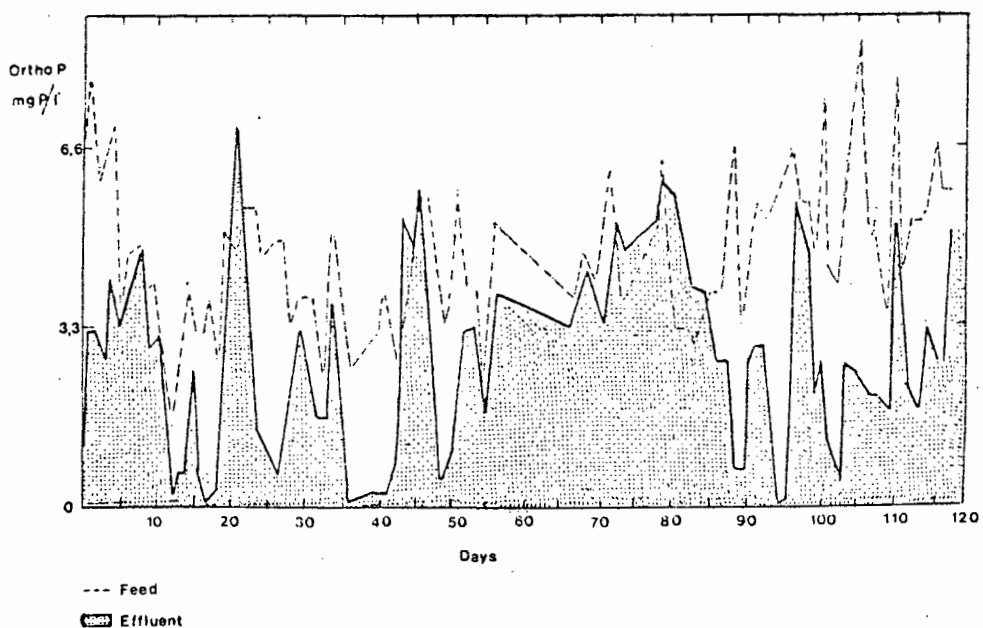


Fig. 3.8. Phosphorus removal data associated with an enlarged anaerobic zone (6 aerators off) at the head of the first aeration basin

Johannesburg Alexandra Plant

-Over this period the average removal of ortho-phosphorus was approximately 1,9 mg P/l. Nitrification efficiency deteriorated slightly - compare the ammonia effluent concentration in Table 3.5. with that in Table 3.6.

Table 3.6.

Average analysis of feed and effluent when 6 aerators were off at the head of the aeration basin

	Feed	Effluent
COD (mg/l)	540	58
TKN (mg N/l)	38	-
Ammonia as N (mg N/l)	17	6,1
Nitrate as N (mg N/l)	-	1,9
Ortho-phosphate as P (mg P/l)	4,7	2,8
Suspended solids (mg/l)		23
MLSS (mg/l)	3 900	

5.2. Accumulation of sludge in the secondary clarifier

The sporadic nature of the phosphorus removal, when six aerators were switched off in the main basin highlighted the unstable nature of the conditions created in the anoxic zone. It was then hypothesized that if a fixed quantity of sludge could be brought into a state of 'deep anaerobiosis' a more consistent phosphorus removal might be accomplished. To induce such a continuous state of anaerobiosis the underflow recycle was reduced so that sludge accumulated in the settling tank. By regulating the recycle appropriately the settling tank could be maintained approximately half full of sludge - Fig. 3.9.

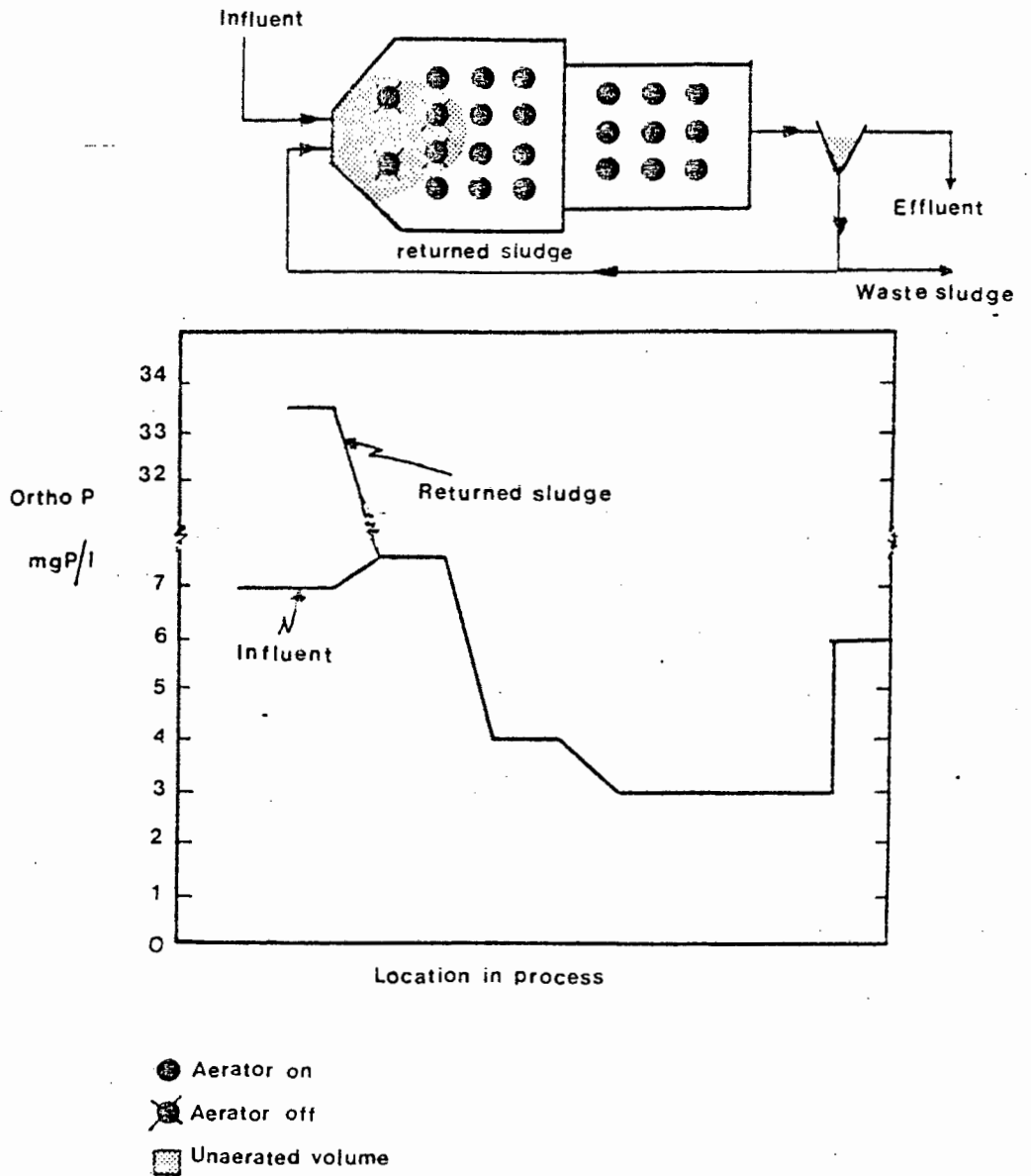


Fig. 3.9. Johannesburg Alexandra Plant
 Phosphorus profile when sludge stored temporarily in final clarifier

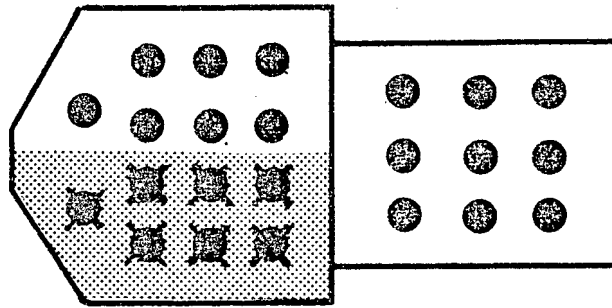
In monitoring the plant while operating in this mode a series of phosphorus determinations on filtered samples was carried out along the centre line of the plant and in the recycle (i.e. equivalent to the bottom of the settling tank). The results are shown in Fig. 3.9. It is clear that whereas very little release had taken place in the anoxic zone per se, significant release did take place in the settling tank. It is also evident that the stress condition caused substantial excess uptake of phosphorus in the aerobic region of the plant. Indeed 4 mg/l of phosphorus was taken up. Unfortunately in the settler the phosphorus was again partially released so that the effluent phosphorus was only 1 mg/l less than the influent. The mode of operation therefore did not provide for a significant removal of phosphorus.

5.3. Storage of sludge on the floor of the aeration basin

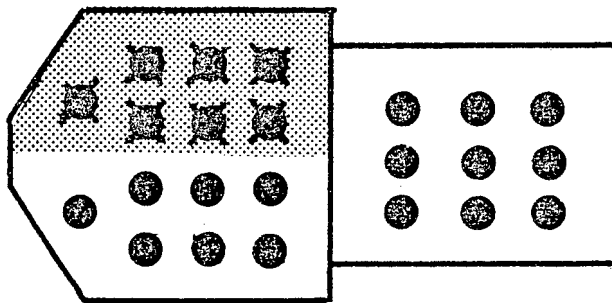
It was hypothesised that release of phosphorus in the sludge could be induced by allowing accumulation of sludge to form on the base of the tank by switching off some of the aerators for a fixed period to test this approach. All the aerators on one side of the first aeration basin were switched off to encourage settlement of sludge on the floor. Sludge accumulated for 24 hours, after which it was resuspended by switching on the aerators, at the same time switching off those in the other half of the basin (see Fig. 3.10.)

This procedure was repeated daily.

The response of the plant, for phosphorus and nitrate, as measured along the centre line of the plant, is plotted in Fig. 3.11.



First day's operation



Second day's operation

- aerator on
- ⊗ aerator off
- ▨ un-aerated volume

Fig. 3.10. Johannesburg Alexandra Plant

Creation of an anaerobic zone in each half of the first aeration basin by sequential turning off of aerators

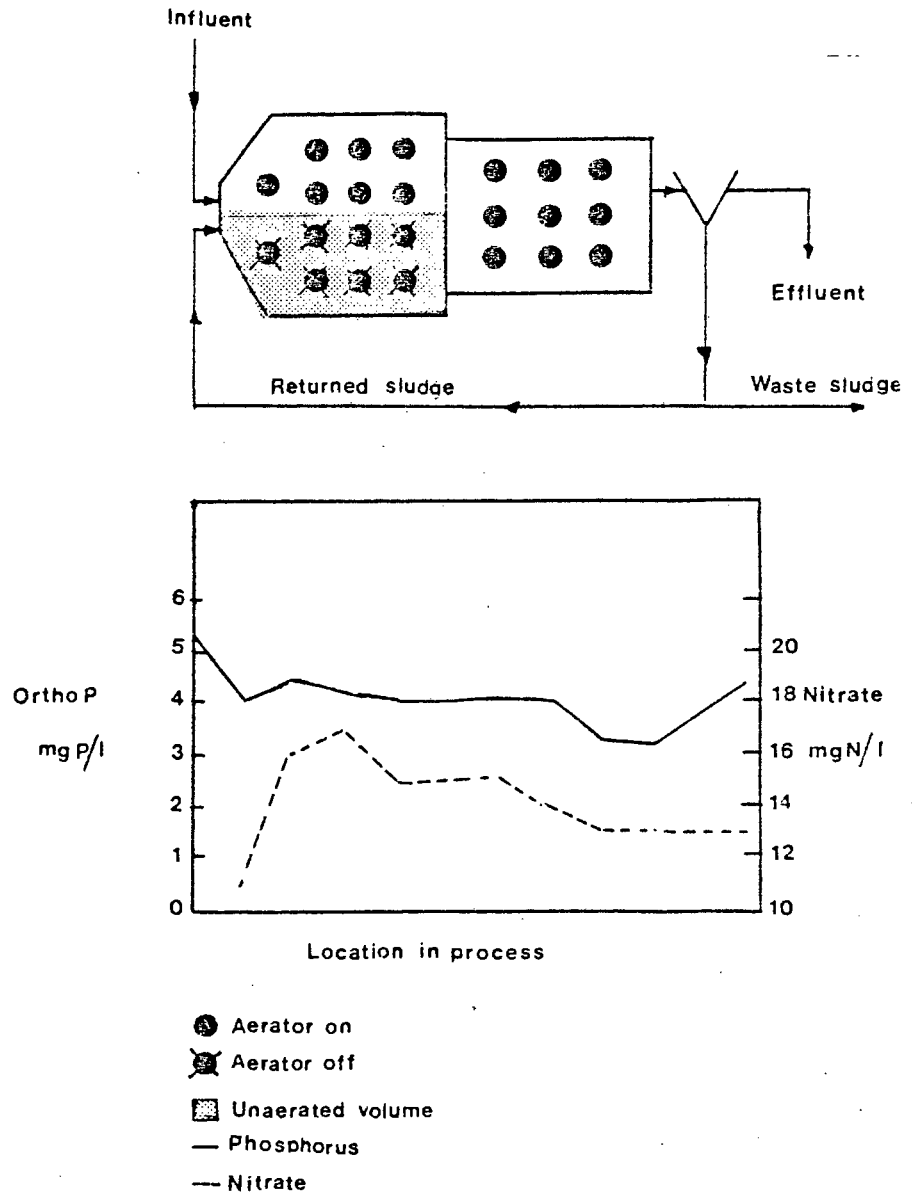


Fig. 3.11. Johannesburg Alexandra Plant

Phosphorus profile across plant with anaerobic zone on side of first basin

Clearly phosphorus removal was negligible and the nitrate in the effluent was considerably higher than when aerators were switched off around the influent point. From observation it would appear that when the aerators were switched off on one side of the plant the intermixing of the aerated and unaerated sections was relatively minor and this possibly was the reason for the low nitrate removal. It is evident that this mode of operation has little to recommend it both for nitrogen and phosphorus removal.

6. GENERAL CONCLUSIONS

The work on the Alexandra plant points to the following conclusions:

- (i) By switching off aerators around the point of entry of the influent and recycle flows denitrification is readily achieved.
- (ii) The mass of nitrate removed appears to increase as the number of aerators that are switched off is increased. The number of aerators that can be switched off can be increased only to that number where the nitrification becomes affected. In the Alexandra plant when 6 aerators were switched off nitrification efficiency commenced to decline. On any plant this limit will need to be determined by trial and error because of the difficulty in quantifying the fractional volume of the plant that becomes anoxic when one or more aerators are switched off.
- (iii) If sufficient aerators are switched off (6 in the Alexandra plant) apparently an anaerobic region can be developed of such intensity that phosphorus release will be observed and some excess phosphorus removal obtained. Again the anaerobic

region can not be defined and appears to fluctuate, resulting in very variable phosphorus uptake. In the Alexandra plant phosphorus release commenced with 6 aerators off, but at the same time the nitrification efficiency commenced to decline; phosphorus removal therefore may be constrained by the requirement that good nitrification must be maintained.

- (iv) The operational modification in which sludge was stored in the settling tank, gave rise to phosphorus release but due to the fact that the settling tank also served as the final separator the phosphorus release to the effluent nullified the excess uptake in the plant. It would appear therefore that this method also has little value.
- (v) Storage of sludge on the base of the aeration tank by switching off sections of the aerators did not appear to work - phosphorus removal was minimal and denitrification efficiency declined.
- (vi) A general conclusion on the work on the Alexandra plant is that in most activated sludge plants it should be possible to achieve fair to excellent removal of nitrate by creating an unaerated region around the area where the influent and recycle flows meet. The degree of nitrogen removal needs to be found by experiment, the limiting size of the region being given either by the point at which nitrification efficiency commences to decline or, the reduced aeration capacity no longer can supply sufficient oxygen for carbonaceous energy degradation and nitrification.

CHAPTER FOUR

THE GOUDKOPPIE PLANT

1. INTRODUCTION

The Goudkoppie Works was proposed when it became apparent that the Klipspruit Works, the oldest in the City, could not be extended any further. The new works was to be constructed adjacent to the biological filter plant at Klipspruit, as shown in Fig. 1.1.

At the same time a new tunnel sewer was to be constructed to replace two older sewers draining the central business area of Johannesburg. Most of the flow to the new works was of domestic origin, although some light industries and the City Gasworks also discharged into this line. Effluent from the Gasworks, although small in volume, was of very high organic strength and high in ammonia concentration.

The site chosen was sufficiently large to permit the construction of works ultimately to treat 650 Ml/d. Design objectives included the production of high quality effluent low in nitrogen content, readily amenable for reuse without further treatment. Treatment was to take place in three identical modules each with the capacity to treat 50 Ml/d. Each module was to be provided with adequate metering facilities to permit large scale experiments to be carried out on the plant in order to determine the optimum mode of operation. The concrete shell of a fourth 50 Ml/d unit also was to be constructed but not to be equipped with any aerators or other auxiliary appliances. The intention here was to modify and complete this unit at some later date depending on the outcome of the experimental program on the fully equipped units.

At the time of construction there was uncertainty regarding the best mode of treating the solids arriving at the works, i.e. by the normal anaerobic digester route or via an aerobic digestion route, the latter, if necessary, could be carried out in the fourth unfinished basin. Pending a final decision on sludge treatment it was decided that all sludge generated in the first 150 Ml/d stage would be pumped to the adjacent Klipspruit Works for treatment in existing digesters there.

It was during the period of construction that the National Institute for Water Research showed, at pilot plant scale, that enhanced biological uptake of phosphorus could be achieved in Bardenpho type plants provided that the activated sludge was exposed to some degree of anaerobiosis during the process. The experiments at the Alexandra plant and subsequently at Olifantsvlei Works (Venter, Halliday & Pitman, 1978) confirmed that phosphorus removal could take place on large plants provided that the active solids are passed through an anaerobic zone placed ahead of the main aeration basin. Based on this evidence, albeit somewhat meagre, a decision was taken to modify the Goudkoppie Works while under construction by providing an anaerobic zone at the influent end of the activated sludge plant. Physical constraints such as the presence of existing pipes limited the size of basin that could be added. This resulted in the construction of a zone having a retention time of one hour based on the ADWF of incoming sewage. In terms of the criteria available at that time the volume of the anaerobic reactor can be considered on the low side.* (Barnard, 1975, proposed a nominal retention time of $1\frac{1}{2}$ to 2 hours.)

* The additional cost of this modification was about R250 000, compared to the total cost of the works of R10,7 million. This amounted to 2,4 per cent.

- By incorporating the anaerobic reactor the Goudkoppie plant became the first large plant in the world designed specifically for the biological removal of both nitrogen and phosphorus. It was only natural that the plant subsequently would be the subject of evaluation to determine the degree to which the plant fulfilled the design expectations.

2. DESIGN FEATURES

The plant, as constructed, provided for primary and secondary screening, degritting, primary sedimentation using six tanks, and flow balancing, these to serve three identical modules in parallel each designed in conformity with the Phoredox system. A diagrammatical sketch of the plant is shown in Fig. 4.1., detailed design data are given in Table 4.1. Some of the design features are discussed in more detail below:

- (i) Primary sedimentation followed by sludge digestion was required on this plant as the City Council was under contract to supply AECI with sludge gas from the adjacent Klipspruit Works for conversion to cyanide for use on the gold mines; this contract was likely to be extended to the Goudkoppie Works.

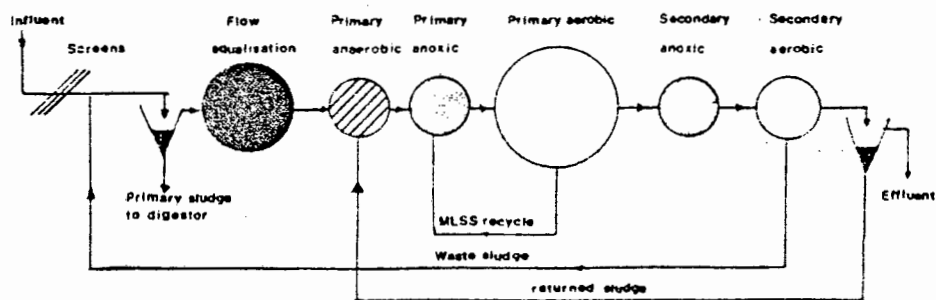


Fig. 4.1. Goudkoppie Works Basin Configuration

Table 4.1.

GOUOKOPPIE PLANT DESIGN DATA

Flow / module	(Ml/d)	50				
Raw sewage composition	BOD (mg/l)	400				
	TKN (mg N/l)	60				
Settled sewage composition	BOD (mg/l)	300				
	TKN (mg N/l)	50				
Sewage temperature	(°C)	16	15	12		
BOD loading	(kg BOD/kg MLSS)	0,16	0,15	0,11		
MLSS	(g/m ³)	3,5	3,7	5,1		
Sludge age	(days)	8,4	9,1	13,9		
Volume	(1) Anaerobic (m ³)	2 080				
	(2) Primary anoxic (m ³)	4 800				
	(3) Primary aerobic (m ³)	14 700				
	(4) Secondary anoxic (m ³)	4 800				
	(5) Secondary aerobic (m ³)	2 700				
	Total (m ³)	29 080				
Basin number		1	2	3	4	5
Nominal retention time	(h)	1	2	7	2	1
Number of aerators		-	-	12	-	2
Power / aerator	(kW/aerator)			110		45
Oxygenation capacity	(kg O ₂ /h per aerator)			205		74
Oxygen transfer efficiency				1,8		1,5
Number of mixers		2	4	-	4	-
Power per mixer	(kW)	11	22,5	-	15	-
Number per module		4				
Average nominal daily flow to secondary sedimentation tank (without recycle)	(m ³ /h)	520				

Table 4.1. continued

Secondary sedimentation

Peak factors based on nominal flow

Dry weather peak	1,0
Wet weather peak	1,6

Solid flux based on nominal flow ($\text{kg/m}^2/\text{h}$)

Winter	2,5
Summer	3,5

Solid inflow based on nominal flow (kg/m^3)

Winter	5,1
Summer	3,5

Surface loadings based on mean flow (m/h)

Winter	0,5
Summer	1,0

Surface area required per tank
based on mean flow (m^2)

Winter	1 040
Summer	832

Diameter of tank (m) 36,2

Volume of tank (m^3) 2 360

Side wall depth (m) 2,4

(ii) The balancing tank was designed to minimise the cyclic variation in flow and load, and was located in line between the primary sedimentation tanks and the activated sludge process. Since the influent to this tank was to be settled sewage it was hypothesised that settlement of solids would be minimal, and by installing dwarf walls a flushing action would be induced at low flow; this in turn should remove any solids which might have settled, hence, no mixers were installed. However, should it be found subsequently that settlement is a problem provision was made for sparges, positioned along the wall closest to the point of entry, to be switched on to resuspend the solids.

The balancing tank scheme was justified by the expected benefits. These were, smaller final settling tanks, smaller hydraulic connections and lower peak oxygen aeration capacity. The estimated saving was considerable, approximately R1 000 000 (Keay and Pitman, 1980)

(iii) The activated sludge process was based on the Phoredox system (Fig. 2.8.) It consisted of five completely mixed reactors in the sequence: anaerobic - primary anoxic - primary aerobic - secondary anoxic - secondary aerobic. Some of the more important features in the process design are:

- * All the zones are physically separated by baffles containing suitable openings to permit flow continuity, the one exception being a solid wall with surface overflow between the primary aerobic and secondary anoxic reactors in order to allow for the installation of a movable weir for oxygen control.

- * The MLSS is recycled from the effluent end of the primary aeration zone back to the influent point of the primary anoxic reactor at recycle ratios (recycled flow/ feed flow) that can be varied from zero to 16.
- * The aerobic reactors are aerated using 12 X 110 kW surface aerators in the primary aeration basin and 2 X 45 kW aerators in the secondary aeration basin.
- * The sludge age is controlled by wasting a fixed volume of mixed liquor each day from the secondary aeration zone i.e. hydraulic control of the sludge age is applied.

(iv) Final clarifiers , of which there are four per module, each with a diameter of 36,2 m and a very shallow side wall depth of only 2,4 m . Sludge removal is by means of a rotating bridge carrying five syphons.

(v) Flow metering

This plant has metering facilities at the following points:

- * Feed flow from the balancing tank to the process is automatically controlled to a selected flowrate by activation of a Rotork valve. The flowrate is monitored by a flume type flow meter. Provided the balancing tank does not empty during the day or overflow, a fixed volume of feed passes into the process each day.

- * Underflow sludge from the final clarifiers to the anaerobic zone is measured by a flume type flow meter and controlled by means of a valve which in turn adjusts the volume of sludge syphoned from the clarifier floor.
- * Waste sludge control from basin 5 is exercised through a hand operated valve and the volume measured by means of a flow meter similar to that for the underflow

The built in meters were checked against a portable standardised meter and found to be accurate to within 10 percent of the indicated value.

(vi) Dissolved oxygen control

Two forms of control are available:

- * Manual control achieved by switching aerators in and out as required.
- * Automatic control can be activated by any one of a number of dissolved oxygen probes in the primary aeration basin. The oxygen input is increased or decreased by raising or lowering a level control weir at the end of this basin, thereby increaseing or decreasing the depth of immersion of the aerators. Experience with this system has shown that it is not effective or reliable; for the series of experiments hereafter it was necessary to use only the manual control method.

3. PLANT PERFORMANCE EVALUATION

3.1. Objectives

The objectives of the experimental programme were:

- (1) To trace the response of a set of selected parameters in each reactor under the conditions of flow and load set on the process
- (2) To compare the observed performance with predictions by the general activated sludge model developed at the University of Cape Town.

3.2. Conditions of test

- (i) Flow to the experimental module was to be constant and limited to 30 Ml/d. Compared with mean design flow - 50 Ml/d the flow to the plant was still well below the design flow.
- (ii) Sludge age was to be maintained in the range 20 - 25 days which, according to Pitman (1974) , was likely to give rise to a rapidly settling sludge. Notwithstanding this provision immediately prior to the commencement of this experiment, the sludge volume index (S V I) was in the range 200 - 300, so that the value of this provision is to be questioned.
- (iii) Physical details of the process layout are given in Fig. 4.2.(a) and Table 4.2.
- (iv) The MLSS recycle (s) ratio was fixed at 6:1 .
- (v) The underflow recycle sludge recycle was fixed at 0,9 :1 .

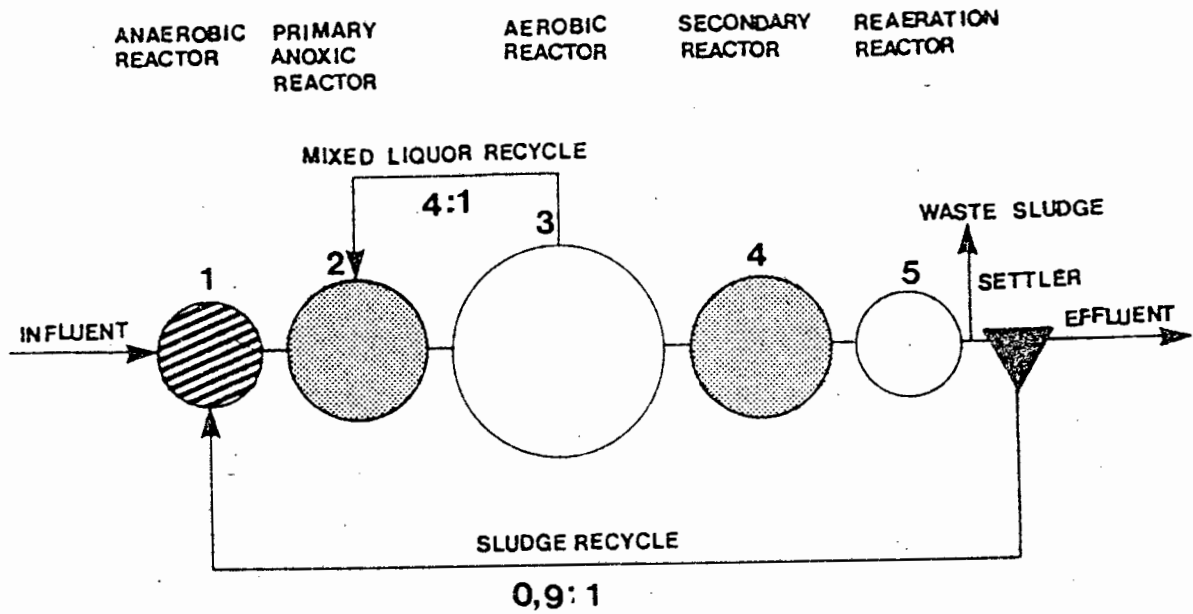


Fig. 4.2.(a) The process layout of the Goudkoppie plant during the experimental investigations

Table 4.2.

Physical details of the Goudkoppie plant during the experimental investigations

	Anaerobic	Primary Anoxic	Primary Aerobic	Secondary Anoxic	Secondary Aerobic
Reactor number	1	2	3	4	5
Volume (m ³)	2 080	4 800	14 700	4 800	2 700
Nominal retention time (d)	0,069	0,160	0,490	0,160	0,090
Mass fraction	0,072	0,165	0,505	0,165	0,093

- (vi) - Dissolved oxygen concentrations in both aerobic reactors were to be maintained in excess of 2 mg/l . Under the proposed conditions of low flow and load this concentration could be achieved with only half the total number of aerators operational, as shown in fig. 4.2.(b)

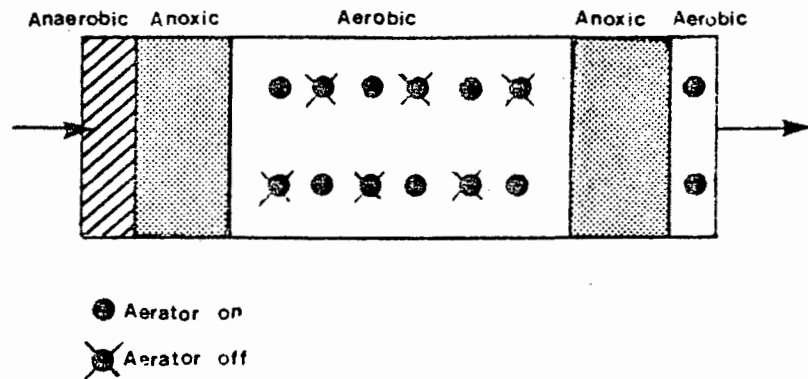


Fig. 4.2.(b) Operational aerators during experimentation
on Goudkoppie Works

- (vii) Sampling: In each reactor "grab" samples were to be taken near the end of the reactor because these would reflect the concentrations passing into the subsequent reactor. The samples were processed by filtering immediately through coarse filter paper into a bottle containing 1ml of 0,85% (mass/mass) mercuric chloride to inhibit further bacterial action. Samples were stored at ambient temperature and collected daily for laboratory analysis. Sampling of the feed and effluent were done by automatic samplers. The sampling pipework was flushed with the liquid being sampled immediately prior to collecting the sample. The influent

samples were to be collected immediately prior to the inlet point. The effluent samples were to be taken at the confluence point of the effluent from all four clarifiers. Individual one litre samples were taken, treated with mercuric chloride and stored in separate plastic bottles until analysed.

- (viii) Oxygen utilisation measurements: Preliminary tests on oxygen utilisation taken along the longitudinal centre line of the reactors, at each reactor, indicated an oxygen utilisation rate profile high at the influent end of the reactor and falling to lower values at the effluent end, indicating that completely mixed conditions were not present in the reactors. The profile varied as the flow and load on the reactor varied. Eventually it was decided that the most representative assessment of the mean behaviour of the reactors would be achieved by testing at three points along the centre line of the reactors; between the first and second, third and fourth, and fifth and sixth, and averaging the results. The detailed procedure followed during a utilisation rate test is set out in Appendix A4 .

4. PERFORMANCE DATA

Sampling of the feed, effluent and contents of each zone was carried out at two - hourly intervals for a period of one week and samples analysed for COD, TKN, ammonia, nitrate and phosphorus. Oxygen utilisation rate tests were also taken at the same time. Unfortunately the results of three of the seven days had to be discarded due to the malfunction of one of the samplers.

The results from the remaining four days, at each time point, were averaged and are listed in Appendix A5 . Plots of these averages are shown in Figs. 4.3. to 4.10. as follows:

- Fig. 4.3. Unfiltered COD concentration in the feed and filtered effluent and reactor contents.
- Fig. 4.4. Unfiltered TKN concentration in the feed and filtered effluent
- Fig. 4.5. Total phosphorus concentration in the unfiltered feed and filtered effluent
- Fig. 4.6. Oxygen utilisation rate in the main aeration basin
- Fig. 4.7. Soluble COD concentration in filtered samples in each basin
- Fig. 4.8. Soluble ammonia concentration in filtered samples in each basin
- Fig. 4.9. Nitrate concentration in filtered samples in each basin
- Fig. 4.10. Soluble total phosphorus concentration in filtered samples in each basin

The consolidated daily average results over 24 hours of the above data are given in Table 4.3.

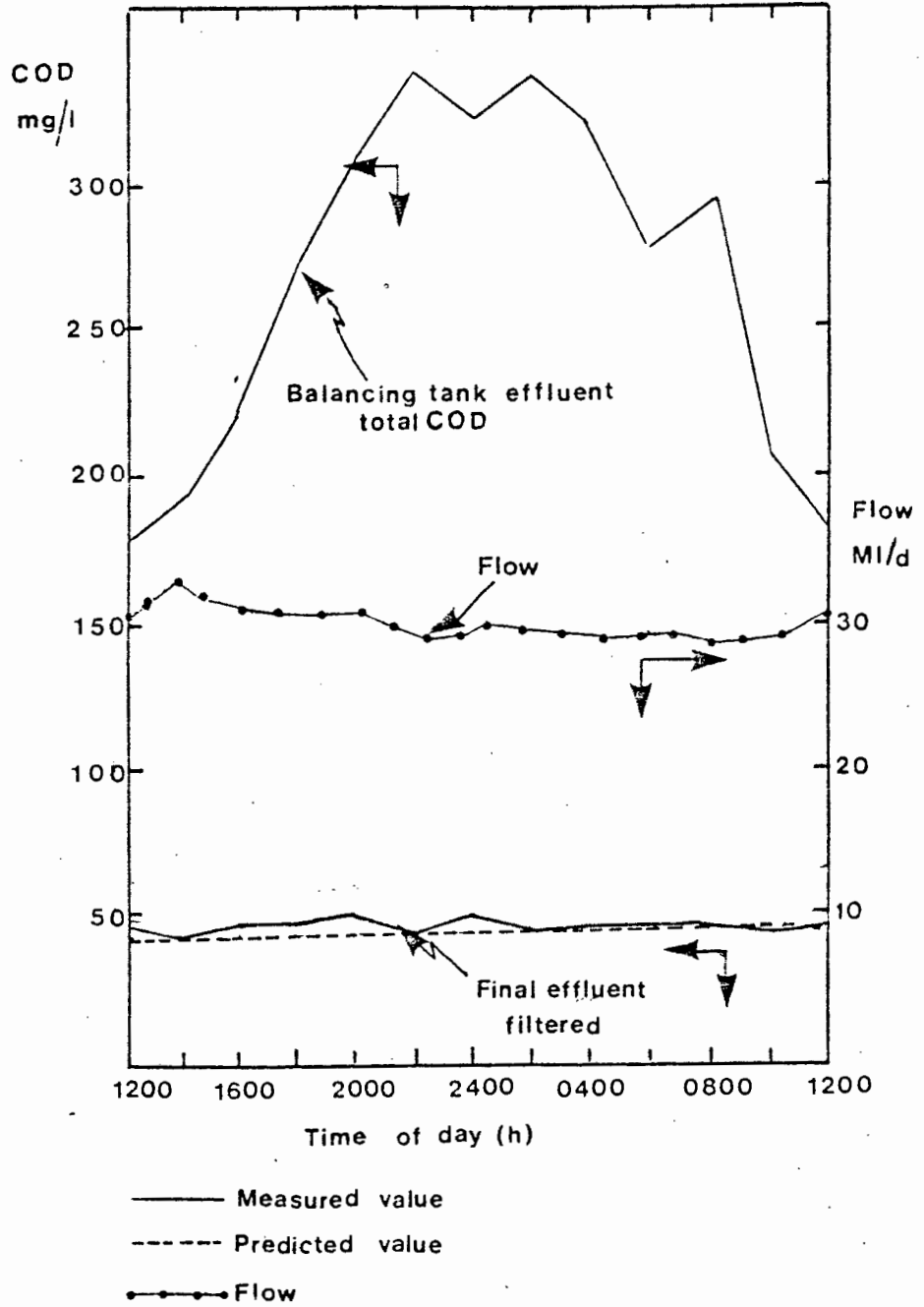


Fig. 4.3. Goudkoppie Plant

Total COD concentrations in the feed and filtered COD in the effluent together with predicted values.

In addition the flow to the process during the test period is shown

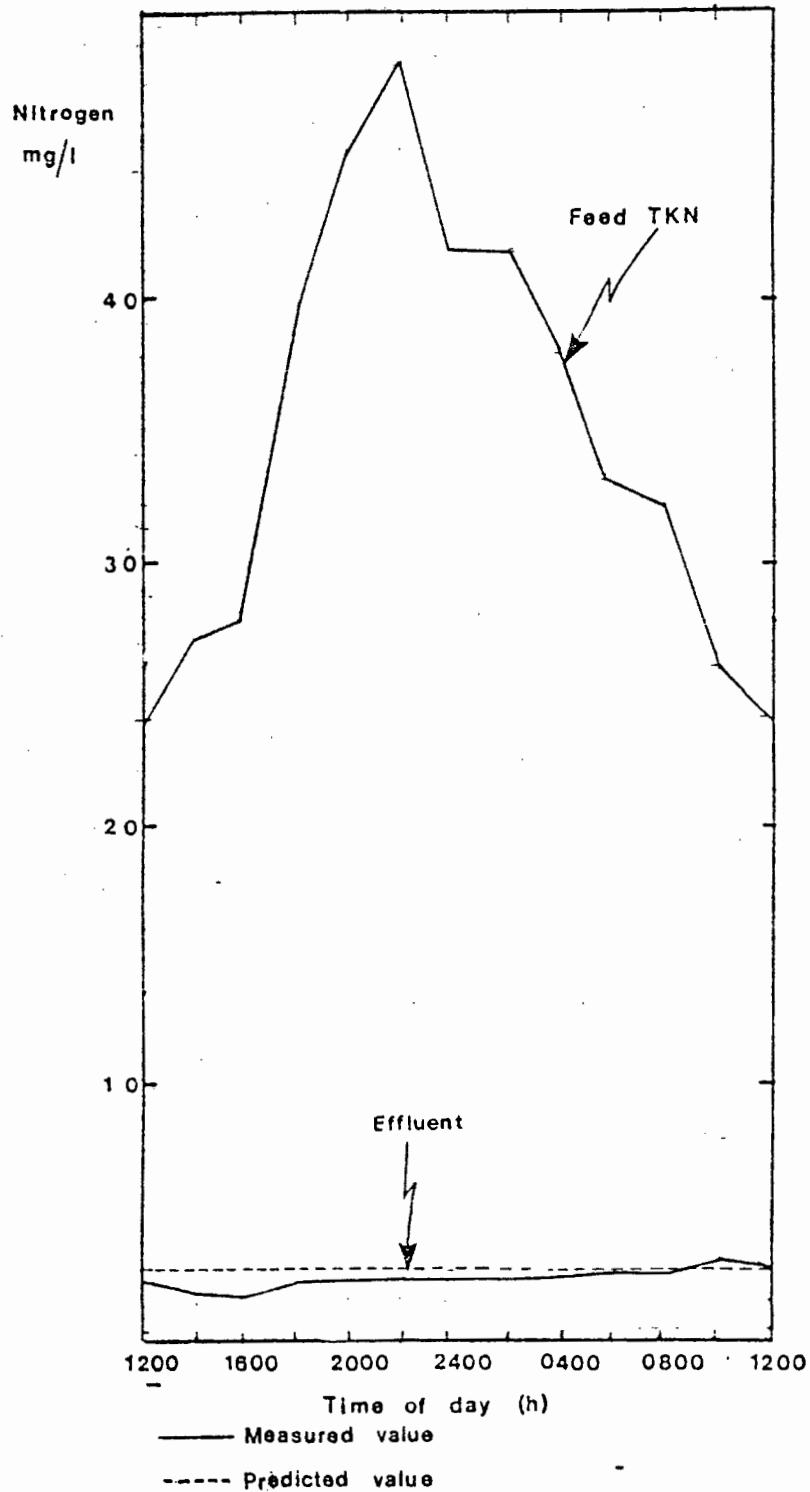


Fig. 4.4. Goudkoppie Plant

TKN concentration in the feed and filtered TKN in the effluent together with predicted values for the effluent

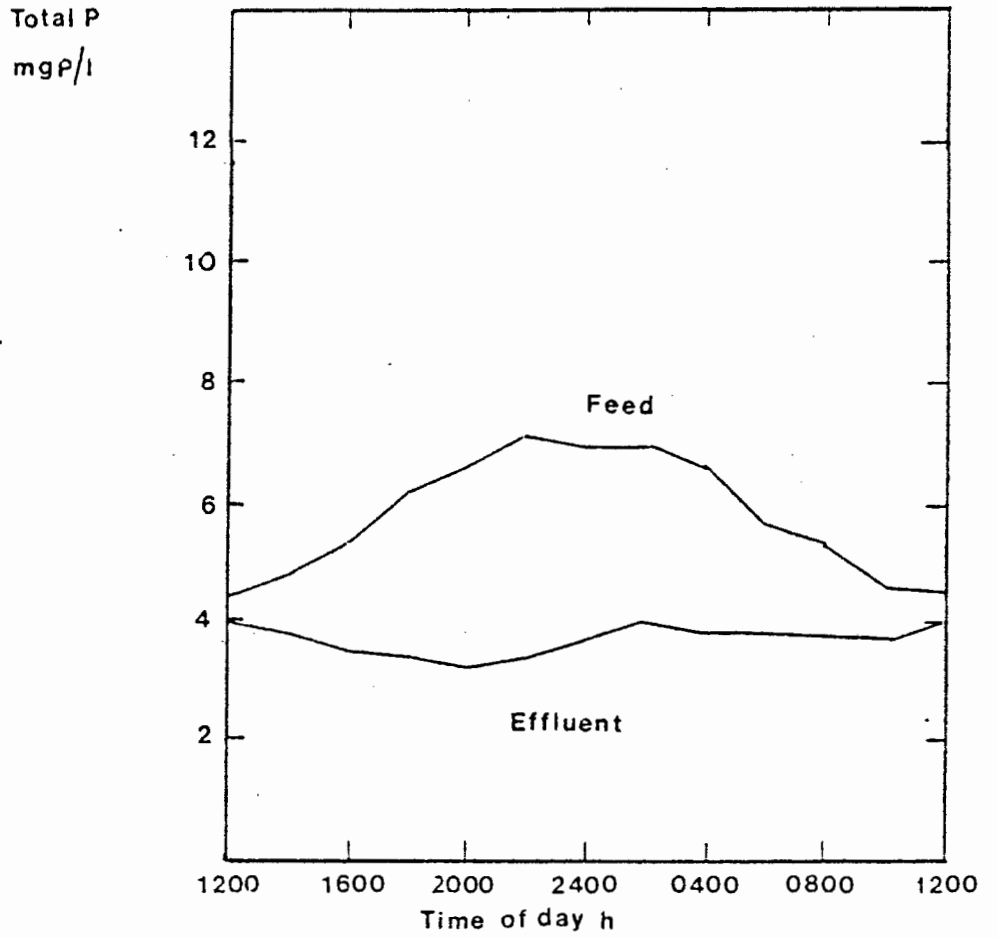


Fig. 4.5. Goudkoppie Plant
Total phosphorus concentrations in the feed
and filtered effluent

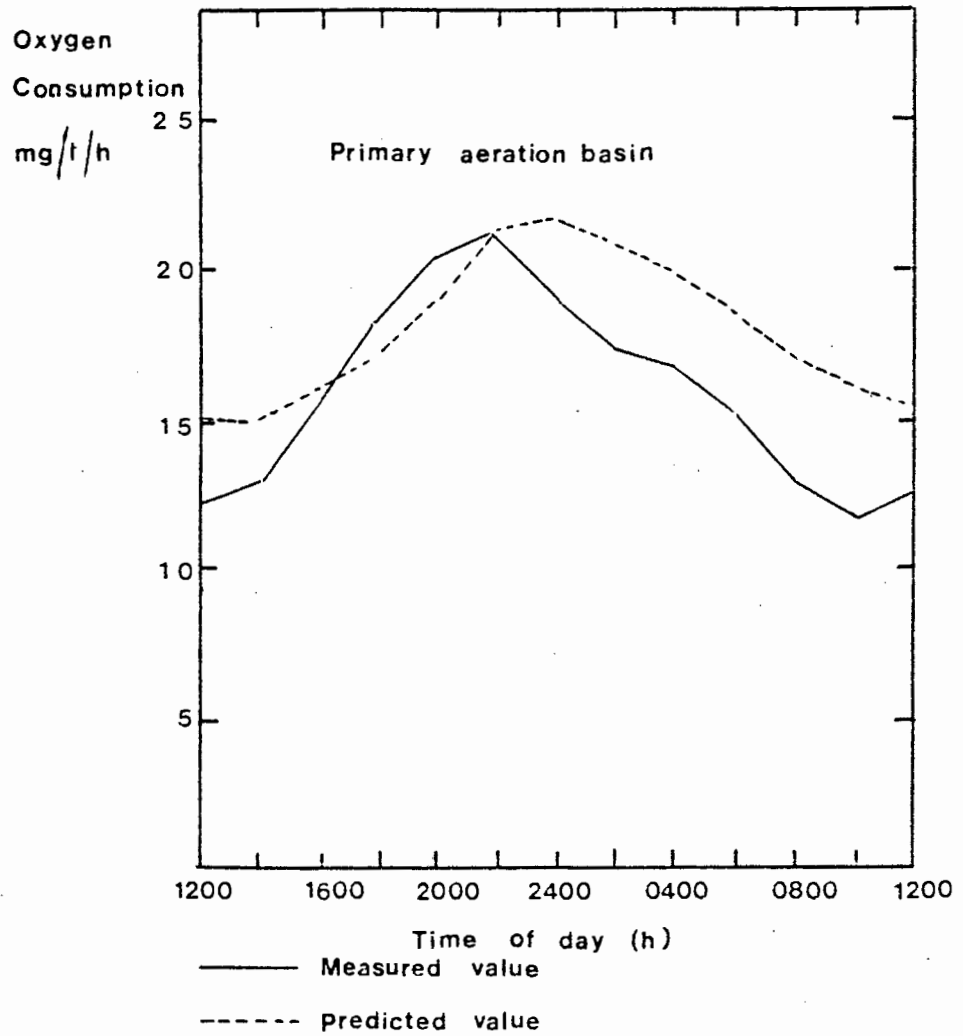
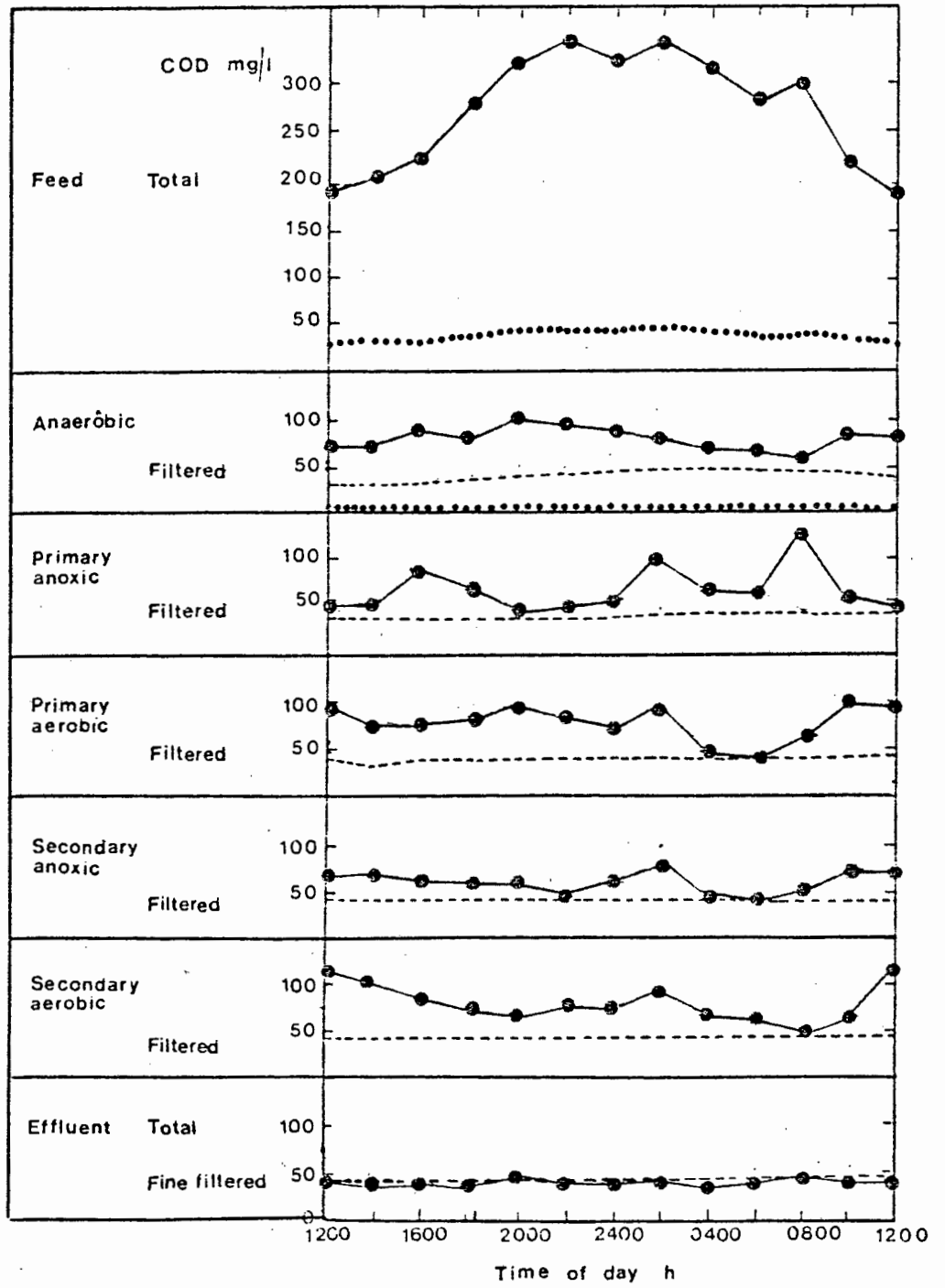


Fig. 4.6. Goudkoppie Plant
 Predicted and measured oxygen consumption rates
 in the primary aeration basin



●—●—●— measured values
 - - - - - predicted values
 ●—●—●— predicted values of readily biodegradable COD

Fig. 4.7. Goudkoppie Plant
 Measured filtered COD and predicted soluble COD concentrations in each reactor of the system

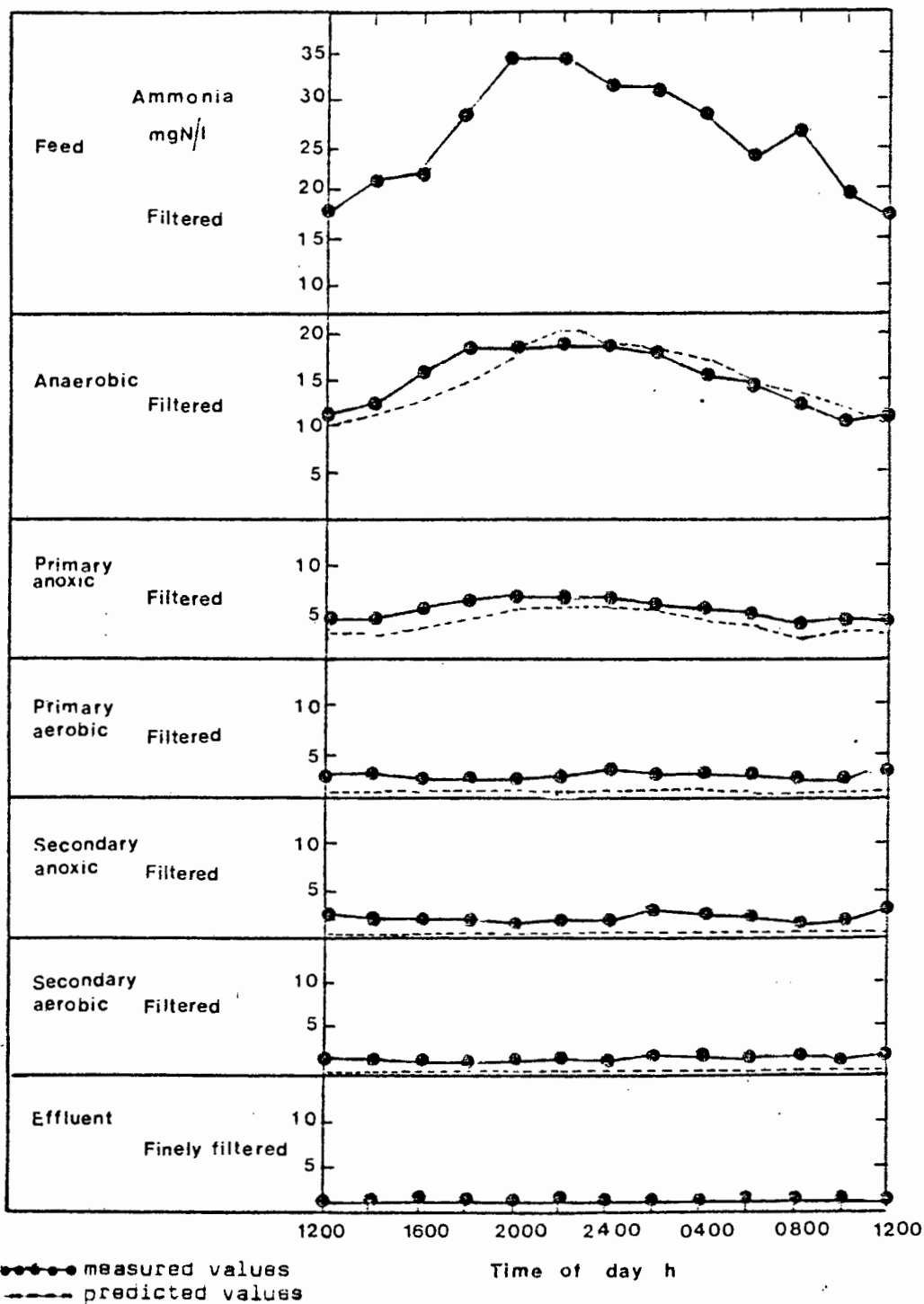


Fig. 4.8. Goudkoppie Plant
 Measured filtered and predicted ammonia concentrations
 in each reactor of the system

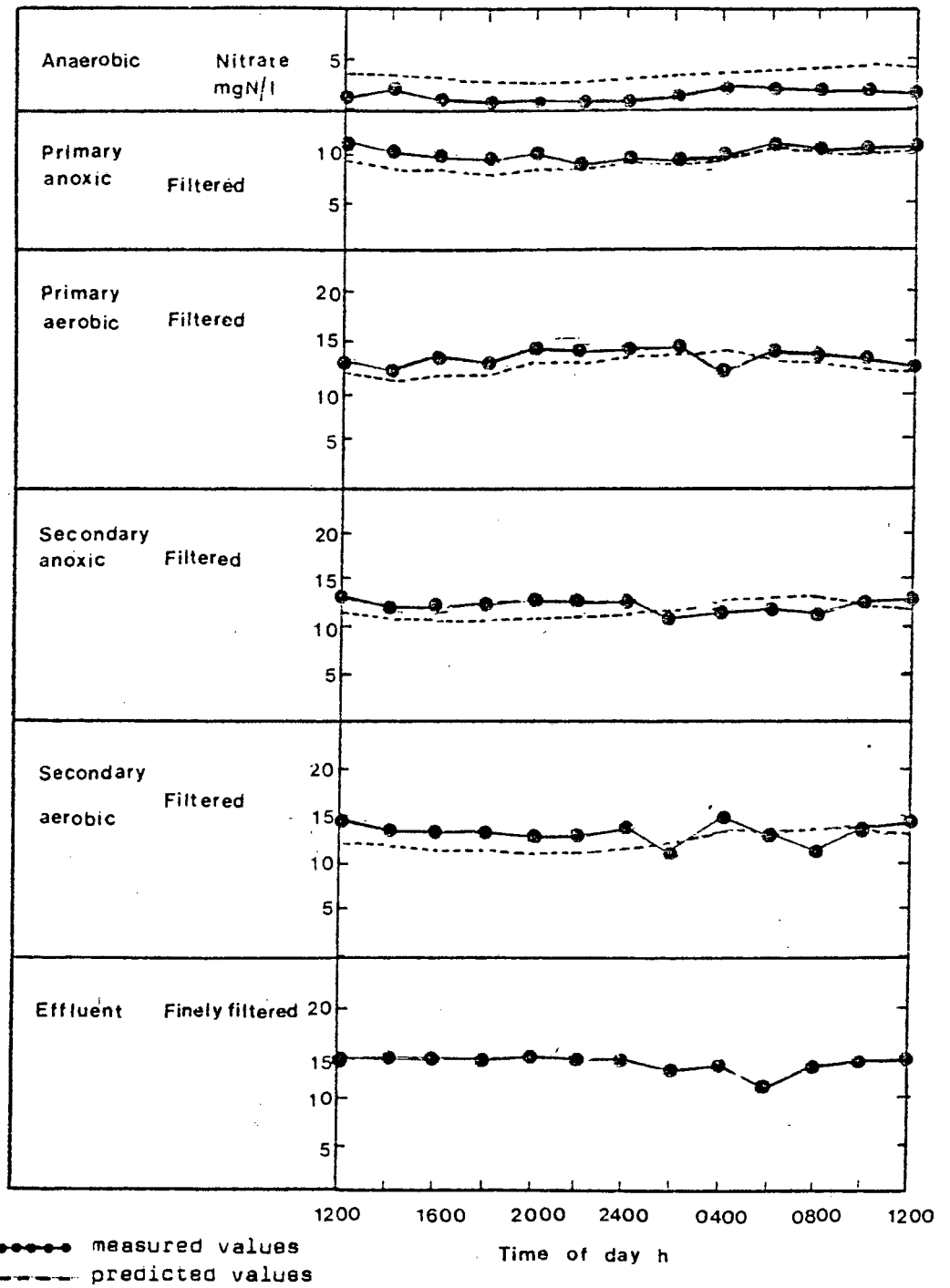


Fig. 4.9. Goudkoppie Plant
 Measured filtered and predicted nitrate concentrations
 in each reactor of the system

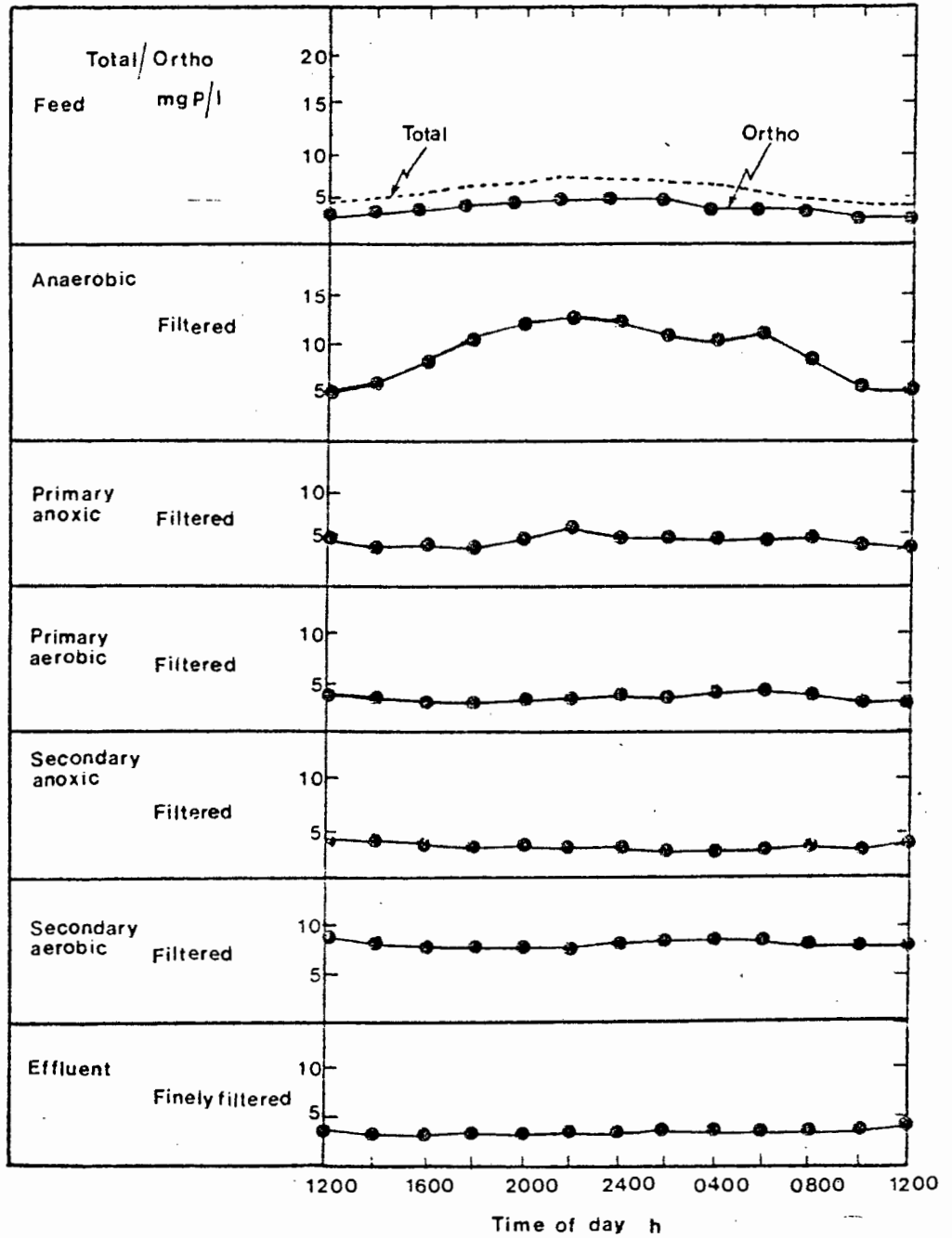
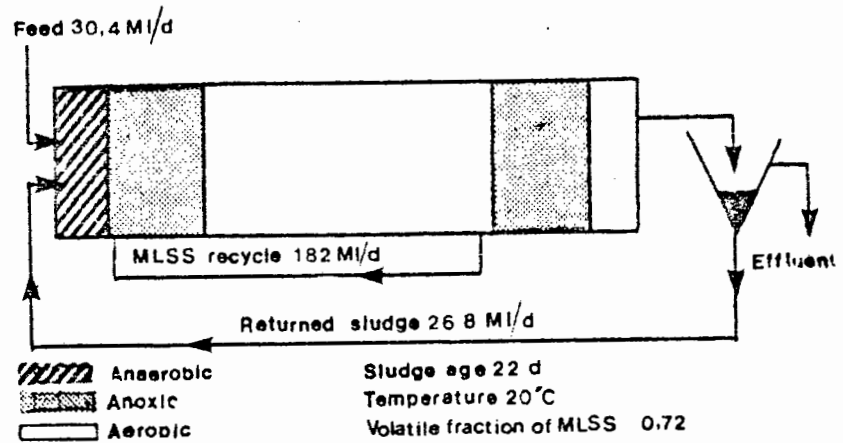


Fig. 4.10. Goudkoppie Plant

Unfiltered total phosphorus concentration in influent and filtered orthophosphate concentrations in the reactors and effluent

Table 4.3.

The average performance of the Goudkoppie Plant during the experimental period



	Feed	Anaerobic	Primary Anoxic	Primary Aerobic	Secondary Anoxic	Secondary Aerobic	Effluent *
Total COD	270						63
Soluble COD	170	77	67	73	55	72	43
S _{bs}	29						
1xN as N	35						2,0
Ammonia as N	26	15	5,1	1,9	1,6	1,5	1,5
Nitrate as N		1,1	9,8	13,4	12,5	13,3	13,6
Total P	5,8						3,7
Soluble P	3,6	10	5,0	3,9	3,6	3,2	3,6**
Suspended solids				1 940			
Oxygen consumption mg/l/h				16,2			

Unless otherwise stated all results expressed as mg/l

- * Effluent samples were filtered through Whatman 542 filter paper while the reactor samples were filtered through coarse filter paper
- ** Reactor samples were filtered and test by means of the orthophosphate test

S_{bs} = readily biodegradable COD concentration mg/l

4.1. Data validity

Before testing the plant data against model predictions it is necessary to check if the data is acceptable. A powerful check is to verify if mass balances over the plant, of the COD and nitrogen, are achievable within reasonable limits. Ekama and Marais (1978) concluded from a number of investigations that balances are acceptable if these fall within the range 96 to 104 percent. Their investigations, however, were on laboratory and pilot scale facilities where close control was exercised over the input to the plant and its operation.

Clearly control over a large scale plant can never be of the level attainable on a laboratory or pilot scale plant. At the Goudkoppie plant every endeavour was made to achieve as high a level of control as possible, by having an equalization tank and setting a consistent flow rate to the test modules. Even so the masses of COD and TKN did fluctuate from day to day. These fluctuations can be smoothed out by testing the plant over a number of days and averaging out the results, to obtain a mean behaviour. It was for this reason that the experimental period extended over 7 days (of which, we have seen, four days were acceptable)

Utilising the averaged daily values over the days of acceptable readings, the mass balances achieved on the COD and nitrogen were 99,8 and 107 percent recoveries respectively.* These balances very likely are as good as can be expected on such a large plant and indicate inter alia that the sampling procedure was consistent and the testing accuracy acceptable.

*For details of the procedures followed to make these balances, see Appendix A6 .

5. MODEL PREDICTIONS

To test the validity of the bisubstrate death - regeneration model, as developed in its final form by van Haandel, Ekama and Marais (1981), it is necessary to establish values for certain parameters utilised in the model. These are, firstly, the sewage characteristics which are specific to the particular waste flow, and secondly, the kinetic parameters; the latter in turn being subdivided into two groups:

- (i) those that change with different waste water sources
- (ii) those that are constant from one waste source to another

5.1. Sewage characteristics

Parameters defining the physical and chemical characteristics of the particular waste flow are the following:

- (1) Total influent COD S_{ti} (mg O/l) and total influent TKN (mg N/l) together with the total flow.
- (2) Temperatures in the reactors
- (3) Magnitudes of various influent COD fractions:

- (a) Soluble unbiodegradable COD f_{us} i.e.

$$S_{us} = f_{us} \cdot S_{ti}$$

- (b) Particulate unbiodegradable COD, f_{up} i.e.

$$S_{upi} = f_{upi} \cdot S_{ti}$$

- (c) Biodegradable COD, f_{bi} defined by

$$f_{bi} = (1 - f_{usi} - f_{upi})$$

$$\text{i.e. } S_{bi} = f_{bi} \cdot S_{ti}$$

- (d) The biodegradable COD S_{bi} in turn is subdivided into two fractions, easily biodegradable COD, f_{bsi} defined by

$$\begin{aligned}
 S_{bsi} &= f_{bsi} \cdot S_{bi} \\
 &= f_{bsi} \cdot f_{bi} \cdot S_{ti} \\
 &= f'_{bsi} \cdot S_{ti}
 \end{aligned}$$

and particulate biodegradable COD f_{fpi} i.e.

$$\begin{aligned}
 S_{bpi} &= f_{bpi} \cdot S_{bi} \\
 &= (1 - f_{bsi}) S_{bi} \\
 &= f_{bpi} \cdot f_{bi} \cdot S_{ti} \\
 &= f'_{bpi} \cdot S_{ti}
 \end{aligned}$$

(4) Magnitudes of various TKN influent fractions

(a) Soluble unbiodegradable TKN f_{nui}

$$N_{ui} = f_{nui} \cdot N_{ti}$$

(b) Free and saline ammonia f_{nai}

$$N_{ai} = f_{nai} \cdot N_{ti}$$

(c) Organic nitrogen

$$N_{orgi} = (1 - f_{nui} - f_{nai}) \cdot N_{ti}$$

With regard to the various fractions above, these are not all independent for example, once f_{nu} and f_{na} are known f_{orgi} can be determined from $(1 - f_{nai} - f_{nui})$ and so on.

One parameter that can be of crucial importance, the Total Alkalinity, was not measured. Total alkalinity is the principal parameter forming the precipitate in an activated sludge plant. If the alkalinity in the reactor drops below 35 mg/l CaCO_3 then the pH becomes unstable and may either drop permanently to values below pH = 7 or fluctuate between 6 and 7,5 depending on nitrification.

Nitrification abstracts alkalinity and if it is reduced to less than 35 mg/l CaCO_3 the pH may drop below 7 which in turn inhibits nitrification causing the pH to rise again, hence the fluctuation. At Goudkoppie the alkalinity was relatively high (100 mg/l CaCO_3) in the reactors, consequently, the pH remained stable between 7 and 8 (mean 7,4) and nitrification was not affected. Hence alkalinity measurements were not essential.

We shall now consider briefly the measurement of the key parameters on the Goudkoppie plant.

5.1.1. Influent COD and TKN concentration and reactor temperature

The influent COD and TKN concentrations were measured at 2 hourly intervals over the entire period of the investigation. At the same time the temperature was measured in the main aeration basin because in this basin nitrification takes place and the nitrification rate is most sensitive to temperature.

5.1.2. Soluble unbiodegradable COD fraction (f_{Usi})

For a plant operating at long sludge ages, the filtered effluent COD concentration (provided a fine filter is used) is virtually identical to the soluble unbiodegradable COD concentration. From the average data over the day (Table 4.3.) the average soluble COD in the effluent was 42 mg/l ; the total influent COD , 272 mg/l . Therefore, $f_{\text{Usi}} = 42/272 = 15$ percent. This value normally is approximately 10 percent of the raw sewage COD value but at Goudkoppie an appreciable fraction of the particulate COD is removed in the primary settler,

and a further fraction is removed in the balancing tank ;
 in contrast, the soluble unbiodegradable fraction, f_{us} ,
 is not affected. Consequently the unusually high f_{us}
 was not unexpected.

5.1.3. Unbiodegradable particulate COD fraction (f_{upi})

To determine f_{upi} reliably requires an extensive series of tests on the same waste over a number of sludge ages. The value of f_{upi} is then determined by fitting the equation for MLVSS / unit COD against the observed data. In doing this fit the specific yield constant (Y) and endogenous respiration rate constant (b) are assumed to remain at their standard values for the temperature in the plant (these appear to maintain their values for widely divergent waste characteristics). The best fit then is obtained by inserting a series of values for f_{upi} in the equations describing the process, the final equation being that describing the MLVSS / unit COD. The value of f_{upi} giving the best overall fit over all the sludge ages forms an estimate of f_{upi} . This value is then checked by utilising it to predict the oxygen utilisation rate and comparing it with the observed rates at different sludge ages. This procedure has been found to give very good estimates of f_{upi} when tested on influents (glucose) where it is known that f_{upi} equals zero.

In the investigation of Goudkoppie it was not possible to do the series of tests necessary to determine f_{upi} reliably. The only procedure possible was to accept the concentration of MLVSS in the plant, and knowing the mass COD feed/day ,

the value of f_{upi} found by trial and error to give the best prediction of the MLVSS in the plant. This value was verified to a degree by the close correspondence between the simulated mean OUR against the mean observed OUR.

5.1.4. Easily biodegradable COD fraction (f_{bs})

The value of f_{bs} can differ depending on the source of the sewage. At Goudkoppie a further complication is that the biological action of the sludge layer in the balancing tank may either enhance f_{bs} (by solubilisation of particulate COD) or deplete it by fermentation. The only reliable approach was to conduct a test to determine the value of f_{bs} directly. At the time this investigation was undertaken the test for the estimation of f_{bs} had not yet been developed. Subsequently Dold, Ekama and Marais (1980) provided an experimental procedure by means of which f_{bs} could be determined. (The reader is referred to van Haandel and Marais (1981) for a detailed account on how to conduct such a test.) A summary of this test is given in Appendix A7. When the test became available, a series of tests to determine f_{bsi} was undertaken on the balancing tank effluent. The readily biodegradable COD was found to be approximately 29 mg COD/l. Taking into account the f_{usi} and f_{upi} fractions the biodegradable COD equals $(1 - f_{usi} - 1,48 f_{upi}) S_{ti} = 0,67 S_{ti} = 0,67 \cdot 272 = 182$ mg COD/l. Hence $S_{bs} = 29/182 = 0,16$. Since the operation and influent conditions had remained very stable for a long period the value for $S_{bsi} = 0,16$ was accepted to have applied also over the period the plant was evaluated.

5.1.5. Fraction of TKN in ammonia form (f_{na})

This fraction may differ between waste flows from different sources. Long sewer lines and any other conditions allowing degradation of the proteinaceous material will increase the fraction. For a particular Works it is determined by measuring the free and saline ammonia and TKN of the unfiltered influent. From such tests f_{na} was found to be approximately 0,72 for the effluent from the equalisation tank.

5.2. Constants specific to a waste flow

The magnitude of the maximum specific growth rate of the nitrifiers, μ_{nm} , although a kinetic "constant" can differ greatly in value between sewage sources. Van Haandel, Dold and Marais (1982) have measured values for μ_{nm} at 20 C in the range 0,33 to 0,65 d^{-1} ; this range is ascribed in part to the magnitude of the industrial component in the waste flow, usually the larger the component the lower the value of the constant. Accurate determination requires a specialised series of test (Sehayek 1981). However, by curve fitting to the observed TKN and nitrate (and to the alkalinity response) quite good estimates of μ_{nm} can be obtained. Fortunately when the sludge age is very long (i.e. 22 days), the conversion to nitrate is near completion at 20 C even for $\mu_{nm} = 0,33$, and the nitrate concentration becomes relatively insensitive to μ_{nm} . Under such conditions any reasonable value can be utilised in the simulation model. This was checked in the simulations on Goudkoppie, for μ_{nm} values ranging from 0,33 to 0,5. All the simulations indicated very little difference in the effluent ammonia behaviour.

Eventually a value of $\mu_{nm20} = 0,5$ was selected, primarily because it was felt that the industrial component in the waste flow to Goudkoppie would not have affected the μ_{nm} adversely.

The values of the influent characteristics specified for the waste water at Goudkoppie at the time of the investigation are listed in Table 4.4.

Table 4.4.

Influent characteristics specific to the balancing tank effluent at the time of investigation at Goudkoppie Works

Name	Symbol	Units	Value
Fraction of soluble unbiodegradable COD	f_{us}	mg COD/mg COD	0,15
Fraction of particulate unbiodegradable COD	f_{up}	mg VSS/mg COD	0,12
Fraction of easily biodegradable COD	f_{bs}	mg COD/mg bio-degradable COD	0,16
Easily biodegradable COD concentration	S_{bs}	mg COD/l	29
Fraction of TKN of feed in ammonia form	f_{na}	mg N/mg N	0,72
Specific growth rate constant for nitrifiers	μ_{nm}	d^{-1}	0,5

5.3. Constants invariant for different waste flows

The constants in this group are listed in Table 4.5. Experience in simulating response of processes indicates that these constants do not vary in any significant way and have been used unaltered by Marais and his groups for all their simulations.

Table 4.5.

Kinetic constants invariant for different waste flows

Name	Symbol	Units	Value
Substrate adsorption rate coefficient	K_{a20}	l/mg VSS/d	0,135
Maximum specific growth rate coefficient for heterotrophic organisms utilising stored particulate substrate	K_{mp20}	mg COD/mg VSS/d	3,00
Maximum specific growth rate coefficient for heterotrophic organisms utilising soluble substrate	K_{ms20}	mg COD/mg VSS/d	20,00
Saturation coefficient for heterotrophic cell synthesis from stored particulate substrate	K_{sp20}	mg COD/l	0,04
Saturation coefficient for heterotrophic cell synthesis from soluble substrate	K_{ss20}	mg COD/l	5,0
Specific endogenous respiration rate for heterotrophic organisms	b_{h20}	/d	0,62
Yield coefficient for heterotrophic organisms	Y_{h20}	mg VSS/mg COD	0,45
Unbiodegradable fraction of the active organisms	f	mg VSS/mg VSS	0,08
Nitrogen content of the sludge	f_n	mg N/mg VSS	0,10
COD equivalent of volatile solids	P	mg COD/mg VSS	1,48
Fraction of nitrogen released through endogenous respiration in the organic form	f_{oe}	-	1,00
Fraction of nitrogen required for cell synthesis in the organic nitrogen form	f_{os}	mg N/mg VSS	0,00
Maximum fraction of stored solids with respect to active mass	f_{ma}	mg VSS/mg VSS	1,00
Organic nitrogen to ammonia conversion rate coefficient	K_{r21}	/mg VSS/d	0,015
General parameter for pH dependency coefficients	adsorption on heterotrophic organisms	h_a	1,200
	synthesis by heterotrophic organisms	h_s	1,100
	endogenous respiration by heterotrophic organisms	h_e	1,029
Saturation coefficient for <u>Nitrosomonas</u>	K_{n20}	mg N/l	1,00
Specific endogenous rate for nitrifiers	b_{n20}	/d	0,04
Yield coefficient for nitrifiers	Y_{n20}	mg VASS/mg N	0,10
General parameter for pH dependency coefficients	synthesis by <u>Nitrosomonas</u>	h_s	1,123
	endogenous respiration by <u>Nitrosomonas</u>	h_e	1,029

5.4. Simulated results

The simulated results of the Goudkoppie plant were obtained using the constants listed in Tables 4.4. and 4.5. and the input data , flow COD and TKN as listed in Table 4.6.

Table 4.6.

Goudkoppie Works

Input flow and COD and TKN concentrations used in the model predictions

Time	Flow m ³ /d	COD mg/l	TKN mg N/l	Time	Flow m ³ /d	COD mg/l	TKN mg N/l
1200	31 500	180	24,0	0100	30 200	330	41,8
1300	32 100	188	26,0	0200	30 200	337	41,5
1400	32 700	195	27,3	0300	29 800	327	39,8
1500	32 200	209	27,8	0400	29 800	320	38,0
1600	31 800	222	28,3	0500	29 100	305	35,7
1700	31 500	244	32,0	0600	29 100	278	32,5
1800	31 600	273	39,3	0700	29 200	284	32,2
1900	31 600	290	43,6	0800	29 300	295	32,0
2000	31 700	310	46,1	0900	29 400	250	30,0
2100	30 800	320	47,7	1000	29 700	207	26,0
2200	29 500	342	47,8	1100	30 200	193	25,0
2300	29 800	330	46,0	1200	31 500	180	24,0
2400	30 300	320	42,0				

The simulated results are shown plotted in Figs. 4.3. and 4.4. and 4.6. to 4.9. and listed in tabular form in Tables 4.7. to 4.9.

Table 4.7.

Goudkoppie Plant

Comparison of predicted and experimental COD concentrations in each reactor

Time of Day	Feed		POSITION WITHIN PLANT						Effluent	
			Anaerobic	Primary Anoxic	Primary Aeration	Secondary Anoxic	Secondary Aeration			
	Total Soluble	Soluble	Soluble	Soluble	Soluble	Soluble	Soluble	Total Soluble		
h			E P	E P	E P	E P	E P			
1200	180	110	73 36	40 41	98 42	63 42	110 42	69 46		
1400	190	110	72 35	56 39	76 40	63 41	100 41	63 41		
1600	220	130	84 37	83 39	72 39	55 40	89 40	61 45		
1800	270	150	74 41	69 39	76 39	54 39	66 39	58 43		
2000	310	170	100 46	45 39	88 39	55 39	57 39	60 50		
2200	340	190	94 49	49 40	79 40	45 39	73 39	68 39		
2400	320	210	90 49	52 41	67 41	56 40	69 40	63 48		
0200	340	210	76 49	91 42	89 41	79 41	86 41	56 39		
0400	320	210	67 49	61 43	43 42	42 42	56 41	57 41		
0600	280	210	67 47	62 43	37 43	37 43	51 42	63 43		
0800	300	170	58 46	110 43	56 43	47 43	43 43	68 44		
1000	210	180	74 41	65 42	95 43	58 43	61 43	60 41		
Ave.	270	170	77 44	67 41	73 41	55 41	72 41	63 43		

- Notes:
1. All results expressed in mg O/l
 2. E denotes experimental value obtained
 3. P denotes predicted values
 4. All reactor samples were filtered through coarse filter paper
 5. Effluent samples were filtered through Whatman 542 filter paper

Table 4.8.

Goudkoppie Plant

Comparison of predicted and experimental ammonia concentrations in each reactor. In addition the TKN of the feed is given

Time of Day	Feed		POSITION WITHIN PLANT					Effluent	
			Anaerobic	Primary Anoxic	Primary Aeration	Secondary Anoxic	Secondary Aeration		
	TKN	NH ₃	Soluble	Soluble	Soluble	Soluble	Soluble	TKN	NH ₃
			E P	E P	E P	E P	E P		
1200	24	18	11 10	4,3 3,0	2,1 0,3	2,0 0,5	1,7 0,1	1,9	1,7
1400	27	21	12 11	4,2 3,2	2,0 0,3	1,7 0,6	1,5 0,1	1,7	1,5
1600	28	22	15 12	5,1 3,3	1,9 0,4	1,7 0,6	1,4 0,1	1,6	1,4
1800	39	29	18 14	5,9 3,6	1,4 0,4	1,5 0,6	1,2 0,1	1,9	1,2
2000	46	34	18 18	6,4 4,7	1,5 0,6	1,1 0,8	1,4 0,1	1,8	1,4
2200	48	34	18 19	6,3 5,2	1,6 0,7	1,4 0,9	1,4 0,1	1,9	1,4
2400	42	37	18 18	6,4 5,1	2,8 0,7	1,5 0,9	1,4 0,1	1,8	1,4
0200	42	31	17 17	5,5 4,7	2,1 0,6	1,3 0,9	1,6 0,1	2,0	1,6
0400	38	28	15 16	3,0 4,4	2,4 0,6	1,9 0,8	1,7 0,1	1,9	1,7
0600	33	24	14 14	4,9 4,1	2,0 0,5	1,5 0,7	1,5 0,1	2,4	1,5
0800	32	26	12 13	3,7 4,0	1,8 0,4	1,3 0,6	1,7 0,1	2,1	1,7
1000	26	20	10 11	4,0 3,3	1,8 0,4	1,4 0,6	1,6 0,1	2,8	1,6
Ave.	35	26	15 14	5,0 4,0	2,0 0,5	1,5 0,7	1,5 0,1	2,0	1,5

- Notes:
1. All results expressed as mg N/l
 2. E denotes experimental value
 3. P denotes predicted value
 4. Reactor samples were filtered through coarse filter paper
 5. Effluent samples were filtered through Whatman 542 filter paper

Table 4.9.

Goudkoppie Plant

Comparison of predicted and experimental nitrate concentrations
in each reactor

Time of Day h	Feed	POSITION WITHIN PLANT					Effluent
		Anaerobic	Primary Anoxic	Primary Aeration	Secondary Anoxic	Secondary Aeration	
		Soluble	Soluble	Soluble	Soluble	Soluble	
		E P	E P	E P	E P	E P	
1200		1,3 3,2	10,0 9,2	12,5 12,0	13,0 11,6	13,8 12,6	14,0
1400		1,5 3,0	9,8 8,8	12,0 14,6	12,0 11,3	12,8 12,1	14,0
1600		0,7 2,7	9,6 8,5	13,0 11,3	11,4 10,9	12,8 11,8	14,0
1800		0,6 2,5	9,5 8,3	13,3 11,2	12,3 10,8	13,0 11,5	13,6
2000		0,5 4,7	9,7 8,2	14,0 11,4	12,8 10,9	13,3 11,5	14,0
2200		0,7 4,1	9,2 8,6	14,3 12,1	12,8 11,5	13,3 11,9	14,0
2400		0,5 4,4	9,9 9,1	13,8 12,8	12,8 12,2	13,5 12,6	14,0
0200		0,7 2,7	4,6 9,6	16,0 13,3	10,7 12,7	11,0 13,2	12,0
0400		1,3 2,9	9,1 9,9	8,7 14,4	11,5 14,9	14,4 13,5	13,6
0600		1,4 3,1	10,5 10,5	14,3 13,3	12,0 12,9	13,3 13,6	10,6
0800		1,7 3,2	10,6 9,9	13,8 13,0	8,7 12,6	9,7 13,4	13,0
1000		1,7 3,2	10,1 9,6	13,5 12,6	13,0 12,3	13,8 13,0	14,0
Average		1,1 3,0	9,4 9,1	12,3	12,5 11,9	13,3 12,6	13,5

- Notes:
1. All results expressed as mg N/l
 2. E denotes experimental value
 3. P denotes predicted value
 4. Nitrate is absent from feed
 5. Reactor samples were filtered through coarse filter paper
 6. Effluent samples were filtered through Whatman 542 filter paper

6. DISCUSSION

6.1. Model predictions

In comparing the observed and predicted response one can proceed by examining the daily mass response and the cyclic response separately.

6.1.1. Daily mass response

The response of the plant, averaged over the day, observed and predicted are shown in Table 4.10.

Table 4.10.

Goudkoppie Works

Comparison of average predicted effluent results, oxygen consumption rate and MLSS concentration, with those obtained experimentally

Test	Average Model prediction	Average Measured value
COD (mg O/l)	40,8	43,0
TKN (mg N/l)	3,0	2,0
Ammonia (mg N/l)	0,1	1,5
Nitrate (mg N/l)	12,6	13,5
Oxygen consumption main basin (mg/l/h)	17,7	16,2
MLSS (mg/l)	1950	1940

Examining Table 4.10., the agreement is remarkably close, the maximum relative difference occurring in the oxygen utilisation rate (OUR) in the main aeration basin where the predicted exceeds the observed by $(17,7 - 15,8) = 1,9$ mg/l/h i.e. about 11 percent. With such a small difference, it is indeed difficult to identify any single significant source of error, but the most likely source is the experimental measurement of OUR ; it is sensitive to the location of the sampling point because the basin is not completely mixed ; although this was countered by taking the average from three locations the likelihood of obtaining the true average is small.

6.1.2. Cyclic_response

Again the predicted and observed response appears to be excellent. The first, and main difference, arises in the oxygen utilisation rate (Fig. 4.6.) where the two curves are slightly out of phase. This, very likely, could be the result of the fact that the model treats the main aeration basin as a single completely mixed unit. More likely, the basin should be treated as a semi - plug flow system by, for example, subdividing the unit into two reactors in series with perhaps a recycle between them. However it is doubtful if any further refinement in the prediction is of real value.

A second and less important difference between the predicted and experimental values is to be noted in Table 4.7. The observed COD concentration leaving the second aeration tank appears to be higher than that leaving the final clarifier, whereas one would expect these values to be almost identical.

Furthermore, the predicted COD concentration is virtually identical to that observed on the effluent from the clarifiers. The higher value in the aeration tank is attributed to the use of a different filter paper ; the tank samples were filtered through coarse filter paper whereas the effluent samples were filtered through Whatman 542 . The latter paper is a very fine one that will remove additional fine particulate material to that by the coarse paper. Consequently the fine paper gave COD concentrations lower than that using a coarse paper. The fine filter paper can be expected to be nearer the soluble COD fraction predicted by the model, and this indeed is the case in Fig. 4.7. It would appear that in future research, for the estimation of the soluble COD fraction in the reactor contents and the effluent, fine filter paper is mandatory, or, the sample should be centrifuged. However even if the fine filter paper is used, the filtered fraction in the sample from the aerobic reactor is unlikely to conform to the predicted value, for in this reactor an appreciable fraction of the soluble COD may consist of larger molecules or colloids that can pass through a fine filter paper.

With the exception of the above two differences, which are not considered serious, the simulated response by the model under the input data is in very good agreement with that observed. The general model, therefore, appears to provide a reliable means for predicting the behaviour of plants at the design stage, to obtain a rapid assessment of operating plants and to investigate the effect on the response if the flow and load conditions on an existing plant are changed.

7. NITROGEN REMOVAL

Nutrient removal from the Goudkoppie plant was poor. From Table 4.3. the average total nitrogen in the feed of 35 mg N/l was reduced to 15,5 total nitrogen ($\text{NO}_3\text{-N}$ plus $\text{NH}_3\text{-N}$), and the average total phosphorus in the feed of 5,8 mg P/l to 3,7 mg P/l. The question arises as to whether this poor performance was due to inadequate design, or whether the process was working at maximum efficiency but the input of nutrients was so high that it was not possible to remove them substantially.

To resolve these questions it is useful to refer to the theories on which nitrification / denitrification and excess phosphorus removal are based. From the work of van Haandel, Ekama and Marais (1981) the mass of nitrate formed (nitrification capacity), is approximately proportional to the influent TKN concentration. The concentration of nitrate denitrified (denitrification capacity), for any specific influent is approximately proportional to the influent COD concentration. Consequently, very roughly, the TKN/COD ratio provides an approximate assessment of the ratio of nitrification to denitrification capacities. The TKN/COD ratio therefore forms a useful reference parameter against which the likelihood of complete nitrogen removal of a specific plant can be estimated. It can be readily appreciated that if the COD is fixed, then for a specific configuration and sludge age, the denitrification capacity is fixed, and the process will have the capability of reducing the nitrate formed only if the TKN/COD ratio does not exceed an upper critical limit. For example, van Haandel et al (1981) showed that, for raw sewage, an optimal Bardenpho design can not attain complete denitrification if the TKN/COD ratio exceeds about 0,09 and in the Phoredox process, an optimal design can not give complete denitrification

if the TKN/COD ratio is equal to or greater than about 0,07 to 0,08 . These ratios refer to raw sewage. For settled sewage the ratios are slightly different principally because in primary settling, the easily biodegradable COD is only slightly affected, and magnitude of this fraction has a significant effect on denitrification. It is also evident from the work of van Haandel et al (1981) that the primary anoxic reactor is significantly more efficient in nitrate removal per unit volume than the secondary anoxic reactor. Consequently, for a specified TKN/COD the primary anoxic reactor should be increased in volume at the expense of the second anoxic reactor until zero nitrate is attained, the total anoxic mass fraction remaining constant. When the TKN/COD ratio attains a particularly high value this procedure will show that the volume of the second anoxic reactor eventually becomes zero. At any higher TKN/COD ratio it will be impossible to achieve complete nitrate removal for the fixed total anoxic mass fraction. Consequently in a particular plant, if nitrate is present in the effluent and that plant has a second anoxic reactor, the removal of nitrate could have been improved by increasing the primary anoxic reactor at the expense of the secondary anoxic reactor.; to express it differently, in an existing plant, if nitrate is present in the effluent and a second anoxic reactor is present, then the plant has not been optimally designed. The Goudkoppie plant, judged in terms of this criterion can be taken to have a design not optimal for the influent as received.

The relatively poor reduction in nitrate achieved in the second anoxic reactor is evident from the nitrate data in Table 4.3. The difference between the influent and effluent in this reactor is 8,9 (NO₃-N)/l . With respect to the influent doing a mass balance

the removal is 0,9 . Total flow / nominal flow = 0,9 (30,4 + 26,8)/30,4 = 1,7 mg (NO₃-N) . A similar balance on the primary anoxic reactor gives (13,5 . 26,8 + 182 . 13,4) - (30,4 + 182 + 26,8) . 9,8 / 30,4 = 15,0 mg (NO₃-N) removed.

Was the design optimal with regard to the assumed influent characteristics (Table 4.1.) ? To test this, the plant response was simulated using the general model taking as sewage inputs the COD and TKN values assumed in the original design, and the same flow value to the modules as for the experimental investigations, 30 Ml/d . (COD was obtained from the ratio BOD : COD = 1 : 2) . The results of this simulation are summarised in Table 4.11.

Table 4.11.

The predicted performance of the Goudkoppie
Plant under design load conditions

	Feed	Anaerobic	Primary Anoxic	Primary Aeration	Secondary Anoxic	Secondary Aeration
COD	600	150	91	90	90	90
TKN as N	50	32,6	10,9	3,8	5,6	2,9
Ammonia as N		30,9	8,6	1,4	3,1	0,5
Nitrate as N		0	1,9	7,6	1,2	3,4
Oxygen consumption rate mg/l/h				47,5		
MLSS				6 500		

Note: Flow 30 Ml/d

An inspection of Table 4.11. indicates that nitrate removal would have been virtually complete, and in terms of the theory for excess phosphorus removal, high removals would have been achieved.

However, the influent TKN to COD ratio assumed in the design was completely uncharacteristic of the value to be expected in settled sewage. The assumed TKN/COD ratio is $50/600 = 0,083$, whereas more characteristically this ratio should be 0,1 or higher. The COD concentration assumed in the design appears to be on the high side - it would indicate that the designers expected an influent COD of approximately 800 mg/l .

The analysis above points to an important conclusion: An appropriate design can be made only if the sewage characteristics, in particular the TKN/COD ratio, are known within reasonable limits. Should the TKN/COD ratio be more adverse than that assumed in the design, the process may fail in its objective.

These comments apply particularly to the Phoredox process. Once the process cannot fulfill its design objectives by virtue of a high TKN/COD ratio there is, in fact, no operational procedure whereby it is possible to make the plant operate in an improved fashion, with regard to phosphorus removal for a specific sewage characteristic under which it has shown failure.

8. PHOSPHORUS REMOVAL

Siebritz, Ekama and Marais (1982) concluded that the mass of phosphorus removed by a Phoredox process is dependent upon three factors:

- (i) The magnitude of the readily biodegradable COD in excess of 25 mg/l in the anaerobic reactor.
- (ii) The actual retention time in the anaerobic reactor
- (iii) The fractional mass of sludge in the system that is recycled through the anaerobic reactor each day.

Of these three factors the first only is related to the influent COD, but the concentration of readily biodegradable COD (S_{bsa}) in the anaerobic reactor depends also on the nitrate recycled to this reactor via the underflow recycle. For every 1 mg/l nitrate entering the anaerobic reactor 8,6 mg/l of S_{bsa} are used.

Taking these criteria into account there are two reasons why the phosphorus removal is so low ; .

- (i) The high nitrate concentration (13,5 mg N/l) in the underflow would remove 116 mg/l of easily biodegradable COD . The influent readily biodegradable COD concentration S_{bs} is 29 mg/l (see Table 4.4.) and consequently no S_{bsa} would be present in the anaerobic reactor and no excess phosphorus removal could be expected.
- (ii) Even if the nitrate in the recycle was zero this plant would still not have removed phosphorus, for the concentration in the anaerobic reactor with a 0,9:1 underflow recycle would be $29 / (0,9 + 1) = 15,3$ which is still less than the critical 25 mg/l required for excess phosphorus removal. It would appear therefore that the Phoredox process is inappropriate to give excess phosphorus removal for this influent. This conclusion is borne out by a study of the past history of the plant over the two and a half years of operation prior to this investigation - excess phosphorus removal was observed only sporadically (see Fig. 4.11.) even though the plant was operated with deficient oxygen input so that nitrification was only partial (the average recycle nitrate concentration was 4,5 mg N/l)

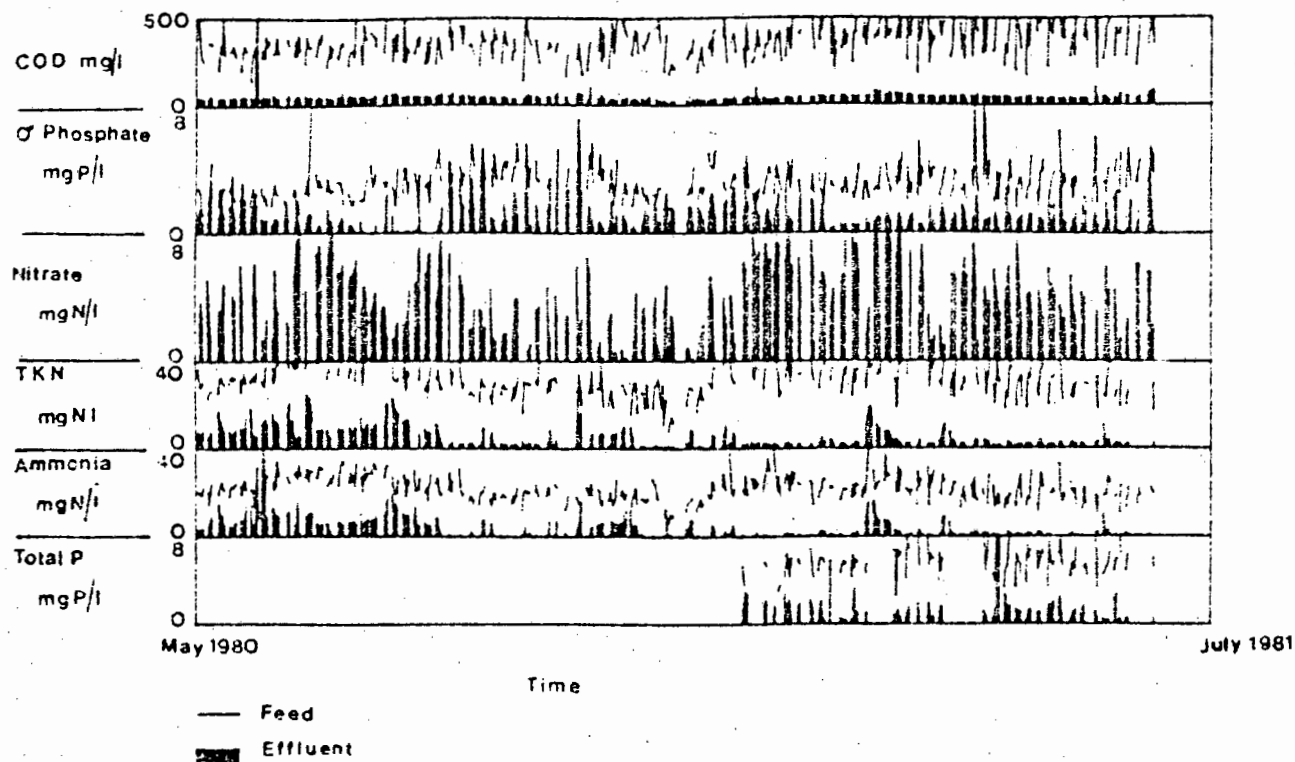


Fig. 4.11. The Goudkoppie plant

The feed and effluent concentrations of some of the main parameters from the start up of the plant until the balancing strategy was applied.

9. POSSIBLE MODIFICATION TO THE GOUDKOPPIE PLANT TO IMPROVE NUTRIENT REMOVAL UNDER THE EXISTING FLOW AND LOAD

In view of the poor nitrogen and phosphorus removal attained in the Goudkoppie plant it was decided in July 1980 to investigate possible ways of improving plant performance. Two approaches to the problem of improving plant performance were considered :

- (i) Changing process configuration from a Phoredox to a Modified UCT process

(ii) Bypassing the equalisation tanks

(iii) Changing the operating mode of the equalisation tank

In each case the process model was utilised in the analysis of the effects of different changes on plant performance. The model serves a very useful purpose in this respect because the accuracy of predictions obtained from the model can be regarded with confidence ; this aspect has become apparent from the very close correspondence between observed and predicted results presented in the previous section.

9.1. Changing to a Modified UCT Process configuration

Siebritz, Ekama and Marais (1980) have recommended the use of the Modified UCT process to a single sludge system where, firstly nutrient removal is required, and secondly the feed TKN/COD ratio is high. It would appear that the process would lend itself to the conditions which exist at the Goudkoppie plant provided the sewage characteristics were more favourable ; the S_{bsi} of the feed to the process is so low that it is doubtful whether any process modification would significantly improve the mass of phosphorus removed. Nevertheless in this section the feasibility of modifying the Goudkoppie plant from a Phoredox process to a Modified UCT process was investigated.

For optimal nitrogen removal in the Modified UCT process, Siebritz et al (1980) have recommended the anoxic mass fraction and recycled flows given in Fig. 4.12. The Goudkoppie plant could be changed to incorporate these recommendations but it would require certain expensive modifications. The main modification would be to increase the size of the primary anoxic reactor by converting the existing secondary anoxic reactor into a new second primary anoxic reactor. With this change the mass fractions of the various zones then would be approximately within practical limits, the desired values given in Fig. 4.12. (see Table 4.12.)

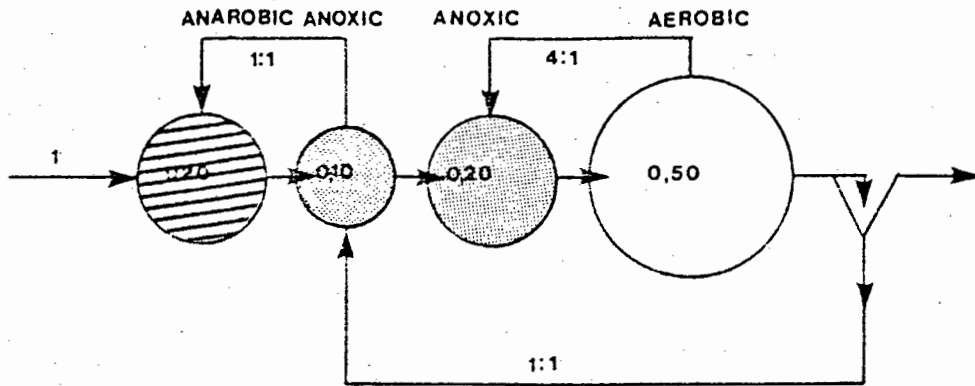


Fig. 4.12. Layout of the Modified UCT Process

Note: Values given inside reactor denote mass fraction

Table 4.12.

The mass fractions of the various zones when the secondary anoxic reactor is converted into a primary anoxic reactor

	Bardenhoo Goudkoppie	UCT Goudkoppie
Anaerobic	0,07	0,20
Primary anoxic 1	0,17	0,10
2	0,17	0,20
Aerobic	0,51	0,50
Secondary anoxic	0	0
Re aeration	0,09	0

A more detailed explanation of the changes required to convert the Goudkoppie plant from a Bardenpho to a UCT process is given below :

- (i) With relatively minor changes the primary and secondary anoxic reactors could be interconnected using the existing MLSS recycle channels; the activated sludge could be made to flow by gravity from the existing primary anoxic to the existing secondary anoxic reactors. Using a new recycle pump the sludge from the former secondary anoxic reactor could be pumped back to the primary anoxic reactor. In this manner the two reactors would be interconnected.
- (ii) The MLSS recycle ratio could be controlled by allowing the desired proportion of effluent to flow from the main aeration basin into the newly created primary anoxic reactor. (Since there would be a head loss between these two reactors no pumping would be required.) The remainder of the effluent from the main aeration basin could bypass the new primary anoxic reactor and flow into the old re-aeration reactor. Again no pumping is required as there is a head loss between the two reactors, but a new pumping main is required.
- (iii) The Modified UCT process requires that the returned sludge be returned to the primary anoxic reactor and not to the anaerobic reactor as is the case at present. Such a modification would necessitate additional pipework ; no additional pumping would be required.
- (iv) The MLSS in the first primary anoxic reactor should be recycled back to the anaerobic reactor; one of the redundant

- MLSS recycle pumps could be used for this purpose in conjunction with a new pumping main.

A summary of the suggested modifications is depicted in Fig. 4.13. together with the relative mass fractions of each reactor.

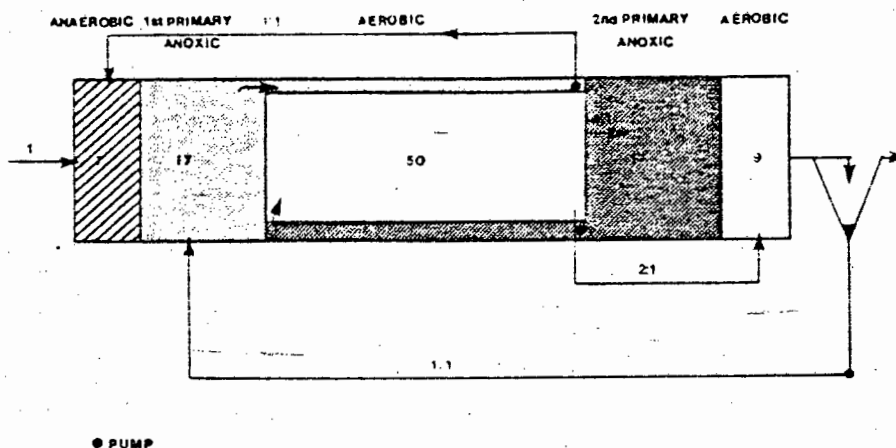


Fig. 4.13. Proposed modification of the Goudkoppie Plant to a Modified UCT Process

Such modifications as proposed above would involve considerable expense and to establish whether such expenditure is justified the data obtained during the intensive test period discussed earlier is used in the model to predict the performance of the plant incorporating the Modified UCT process. The constants used in the model prediction were the same as those used in the initial test run - see Tables 4.4. and 4.5. , and the input flow , COD and TKN data the same as given in Table 4.5.

The average results predicted by the model for the Goudkoppie plant when operating a modified UCT mode are given in Table 4.13. and the dynamic predictions of soluble COD, ammonia and nitrate in each reactor are given in Figs. 4.14. , 4.15. and 4.16. respectively.

Table 4.13.

Predicted performance of Goudkoppie plant when changed to a Modified UCT Process and fed with balancing tank effluent

Test	Feed	Anaerobic	First Primary Anoxic	Second Primary Anoxic	First Aerobic	Second Aerobic
Total COD	270	53	42	41	41	41
Soluble easily biodegradable COD	29	11,7	0,8	0	0	0
TKN as N	35					
Ammonia as N	25	19,9	13,3	4,9	0,5	0,1
Nitrate as N	0	0	1,1	7,0	11,2	11,8
MLSS					1 740	
Oxygen consumption rate mg O ₂ /l/h					18,1	

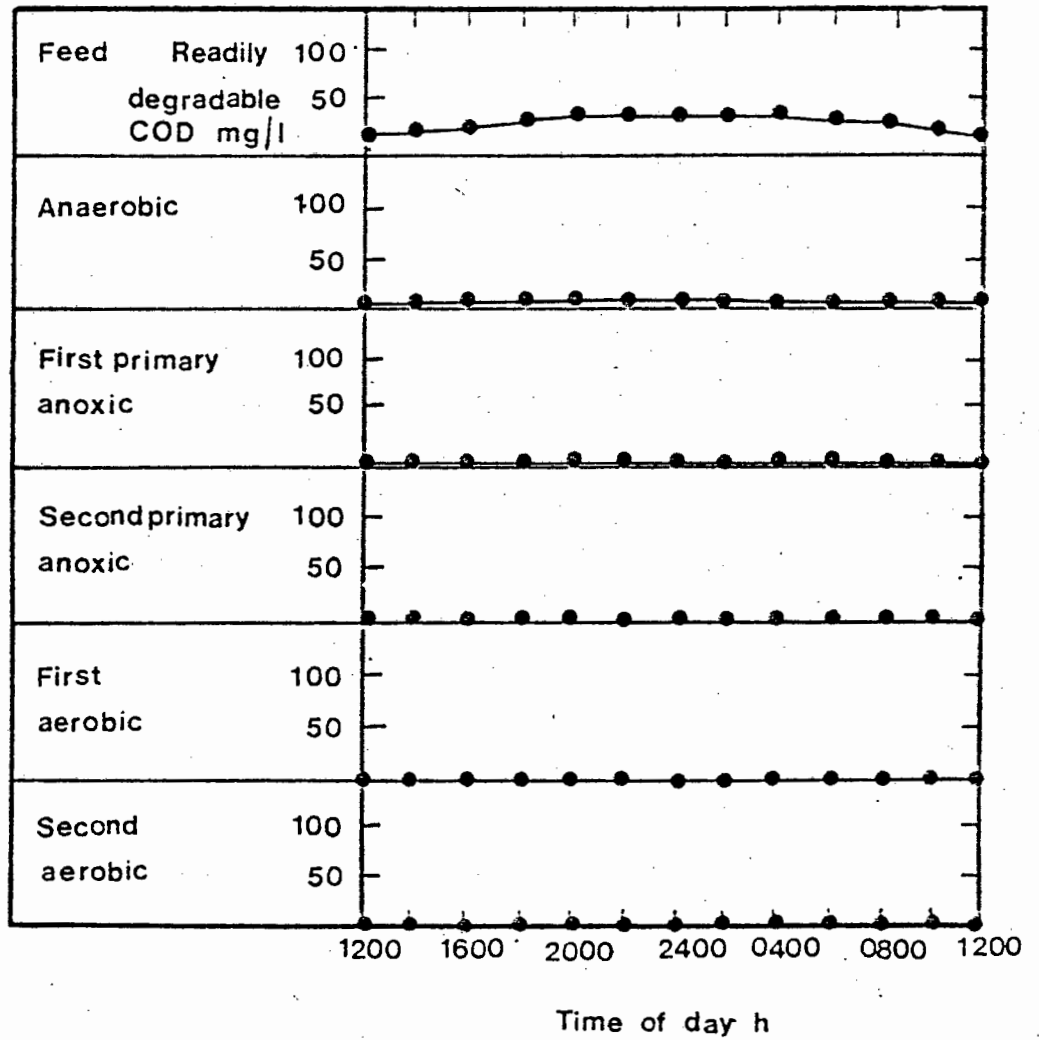


Fig. 4.14. Process configuration : Modified UCT

The predicted readily biodegradable COD concentrations when the plant is fed with balancing tank effluent

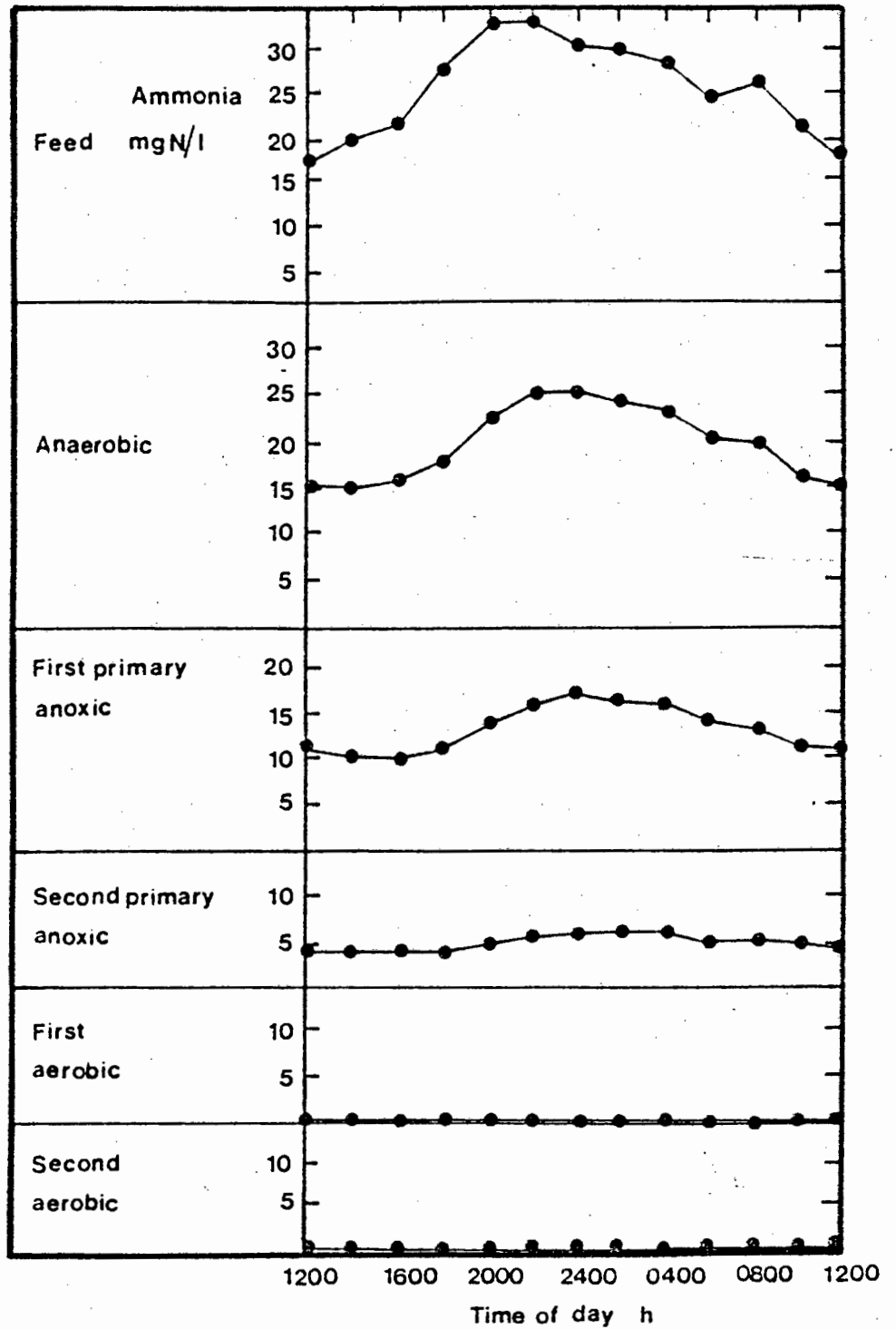


Fig. 4.15. Process configuration : Modified UCT

The predicted ammonia concentrations when the plant is fed with balancing tank effluent

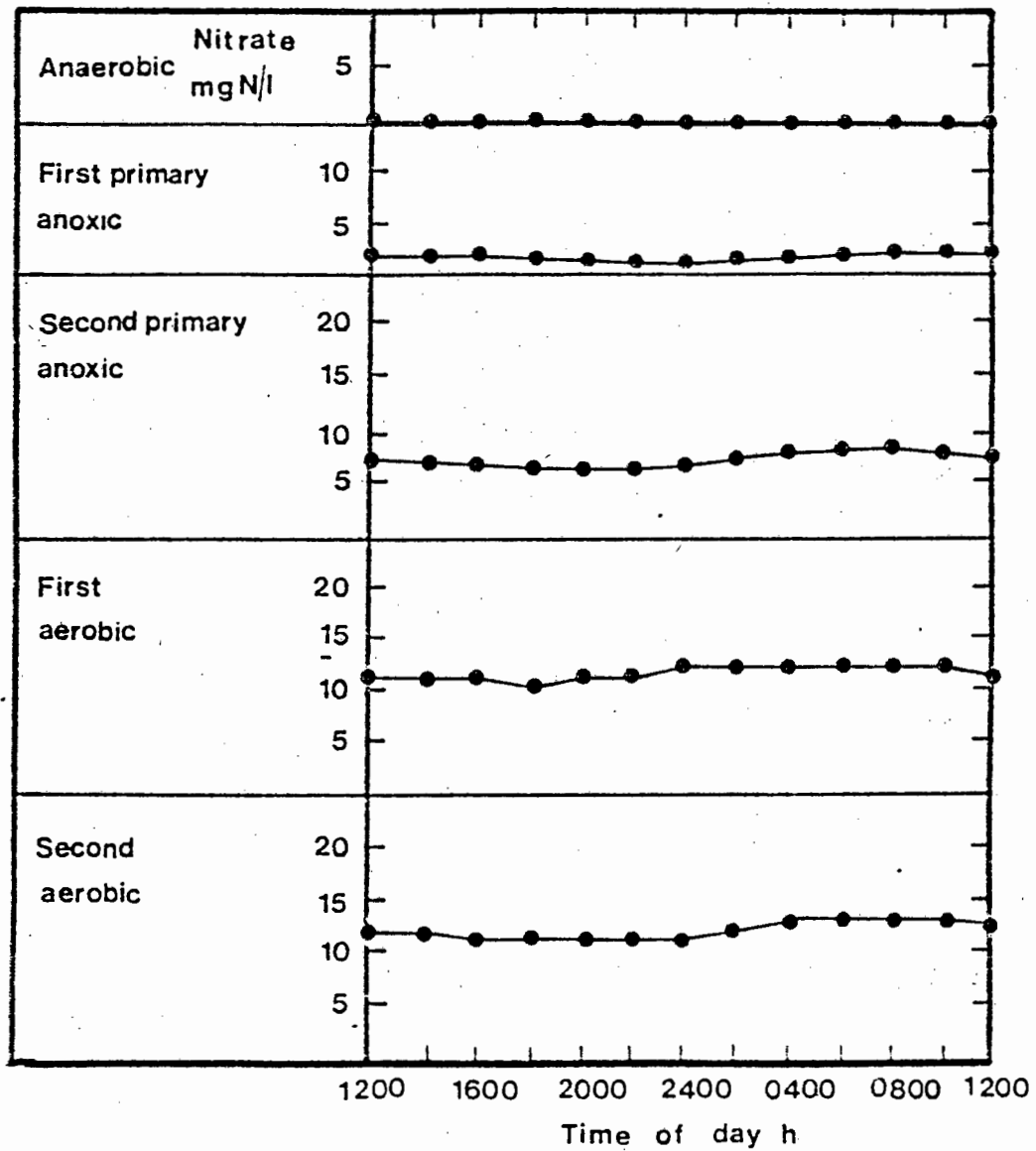


Fig. 4.16. Process configuration : Modified UCT

The predicted nitrate concentrations when the plant is fed with balancing tank effluent

9.1.1. Discussion

From the analysis above it was apparent that ;

(i) The nitrate concentration in the first and second anoxic reactors is 1,1 and 7,0 mg N/l respectively, which is not acceptable both from a nitrogen and phosphorus removal point of view. This lack of removal may be attributed to:

- * the high TKN/COD ratio
- * the low readily biodegradable COD concentration of the feed to the process, 29 mg/l

(ii) As the average readily biodegradable COD concentration in the anaerobic reactor is estimated to be approximately 12 mg COD , no release of phosphorus is expected, with the result that no or little excess phosphorus uptake is likely to occur.

Both the above factors show clearly that, for this particular situation, there is no merit in considering changing the process configuration from that of the Phoredox to the Modified UCT process. With regard to nitrogen removal, performance with the Modified UCT process would be marginally only better than that attained with the present Phoredox configuration. The cost of implementing the change in configuration (in the region of R300 000) would definitely not be justified in the light of the slight improvement in effluent quality.

Comparison of the performance of the Goudkoppie plant (either in a Phoredox or Modified UCT configuration) with the performance of other plants clearly showed that the principal cause of poor nutrient removal was the low fraction of easily biodegradable COD entering the process. An investigation into the causes for the relatively low value led to the second approach to improving plant performance i.e. changing the mode of operation.

9.2. Bypassing the settling tank

A schematic representation of the flow entering the plant is shown in Fig. 4.1. Raw sewage after screening and grit removal enters the primary sedimentation where approximately 45% of the influent COD is removed in an underflow stream which constitutes only two percent of the inflow. Overflow from the primary sedimentation tank passes to the biological process via the balancing tank. Overflow from the primary sedimentation tank passes to the biological process via the balancing tank.

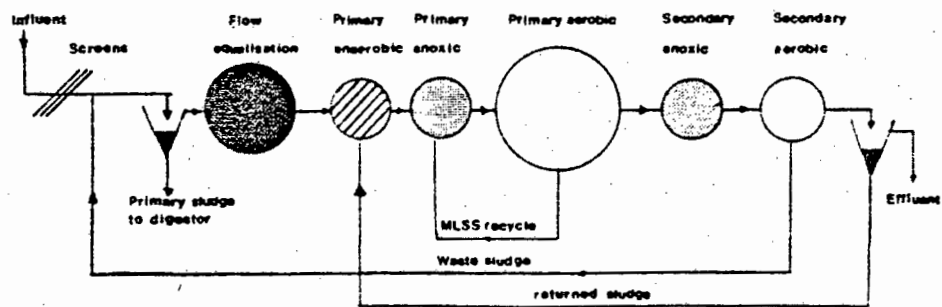


Fig. 4.1. Goudkoppie Plant
Basin configuration

Measurement over a period of time of the sewage characteristics of the unsettled (raw), settled and equalised flows are given in Table 4.14. The concentration of easily biodegradable COD S_{bsi} can be expressed as a fraction of the biodegradable COD f_{bs} by taking due account of f_{us} and f_{up} in these influents. The S_{bs} was estimated and is also given in Table 4.14.

Table 4.14.

Sewage characteristics of raw, settled and equalised flows at the Goudkoppie plant.

	Symbol	Raw Sewage	Settled Sewage	Balancing tank effluent
Fraction of readily biodegradable COD (mg COD/mg biodegradable COD)	f_{bs}	0,19	0,36	0,16
Fraction of particulate unbiodegradable COD (mg VSS/mg COD)	f_{up}	0,10	0,08	0,12
Fraction of soluble unbiodegradable COD (mg COD/mg COD)	f_{us}	0,06	0,08	0,10
Easily biodegradable COD concentration (mg COD/mg COD)	S_{bs}	114	114	29

The following should be noted regarding the results given in Table 4.14.

- (i) Influent raw sewage had a f_{bs} value of 0,19 mg COD/mg COD. This is only marginally less than that encountered at other works in South Africa ; usually $f_{bs} \approx 0,24$.
- (ii) Approximately 45% of the raw influent COD was removed in the primary sedimentation tanks; and the COD removed consisted principally of particulate matter; however this mass of COD is removed in only 2% of the flow. Therefore it can be expected that the readily biodegradable fraction with

respect to the total COD in settled sewage should be almost double that of the raw waste water. This was borne out by the results which showed that f_{bs} increased from 0,19 to 0,36 mg COD/mg biodegradable COD in passing through the primary sedimentation tank.

- (iii) The value of f_{bs} for settled sewage and for balancing tank effluent should be similar i.e. very little change in the value of f_{bs} should be expected because the function of the un-aerated balancing tank merely is to attenuate cyclic flow and load rate variations. However, the results show that the value of f_{bs} was reduced considerably (by 56 percent). In addition, it was found that there was approximately a 20 percent mass loss of COD across the balancing tank. These aspects were a direct consequence of the operational procedure applied to the balancing tank, and merit further discussion.

The Goudkoppie balancing tank is unmixed ; therefore settlement of sewage solids on the base of the tank is to be expected. If the tank is not drained regularly there will be a large accumulation of solids in the tank. In practice, this was found to be the case ; in fact, the accumulation of solids was so large that plant operators specifically avoided draining the tank because, if this happened, the COD concentration in the stream leaving the tank rose to as high as 6 000 mg/l . This led to operational problems in the down stream process. This operational approach had severe consequences as regards the performance of the Goudkoppie Plant;

- * The accumulation of solids on the tank floor led to a substantial degree of anaerobic activity in this sludge. Evidence of this was the gas bubbles which rose to the

surface of the balancing tank together with masses of sludge. The net result was that the f_{bs} reduced significantly*. In addition the TKN/COD ratio was increased from 0,10 to 0,13 because there was only a marginal loss of TKN across the tank. The consequence of both the above phenomena was the poor nitrogen and phosphorus removal in the subsequent activated sludge plant discussed earlier.

To avoid the reduction in f_{bs} so as to enhance plant performance two approaches were considered:

- (i) Bypassing the balancing tank
- (ii) Operating the balancing tank so as to ensure no accumulation of solids on the floor.

The first approach was unlikely to be practical at the Goudkoppie plant. Firstly, considerable expense would be required to bypass the balancing tank, and secondly the plant was designed to handle only the equalised (or mean) flow rate i.e. smaller final clarifiers, larger peak load concentration which would result in peak aeration capacities which could be greater than the maximum oxygen input allowed for in the design. In addition, operation under

* At one stage it was thought that the anaerobic activity would, in fact increase f_{bs} ; it was hypothesised that anaerobic degradation of particulate matter would increase the amount of readily biodegradable material which in turn would improve the phosphorus removal capacity. However, with the long retention time of the solids on the floor, the fermentation was more than likely taken to completion with methane and not volatile acids as the end product. The gas produced was not analysed because at that time methane fermentation was not suspected

unequalised inputs would lead to major operational and control problems. However to illustrate the effect which an increase in f_{bs} would have on the performance, it was decided to simulate the behaviour of the plant having as input the settled sewage flow before entry into the balancing tank i.e. there would be no reduction in either the readily biodegradable fraction of COD or the mass of COD leaving the primary sedimentation tanks. The kinetic parameters used were the same as those used previously, given in Table 4.5., while those kinetic parameters specific to this settled sewage are given in Table 4.15. The plant operating conditions are the same as discussed in section 3.2. with the exception of the flow to the process which would not be equalised. The input data with respect to flow, COD and TKN concentrations of the feed to the process (settled sewage) is listed in Table 4.16.

The results of this simulation are shown in Table 4.17. and in Figs. 4.17., 4.18. and 4.19.

Table 4.15.

Kinetic parameters specific to settled sewage at
Goudkoppie plant when balancing tank bypassed

Name	Symbol	Units	Value
Fraction of soluble unbiodegradable COD	f_{us}	mg COD/mg COD	0,08
Fraction of particulate unbiodegradable COD	f_{up}	mg VSS/mg COD	0,08
Fraction of easily biodegradable COD	f_{bs}	mg COD/mg bio-degradable COD	0,36
Easily biodegradable COD concentration	S_{bs}	mg COD/l	113
Fraction of TKN of feed in ammonia form	f_{nu}	mg N/mg N	0,76
Specific growth rate for nitrifiers	μ_{nm}	d^{-1}	0,5

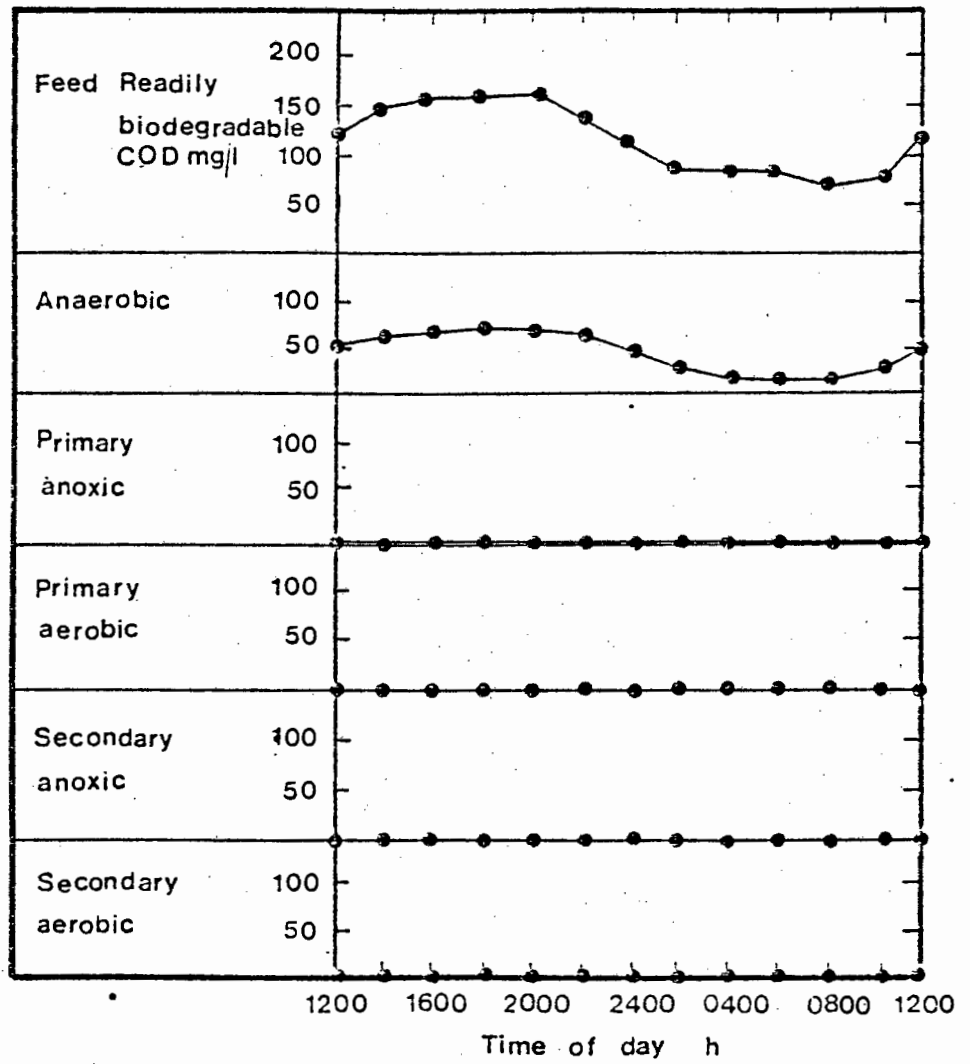


Fig. 4.17. Process configuration : Phoredox
 Predicted readily biodegradable COD concentrations
 when the balancing tank is bypassed

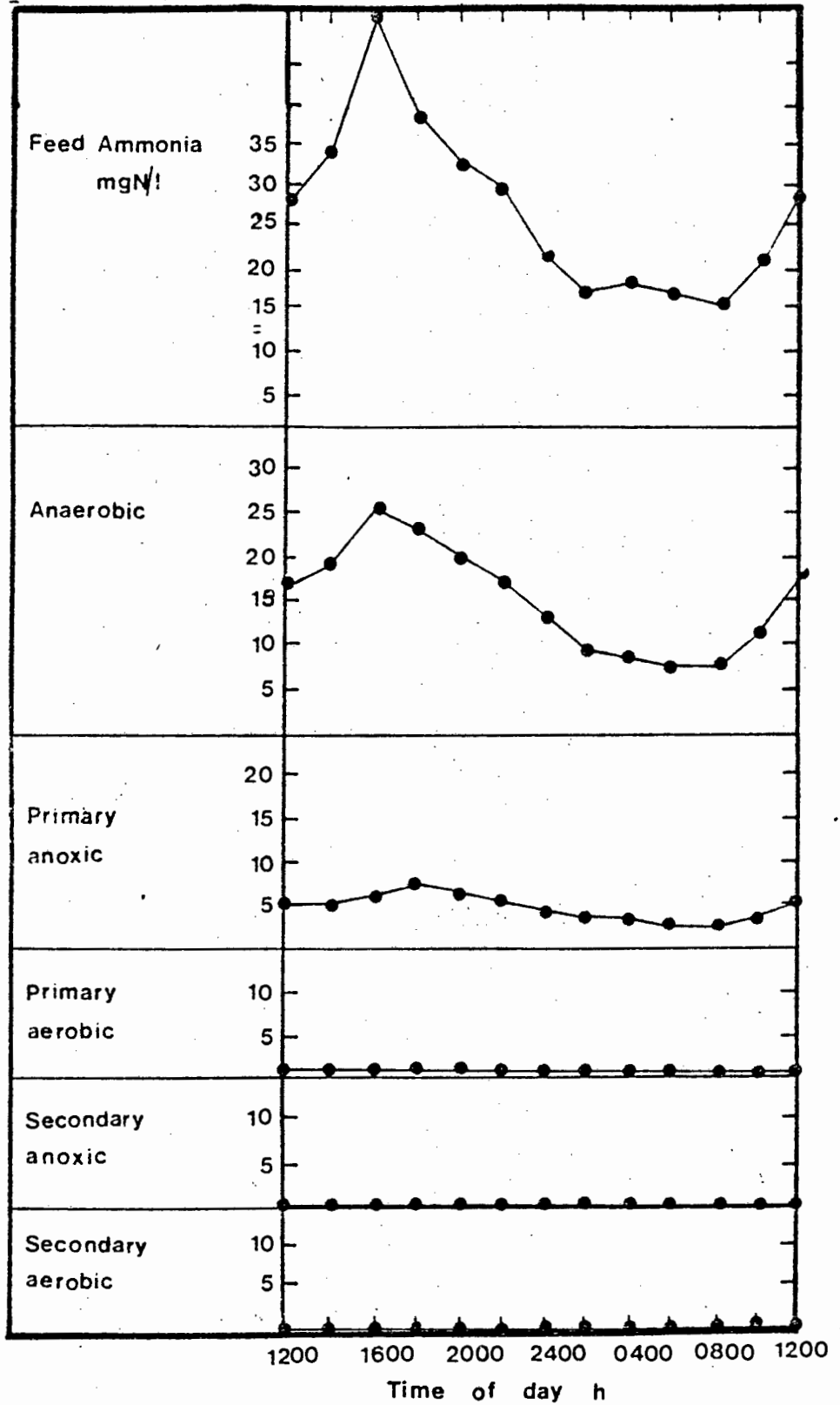


Fig. 4.18. Process configuration : Phoredox
 Predicted ammonia concentration when the
 balancing tank is bypassed

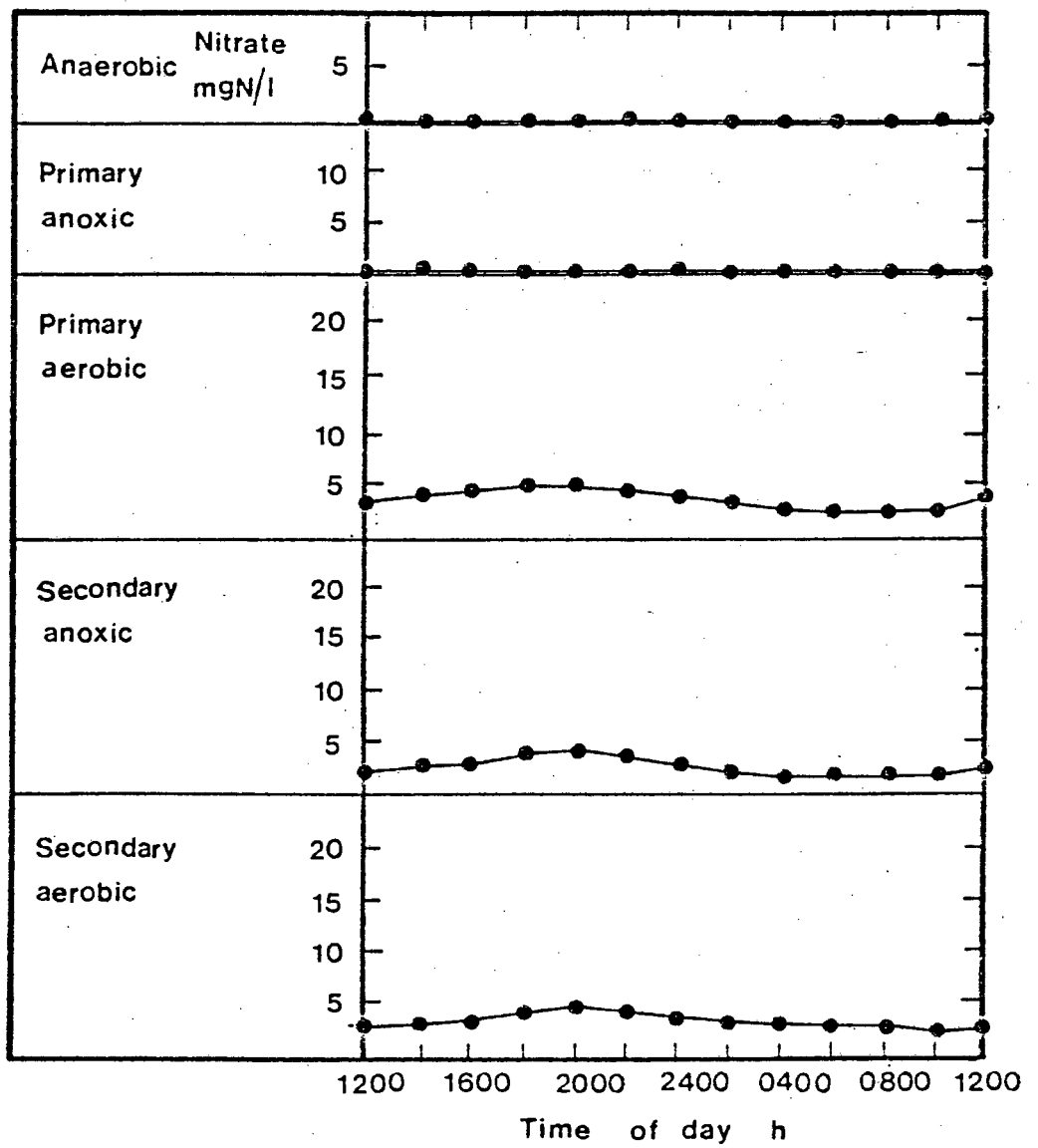


Fig. 4.19. Process configuration : Phoredox

Predicted nitrate concentration when the
balancing tank is bypassed

Table 4.16.

Input flow and COD and TKN concentrations used in the model to predict the behaviour of the process if the balancing tank were bypassed

Time	Flow m ³ /d	COD mg/l	TKN mg N/l	Time	Flow m ³ /d	COD mg/l	TKN mg N/l
1200	51 800	420	37	0100	25 900	330	25
1300	46 100	460	41	0200	23 300	280	22
1400	40 300	500	45	0300	20 600	280	23
1500	40 300	515	43	0400	17 900	280	24
1600	40 300	530	60	0500	17 600	285	23
1700	43 200	530	45	0600	17 300	290	23
1800	46 100	530	50	0700	22 200	255	21
1900	41 800	545	46	0800	27 100	220	19
2000	37 400	560	42	0900	38 000	240	24
2100	37 400	520	41	1000	49 000	260	28
2200	37 400	480	39	1100	50 400	340	32
2300	33 100	430	32	1200	51 800	420	37
2400	28 800	380	28				

Table 4.17.

Predicted performance of the Goudkoppie Plant when the balancing tank is bypassed

Test	Feed	Anaerobic	Primary Anoxic	Primary Aerobic	Secondary Anoxic	Secondary Aerobic
Total COD	330	77,8	36,6	33,5	33,5	33,5
Soluble easily biodegradable COD	113	44,7	3,1	0	0	0
TKN as N	34					3,1
Ammonia as N	25,7	14,7	4,2	0,5	0,9	0,1
Nitrate as N	0	0	0	3,0	2,1	3,0
MLSS				3 060		
Oxygen utilisation rate mg O ₂ /l/h				27,4		

From these plots several factors are of interest. When the balancing tank is bypassed, the model predicts :

- (i) Good nitrification in the primary aerobic reactor with ammonia concentrations of approximately 0,5 mg N/l leaving the reactor.
- (ii) Denitrification improves dramatically ; the mean effluent nitrate declines from 13,5 mg N/l for the equalised effluent to approximately 3 mg N/l if the bypass is instituted.
- (iii) Even with a recycle nitrate of 3 mg N/l discharging to the anaerobic reactor, the concentration of readily biodegradable COD (S_{ba}) in the anaerobic reactor increases ; zero when the balancing tank is in line to 45 mg/l when the balancing tank is bypassed. The latter concentration is well above that suggested by Siebritz et al (1980) (25 mg/l) necessary to cause good phosphorus release and effective phosphorus removal.

Since the model predictions do not consider the relatively high dissolved oxygen concentrations in the feed entering the process, the 45 mg/l readily biodegradable COD in the anaerobic reactor predicted by the model would be lower than that hypothesised. If, for example, the feed contained 4 mg/l dissolved oxygen, then $2,86 \times 4 = 11,4$ mg/l of readily biodegradable COD would be utilised by this oxygen, thus reducing the readily biodegradable fraction to approximately 34 mg/l ; however this is well in excess of the limit suggested by Siebritz et al (1980) . Consequently appreciable phosphorus removal would be expected. Since bypassing the balancing tank was not a practical proposition, the discussion above is only hypothetical. However the results serve the useful purpose of illustrating the positive effect of ensuring no reduction in f_{bs} .

9.3. Modification to operating the settling tank

An avenue to improve the phosphorus removal propensity would be to avoid the reduction of f_{DS} in the balancing tank. It was hypothesised that if the operating mode of the balancing tank was changed such that solids would not accumulate on the floor of the tank, then possibly fermentation of the sludge might not take place, and the readily biodegradable COD concentration could preserve its value in the influent to the tank. At that time it became possible to test the validity of this hypothesis. An on-line microprocessor - based equalisation tank control strategy, developed by Dold, Buhr and Marais (1982), was implemented at the balancing tank at Goudkoppie, and operated very successfully. Amongst the important desirable features achieved by the operation of on-line control balancing tank strategy were :

- (i) The efficiency of equalisation attained under control strategy operation was far superior to that achieved when the balancing tank was operated manually. Previously, problems of balancing tank overflows were encountered almost weekly, whereas under the control strategy operation this did not occur. The improved equalisation performance also reduced operation and control problems on the downstream plant substantially e.g. aeration control, settling tank control etc.
- (ii) Under control strategy operation, the problem of solids accumulation in the balancing tank was also overcome. This was achieved by a feature of the control strategy operation, that each day the tank level was regulated to drop sufficiently to ensure flushing of any accumulated solids. As a consequence, anaerobic degradation of the settled material no longer took place in the balancing tank, with the result

- that there was no significant decrease in the value of f_{bs} across the tank and, during the hold up, the mass loss of the total COD did not exceed one percent (As opposed to a 20 percent mass loss encountered under manual balancing tank operation); the net effect of these two aspects was that there was no significant increase in the TKN/COD ratio across the balancing tank.

10. THEORETICAL EVALUATION OF THE EFFECT ON NUTRIENT REMOVAL WHEN THE BALANCING TANK CONTROL STRATEGY WAS APPLIED

The successful operation of the balancing tank under equalisation control strategy operation (Dold, Buhr and Marais, 1982) prompted a simulation investigation on the possible effects on the nutrient removal capacity of the plant. For the simulation the input data are listed below in Table 4.18.

Table 4.18.

Input data for simulation under equalisation control strategy

Parameter	Symool	Units	Value
Total COD	S_{ti}	mg/l	390
TKN	N_{ti}	mg N/l	34
Underflow recycle ratio	s		1:1
Mixed liquor recycle ratio	a		4:1
Sludge age	R_s	d	22
Fraction of unbiodegradable particulate COD	f_{up}	mg VSS/mg COD	0,08
Fraction of soluble unbiodegradable COD	f_{us}	mg COD/mg COD	0,08
Fraction of readily biodegradable	f_{bs}	mg COD/mg biodegradable COD	0,36
Flow		ml/d	35
Temperature	T	C	20

The other required constants and physical parameters are the same as those listed in Tables 4.4. and 4.5. The process configuration and volume of reactors are listed in Fig. 4.2.a

The simulated results using the above input data are given in Table 4.19.

Table 4.19.

Input data for simulation under equalisation control strategy operation

	Feed	Anaerobic	Primary Anoxic	Primary Aerobic	Secondary Anoxic	Secondary Aerobic	Effluent
Total COD	370	77,8	36,6	35,5	33,5	33,5	33,5
Soluble COD	114						
S_{bs}	113	44,7	3,1	0	0	0	0
TKN as N	34					3,1	3,1
Ammonia as N	25,7	14,7	4,2	0,5	0,9	0,1	0,1
Nitrate as N	0	0	0	3,0	2,1	3,0	3,0
Total P							
Soluble P							
Suspended solids				3 060			
Oxygen consumption mg/l/h				27,4			

Unless otherwise stated all results expressed as mg/l

The theoretical concentration of phosphorus which could be expected to be removed by the plant when operating under the conditions above may be estimated as follows :

(i) Biodegradable COD concentration S_{bi}

$$\begin{aligned}
 S_{bi} &= (1 - f_{us} - 1,48 f_{up}) S_{ti} & (11) \\
 &= (1 - 0,08 - 1,48 \cdot 0,08) 394 \\
 &= \underline{315 \text{ mg COD/l}}
 \end{aligned}$$

- (ii) Readily biodegradable COD concentration in the anaerobic reactor S_{bsa}

$$\begin{aligned}
 S_{bsa} &= (S_{bi} f_{bs} - 8,6 S N_{nr}) / (1 + S) & (13) \\
 &= (315 \cdot 0,36 - 8,6 \cdot 1 \cdot 3,0) / (1 + 1) \\
 &= (113 - 25,8) / 2 \\
 &= \underline{43,6 \text{ mg COD/l}}
 \end{aligned}$$

- (iii) The mass concentration of phosphorus that will be removed and the anaerobic mass fraction is expressed by the phosphorus removal propensity factor P_f ;

$$\begin{aligned}
 P_f &= (S_{bsa} - 25) f_{xa} & (14) \\
 &= (43,6 - 25) 0,07 \\
 &= 1,30
 \end{aligned}$$

The propensity factor is related to the percentage concentration of phosphorus with respect to the volatile active suspended solids VASS , , as follows :

$$\begin{aligned}
 &= 0,35 - 0,29 e^{-0,242 P_f} \\
 &= 0,35 - 0,29 e^{-0,242 \cdot 1,3} \\
 &= 0,14
 \end{aligned}$$

Having estimated α , the concentration of phosphorus which can be removed is estimated using equation (15)

$$\begin{aligned}
 \Delta P &= S_{t1} \left((1 - f_{us} - P f_{up}) Y_h (\alpha + f_p f_b R_s) \right) / & (15) \\
 &\quad \left[(1 + b R_s) + f_p f_{up} / \rho \right] \\
 &= 394 \left((1 - 0,08 - (1,48 \cdot 0,08)) 0,45 \right. \\
 &\quad \left. (0,14 + (0,015 \cdot 0,2 \cdot 0,24 \cdot 22)) \right) / \\
 &\quad \left[(1 + 0,24 \cdot 22) + (0,015 \cdot 0,08) / 1,48 \right] \\
 &= 394 (0,009) \\
 &= \underline{3,5 \text{ mg P/l}}
 \end{aligned}$$

Hence provided the influent characteristics are given in Table 4.18. under equalisation control strategy, the Goudkoppie plant could be expected to remove 3,5 mg P/l .

It is important to note that this removal is subject to the nitrate concentration in the effluent remaining at or below 3 mg N/l . If the nitrate concentration should increase the removal will decrease. In effect the plant performance will be critically affected by the nitrate concentration. Clearly such a situation may be tolerated, but is not desirable for stability in operation.

Based on the consideration expressed above that if the nitrate in the effluent could be limited to about 3 mg N/l , phosphorus removal is achievable, the plant was operated with some of the aerators switched off near the point discharging the flow from the primary anoxic reactor. The aeration capacity in the rest of the reactor was appropriately increased so that nitrification could proceed within this region.

During the experimental period the flow was 40 Ml/d and the sewage characteristics were as listed in Table 4.18. The mean observed effluent quality under these conditions is listed in Table 4.20. and the daily response of the plant shown in Fig. 4.20.

Table 4.20.

Feed and effluent concentrations when the Goudkoppie plant was operated with equalisation control strategy.

Test	Feed	Effluent
Total COD	300	50
TKN as N	32	9,5
Ammonia as N	21	8,5
Nitrate as N	-	1,0
Total phosphorus as P	6,1	1,0
Readily biodegradable COD	86	-

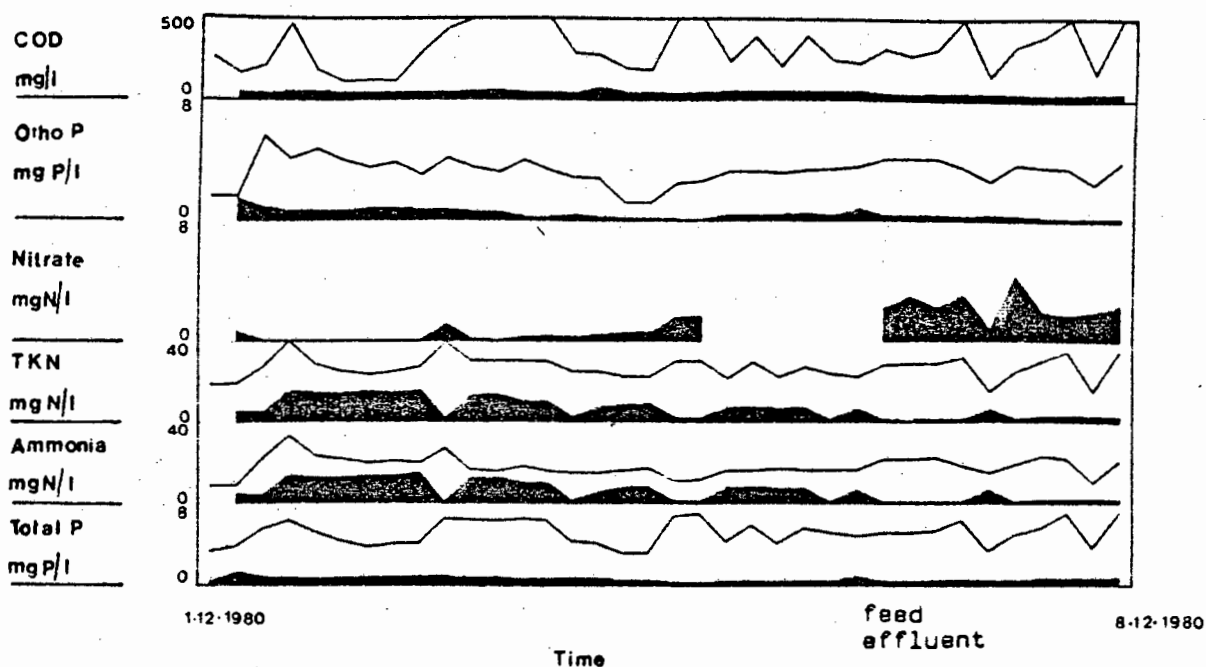


Fig. 4.20. Goudkoppie Works

The feed and effluent concentrations of the more important parameters when operating under equalisation control strategy

The theoretical phosphorus removal i.e. ($P = 3,0 \text{ mg P/l}$) was lower than the mean removal observed ($P = 3,5 \text{ mg P/l}$); but because the plant was now no longer operated in a definable sense kinetically the removal could deviate from the predicted. At least the theory did indicate successfully the direction to be taken to achieve phosphorus removal. Examining Fig. 4.21., the phosphorus removal remained relatively constant over the period of the experimentation. which replaced the unstable removal situation of before. Indeed, in as far as the phosphorus removal aspects on the Goudkoppie plant is concern concerned, the removal is probably near the maximum possible theoretically, and no other operational procedure is likely to improve it. However this efficiency was attained only by curtailing the aeration capacity

at the beginning of the main aeration basin, thereby effectively increasing the anoxic zone. It was also possible to operate in this fashion because the remaining mass fraction was large enough to allow nitrification to take place. i.e. this mode of operation will be limited by the load to the plant.

CHAPTER FIVE

NORTHERN WORKS

1. INTRODUCTION

As the name implies, this plant treats the waste water discharged in the watershed draining the north of Johannesburg. Originally there were three waste water treatment plants serving this area ; Bruma, Cydna and Delta Works (see Fig. 1.1.) However by 1950 these plants were surrounded by residential areas and were overloaded. As a result a new treatment plant, the Northern Works, was constructed on the banks of the Jukskei river, approximately 25 km from the city centre (see Fig. 1.1.) . This plant is a metropolitan one treating, not only waste water that flowed to the old Bruma, Cydna and Delta plants, but also those derived from the satellite towns surrounding Johannesburg in the northern drainage basin.

The first Northern Works, commenced in 1961, comprised two identical units each with primary sedimentation, biological filtration and sludge digestion. The effluent from these plants discharged into the Jukskei river either directly or indirectly via Kelvin Power Station (where the effluent was used as cooling water). These discharges were high in nitrate and phosphate, and at times, caused severe eutrophication problems in Hartebeespoort Dam located downstream of the effluent discharge points.

By the late 1970's both these plants were overloaded and, as a result, Northern Works Unit 3 was designed, constructed and commissioned by the end of 1979. This new plant, located alongside the existing plants, is of a similar type to that at Goudkoppie, discussed in Chapter 4.

Table 5.1

NORTHERN WORKS DESIGN DATA

Flow / module	(Ml/d)	50				
Raw sewage composition	BOD (mg/l)	380				
	TKN (mg N/l)	46				
Settled sewage composition	BOD (mg/l)	265				
	TKN (mg N/l)	30				
Sewage temperature	(°C)	13				
BOD loading	(kg BOD/kg MLSS)	0,401				
MLSS	(g/m ³)	5 250				
Sludge age	(days)	25				
Volume	(1) Anaerobic (m ³)	2 230				
	(2) Primary anoxic (m ³)	4 575				
	(3) Primary aerobic (m ³)	15 300				
	(4) Secondary anoxic (m ³)	4 575				
	(5) Secondary aerobic (m ³)	2 500				
Total	(m ³)	29 180				
Basin number		1	2	3	4	5
Nominal retention time (h)		1	2	7	2	1
Number of aerators		-	-	10	-	3
Power / aerator (kW/h)				90		22
Oxygenation capacity (kg O ₂ /h per aerator)				186		45
Oxygen transfer efficiency (kg O ₂ /kWh)				2,03		1,94
Number of mixers		2	4	-	4	-
Power per mixer (kW)		15	15	-	22	-
Number per module		3				
Design nominal daily flow to secondary sedimentation tank (without recycle) (m ³ /h)		700				

Table 5.1. continuedSecondary sedimentation

Peak factors based on nominal flow

Dry weather peak 1,2

Wet weather peak 1,2

Solid flux based on nominal flow (kg/m²/h)

Winter 4,6

Summer 5,5

Surface loadings based on mean flow (m/h)

Winter 0,61

Summer 0,95

Diameter of tank (m) 38

Volume of tank (m³) 3 740

Side wall depth (m) 3,25

2. DESIGN FEATURES

The design of this plant provided for primary and secondary screens, degritting, primary sedimentation using 8 rectangular tanks, flow balancing tanks and 3 activated sludge modules in parallel, each designed as a Phoredox process. A diagrammatic sketch of the plant is shown in Fig. 5.1. and the details of the design data are given in Table 5.1.

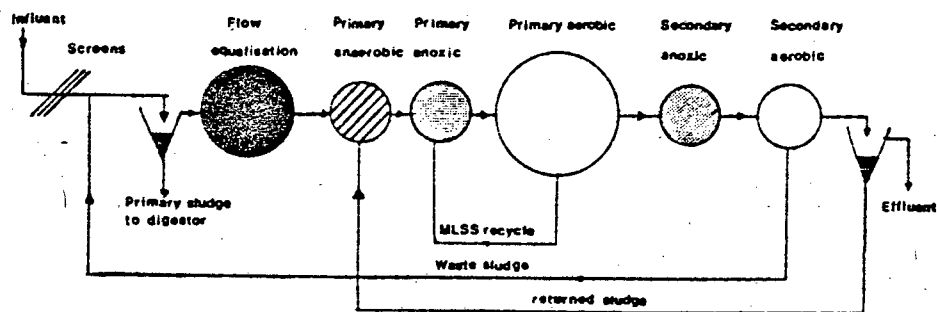


Fig. 5.1. Northern Works basin configuration

Some design features are discussed below, and, in particular, are compared with those at Goudkoppie.

- (i) Primary sedimentation The tanks at Northern Works are rectangular whereas those at Goudkoppie are circular. The design upward velocity in these tanks on both plants is 1,5 m/h. However both plants were hydraulically underloaded during the experimental period and the upward velocities on both plants were lower.

- (ii) Balancing tanks : the design includes three balancing tanks, one for each of the three modules, not interconnected in any way i.e. they are completely separate units, each balancing tank feeding to a specific module. (This differs from the design of the balancing facility at Goudkoppie Works where one balancing tank serves all four modules)). At Northern Works the balancing tanks are designed as long rectangular basins with the floors sloping sharply to the longitudinal axis of the tank, to a channel running along the same axis of the tank. Any solids deposited while the tank is full are flushed out when the balancing tank empties.
- (iii) Process units : The process units are identical and based on the Phoredox process. Each consists of five completely mixed reactors in sequence : anaerobic - primary anoxic - primary aerobic - secondary anoxic - secondary aerobic . The important features in the process design are very similar to those listed for the Goudkoppie plant with the following exceptions :
- * The maximum MLSS recycle ratio is less than at Goudkoppie, i.e. 10 : 1 , compared with 16 : 1 of Goudkoppie.
 - * In the design of the oxygenation requirements at the Northern Works, cognizance was taken of the BOD utilised during denitrification, with the result that the overall available oxygenation capacity, per influent BOD in the primary aeration basin is less than that at the Goudkoppie plant i.e. $900/1\ 325 = 0,68$ as against $1\ 320/1\ 410 = 0,94$ kg O/kg BOD/day
- (iv) Final clarifiers : There are three per module, each with a diameter of 38 m and a side wall depth of 3,25 m .

- Sludge is removed by means of 10 syphons located on a rotating bridge spanning the diameter of the clarifier. The mixed liquor enters the clarifier at the bottom, in the centre, in a tangential flow pattern. In the initial design no baffles, to reduce the kinetic energy of the incoming flow, were provided ; considerable shortcircuiting was observed . In order to counter this, horizontal and vertical baffles were fitted which effectively have reduced this problem.

(v) Flow metering : This plant has flow metering facilities at the following points :

- * Feed flow to the process from the balancing tank is automatically controlled to a selected flow rate by a Rotork valve activated by a magnetic flow meter. Hence, provided the balancing tank does not empty over the day, an approximately fixed volume of flow should pass into the process each day.
- * Sludge wastage, to control sludge age, is wasted directly from the secondary aeration basin. The volume wasted each day is metered by means of a magnetic flow meter.

No metering facility is provided in the returned sludge line consequently underflow sludge flow rate has to be measured indirectly by means of a solids mass balance across the secondary clarifier.

3. PLANT PERFORMANCE EVALUATION

3.1. Objectives

Once the plant operated under stable conditions for a period of approximately 3 sludge ages, it was planned to carry out a detailed investigation of the plant performance similar to that of the Goudkoppie plant :

- * An indepth evaluation of the performance of each zone
- * Estimation of oxygen consumption rates under different conditions in the primary aerobic reactor.
- * Overall appraisal of nitrification / denitrification efficiency
- * Assessment of phosphorus removal capability
- * Comparison of actual performance data with predictions obtained making use of the mathematical model developed by the University of Cape Town.

3.2. Conditions of test

- (i) Flow to the experimental module was fixed at a constant 40 Ml/d (compared with design flow 50 Ml/d) - the total flow to the plant was still below the design flow.
- (ii) Sludge age was monitored at 27 days which, as for the Goudkoppie plant, was presumed to give rise to a rapid settling sludge.
- (iii) Physical details of the process layout are given in Fig 5.2.(a) and Table 5.2.

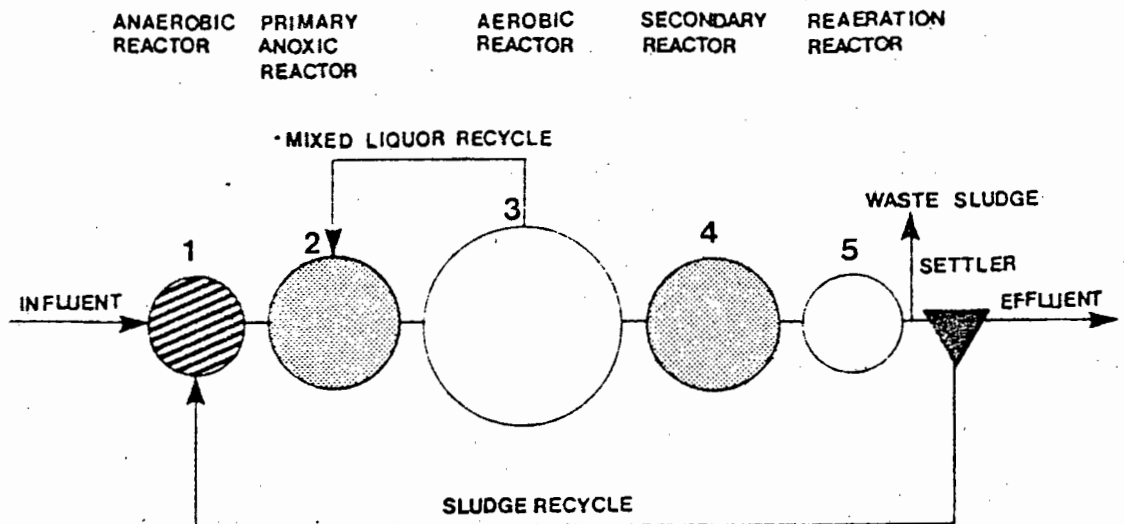


Fig. 5.2.(a) The process layout at Northern Works

Table 5.2.

Physical details of the process layout at the Northern Works

	Anaerobic	Primary Anoxic	Primary Aerobic	Secondary Anoxic	Secondary Aerobic
Reactor number	1	2	3	4	5
Volume (m ³)	2 230	4 575	15 300	4 575	2 500
Nominal retention time (d)	0,056	0,114	0,383	0,114	0,063
Mass fraction	0,076	0,157	0,524	0,157	0,086

- Note
1. MLSS recycled from reactor 4 to reactor 2, ratio 4:1
 2. Sludge return recycled from settler to reactor 1, ratio 0,4:1

- (iv) Dissolved oxygen concentration in both aerobic reactors was maintained in excess of 2 mg/l . Under the existing conditions of low flow and load this concentration could be achieved with only half the total number of aerators operational, as shown in Fig. 5.2.(b) .

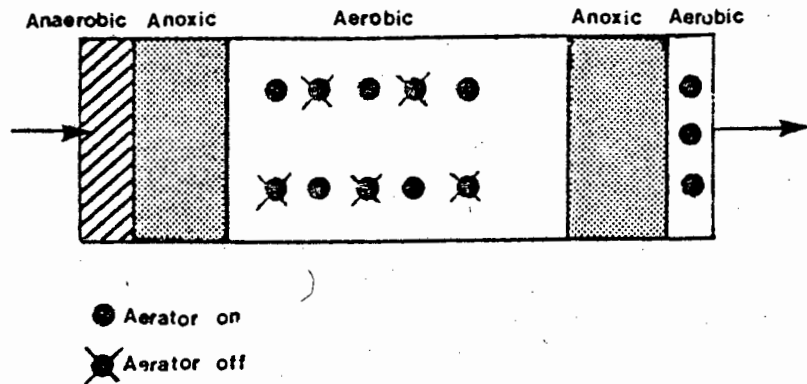


Fig. 5.2.(b) Operational aerators during experimentation

- (v) Sampling procedures during this investigation were the same as those in the Goudkoppie experiment, with one important exception : The filter paper, when filtering the grab sample from each reactor, was Whatman 542 and not the coarse paper used in the Goudkoppie tests - it was evident from the Goudkoppie experimental results on the effluent that when fine filter paper was used, the fine filter filtrate corresponded more closely to the soluble COD than the filtrate of the rough filter paper.
- (vi) Measurements of the oxygen utilisation rate were done by means of a respirometer identical to the type used at Goudkoppie (see Appendix A4) . During the experimental period the respirometers were positioned in the following positions ; between the first and second aerator bridges, in line with the third aerator bridge, and between the fourth and fifth bridges .

3.3. Performance data

Sampling of the influent, the effluent and the contents of each reactor was carried out at two hourly intervals over a period of four days, and samples analysed for COD, TKN, ammonia, nitrate and phosphate. The oxygen utilisation rates were measured at the same time. Unfortunately the results of three of the seven days had to be discarded due to the malfunction of one of the respirometers. The results of the remaining data at each time point are listed in Appendix A8. These were averaged and are listed in Table 5.3. Plots of the average values are shown in Figs 5.3. - 5.10.

- Fig. 5.3. Unfiltered COD concentration in the feed and filtered effluent and reactor contents
- Fig. 5.4. Unfiltered TKN concentration in the feed and filtered effluent
- Fig. 5.5. Total oxygen utilisation rate in the main aeration basin
- Fig. 5.6. Total phosphorus concentration in the unfiltered feed and filtered effluent
- Fig. 5.7. Soluble COD concentration in filtered samples in each basin
- Fig. 5.8. Soluble ammonia concentration in filtered samples in each basin
- Fig. 5.9. Nitrate concentration in filtered samples in each basin
- Fig. 5.10. Soluble phosphorus concentration in filtered samples

The reliability of the data was checked, as discussed earlier, by conducting mass balances over the reactor on COD and nitrogen.

(For details on the procedure see Appendix A9) .

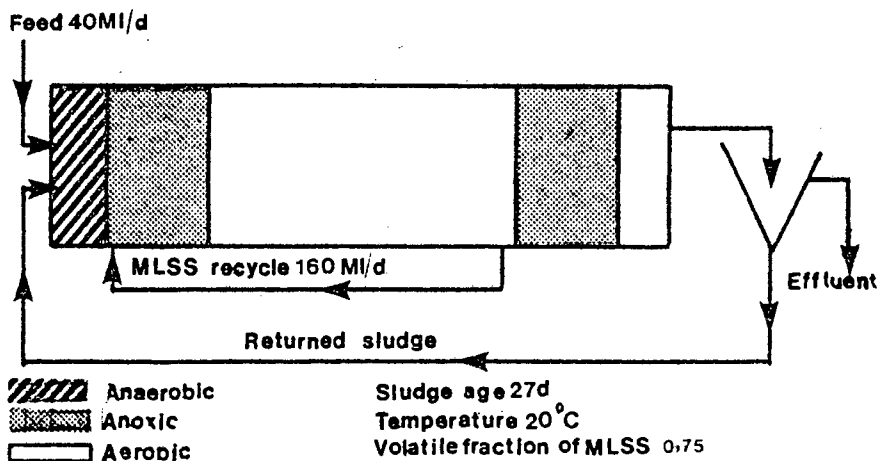
Mass balances achieved are $M_{TKN} = 99$ percent recovery

$M_{COD} = 90$ percent recovery

These mass balances are very good indicating that the data are acceptable.

Table 5.3.

The average performance of the Northern Works during the experimental period



	Feed	Anaerobic	Primary Anoxic	Primary Aerobic	Secondary Anoxic	Secondary Aerobic	Effluent
Total COD	240						46
Soluble COD		72	45	42	34	38	38
S_{bs}	33						
TKN as N	39						2,1
Ammonia as N	24	16	6,8	0,7	0,5	0,0	0,0
Nitrate as N		4,3	10	16	16	16	17
Total P	9,7						7,6
Soluble P	6,5	6,9	6,6	6,5	6,6	6,5	6,5
Suspended solids				1 440			
Oxygen consumption mg/l/h				24,1			

Unless otherwise stated all results expressed as mg/l

- * All reactor and effluent samples were filtered through Whatman 542 filter paper
- ** Reactor samples were filtered and tested by means of the orthophosphate test
- *** S_{bs} = readily biodegradable COD concentration mg COD/l

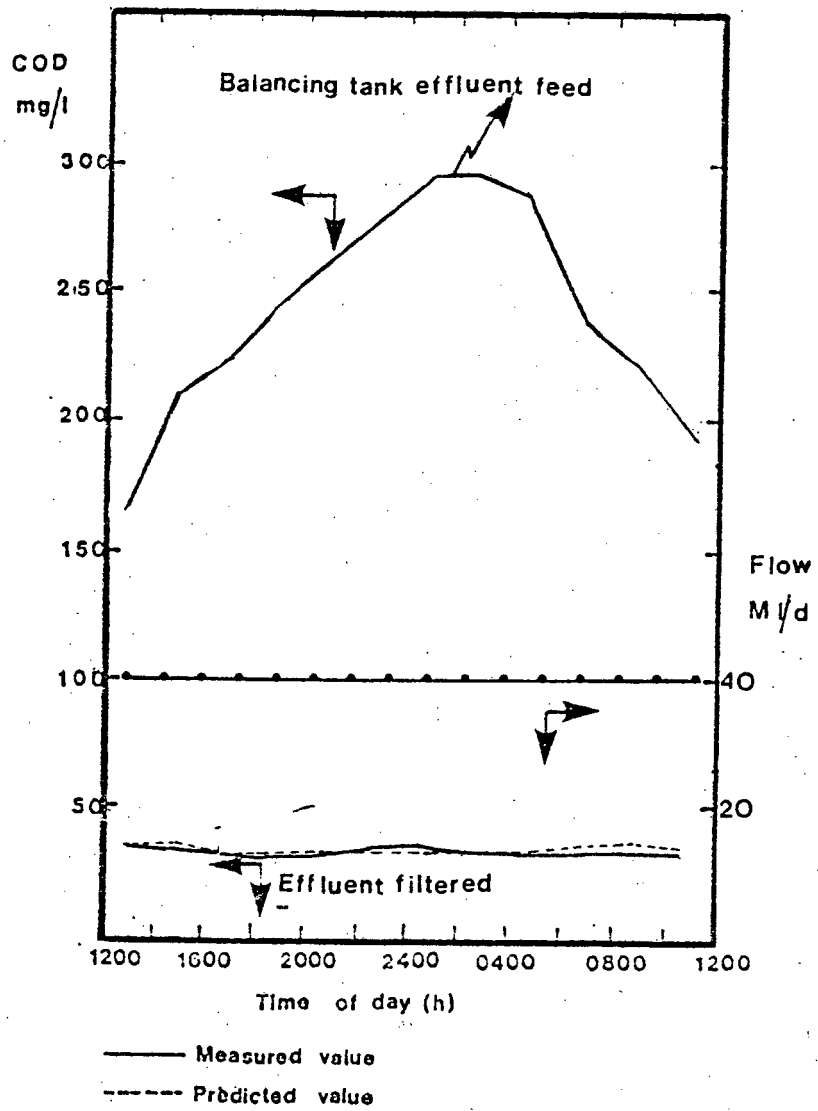


Fig. 5.3. Northern Works

Total COD concentration in the feed and filtered COD in the effluent together with predicted values. In addition the flow to the process during the test period is shown.

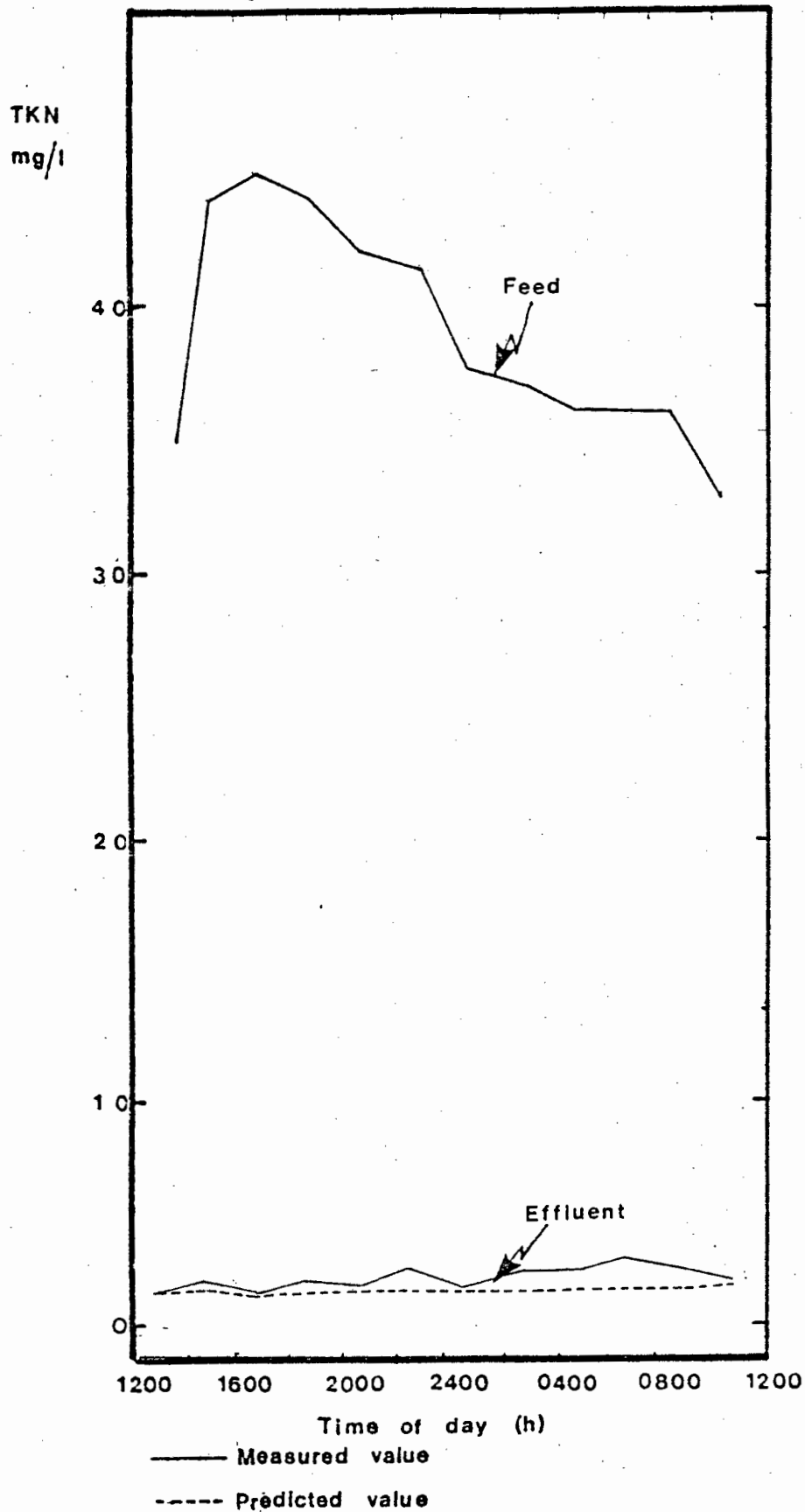


Fig. 5.4. Northern Works
TKN concentration in the feed and filtered TKN in the effluent together with predicted values for the effluent

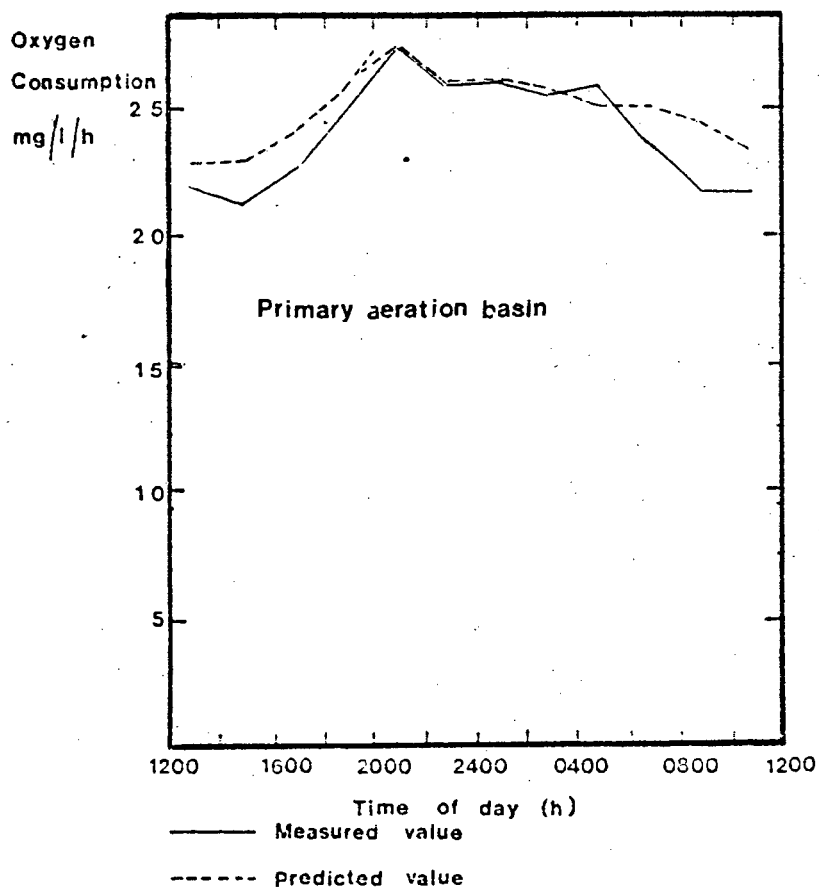


Fig. 5.5. Northern Works
Predicted and measured oxygen consumption
rates in the primary aeration basin.

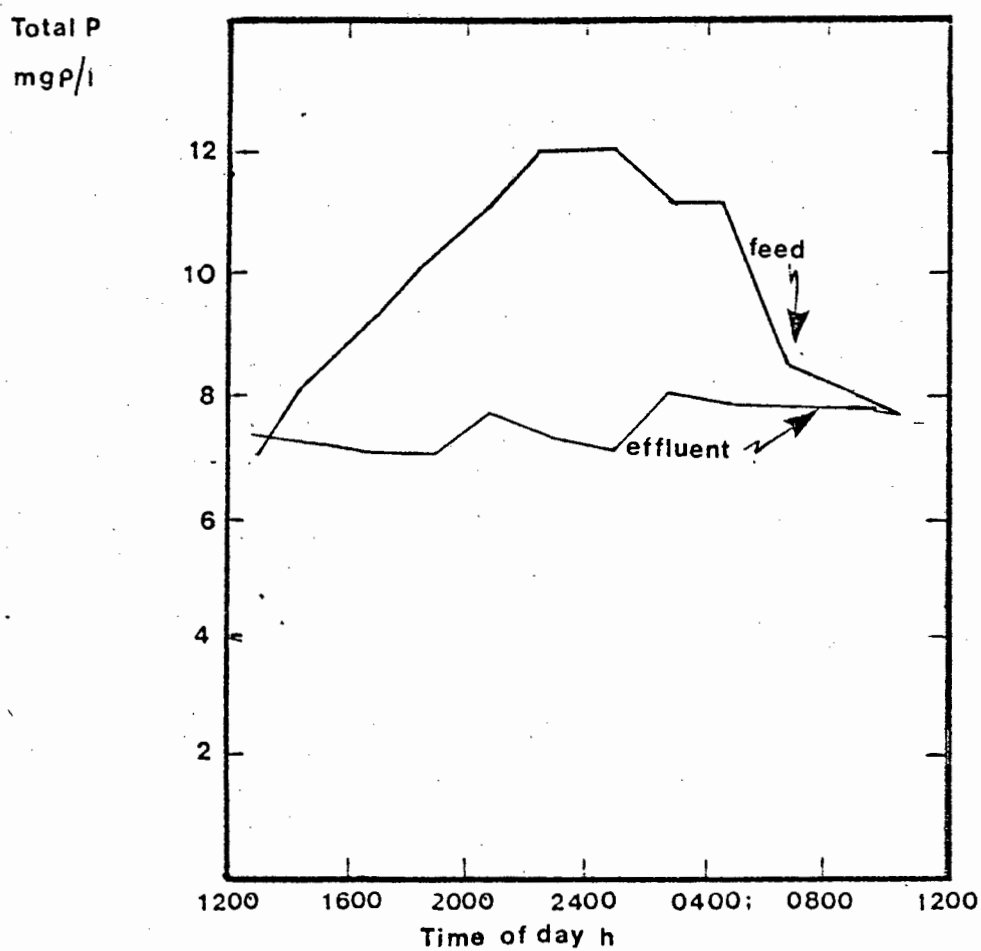
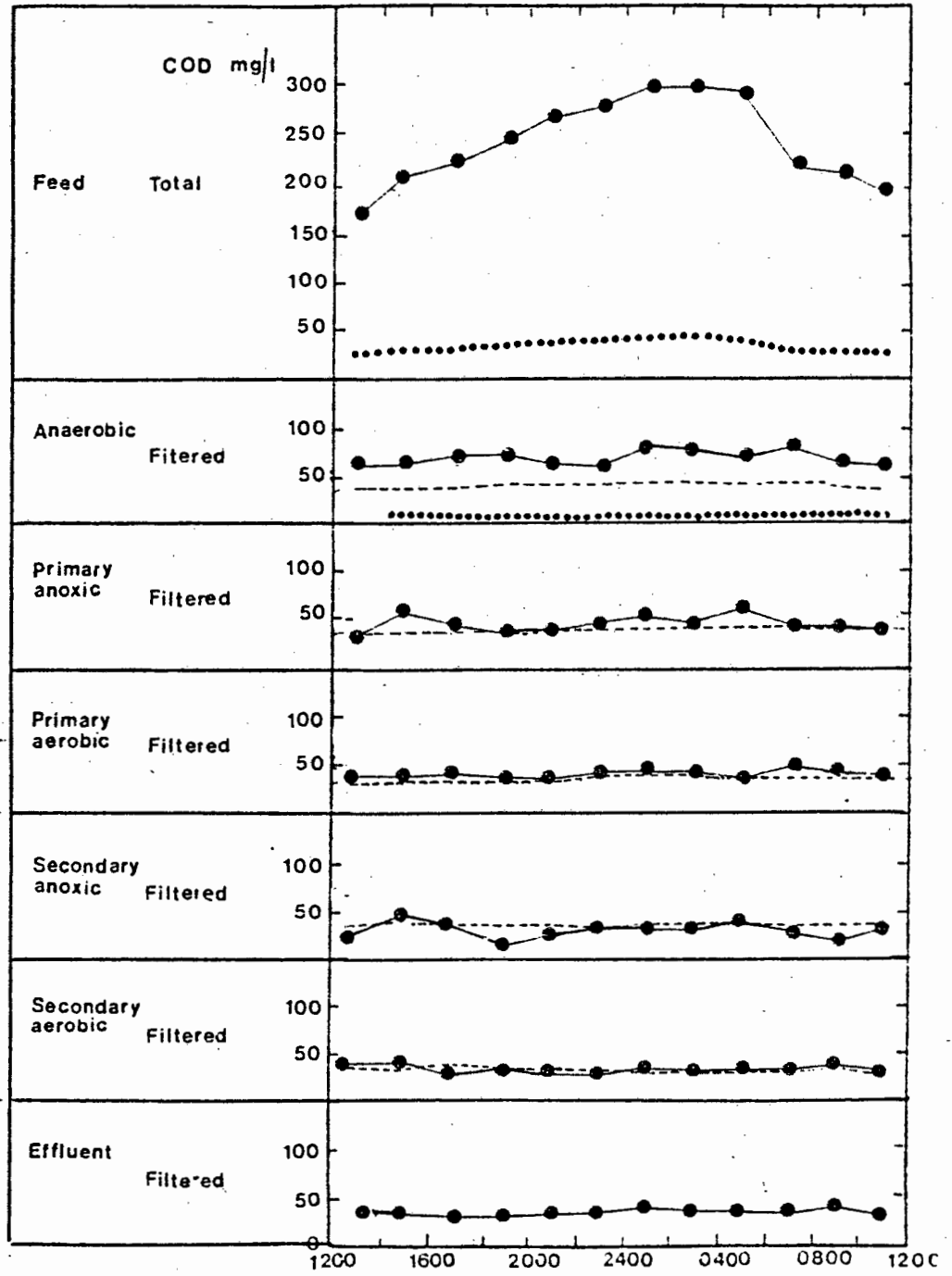


Fig. 5.6. Northern Works

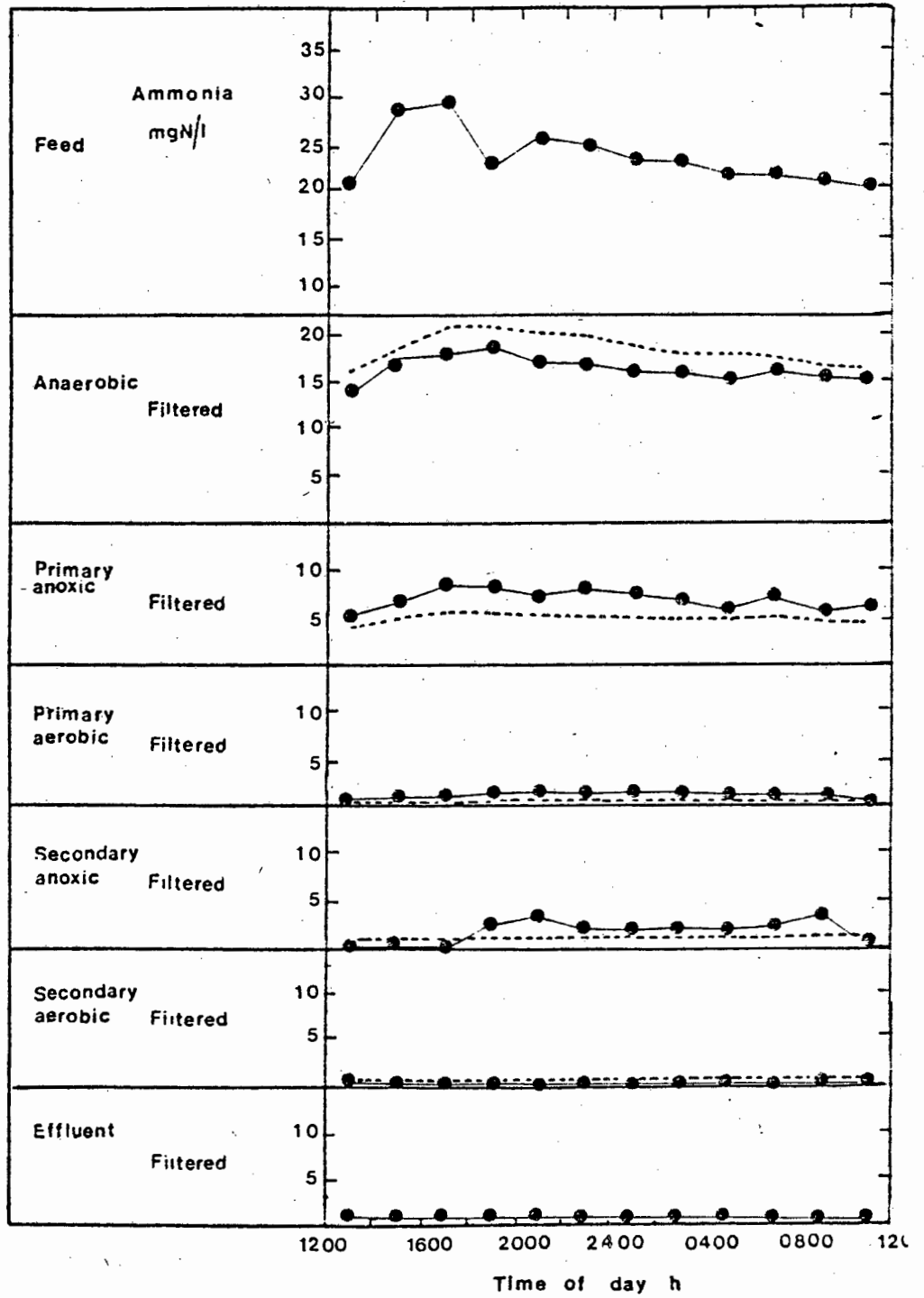
Total phosphorus concentration in the feed and filtered effluent.



●●●●●●●● measured values
 - - - - - predicted values
 predicted values of readily biodegradable COD

Fig. 5.7. Northern Works

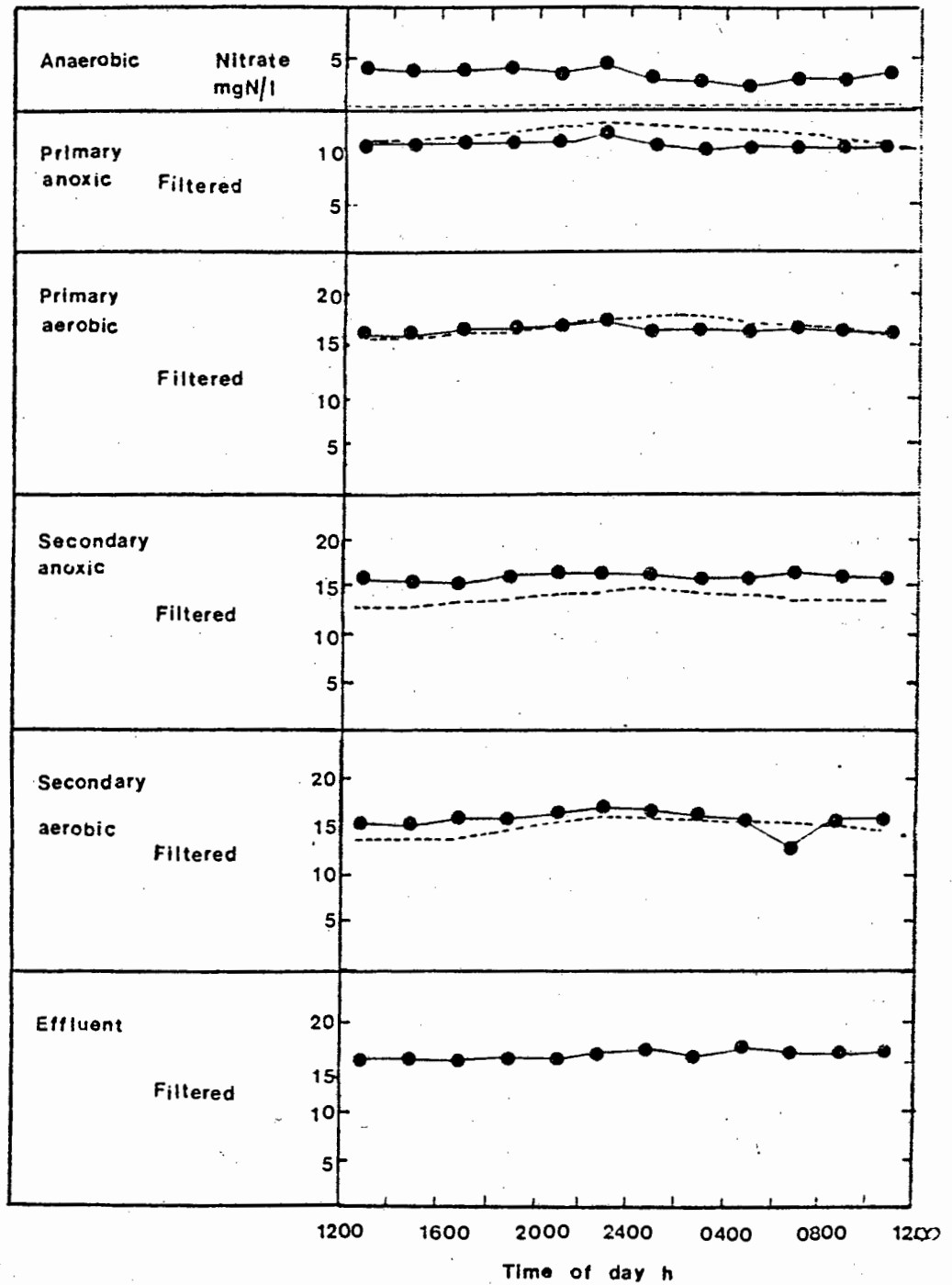
Measured filtered COD and predicted soluble COD concentrations in each reactor of the system



●●●●● measured values
 - - - - - predicted values

Fig. 5.8. Northern Works

Measured filtered and predicted ammonia concentrations in each reactor of the system



●●●●●●●● measured values
 - - - - - predicted values

Fig. 5.9. Northern Works

Measured filtered and predicted nitrate
 concentrations in each reactor of the system

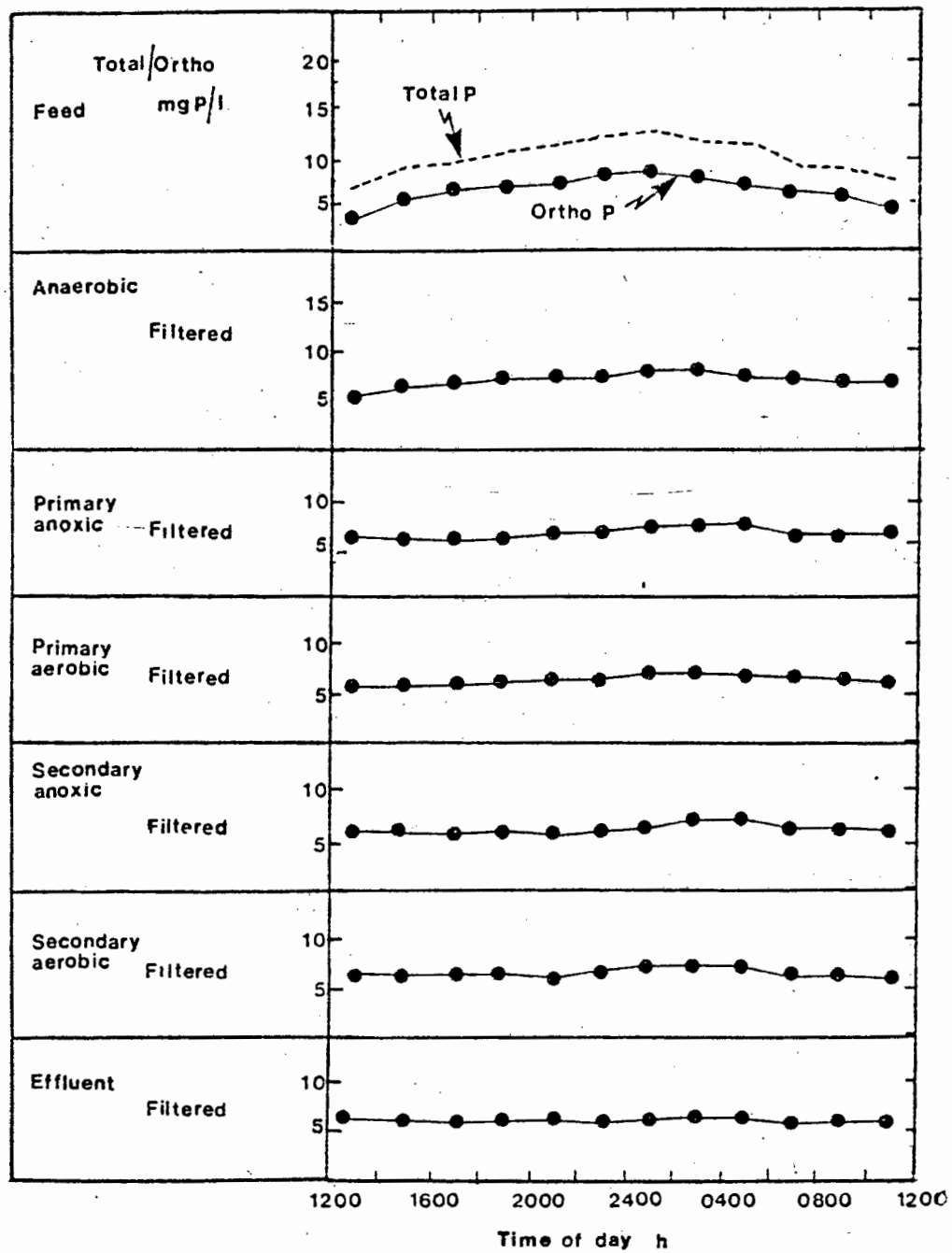


Fig. 5.10. Northern Works
 Unfiltered total phosphorus concentration in the influent and filtered orthophosphorus concentration in each reactor of the system

4. EVALUATION OF EXPERIMENTAL PLANT RESPONSE

The response of particular interest is that concerned with nitrogen and phosphorus removal. Removal of nitrogen and phosphorus over the test period was relatively poor when one considers that the plant was designed specifically to accomplish the elimination of these two parameters. The total nitrogen concentration reduced from 39 mg N/l in the feed to 19,1 mg N/l in the effluent, a removal of only 19,9 mg N/l, and the total phosphorus from 9,7 mg P/l to 7,6 mg P/l i.e. 2,1 mg P/l only (see Table 5.3.)

The two most likely reasons for the poor nitrogen removal are ; the TKN/COD ratio, which was very high, equal to 0,16 (Table 5.3.) and, the fraction of easily biodegradable COD which was very low, equal to 0,16 mg COD/mg biodegradable COD (Table 5.6.) The TKN/COD ratio of the feed (equalised settled sewage) during the test period of 0,16 was considerably higher than that normally expected from a settled sewage i.e. 0,10 approximately. The nitrification capacity in consequence was high per unit COD input. In contrast, the easily biodegradable fraction was low (33 mg/l) and this fraction contributes significantly to the denitrification capacity of the system. The consequence of the combination of high nitrification and low denitrification capacities was the high nitrate in the effluent. Due to the high nitrate concentration in the underflow recycle (17,0 mg N/l) , there would have been no readily biodegradable COD (S_{bsa}) in the anaerobic reactor, because it would have the capacity to remove $8,6 \cdot 17,0 \cdot 0,4 = 58,5$ mg COD/l influent readily biodegradable COD, and the concentration of readily biodegradable COD in the influent is only 33 mg/l (Table 5.3.) Consequently there would have been no S_{bsa} in the anaerobic reactor and the phosphorus removal would be minimal. (Note when a primary

anoxic reactor is present α (equation 15) has a minimal value of $\pm 0,06$, according to Siebritz, Ekama and Marais (1982), as against 0,025 for a pure aerobic system.)

Evaluating the plant design, it is clear that with the influent as received, it is not possible, in terms of the excess phosphorus removal theory, for this plant to remove phosphorus by the excess phosphorus removal mechanisms under any circumstances, because the influent readily biodegradable COD concentration (S_{bs}) is too low; even if complete denitrification was attainable the concentration of S_{bsa} in the anaerobic reactor would be $S_{bs} / (1 + S) = 33 / (1 + 0,4) = 24 \text{ mg } S_{bsa}/\text{l}$, which is less than the 25 mg/l necessary to initiate excess phosphorus removal. With nitrates present in the recycle the S_{bsa} in the anaerobic reactor reduces to zero and no possibility exists for excess removal.

With regard to nitrogen removal, nitrate was present throughout the system (see Table 5.3.). This implies that the anoxic reactors were loaded with nitrate above their denitrification capacities. Under these circumstances it has been shown by van Haandel et al (1981) that the maximum removal of nitrate will be achieved if the secondary anoxic zone is excluded from the system, and its volume combined with that of the primary anoxic reactor, because the latter is much more efficient than the former. This conclusion is verified by the experimental results; Doing a mass balance of nitrate over the primary and secondary anoxic zones respectively indicates the removal in the primary anoxic zone is 16,8 mg N/l and that in the secondary anoxic zone is nil. Clearly from the above analysis the process design is not optimal for this influent.

Although the design is not optimal for the influent received by the plant, it may be that for the assumed characteristics (Table 5.1.), the plant would have been adequate. To test this, the general model was used to simulate the plant performance taking as sewage inputs the COD and TKN assumed in the original design (Table 5.1.) (It is assumed that $BOD : COD = 1 : 2$) . The results of this simulation are summarised in Table 5.4.

Table 5.4.

The predicted performance of the Northern Works under design load conditions, treating settled sewage.

	Feed	Anaerobic	Primary Anoxic	Primary Aerobic	Secondary Anoxic	Secondary Aerobic
COD	530	148	83	90	50	80
TKN as N	30	21,3	8,3	3,6	6,4	3,6
Ammonia as N		18,1	5,7	0,8	3,9	0,2
Nitrate as N		nil	0,1	3,0	nil	2,0
Oxygen consumption rate mg/l/h				44,2		
MLSS				4570		

Evidently the nitrate would have been near zero in the effluent and consequently the nitrate in the underflow recycle would have been minimal if the normal situation existed in that $S_{bsi} \approx 0,24 S_{ti}$ (where S_{ti} = settled sewage), the S_{bi} would have equaled 120, and $S_{bsa} \approx 50$. Under these conditions the removal potential of phosphorus would have been approximately 10 mg/l (in terms of the excess phosphorus removal theory) and an effluent phosphorus of near zero should have been achievable.

5. MODEL PREDICTIONS AND COMPARISON WITH PERFORMANCE DATA

5.1. Sewage characteristics

To test the validity of the general model to describe the behaviour of the Northern Works process, it was necessary first to estimate the characteristics of the influent waste water, in this case, the discharge from the balancing tank.

(i) Soluble unbiodegradable COD fraction , f_{us}

This fraction, as shown before, is estimated from the filtered effluent from a plant with a long sludge age. The average filtered effluent COD in this case was 38 mg/l ; the total influent COD concentration was 240 mg/l , hence

$$f_{us} = 38 / 240 = 0,16 \text{ mg COD/mg COD}$$

(ii) Unbiodegradable particulate COD fraction , f_{up}

For the reasons already discussed in the investigation of the Goudkoppie plant, the value of this constant was estimated by trial and error, curve fitting procedure to give

$f_{up} = 0,01 \text{ mg VSS/mg influent COD}$ This value is considerably lower than that found for Goudkoppie ($f_{up} = 0,12$) ; possible reasons for the difference are :

- * The primary sedimentation tanks at Northern Works were particularly efficient, whereas those at Goudkoppie were not. The efficiency of the primary tanks can be seen from Table 5.5. , where various parameters are compared before and after the primary settling tank. (The values listed in Table 5.5. are the annual average values for the first year of operation). Principally for this reason the unbiodegradable particulate fraction entering the process would be smaller than the values found at Goudkoppie Works.

Table 5.5.

Average annual analysis of raw sewage and settled sewage

Test	Raw Sewage	Settled Sewage
COD	510	330
TKN as N	37	34
Ammonia as N	19	22
Total phosphorus	8,4	7,3
Suspended solids	190	66
TKN/COD	0,073	0,103

(iii) Fraction of TKN in the ammonia form, f_{nu}

This fraction was found by analytical measurement of the TKN and ammonia concentrations in the influent to the process and found to be approximately $24/39 = 0,62$ mg N/mg N

(iv) Readily biodegradable COD fraction of the feed, f_{bs}

The procedure for measurement of this fraction in raw and settled sewage and balancing tank effluent is set out in Appendix A7, and the values measured are listed in Table 5.6.

Table 5.6.

Estimation of the readily biodegradable fraction of raw sewage, settled sewage and balancing tank effluent over the test period

	Symbol	Raw Sewage	Settled Sewage	Balancing tank effluent
Fraction of readily biodegradable COD (mg COD/mg biodegradable COD)	f_{bs}	0,15	0,23	0,15
Concentration of readily biodegradable COD (mg/l)	S_{bs}	57	66	33

Note: In order to express as S_{bs} the following constants were used ;

Raw sewage $f_{us} = 0,12$ $f_{up} = 0,09$

Settled sewage $f_{us} = 0,12$ $f_{up} = 0,01$

Balancing tank effluent $f_{us} = 0,12$ $f_{up} = 0,01$

The increase in the fraction of readily biodegradable COD in the primary sedimentation tank effluent, compared to that in raw sewage is to be expected since a considerable mass of particulate COD is removed in the primary sedimentation tanks, whereas virtually no readily biodegradable COD is removed. The decrease in readily biodegradable COD concentration across the balancing tank followed the same trend as across the tank at the Goudkoppie Works where the tank was operated not to flush the solids. At Northern Works, the balancing tanks are not emptied, so that sludge accumulates and fermentation will similarly reduce the readily biodegradable COD concentration. The following remarks are important re the fraction of readily biodegradable COD in the raw influent, the settled and balancing tank effluents :

- * The readily biodegradable fraction of the raw influent was correspondingly lower than that at the Goudkoppie Works i.e. 0,15 (mg COD/mg biodegradable COD) as against 0,19 (mg COD/mg biodegradable COD) at Goudkoppie Works. One explanation for this decrease is that excess sludge generated at the Alexandra plant is returned to the sewer and forms part of the raw sewage being treated at Northern Works ; any active mass content in raw sewage will cause oxygen, entering the sewer, to be utilised immediately, by the active material, to metabolise the readily biodegradable COD at the rate 1 mg O₂ metabolises approximately 3 mg readily biodegradable COD .

- * The increase in the readily biodegradable fraction across the primary sedimentation tank again is to be expected because particulate COD is removed but not soluble readily biodegradable COD .
- * At the Northern Works, over a day's operation, the balancing tanks did not empty sufficiently to flush the sludge accumulating on the sloping floor of the tank, so that very likely fermentation in these tanks caused the fraction of readily biodegradable COD to be reduced during the tank holdup. To check if settlement and COD reduction in fact did occur in the tank, the COD concentration at the inlet and outlet, and the flow, were monitored every two hours for one week. Using this data (see Appendix A10) , the loss in COD mass per day was calculated and is given in Table 5.7.

Table 5.7.

The mass of COD lost across the balancing tank

	kg COD/d
Average mass of COD entering the balancing tank	13 834
Average mass of COD leaving balancing tank	11 450
Loss of COD	2 384

From Table 5.7. a considerable loss (± 17 percent) occurred in the waste flow during its passage through the tank - a figure very similar to that obtained at Goudkoppie Works (± 20 percent) . Doubtless, the accumulation of this COD on the floor caused the reduction in the readily biodegradable COD concentration.

A summary of the values of these and other constants necessary for simulation is set out in Tables 5.8.(a) and 5.8.(b) .

Table 5.8.(a)

Kinetic parameters specific to the balancing tank effluent at the time of this investigation at Northern Works

Name	Symbol	Units	Value
Fraction of soluble unbiodegradable COD	f_{us}	mg COD/mg COD	0,12
Fraction of particulate unbiodegradable COD	f_{up}	mg VSS/mg COD	0,01
Fraction of easily biodegradable COD	f_{bs}	mg COD/mg bio-degradable COD	0,16
Easily biodegradable COD concentration	S_{bs}	mg COD/l	32
Fraction of TKN of feed in ammonia form	f_{nu}	mg N/mg N	0,62
Specific growth rate for nitrifiers	μ_{mn}	d^{-1}	0,50

Table 5.8.(b)

Kinetic constants invariant for different waste flows

Name	Symbol	Units	Value
Substrate adsorption rate coefficient	K_{a20}	1/mg VSS/d	0,135
Maximum specific growth rate coefficient for heterotrophic organisms utilising stored particulate substrate	K_{mp20}	mg COD/mg VSS/d	3,00
Maximum specific growth rate coefficient for heterotrophic organisms utilising soluble substrate	K_{ms20}	mg COD/mg VSS/d	20,00
Saturation coefficient for heterotrophic cell synthesis from stored particulate substrate	K_{sp20}	mg COD/l	0,04
Saturation coefficient for heterotrophic cell synthesis from soluble substrate	K_{ss20}	mg COD/l	5,0
Specific endogenous respiration rate for heterotrophic organisms	b_{h20}	/d	0,62
Yield coefficient for heterotrophic organisms	Y_{h20}	mg VSS/mg COD	0,45
Unbiodegradable fraction of the active organisms	f	mg VSS/mg VSS	0,08
Nitrogen content of the sludge	f_n	mg N/mg VSS	0,10
COD equivalent of volatile solids	P	mg COD/mg VSS	1,48
Fraction of nitrogen released through endogenous respiration in the organic form	f_{oe}	-	1,00
Fraction of nitrogen required for cell synthesis in the organic nitrogen form	f_{os}	mg N/mg VSS	0,00
Maximum fraction of stored solids with respect to active mass	f_{ma}	mg VSS/mg VSS	1,00
Organic nitrogen to ammonia conversion rate coefficient	K_{r21}	/mg VSS/d	0,015
General parameter for pH dependency coefficients	adsorption on heterotrophic organisms	h_a	1,200
	synthesis by heterotrophic organisms	h_s	1,100
	endogenous respiration by heterotrophic organisms	h_e	1,029
Saturation coefficient for <u>Nitrosomonas</u>	K_{n20}	mg N/l	1,00
Specific endogenous rate for nitrifiers	b_{n20}	/d	0,04
Yield coefficient for nitrifiers	Y_{n20}	mg VASS/mg N	0,10
General parameter for pH dependency coefficients	synthesis by <u>Nitrosomonas</u>	h_s	1,123
	endogenous respiration by <u>Nitrosomonas</u>	h_e	1,029

5.2. - Simulated results

The simulated results of the Northern Works were obtained using the constants as listed in Tables 5.8.(a) and 5.8.(b) and the input data :- feed flow rate, COD and TKN concentrations as listed in Table 5.9.

Table 5.9.

Input flow and COD and TKN concentrations used in model predictions

Time	Flow Ml/d	COD mg/l	TKN mg N/l	Time	Flow Ml/d	COD mg/l	TKN mg N/l
0700	40,0	238	36	2000	40,0	252	43
0800	40,0	230	36	2100	40,0	260	42
0900	40,0	220	36	2200	40,0	265	41
1000	40,0	208	34	2300	40,0	270	41
1100	40,0	193	33	2400	40,0	280	40
1200	40,0	183	34	0100	40,0	293	48
1300	40,0	170	35	0200	40,0	293	38
1400	40,0	190	40	0300	40,0	293	37
1500	40,0	210	44	0400	40,0	287	36
1600	40,0	217	44	0500	40,0	283	35
1700	40,0	223	45	0600	40,0	260	36
1800	40,0	235	45	0700	40,0	238	36
1900	40,0	245	44				

The simulated results are listed in tabular form in Tables 5.10. , 5.11. and 5.12. , and are shown plotted in Figs 5.3. to 5.5. and 5.7. to 5.9:

Table 5.10.

Comparison of predicted and experimental COD concentrations
in each zone of the Northern Works Plant

Time of Day	Feed		POSITION WITHIN PLANT										Effluent	
			Anaerobic		Primary Anoxic		Primary Aeration		Secondary Anoxic		Secondary Aeration			
h	Total	Soluble	Soluble	Soluble	Soluble	Soluble	Soluble	Soluble	Soluble	Soluble	Soluble	Total	Soluble	
			E	P	E	P	E	P	E	P	E	P		
1500	170	110	63	32	38	35	44	36	32	37	41	38	44	38
1600	210	130	65	34	52	34	40	35	41	35	43	37	45	38
1700	220	150	70	36	42	34	41	34	35	35	33	36	38	35
1900	240	160	77	38	39	35	38	34	29	35	35	35	41	34
2100	260	170	71	40	40	35	40	35	33	35	34	35	49	34
2300	270	180	69	42	48	36	45	35	34	35	35	35	45	39
0100	290	190	83	44	51	37	44	36	35	35	39	35	49	43
0300	290	200	81	46	47	38	43	37	36	36	38	36	48	35
0500	280	180	75	45	54	39	38	38	39	37	38	37	44	36
0700	240	170	79	41	44	39	48	38	31	38	38	37	50	36
0900	220	160	69	38	44	38	45	38	29	38	40	38	57	39
1100	190	150	63	35	41	37	43	38	31	38	39	38	45	36
Ave.	240	160	72	39	45	36	42	36	34	36	38	36	46	37

- Note:
1. All results expressed as mg/l
 2. E denotes experimental value
 3. P denotes predicted value
 4. Reactor and effluent samples were filtered through Whatman 542 filter paper

Table 5.11.

Comparison of predicted and experimental ammonia concentrations
in each zone of the Northern Works Plant

Time of Day	Feed		POSITION WITHIN PLANT										Effluent	
			Anaerobic		Primary Anoxic		Primary Aeration		Secondary Anoxic		Secondary Aeration			
	TKN	NH ₃	Soluble	Soluble	Soluble	Soluble	Soluble	Soluble	Soluble	Soluble	TKN	NH ₃		
			E	P	E	P	E	P	E	P	E	P		
1300	35	21	14	16	5,9	4,8	0,6	0,3	0,0	1,7	0,0	0,2	2,0	0,0
1500	44	28	18	19	6,6	5,4	0,7	0,4	0,0	1,8	0,0	0,2	2,2	0,0
1700	45	29	18	22	7,7	6,1	0,7	0,5	0,0	1,9	0,0	0,2	2,1	0,0
1900	44	23	19	21	7,6	6,3	0,8	0,5	0,6	1,9	0,0	0,2	2,1	0,0
2100	42	26	17	20	7,0	6,1	1,0	0,4	0,9	1,9	0,0	0,2	1,9	0,0
2300	41	25	17	20	7,6	5,4	0,9	0,4	0,6	1,9	0,0	0,2	2,2	0,0
0100	38	23	16	19	7,4	5,7	1,4	0,4	1,0	1,9	0,1	0,2	1,7	0,0
0300	37	23	16	18	6,9	5,3	1,1	0,4	0,6	1,8	0,0	0,2	2,3	0,1
0500	36	22	15	18	6,5	5,2	0,8	0,3	0,5	1,8	0,0	0,2	2,2	0,2
0700	36	22	16	17	6,6	5,2	0,4	0,3	0,6	1,8	0,1	0,2	2,6	0,0
0900	36	21	15	16	6,4	5,0	0,5	0,3	0,9	1,8	0,0	0,2	2,5	0,0
1100	33	20	15	16	6,0	4,8	0,2	0,3	0,0	1,8	0,1	0,2	2,3	0,0
Ave.	39	24	16	18	6,9	5,5	0,8	0,4	0,5	1,8	0,0	0,2	2,1	0,0

- Note:
1. All results expressed as mg N/l
 2. E denotes experimental value
 3. P denotes predicted value
 4. Reactor and effluent samples were filtered through Whatman 542 filter paper

Table 5.12.

Comparison of predicted and experimental nitrate concentrations
in each zone of the Northern Works Plant

Time of Day	Feed	POSITION WITHIN PLANT					Effluent
		Anaerobic	Primary Anoxic	Primary Aeration	Secondary Anoxic	Secondary Aeration	
		Soluble	Soluble	Soluble	Soluble	Soluble	
h		E P	E P	E P	E P	E P	
1300	0	4,4 0,4	10 10,6	16 15,5	16 12,8	16 14,7	16
1500	0	4,4 0,3	10 10,7	16 15,7	16 12,7	15 14,5	16
1700	0	4,6 0,1	11 11,1	17 16,3	16 13,1	16 14,7	16
1900	0	4,7 0,1	11 11,6	17 17,2	16 13,7	16 15,2	16
2100	0	4,2 0,1	10 12,0	17 17,8	16 14,3	16 15,9	16
2300	0	5,0 0,1	11 12,2	14 17,9	16 14,7	17 16,4	17
0100	0	4,0 0,1	10 12,2	17 17,8	16 14,9	17 16,7	17
0300	0	3,9 0,1	9,9 11,9	16 17,5	15 14,7	17 16,6	17
0500	0	3,6 0,1	10 11,5	16 16,9	16 14,3	16 15,3	17
0700	0	4,4 0,1	10 11,2	17 16,4	17 13,8	18 15,8	17
0900	0	4,0 0,1	10 10,9	17 16,0	16 13,4	17 15,3	16
1100	0	4,3 0,2	11 10,7	16 15,7	16 13,8	16 14,9	16
Ave.	0	4,3 0,1	10 11,4	16 15,7	16 13,9	16 15,6	16

- Note:
1. All results expressed as mg N/l
 2. E denotes experimental value
 3. P denotes predicted value
 4. Reactor and effluent samples were filtered through Whatman 542 filter paper

6. DISCUSSION

6.1. Model predictions

The model predictions are analysed under two headings :

- (i) Average daily mass response
- (ii) Cyclic daily mass response

6.1.1. Average daily response

The average daily observed and predicted responses of the plant are shown in Table 5.13. Comparing these it can be seen that the predictive ability of the model was excellent.

Table 5.13.

Northern Works : Comparison of average predicted effluent and other results with those observed

Test	Average Model prediction	Average Measured value
COD (mg O/l)	36	37
TKN (mg N/l)	3,3	2,1
Ammonia (mg N/l)	0,2	0,0
Nitrate (mg N/l)	15,6	16,4
Oxygen consumption main basin (mg/l/h)	25,3	24,1
MLSS (mg/l)	1 542	1 440

6.1.2. Cyclic response

Inspection of Figs 5.3. to 5.5. and 5.7. to 5.9. shows excellent agreement between the predicted and observed response of the most significant parameters thereby indicating that the model predicted the dynamic behaviour of the plant exceedingly well. The differences that are present are so slight that it is unlikely that one could ascribe a specific reason for those differences. Compared to the predictions of the Goudkoppie plant, those at Northern Works were closer to the observed data.

7. OPERATIONAL CHANGES TO ACHIEVE HIGHER PHOSPHORUS AND NITROGEN REMOVAL

At the Goudkoppie plant the existing process configuration was analysed for the settled flow bypassing the balancing tank, and it was shown that phosphorus removal could be achieved provided the denitrification capability was increased by creating an anoxic zone in the aerobic reactor (by switching off appropriately selected aerators). Efficient nitrification was insured by switching in extra aerators in the remaining aerobic zone. It is very likely that a similar operational modification at the Northern Works may also allow phosphorus removal. Although success is not as likely because the fraction of readily biodegradable COD in settled sewage at Northern Works is less than that at Goudkoppie Works. A more fruitful possibility for the Northern Works would be to model the process assuming that raw sewage could be added directly to the bioreactors. This is a feasible proposal for, at Northern Works, in contrast to the situation at Goudkoppie, the primary sludge is not totally committed to gas production. Both at Goudkoppie and Northern Works phosphorus removal

to 1 mg P/l in the effluent is mandatory so that the proposed change in the influent feed composition to the Northern Works process may be an acceptable means of achieving this objective.

Using the model to predict effluent quality when raw sewage is used as the feed to the process, the kinetic constants used were the same as those given in Table 5.8.(b). The operational parameters used were the same as those given in Table 5.8.(a) with the following exceptions :

- (i) Fraction of readily biodegradable COD , f_{bs} was estimated using the method described in Appendix A7 and found to be 0,15 mg COD/mg biodegradable COD
- (ii) Fraction of particulate unbiodegradable COD, f_{up} ;
the same value was used as that determined by Ekama and Marais (1978) for raw sewage i.e. 0,09 mg VSS/mg COD .
- (iii) Fraction of soluble unbiodegradable COD, f_{us} ;
using the annual averages the COD concentration of the feed and effluent were found to be 570 mg/l and 68 mg/l respectively ; hence the fraction could be estimated by assuming that all the effluent COD was soluble and unbiodegradable i.e. $63/570 = 0,12$ mg COD/mg COD .

A summary of the kinetic parameters specific to the raw sewage used in this investigation is given in Table 5.14.

The daily input data for raw sewage flow and load is given in Table 5.15.

The average daily model predictions using the above data are given in Table 5.16. and the daily cyclic variation of biodegradable COD concentration, ammonia and nitrate are given in Figs. 5.11. , 5.12. and 5.13. respectively.

Table 5.14.

Kinetic parameters specific to raw sewage at the
time of this investigation at Northern Works

Name	Symbol	Units	Value
Fraction of soluble unbiodegradable COD	f_{us}	mg COD/mg COD	0,12
Fraction of particulate unbiodegradable COD	f_{up}	mg VSS/mg COD	0,09
Fraction of easily biodegradable COD	f_{bs}	mg COD/mg bio-degradable COD	0,15
Easily biodegradable COD concentration	S_{bs}	mg COD/l	64
Fraction of TKN of feed in ammonia form	f_{nu}	mg N/mg N	0,65
Specific growth rate for nitrifiers	μ_{mn}	d^{-1}	0,50

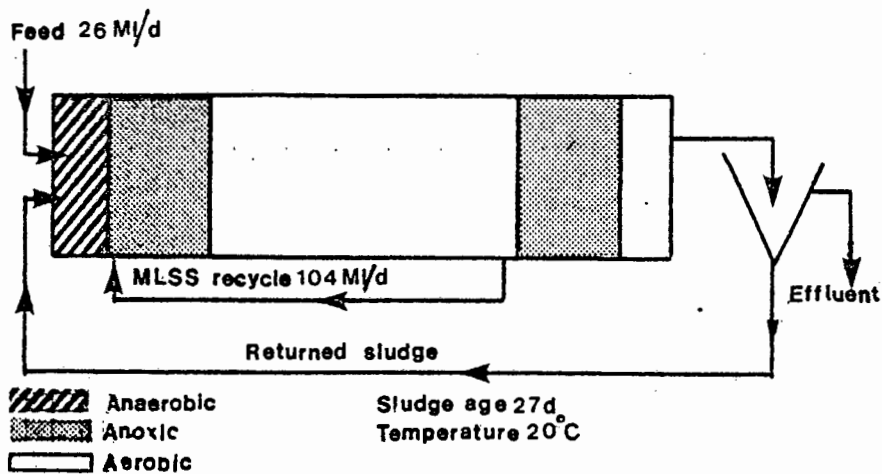
Table 5.15.

Input flow and COD and TKN concentrations used in the model prediction
when raw sewage was fed to the process

Time	Flow Ml/d	COD mg/l	TKN mg N/l	Time	Flow Ml/d	COD mg/l	TKN mg N/l
0700	25,6	520	31	2000	26,4	680	46
0800	24,4	370	31	2100	27,0	460	45
0900	24,4	375	31	2200	27,6	640	44
1000	24,4	380	30	2300	27,0	635	42
1100	25,0	475	51	2400	26,4	630	39
1200	25,6	570	72	0100	26,8	645	42
1300	25,6	595	62	0200	27,2	650	43
1400	25,6	620	53	0300	27,4	675	41
1500	25,0	650	54	0400	27,6	600	41
1600	26,0	680	56	0500	27,2	535	37
1700	26,2	670	51	0600	26,8	470	32
1800	26,4	660	46	0700	25,6	520	31
1900	26,4	620	46				

Table 5.16.

The average predicted performance of the Northern Works when both the balancing tank and primary sedimentation tanks are bypassed



	Feed	Anaerobic	Primary Anoxic	Primary Aerobic	Secondary Anoxic	Secondary Aerobic	Effluent
Total COD	575						
Soluble COD	429	110	70	69	69	69	
S_{bs}	64	41,6	0,9	0	0	0	
TKN as N	44	24,8	8,5	3,5	5,4	3,2	
Ammonia as N	29	22	5,9	0,3	2,8	0,2	
Nitrate as N	0	0	0	5,6	0	2,5	
Total P							
Soluble P							
Suspended solids				3 690			
Oxygen consumption mg/l/h				25,1			

Unless otherwise stated all results expressed as mg/l

- Note :
1. Unless otherwise stated all results expressed as mg/l
 2. S_{bs} - readily biodegradable COD concentration mg COD/l
 3. The underflow sludge return ratio is 0,4:1
 4. The MLSS recycle ratio is 4,0:1

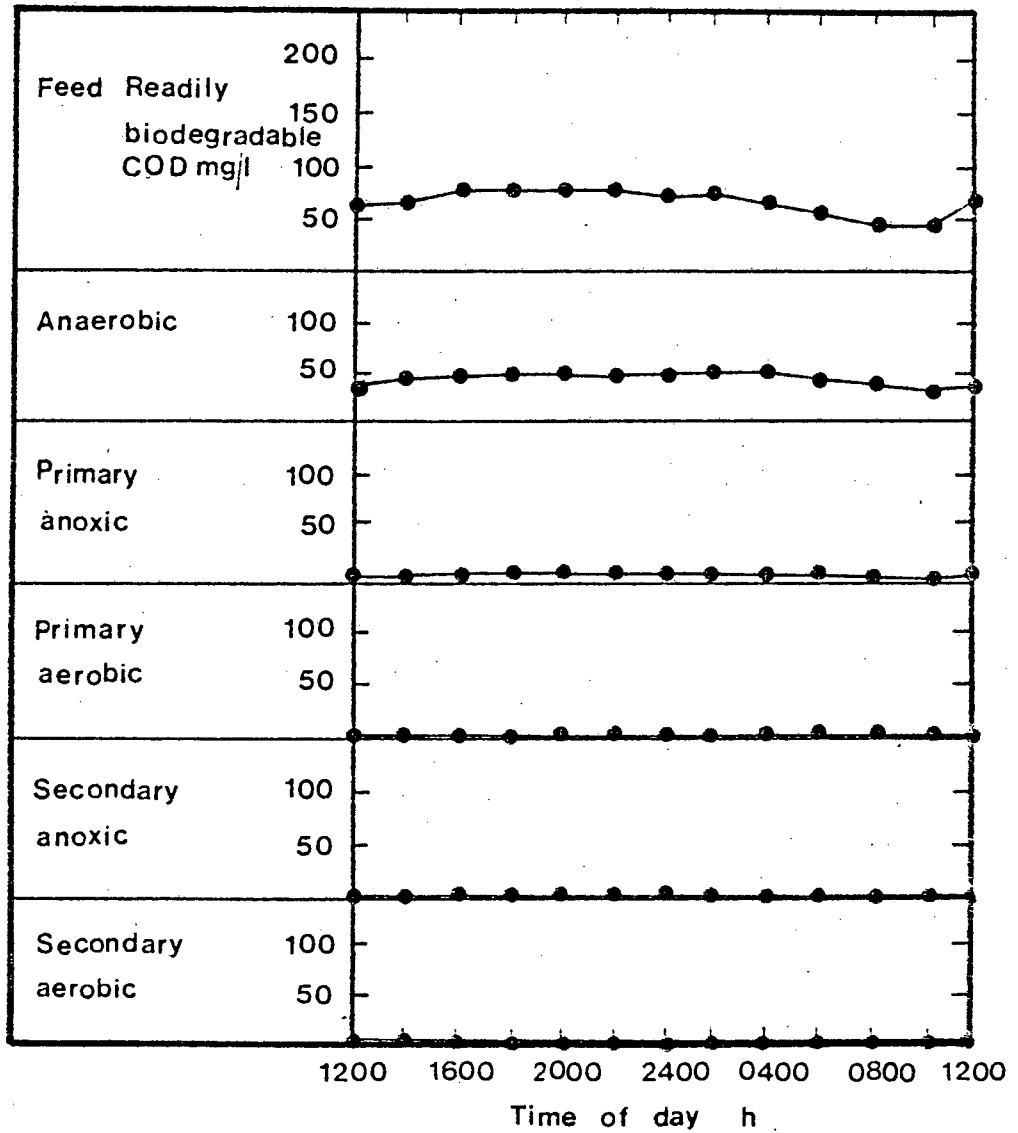


Fig. 5.11. Northern Works

Predicted readily biodegradable COD concentration when both the balancing tank and primary sedimentation tanks were bypassed

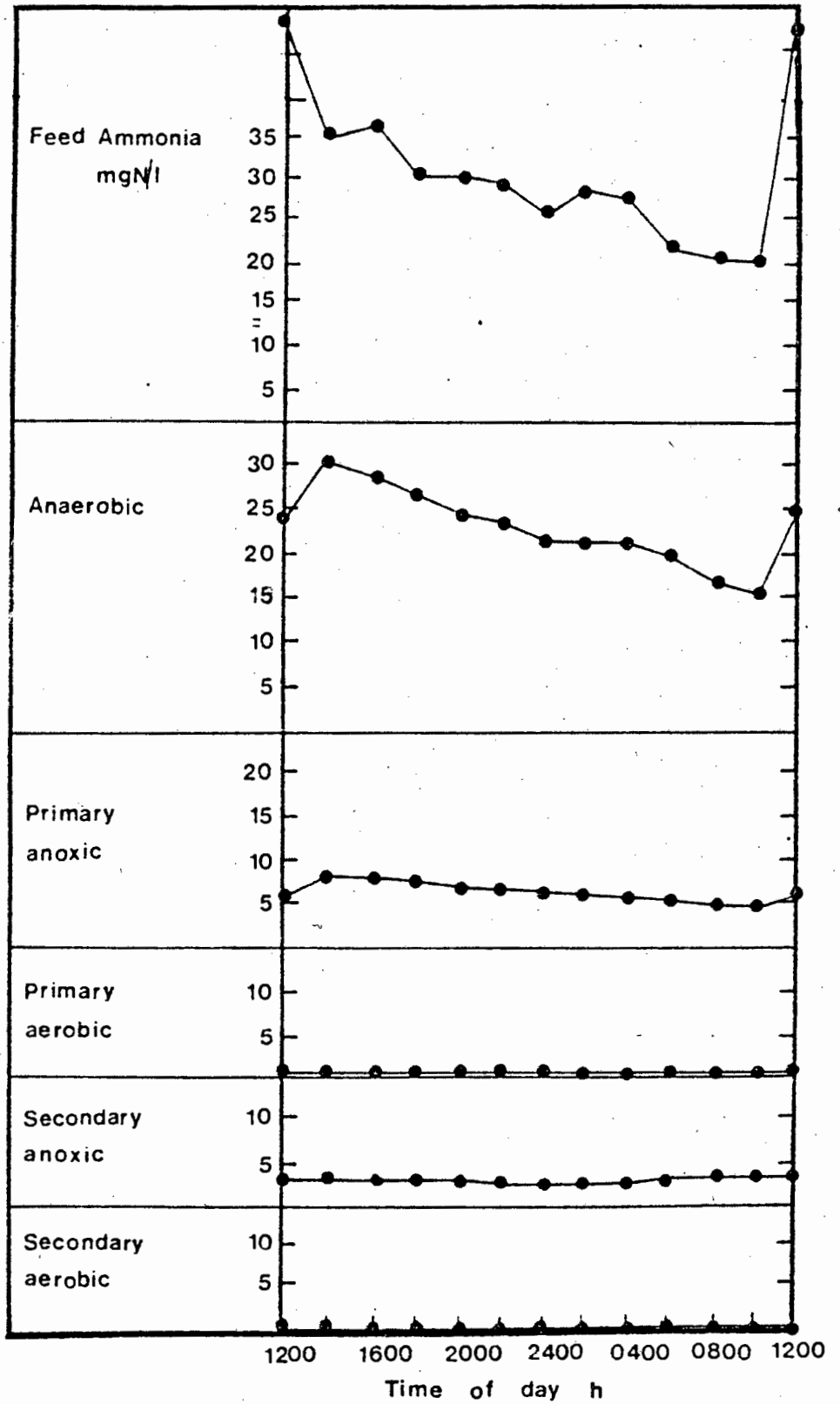


Fig. 5.12. Northern Works

Predicted ammonia concentration when both the balancing tank and primary sedimentation tanks were bypassed.

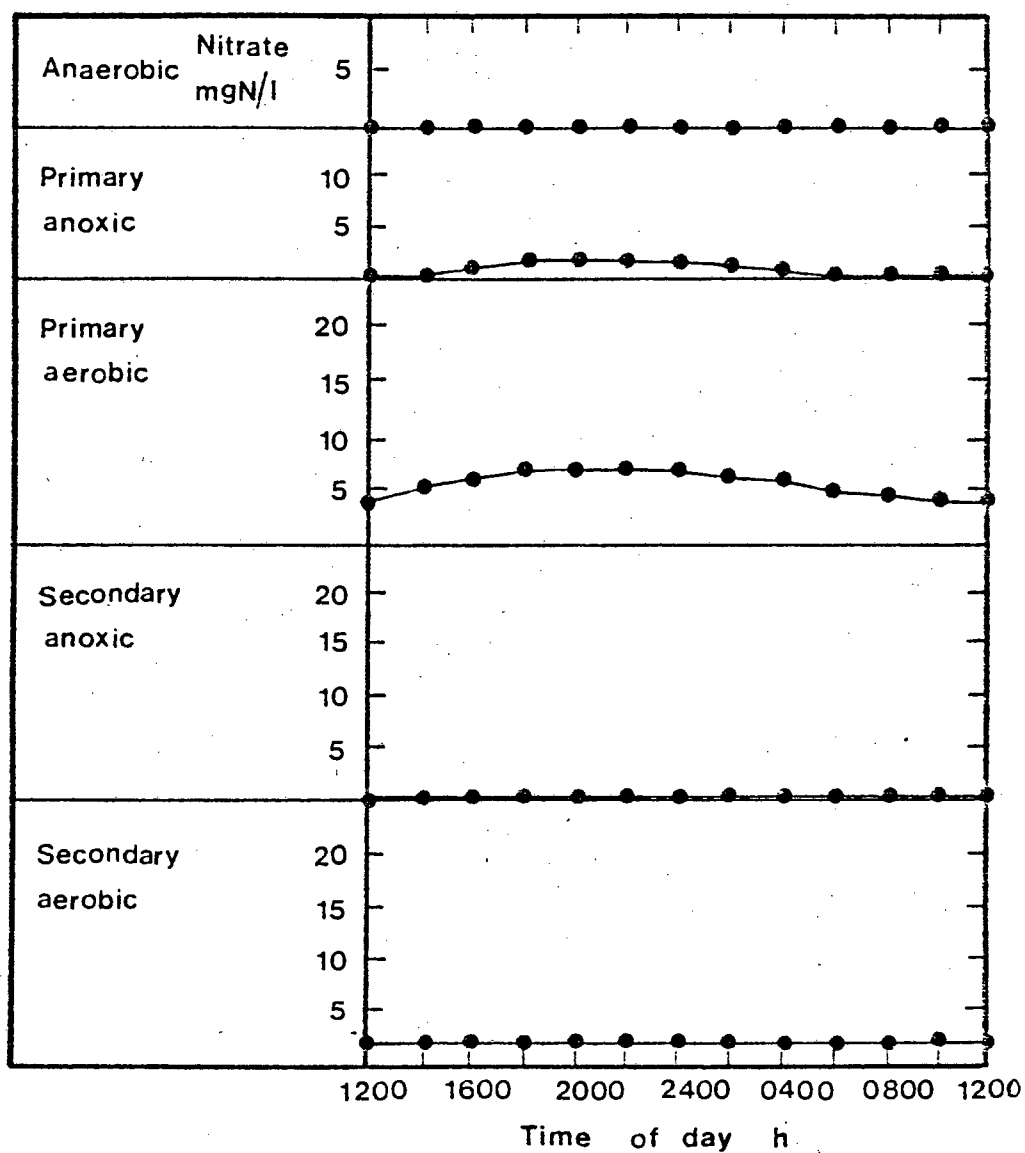


Fig. 5.13. Northern Works

Predicted nitrate concentration when both the balancing tank and primary sedimentation tanks were bypassed

From the average predicted performance (Table 5.16.) it is clear that the Northern Works process as designed would not fulfil its design objectives completely. The effluent total nitrogen would be reduced from an influent concentration of 44 mg N/l to 5,7 mg N/l, the ammonia and nitrate concentrations being 0,2 and 2,5 mg N/l respectively. Ekama, Siebritz and Marais (1982) have shown that if the TKN/COD ratio of raw sewage is not greater than 0,07 - 0,08, the five stage Phoredox process can be designed to bring about virtually complete nitrification for temperatures greater than 14 °C. The simulated data for this plant verifies this conclusion so that, if raw sewage is treated, the five stage process is indicated. When one compares the performance of this plant treating raw sewage with that treating settled sewage after passing through the balancing tank, it highlights the conclusion of Ekama et al (1982) that the process configuration for optimal treatment of a specified influent is governed by the influent sewage characteristics.

The phosphorus removal capacity of this plant can be readily assessed. The concentration of readily biodegradable COD in the anaerobic reactor is 41,6 mg/l. Applying equations (13), (14) and (15) the concentration of phosphorus which can be removed is estimated to be 6,2 mg/l. The measured influent phosphorus concentration is 9,7 mg P/l, which is artificially high due to the recycle of digester liquors from digesters fed with primary and waste activated sludge. From the above, the process should remove 6,2 mg P/l, hence the effluent quality should be approximately 3,2 mg P/l. If the sludge age were reduced from 27 d to 15 d and equations (13) to (15) again applied, the concentration of phosphorus which could then be removed is estimated to be 8,1 mg P/l; which means that the estimated effluent phosphorus concentration would be approximately $9,7 - 8,1 = 1,6$ mg P/l.

Since the plant was designed assuming a feed COD of 600 mg/l (see Table 5.1.) and a flow of 50 Ml/d, the process should have sufficient aeration capacity available to oxidise both the COD and TKN in the raw sewage. A simulated raw sewage feed to the biological reactor can be achieved by re-blending primary clarifier sludge with the effluent from these same clarifiers. In the original design of this plant provision was made to accomplish this. Hence, should this scheme be acceptable, no additional cost need be incurred in modifications to the plant. However, the additional load entering the process would increase the power consumption sharply.

CHAPTER SIX

CONCLUSIONS

The objectives of this dissertation were :

- (i) to investigate whether the operation of an existing large scale activated sludge plant could be modified to induce both nitrogen and excess phosphorus removal. The Johannesburg Alexandra Plant was selected for this purpose.
- (ii) to evaluate the applicability of the nitrification / denitrification bi-substrate death-regeneration activated sludge model of Dold et al (1980) and van Haandel et al (1980) at full scale. For this objective the Johannesburg Goudkoppie and Northern Works were selected for evaluation.

Conclusions drawn from these investigations are summarised as follows:

6.1. Enhanced nutrient removed from conventional activated sludge plants

Experiments carried out at the Alexandra Works showed that it was possible to create an anoxic zone by switching off selected surface aerators at the influent end of the aeration basin. Adjacent operating aerators served to recycle some of the nitrates produced lower downstream in the first aeration basin to the anoxic area. Sludge deposition in the unaerated area was prevented by periodically operating the decommissioned aerators to resuspend the solids.

This simple modification to normal operating procedures proved to be remarkably effective. A necessary prerequisite for success in this regard was that the remaining operational aerators in the primary aerobic basin were capable of sustaining high dissolved oxygen levels to ensure nitrification.

Enhanced phosphorus removal proved to be possible when additional aerators were switched off at the head of the aeration basin, making certain areas of the non-aerated zone completely anaerobic. Sustained successful operation proved to be difficult to maintain and it was dependent on having a sufficient degree of anaerobiosis to induce phosphorus release in the non-aerated area.

Experience gained during these experiments showed that the introduction of non-aerated zones into conventional activated sludge systems may give rise to instability of the mixed liquor if the overall dissolved oxygen level fell below 1,5 to 2,0 mg/l. Copious growths of the floating filamentous organism Nocardia also gave warning of a possible shift in bacterial population dynamics.

6.2. Applicability of model predictions to full scale plant performance

The data presented in this dissertation has shown that the model generated by the University of Cape Town Group could accurately predict the performance of the Johannesburg Goudkoppie Plant. Computed and actual mass balances of selected key parameters showed a very close degree of similarity. This successful practical application of the model promoted confidence in its use, and it was subsequently used to examine the reasons for the short fall in phosphorus removal at the Johannesburg Northern Works, and the likely performance of the city's new Bushkoppie plant.

During the day to day monitoring of the Johannesburg activated sludge plants performance was continually compared with certain key criteria such as TKN/COD ratio, amount of readily biodegradable substrate in the influent etc., and the limits proposed by the UCT group to ensure good plant performance were generally confirmed.

With this reassuring background it became possible to make predictions why many of the nutrient removing activated sludge plants in South Africa as reported by Paepka (1982) were not performing as well as expected. Failure may be attributed to one or a combination of the following :

1. Less than 25 mg/l of readily biodegradable COD in the anaerobic zone

Experience has shown this requirement is independent of the TKN/COD ratio. Under such conditions excess biological uptake of phosphorus will not be possible, even if nitrate addition to the anaerobic zone is limited by discouraging nitrification in the aerobic zone.

Concentrations of readily biodegradable COD in the influent to the biological reactors less than 50 mg/l (to allow for 1:1 dilution in the anaerobic zone) may arise where :

- (a) the strength of the influent sewage is weak i.e. COD is less than 150 - 200 mg/l
- (b) the readily biodegradable COD is depleted due to some biological interaction before the flow reaches the reactors e.g. fermentation in balancing tanks containing deposited solids or where the influent flow originates from septic tanks.

2. TKN/COD ratio such that complete denitrification is impossible

In such cases, even if the required minimum amount of 50 mg/l of readily biodegradable COD is present in the influent, this will all be assimilated by denitrifiers in the anaerobic zone before the necessary free, residual of 25 mg/l can be achieved to induce phosphorus release and subsequent uptake under aerobic conditions. Such conditions exist when $TKN/COD > 0,13$

3. Readily biodegradable COD greater than 50 mg/l and
TKN/COD ratio 0,08 - 0,13

In such cases it is possible to prevent nitrate from entering the anaerobic zone by adopting the UCT process with its auxillary denitrifying reactor.

4. Readily biodegradable COD greater than 50 mg/l and
TKN/COD ratio less than 0,08

Under these circumstances it is likely that the 5 stage Phoredox process will achieve phosphorus removal provided the recycled liquors from the final clarifiers contain zero nitrates. However experience gained at the Johannesburg works has clearly demonstrated that if the measured TKN/COD ratio is higher or the readily biodegradable COD is lower than the values used in the design, nitrates will appear in the effluent and hence in the sludge returned from the clarifier which will in turn adversely affect phosphorus removal. A prior knowledge of the influent sewage characteristics is obviously highly desirable. Should this data not be available and very conservative values be allocated to these parameters at the design stage, plant efficiency will suffer accordingly. If there is any doubt about sewage characteristics the adoption of the Phoredox process appears to be contra-indicated.

5. Lack of dissolved oxygen control

The operation of large scale nutrient removal activated sludge plants has clearly shown the need to match oxygen supply against oxygen requirements. If not adequately controlled excessive aeration may cause the plant not to give the expected phosphorus removal and under aeration may cause inadequate nitrification and the production of sludges with poor settling characteristics.

LIST OF REFERENCES

- ANON., (1971) "Kettings waterhiasint se tier", Landbouweekblad, 53, pp 38 - 39.
- ARVIN, C., CHRISTENSEN, M.H. (1979) "Precipitation of calcium phosphate and pH effects in denitrifying biofilm.", Vatten, 4, pp 329 - 336.
- BARNARD, J.L. (1973) "Biological denitrification.", Wat. Poll. Control, 72, pp 705 - 720.
- BARNARD, J.L. (1974) "Cut P and N without chemicals.", Water and Wastes Engineering, 11, pp 33 - 36.
- BARNARD, J.L. (1975) "Biological phosphorus removal in the activated sludge process. Review and proposals.", Paper presented to the Institute of Water Pollution Control.
- BARNARD, J.L. (1976) "A review of biological phosphorus removal in the activated sludge process.", water S.A., 2, 3, pp 136 - 144.
- BUCHAN, L. (1981) "Location and Nature of accumulated phosphorus in activated sludge.", Ph D thesis, Dept. of Agriculture, Univ. of Pretoria.
- CHRISTENSEN, M.H. and HARREMOES, P. (1972) "Biological denitrification in water treatment.", Rep. 72-2, Dep. San. Engng., Techn. Univ. of Denmark.

- CHRISTENSEN, M.H., HARREMOES, P. and ROED - JENSEN (1975)
 "Combined sludge denitrification of sewage utilizing internal carbon sources." , Presented Conf. on Nitrogen as a Water Pollutant , Copenhagen 1975 .
- CHRISTENSEN, M.H. and HARREMOES, P. (1974) "A literature review of biological denitrification of sewage." , I A W P R Specialised Conference, Copenhagen.
- DAWES, E.A. and SENIOR, P.J. (1973) "Energy reserve polymers in microorganisms." , Adv. in Microbiol. Physiology, 10, pp 135 - 266
- DAVELAAR, D., DAVIES, T.R. and WIECHERS S.G. (1978) "The significance of an anaerobic zone for the biological removal of phosphate from wastewaters." , Water S.A. , 4, 2, pp 54 - 60.
- DE BOICE, J.N. and THOMAS, J.F. (1975) "Chemical treatment for phosphorus control." , J W P C F , 47 , pp 2246 - 2255
- DOLD, P., EKAMA, G.A. and MARAIS, G. vR. (1980) "A general model for the activated sludge process." , Prog. Wat. Tech. 12 , pp 47 - 77.
- DOLD, P., BUHR, H.O. and MARAIS, G.vR. (1982) Dept. of Civil Engng., Univ. of Cape Town.
- DOWNING, A.L., PAINTER, H.A. and KNOWLES, G. (1964) "Nitrification in the activated sludge process." , J. Proc. Inst. Sew. Purif., 64 , 2 , pp 130 - 158.

- EKAMA, G.Å. and MARAIS, G. vR. (1978) "The dynamic behaviour of the activated sludge process." , Research Report No. W27 , Dept. of Civil Engng. , Univ. of Cape Town.
- FERGUSON, J.F., JENKINS, D. and EASTMAN, J. (1973) "Calcium phosphate precipitation at slightly alkaline pH values." , J W P C F , 45 , 4 , pp 620 - 631.
- FUHS, G.W. and CHEN, M. (1975) "Microbiological basis of phosphate removal in the activated sludge process for the treatment of wastewater." , Microbiol. Ecology, 2 , 2 , pp 119 - 138.
- GARBER, W.F. (1972) "Phosphorus removal by chemical and biological mechanisms." Presented at a conference on "Application of new concepts of physical chemical wastewater treatment." , Sept.
- HAROLD, F. (1966) "Inorganic polyphosphates in biology structure metabolism and function." , Bacteriol. Reviews, 30 , pp 772 - 794.
- HOFFMAN, R.J. and MARAIS, G. vR. (1977) "Phosphorus removal in the modified activated sludge process." , Research Report No. W22 , Dept. of Civil Engng. , Univ. of Cape Town.
- KEAY, G. and PITMAN, A.R. (1980) "The role of load balancing in the modified activated sludge plant." , Report CYD 62 , City Health Labs, Johannesburg.
- KUHL, A. (1960) "De biologie der kondensierten anorganischem phosphate." , Ergeb. Biol., 23 , pp 144 - 185.
- KULAEV, I.S. (1975) "Biochemistry of Inorganic polyphosphates." , Rev. Physiol. Biochem. Pharmacol., 73 , p 131.

- LAWSON, E. (1980) Dept. of Botany, University of the Witwatersrand
Private Communication.
- LEVIN, G.V. and SHAPIRO, J. (1965) "Metabolic uptake of phosphorus
by wastewater organisms." , J W P C F , 37 , 6 , pp 800 - 821
- LUDZACK, F.J. and ETTINGER, M.B. (1962) "Controlling operation to
minimise activated sludge effluent nitrogen." , J W P C F ,
34 , pp 920 - 931.
- MARAIS, G.vR. and EKAMA, G.A. (1976) "The activate sludge process:
Steady state behaviour." , Water S.A. , 2 , pp 163 - 200.
- MARSDEN, M.G. and MARAIS, G.vR. (1977) "Role of the primary anoxic
reactor in denitrification and biological P removal." ,
Res. Report No. W19 , Dept. of Civil Engng., Univ. of Cape Town.
- MENAR, A.B. and JENKINS, D. (1970) "Role of phosphorus in wastewater
treatment processes: The enhanced removal of phosphorus by
activated sludge." , Environ. Sci. Technol. , 4 , pp 1115 - 1121.
- McLAREN, A.R. and WOOD, R.J. (1976) "Effective phosphorus removal
from sewage by biological means." , Water S.A. , 2 , 1 , pp 47 - 50.
- MILBURY, W.F., McCAULEY, D. and HAWTHORNE, C.H. (1971) "Operation
of conventional activated sludge for maximum phosphorus removal." ,
J W P C F , 43 , 9 , pp 1890 - 1901

- PAEPKE, B. (1982) "Performance and operation aspects of biological phosphate removal in South Africa." , 11th International Conference I A W P R , April 1982 . Post conference seminar on Phosphate Removal in Biological Treatment Processes.
- PITMAN, A.R. (1974) "Bioflocculation as a means of improving the dewatering characteristics of activated sludge." ,
Wat. Poll. Control., 74 , 6 , pp 688 - 700.
- POTGIETER, D.J.J. (1980) "Biochemiese prosesse wat fosfaat- en stikstofverwydering in die gemodifiseerde geaktiveerde - slyk - proses te weeg bring." , Report Univ. of Pretoria presented to CSIR .
- SAWYER, C.N. (1944) "Biological engineering in sewage treatment." ,
Sew. Wks. J. , 16 , pp 625 - 635.
- SCALF, M.R., PFEFFER, F.M., LIVELY, D. and WITHEROW, J.O. (1969)
"Phosphate removal at Baltimore, Maryland." , Proc. Am. Soc. Civil Eng., 545 , p 817.
- SCOTT, W.E. and STEYN, D.J. (1979) "Chemical control of the water hyacinth on Hartbeespoort Dam." , Collaborative report Water Research Commission, National Inst. for Water Res. C S I R, S.A. and Hydrological Research Inst. Dept. Water Affairs, S.A.

- SEHAYEK, E. and MARAIS, G.vR. (1981) "Kinetics of biological nitrogen removal in single sludge activated sludge process." , Research Report No. W41 , Dept. of Civil Engng., Univ. of Cape Town.
- SIEBRITZ, I.P., EKAMA, G.A. and MARAIS, G.vR. (1980) "Excess biological phosphorus removal in the activated sludge process at warm temperate climates." , Proc. Waste Treatment and Utilisation , 2 , pp 233 - 251. Eds C, W. Robinson, M. Moo-Young and G.J. Farquhar , Pergamon Press, Toronto.
- SIEBRITZ, I.P., EKAMA, G.A. and MARAIS, G.vR. (1982) "A Parametric model for biological excess phosphorus removal." , Presented at I A W P R post conference seminar on phosphate removal, Pretoria. Procs to be published in Water Science and Technology.
- SIMKINS, M.J. and McCLAREN, R.A. (1978) "Consistent biological phosphate removal in an activated sludge plant." , Proc. Wat. Tech. , 10 , 5/6 , pp 433 - 442.
- STERN, L.B. and MARAIS, G.vR. (1974) "Sewage as electron donor in biological denitrification." , Research Report No. W7 , Dept. of Civil ENGNG., Univ. of Cape Town.
- VACKER, D., CONNELL, C.H. and WELLS, W.N. (1967) "Phosphate removal through municipal wastewater treatment at San Antonio, Texas." , J W P C F , 39 , 5 , pp 750 - 771.

- VAN HAANDEL, A.C. and MARAIS, G.vR. (1981) "Nitrification and Denitrification Kinetics in Activated Sludge Process." , Univ. of Cape Town, Dept. of Civil Eng. Report No. W 39 .
- VAN HAANDEL, A.C., EKAMA, G.A. and MARAIS, G.vR. (1979) "The Present Status of Research on Nitrogen Removal. A Model for the Modified Activated Sludge Process." , Presented at Symposium on Nitrogen Removal, Pretoria, South Africa, 17th May 1979 .
- VAN HAANDEL, A.C. , EKAMA, G.A. and MARAIS, G.vR. (1981) "The activated sludge process Part 3 - single sludge denitrification." , Water Research , 15 , pp 1135 - 1152.
- VAN HAANDEL, A.C. , DOLD, P.L. and MARAIS, G. vR. (1982) "Optimisation of nitrogen removal in the single sludge activated sludge process." , 11th International Conference, I A W P R , Cape Town , April 1982.
- VENTER, S.L.V., HALLIDAY, J. and PITMAN, A.R. (1978) "Optimisation of the Johannesburg Olifantsvlei extended aeration plant for phosphorus removal." , Prog. Wat. Tech. , 10 , 1/2 , pp 279 - 292.
- WUHRMANN, K. (1957) "Die dritte reinigungsstufe: Wege und bisherige erfolge in der eliminierung eutrophierende stoffe." , Schweiz. Z. Hydrol. , 19 , pp 409 - 427.
- WATER POLLUTION RESEARCH, (1968) p 160
- YALL, I. , BOUGHTON, W.H. , KNUDSON, R.O. and SINCLAIR, N.A. (1970) "Biological Uptake of Phosphorus by Activated Sludge." , Applied Microbiology , 20, 145 .

APPENDIX A1

LIST OF VOLUTIN FORMING BACTERIA WHICH ACCUMULATE POLYPHOSPHATE

Acetobacter suboxydans	Rhodopseudomonas palustris
Aerobacter aerogenes	Rhodospirillum rubrum
Azotobacter agilis	Serratia marcescens
A. vinelandii	Thiobacillus thioxydans
Bacillus subtilis	Nitrobacter
Bacterium aerogenes	Micrococcus denitrificans
B. cloacae	Staphylococcus denitrificans
B. friedländeri	Chlamydomoda
Caulobacter vibroides	Mucor racemosus
Chlorobium thiosulphatophilum	Claviceps purpurea
Chromatium	Acinetobacter
Clostridium spp	Zoogloea ramigera
Corynebacterium diptheria	Nitrosococcus
C. xerosis	Beggiato
E. coli	
Hydrogenomonas spp	
Mycobacterium spp	
M. chelonae	
M. phlei	
M. smegmatis	
M. thamnaphaeos	
M. tuberculosis	

Kuhl (1960) , Harold (1966) , Fuhs and Chen (1975)

APPENDIX A 2.1.

Raw data obtained when the Alexandra plant was operating
with all the aerators closest to the feed on

Flow Ml/d	Ammonia as N mg/l		TKN as N mg/l		Nitrate as N mg/l	Total COD mg/l		Soluble COD mg/l		Suspended solids mg/l	MLSS mg/l
	F	E	F	E	E	F	E	F	E	E	
34,3	20,6	6,2	38,4	12,0	23,0	500	84,6	315	49,9	40,4	2481
35,3	10,3	4,2	39,4	10,0	18,0	623	96,1	238	80,7	11,6	-
31,0	23,4	8,2	48,2	12,0	19,0	462	88,5	292	61,5	39,2	-
26,5	24,2	5,0	43,4	7,4	19,0	438	69,5	215	80,7	38,8	2240
37,2	21,0	3,0	40,1	9,4	19,0	534	39,8	-	11,9	33,6	2646
37,2	22,0	5,3	44,0	5,3	26,5	571	66,0	160	30,0	42,0	2800
37,7	19,9	8,6	39,1	11,4	27,0	498	72,0	215	34,0	26,0	-
32,4	21,0	3,0	34,0	6,8	26,0	504	60,0	216	20,0	20,8	-
28,2	30,0	3,0	38,0	3,0	29,2	510	88,0	200	39,0	12,4	-
40,3	18,2	5,0	32,0	8,0	21,0	669	97,0	300	66,0	15,0	2422
39,2	10,0	0,5	36,0	11,5	25,0	875	64,0	227	64,0	14,8	2812
41,1	22,8	7,8	39,0	10,0	29,0	1018	89,2	137	78,1	26,4	2624
39,5	23,0	6,7	46,0	11,0	31,9	583	65,0	210	60,0	14,0	2680
21,8	18,4	1,6	41,0	5,6	16,0	382	55,8	250	52,9	20,4	-

* F = feed

E = effluent

APPENDIX A 2.2.

Raw data obtained when the Alexandra plant was operating
with one aerator off in the first basin

Flow Ml/d	Ammonia as N mg/l		TKN as N mg/l		Nitrate as N mg/l	Total COD mg/l		Soluble COD mg/l		Suspended solids mg/l	MLSS mg/l
	F	E	F	E	E	F	E	F	E	E	
22,8	18,4	2,2	-	10,3	21,2	-	-	-	-	44,0	-
20,5	23,6	2,8	35,2	6,2	29,0	1007	116	410	31,0	24,0	-
30,4	24,4	3,0	36,0	6,4	29,0	961	93	470	31,0	64,0	2572
30,8	25,2	5,2	41,0	6,3	13,0	255	-	232	-	32,0	-
29,8	20,0	6,0	32,8	9,2	19,0	492	108	284	73,0	34,0	2218
28,3	20,6	5,4	32,8	9,2	19,0	382	66	312	86,0	20,0	2592
27,6	21,2	4,4	32,8	8,4	17,0	388	-	226	75,0	21,0	-
22,2	30,0	4,1	37,0	6,5	14,1	420	132	306	72,0	32,0	2900
22,3	17,0	3,6	43,4	5,4	12,5	409	73	219	73,0	30,0	2370
30,1	27,6	5,6	41,2	9,6	12,0	-	54	178	-	57,0	2518
28,1	11,8	5,0	43,4	6,8	14,4	567	83	197	31,0	43,0	2396
30,8	21,6	6,8	42,8	6,0	11,6	494	69	320	53,0	32,0	2602
28,7	22,2	6,0	43,2	8,8	7,6	535	71	272	67,0	25,0	2792
28,4	23,6	3,0	35,6	8,8	10,5	437	59	228	51,0	10,0	-
25,6	19,0	3,6	39,0	8,0	14,0	465	59	220	55,5	17,0	-
23,4	36,0	4,3	36,0	8,6	17,8	497	-	110	-	16,0	-
32,6	20,0	4,2	42,0	10,0	18,0	504	-	306	-	27,0	2644

* F = feed

E = effluent

APPENDIX A 2.3.

Raw data obtained when the Alexandra plant was operating
with two aerators off in the first basin

Flow Ml/d	Ammonia as N mg/l		TKN as N mg/l		Nitrate as N mg/l	Total COD mg/l		Soluble COD mg/l		Suspended solids mg/l	MLSS mg/l
	F	E	F	E	E	F	E	F	E	E	
18,0	23,0	1,2	38,0	4,0	10,0	453	67,6	147	55,8	25,6	-
16,6	17,4	2,2	52,0	15,0	13,6	318	47,1	147	32,3	74,4	1738
28,8	11,4	2,6	47,3	10,3	15,0	369	50,2	134	35,7	18,4	1498
30,0	17,4	4,0	31,0	6,6	12,0	433	62,5	216	42,6	25,9	2352
28,9	22,0	3,2	44,0	6,6	11,2	597	42,6	219	36,9	20,3	2294
29,7	25,2	2,0	40,2	7,0	11,6	506	53,9	290	42,6	19,2	2346
30,3	24,0	5,0	41,0	8,1	0,0	523	101	240	85,0	25,0	2970
23,9	24,2	4,0	39,2	8,0	7,8	356	100	222	103	29,1	-
23,0	17,2	2,0	41,2	3,6	7,6	398	81,4	225	40,7	32,7	-
32,4	19,2	1,8	40,0	3,9	7,0	640	98,2	280	61,8	43,6	2502
29,7	14,0	7,5	34,4	-	3,0	544	103	309	96,0	14,0	-
30,7	12,0	6,6	33,0	8,8	7,0	846	69,8	381	41,3	33,2	2252
30,5	20,4	6,0	34,0	8,2	-	434	40,4	214	40,4	16,0	2322
31,6	12,0	5,4	35,0	12,8	5,0	311	47,7	175	40,4	19,2	-
25,1	24,0	5,0	38,0	7,9	5,3	472	22,0	150	37,0	49,0	2350

* F = feed

E = effluent

APPENDIX A 2.4.

Raw data obtained when the Alexandra plant was operating
with the three aerators closest to the feed off

Flow Ml/d	Ammonia as N mg/l		TKN as N mg/l		Nitrate as N mg/l	Total COD mg/l		Soluble COD mg/l		Suspended solids mg/l	MLSS mg/l
	F	E	F	E	E	F	E	F	E	E	
36,1	19,0	3,0	44,0	8,1	11,0	500	26	-	-	30	2100
34,9	18,4	2,8	-	-	2,6	624	68	304	56	12	2844
40,3	17,4	3,6	34,0	6,4	2,2	440	40	216	36	46	3002
31,7	18,4	1,0	33,0	1,8	17,8	448	52	176	32	9	2982
37,9	17,4	2,0	29,6	3,8	6,8	610	44	380	36	40	2088
35,9	10,2	1,6	31,0	5,1	5,6	432	72	104	60	7	3196
36,3	20,6	4,2	44,2	7,6	8,2	704	28	332	13	13	3092
33,8	20,4	4,0	37,0	7,0	4,0	416	-	220	-	10	3338
34,8	20,0	4,0	41,0	4,6	3,5	520	56	210	51	8	3200
34,5	10,3	2,6	36,8	3,0	7,2	400	48	280	44	45	3276
27,8	26,0	0,5	49,0	3,8	11,4	1152	55	223	39	8	3158
36,4	26,0	4,0	-	6,0	9,0	1198	34	503	34	5	3360
35,8	41,0	3,6	-	-	3,0	674	98	-	-	64	3354
30,8	17,0	1,5	24,4	3,0	7,3	420	62	-	-	17	3500
27,3	18,2	2,4	37,0	2,4	8,0	423	55	-	47	11	3328
35,7	16,4	2,0	32,0	-	7,2	-	-	-	-	13	3255
33,1	21,4	3,0	42,8	4,8	-	584	48	260	40	-	-
34,7	20,0	3,0	45,0	5,0	5,0	250	46	-	-	12	3178
38,3	21,0	1,6	37,0	8,0	9,0	632	76	272	52	17	3230
36,3	21,0	4,6	-	-	2,4	552	52	240	52	40	3293
37,6	20,3	4,4	37,6	8,8	-	512	56	280	29	29	3058
37,6	19,2	4,6	43,2	9,0	1,0	792	80	296	64	20	3216
41,3	23,2	4,6	43,0	12,8	7,0	728	72	408	48	14	2908

* F = feed

E = effluent

APPENDIX A 3

EXAMPLE CALCULATION OF MIXED LIQUOR SUSPENDED SOLIDS CONCENTRATION
USING MARAIS EKAMA (1976) EQUATIONS

$$\begin{aligned}
 \text{Flow} &= Q = 34,4 \times 10^6 \quad (\text{l/d}) \\
 \text{Feed COD} &= S_{ti} = 590 \quad (\text{mg/l}) \\
 \text{Volume} &= V = 33,5 \times 10^6 \quad (\text{l}) \\
 \text{Sludge age} &= R_s = 15 \quad (\text{d}) \\
 \text{Volatile fraction of sludge} &= 0,72
 \end{aligned}$$

Process constants

$$\begin{aligned}
 Y &= 0,45 \quad (\text{mg VSS/mg COD}) \\
 b &= 0,24(1,03)^{T-20} \quad (\text{d}^{-1}) \\
 SX_{ii} &= \text{unbiodegradable particulate influent COD} \\
 &= \frac{f_{up} S_{ti}}{1,48} \\
 &= 0,13 S_{ti} \quad (\text{mg COD/l}) \\
 S_u &= \text{unbiodegradable soluble influent COD} \\
 &= f_{us} S_{ti} \\
 &= 0,07 S_{ti} \quad (\text{mg COD/mg VSS})
 \end{aligned}$$

Calculation of the biodegradability of the COD

$$\begin{aligned}
 S_{bi} &= S_{ti} - SX_{ii} - S_u \\
 &= 590 - 0,13 (590) - 0,07 (590) \\
 &= \underline{472 \quad \text{mg/l COD}} \rightarrow
 \end{aligned}$$

Calculation of the mass fractions of the MLSS

$$\begin{aligned}
 \text{Active mass} &= MX_a \\
 &= M \frac{S_{bi} YR_s}{1 + bR_s} \\
 &= 34,4 \times 10^6 \frac{472 \cdot 0,45 \cdot 15}{1 + (0,24 \cdot 15)} \\
 &= 23\,825 \text{ kg/d} \longrightarrow
 \end{aligned}$$

$$\begin{aligned}
 \text{Endogenous mass} &= MX_e \\
 &= MX_a \cdot 0,2 \cdot bR_s \\
 &= 23\,825 \cdot 0,2 \cdot 0,24 \cdot 15 \\
 &= 17\,154 \text{ kg/d} \longrightarrow
 \end{aligned}$$

$$\begin{aligned}
 \text{Inert mass} &= MX_{ii} \\
 &= \frac{0,13 R_s M S_{ti}}{1,48} \\
 &= \frac{0,13 \cdot 15 \cdot 34,4 \times 10^6 \cdot 590}{1,48} \\
 &= 26\,741 \text{ kg/d} \longrightarrow
 \end{aligned}$$

$$\begin{aligned}
 MX_V &= MX_a + MX_e + MX_i \\
 &= 67\,720 \text{ kg/d} \longrightarrow
 \end{aligned}$$

$$\begin{aligned}
 MX_T &= \frac{MX_V}{0,75} \\
 &= \frac{67\,720}{0,75} \\
 &= 90\,293 \text{ kg/d} \longrightarrow
 \end{aligned}$$

$$\begin{aligned}
 X_T &= \frac{90\,293 \times 10^6}{33,5 \times 10^6} \\
 &= 2\,695 \text{ mg/l} \longrightarrow
 \end{aligned}$$

APPENDIX A 4

MEASUREMENT OF OXYGEN CONSUMPTION RATE

In a preliminary exploratory investigation it was found that the oxygen consumption rate was highest at the influent and lowest at the effluent end of the aeration zone, indicating that the basin could not be regarded as completely mixed.

These results indicated that it was necessary to carry out measurements at a number of points along the basin. Three respirometers were therefore built as indicated in Fig. A 1, and positioned at the start, middle and end of the primary aeration basin.

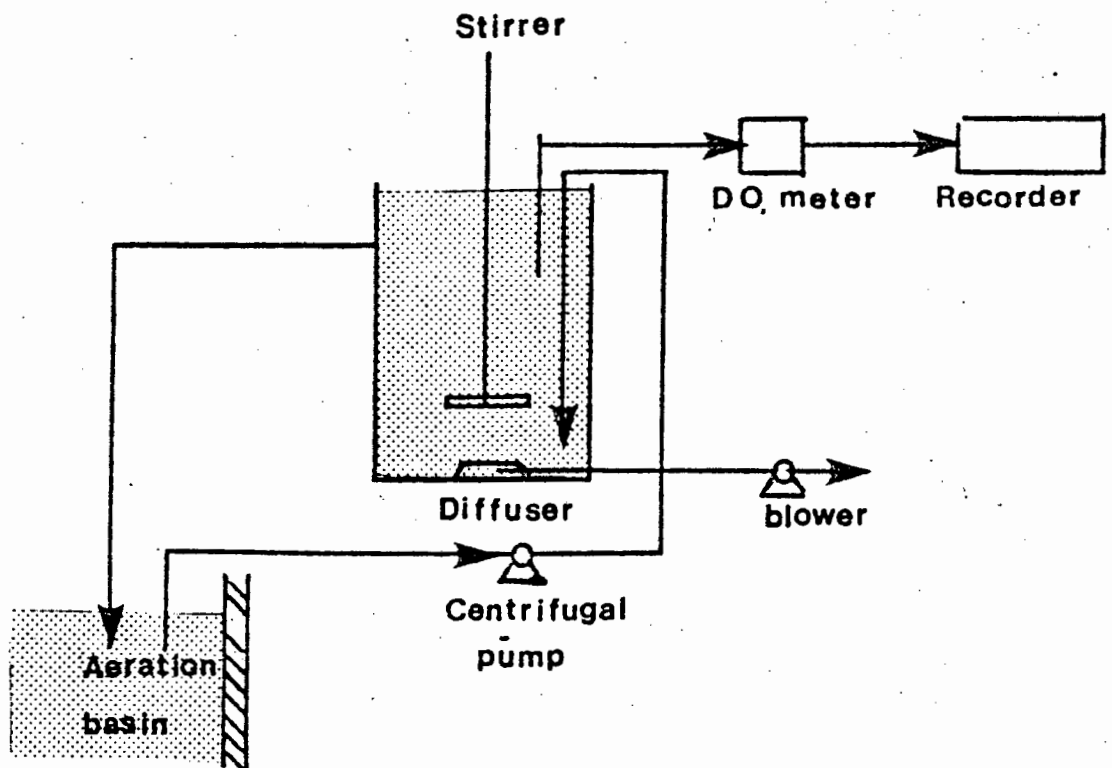


Fig. A 1 Respirometer design

When measuring oxygen consumption in a reactor, the sample must be removed from the reactor, oxygenated and the oxygen reduction with time then measured. This introduces an error as the sample does not receive the influent substrate while it is outside the reactor, possibly resulting in a reduction in the rate of oxygen utilisation. In laboratory scale units this error can be made zero by cutting off the oxygen supply, leaving the influent and effluent flows unaltered and gently stirring the reactor contents. On the plant the error can be minimised by arranging minimum time lapse between collection of the sample and measurement of the rate. To reduce the time effect as much as possible, a special respirometer, as shown in Fig. A 1 was designed containing the following features:

- * A 0,4 m³ tank into which mixed liquor was pumped every half hour, thus permitting determinations to be done.
- * The pumping rate was approximately 10 m³/h so that the respirometer was filled every 2,4 minutes, the liquor displaced being returned to the aeration basin.
- * The pump was so controlled that the volume of the tank was replaced 4 times.
- * During the pumping stage, and for 5 minutes thereafter, the sludge was aerated by compressed air discharging through a fine stone diffuser. The dissolved oxygen concentration at the end of the aeration period was usually above 4 mg/l .
- * The air compressor was then switched off automatically, and the dissolved oxygen meter (coupled to a recorder) switched in.

- * The subsequent decrease in the dissolved oxygen concentration with time was recorded. The gradient of the oxygen - time graph gave the utilisation rate.

- * During the entire operation i.e. taking in a new sample, aerating it, and measuring the oxygen consumption rate, the mixed liquor in the tank was stirred. The stirring rate was designed so that no air was drawn into the mixed liquor by means of vortices.

APPENDIX A5

GOUDKOPPIE PLANT

The raw data obtained during the experimental period

CODE

G1 B E	Balancing tank effluent
G1 Z 12	Anaerobic zone
G1 Z 22	Primary anoxic zone
G1 Z 32	Primary aeration zone
G1 Z 42	Secondary anoxic zone
G1 Z 52	Secondary aeration zone
G1 E 2	Effluent

CHEMICAL OXYGEN DEMAND AS O

	G1BE	G1Z12	G1Z22	G1Z32	G1Z42	G1Z52	G1E2	
DATE	TIME	DAY						
1979-10-21	12H00	SUN	150	-	-	-	41	
1979-10-21	14H00	SUN	110	-	-	-	31	
1979-10-21	16H00	SUN	120	-	-	-	26	
1979-10-21	18H00	SUN	130	-	-	-	26	
1979-10-21	20H00	SUN	210	-	-	-	36	
1979-10-21	22H00	SUN	260	-	-	-	36	
1979-10-21	24H00	SUN	240	-	-	-	41	
1979-10-22	2H00	MON	290	-	-	-	36	
1979-10-22	4H00	MON	240	-	-	-	31	
1979-10-22	6H00	MON	260	-	-	-	26	
1979-10-22	8H00	MON	220	-	-	-	36	
1979-10-22	10H00	MON	130	-	-	-	21	
1979-10-23	12H00	TUE	210	-	-	-	110	
1979-10-23	14H00	TUE	230	-	-	-	97	
1979-10-23	16H00	TUE	230	-	-	-	97	
1979-10-23	18H00	TUE	330	-	-	-	87	
1979-10-23	20H00	TUE	390	-	-	-	92	
1979-10-23	22H00	TUE	450	-	-	-	97	
1979-10-23	24H00	TUE	440	-	-	-	37	
1979-10-24	2H00	WED	410	-	-	-	37	
1979-10-24	4H00	WED	420	-	-	-	97	
1979-10-24	6H00	WED	360	-	-	-	97	
1979-10-24	8H00	WED	350	-	-	-	97	
1979-10-24	10H00	WED	210	-	-	-	100	
1979-10-25	12H00	THU	160	-	-	-	55	
1979-10-25	14H00	THU	190	-	-	-	60	
1979-10-25	16H00	THU	230	-	-	-	71	
1979-10-25	18H00	THU	260	-	-	-	60	
1979-10-25	20H00	THU	320	-	-	-	-	
1979-10-25	22H00	THU	320	-	-	-	71	
1979-10-25	24H00	THU	330	-	-	-	60	
1979-10-26	2H00	FRI	330	-	-	-	44	
1979-10-26	4H00	FRI	380	-	-	-	44	
1979-10-26	6H00	FRI	260	-	-	-	66	
1979-10-26	8H00	FRI	320	-	-	-	71	
1979-10-26	10H00	FRI	270	-	-	-	60	
1979-10-26	12H00	FRI	200	-	-	-	-	
1979-10-26	14H00	FRI	200	-	-	-	-	
1979-10-26	16H00	FRI	260	-	-	-	-	
1979-10-26	18H00	FRI	320	-	-	-	-	
1979-10-26	20H00	FRI	320	-	-	-	-	
1979-10-26	22H00	FRI	340	-	-	-	-	
1979-10-26	24H00	FRI	270	-	-	-	-	
1979-10-27	2H00	SAT	270	-	-	-	-	
1979-10-27	4H00	SAT	240	-	-	-	-	
1979-10-27	6H00	SAT	230	-	-	-	-	
1979-10-27	8H00	SAT	290	-	-	-	-	
1979-10-27	10H00	SAT	220	-	-	-	-	
NO. RESULTS			48	0	0	0	0	35
MEAN			270	-	-	-	-	63
STD. DEVIATION			83	-	-	-	-	28

CHEMICAL OXYGEN DEMAND (SOLUBLE) AS O

			G18E	G1Z12	G1Z22	G1Z32	G1Z42	G1Z52	G1E2
DATE	TIME	DAY							
1979-10-21	12H00	SUN	83	73	93	170	87	190	41
1979-10-21	14H00	SUN	87	120	78	130	51	150	30
1979-10-21	16H00	SUN	73	160	130	87	56	120	20
1979-10-21	18H00	SUN	93	78	120	56	41	56	20
1979-10-21	20H00	SUN	110	150	46	27	46	56	36
1979-10-21	22H00	SUN	110	120	46	71	36	97	21
1979-10-21	24H00	SUN	160	93	46	66	41	46	31
1979-10-22	2H00	MON	140	-	160	200	140	150	26
1979-10-22	4H00	MON	150	62	46	36	-	-	25
1979-10-22	6H00	MON	130	57	46	41	-	-	25
1979-10-22	8H00	MON	150	57	41	41	56	51	30
1979-10-22	10H00	MON	150	52	46	46	46	77	20
1979-10-23	12H00	TUE	97	87	36	100	56	77	66
1979-10-23	14H00	TUE	140	51	41	82	87	87	56
1979-10-23	16H00	TUE	150	67	97	92	56	87	66
1979-10-23	18H00	TUE	150	87	-	120	77	71	71
1979-10-23	20H00	TUE	210	140	-	120	77	56	77
1979-10-23	22H00	TUE	250	97	56	110	56	66	71
1979-10-23	24H00	TUE	290	110	51	97	110	100	77
1979-10-24	2H00	WED	300	140	-	71	87	87	71
1979-10-24	4H00	WED	310	110	77	71	77	77	71
1979-10-24	6H00	WED	290	130	82	46	51	87	66
1979-10-24	8H00	WED	250	32	260	77	61	66	71
1979-10-24	10H00	WED	200	110	87	77	87	92	77
1979-10-25	12H00	THU	110	76	76	76	33	82	27
1979-10-25	14H00	THU	100	54	65	54	38	98	33
1979-10-25	16H00	THU	120	49	54	49	38	60	55
1979-10-25	18H00	THU	120	60	38	60	33	44	49
1979-10-25	20H00	THU	140	38	44	76	38	22	60
1979-10-25	22H00	THU	160	44	54	82	33	38	38
1979-10-25	24H00	THU	170	54	60	49	33	38	44
1979-10-26	2H00	FRI	240	27	76	38	54	49	27
1979-10-26	4H00	FRI	230	38	82	27	16	44	27
1979-10-26	6H00	FRI	210	44	76	33	16	38	49
1979-10-26	8H00	FRI	120	49	98	44	33	22	49
1979-10-26	10H00	FRI	210	32	82	180	60	38	27
1979-10-26	12H00	FRI	140	54	33	33	76	76	49
1979-10-26	14H00	FRI	130	60	38	33	76	76	44
1979-10-26	16H00	FRI	190	60	49	60	71	91	38
1979-10-26	18H00	FRI	220	71	49	65	65	91	33
1979-10-26	20H00	FRI	220	76	44	65	60	92	27
1979-10-26	22H00	FRI	230	110	38	49	54	92	27
1979-10-26	24H00	FRI	230	98	49	54	44	91	38
1979-10-27	2H00	SAT	170	60	38	44	33	54	33
1979-10-27	4H00	SAT	150	54	38	38	33	49	38
1979-10-27	6H00	SAT	160	38	44	27	44	28	33
1979-10-27	8H00	SAT	160	44	38	60	38	33	27
1979-10-27	10H00	SAT	160	49	44	76	38	38	38
NO. RESULTS			48	47	45	48	46	46	48
MEAN			170	77	67	73	55	72	43
STD. DEVIATION			61	34	41	39	24	35	18

AMMONIA AS N

DATE	TIME	DAY	G1BE	G1Z12	G1Z22	G1Z32	G1Z42	G1Z52	G1E2
1979-10-21	12H00	SUN	18	9.7	4.7	3.4	2.8	2.2	2.2
1979-10-21	14H00	SUN	17	12	4.8	2.9	1.6	1.7	1.7
1979-10-21	16H00	SUN	17	18	5.8	3.0	2.0	1.1	1.1
1979-10-21	18H00	SUN	23	18	5.7	0.0	0.8	0.0	0.0
1979-10-21	20H00	SUN	27	17	5.7	0.0	0.9	1.0	1.0
1979-10-21	22H00	SUN	28	16	5.4	0.0	0.8	1.1	1.1
1979-10-21	24H00	SUN	23	15	5.5	4.7	0.0	0.9	0.9
1979-10-22	2H00	MON	25	19	-	-	-	-	-
1979-10-22	4H00	MON	22	12	4.4	3.4	-	-	-
1979-10-22	6H00	MON	23	12	4.3	3.1	-	-	-
1979-10-22	8H00	MON	20	10	2.0	2.6	0.0	1.6	1.6
1979-10-22	10H00	MON	15	8.2	3.3	2.5	1.3	1.1	1.1
1979-10-23	12H00	TUE	16	14	4.1	1.9	1.9	1.5	1.5
1979-10-23	14H00	TUE	20	14	4.0	1.9	1.6	1.4	1.4
1979-10-23	16H00	TUE	22	16	4.7	1.5	1.5	1.5	1.5
1979-10-23	18H00	TUE	25	19	-	1.5	2.1	1.2	1.2
1979-10-23	20H00	TUE	29	21	-	1.5	1.8	1.5	1.5
1979-10-23	22H00	TUE	29	21	5.3	2.1	1.5	1.6	1.6
1979-10-23	24H00	TUE	29	21	5.8	1.5	1.6	1.5	1.5
1979-10-24	2H00	WED	28	19	5.5	1.9	2.1	2.0	2.0
1979-10-24	4H00	WED	27	19	4.9	1.9	1.7	2.0	2.0
1979-10-24	6H00	WED	25	17	4.3	1.5	1.5	1.5	1.5
1979-10-24	8H00	WED	22	15	3.6	1.7	1.5	1.5	1.5
1979-10-24	10H00	WED	19	13	4.4	1.6	1.3	1.6	1.6
1979-10-25	12H00	THU	16	15	3.7	1.4	1.5	1.5	1.5
1979-10-25	14H00	THU	22	15	3.8	1.7	1.7	1.4	1.4
1979-10-25	16H00	THU	22	18	4.6	1.5	1.4	1.4	1.4
1979-10-25	18H00	THU	32	22	5.9	1.9	1.7	1.5	1.5
1979-10-25	20H00	THU	36	23	6.3	2.1	0.0	1.6	1.6
1979-10-25	22H00	THU	37	23	7.0	2.1	1.8	1.4	1.4
1979-10-25	24H00	THU	35	23	7.0	2.1	2.0	1.5	1.5
1979-10-26	2H00	FRI	33	19	5.5	1.7	1.9	0.0	0.0
1979-10-26	4H00	FRI	29	19	5.9	1.9	2.1	1.5	1.5
1979-10-26	6H00	FRI	20	18	5.8	1.9	1.7	1.7	1.7
1979-10-26	8H00	FRI	26	15	4.8	1.5	1.7	1.6	1.6
1979-10-26	10H00	FRI	23	13	4.1	1.4	1.6	1.6	1.6
1979-10-26	12H00	FRI	21	7.0	4.5	1.5	1.6	1.6	1.6
1979-10-26	14H00	FRI	26	7.3	4.3	1.5	2.0	1.6	1.6
1979-10-26	16H00	FRI	27	9.0	5.2	1.5	1.8	1.7	1.7
1979-10-26	18H00	FRI	35	11	6.3	2.0	1.5	2.0	2.0
1979-10-26	20H00	FRI	43	12	7.2	2.2	1.7	1.5	1.5
1979-10-26	22H00	FRI	43	13	7.5	2.0	1.5	1.6	1.6
1979-10-26	24H00	FRI	38	12	7.2	2.7	2.2	1.8	1.8
1979-10-27	2H00	SAT	37	11	-	2.8	2.8	2.8	2.8
1979-10-27	4H00	SAT	32	11	-	-	-	-	-
1979-10-27	6H00	SAT	28	9.8	5.3	1.5	1.3	1.5	1.5
1979-10-27	8H00	SAT	36	8.2	4.5	1.5	2.0	1.9	1.9
1979-10-27	10H00	SAT	23	7.0	4.1	1.8	1.2	2.2	2.2
NO. RESULTS			48	48	43	46	44	44	44
MEAN			26	15	5.1	1.9	1.6	1.5	1.5
STD. DEVIATION			7.1	4.6	1.2	0.8	0.6	0.5	0.5

NITRATE AS N

	G1B6	G1Z12	G1Z22	G1Z32	G1Z42	G1Z52	G1E2		
DATE	TIME	DAY							
1979-10-21	12H00	SUN	-	2.6	11	13	15	16	15
1979-10-21	14H00	SUN	-	3.3	11	12	12	14	16
1979-10-21	16H00	SUN	-	1.6	11	13	9.6	14	16
1979-10-21	18H00	SUN	-	1.3	12	15	12	14	16
1979-10-21	20H00	SUN	-	1.1	13	15	14	14	16
1979-10-21	22H00	SUN	-	1.4	11	15	14	14	16
1979-10-21	24H00	SUN	-	1.9	13	15	14	14	16
1979-10-22	2H00	MON	-	0.0	-	14	15	16	16
1979-10-22	4H00	MON	-	3.7	14	-	-	-	16
1979-10-22	6H00	MON	-	3.9	14	16	-	-	-
1979-10-22	8H00	MON	-	4.7	14	16	-	-	15
1979-10-22	10H00	MON	-	5.1	13	16	16	17	17
1979-10-23	12H00	TUE	-	1.3	11	14	15	15	15
1979-10-23	14H00	TUE	-	0.7	10	14	13	14	14
1979-10-23	16H00	TUE	-	0.4	10	14	13	14	14
1979-10-23	18H00	TUE	-	0.0	-	14	14	14	14
1979-10-23	20H00	TUE	-	0.3	-	14	14	14	14
1979-10-23	22H00	TUE	-	0.3	9.5	15	14	14	14
1979-10-23	24H00	TUE	-	0.0	10	15	14	15	14
1979-10-24	2H00	WED	-	1.9	9.8	14	14	15	13
1979-10-24	4H00	WED	-	0.0	9.5	14	14	14	13
1979-10-24	6H00	WED	-	0.4	9.3	14	13	14	13
1979-10-24	8H00	WED	-	0.5	9.6	13	12	14	13
1979-10-24	10H00	WED	-	0.4	9.6	13	13	14	13
1979-10-25	12H00	THU	-	0.8	9.2	12	12	13	12
1979-10-25	14H00	THU	-	0.6	9.3	10	12	12	12
1979-10-25	16H00	THU	-	0.0	8.6	13	12	12	12
1979-10-25	18H00	THU	-	0.6	8.3	13	12	12	11
1979-10-25	20H00	THU	-	0.0	7.8	14	11	13	12
1979-10-25	22H00	THU	-	0.6	8.6	14	12	13	12
1979-10-25	24H00	THU	-	0.0	8.5	13	12	13	12
1979-10-26	2H00	FRI	-	1.1	8.7	14	12	14	9.7
1979-10-26	4H00	FRI	-	1.4	8.8	14	12	13	12
1979-10-26	6H00	FRI	-	0.8	9.1	13	12	13	12
1979-10-26	8H00	FRI	-	0.8	9.5	13	11	13	11
1979-10-26	10H00	FRI	-	0.8	8.9	12	11	12	12
1979-10-26	12H00	FRI	-	0.5	8.9	11	10	11	-
1979-10-26	14H00	FRI	-	-	8.7	12	11	11	-
1979-10-26	16H00	FRI	-	1.0	8.6	12	11	11	-
1979-10-26	18H00	FRI	-	-	8.1	11	11	12	-
1979-10-26	20H00	FRI	-	0.7	8.3	13	12	12	-
1979-10-26	22H00	FRI	-	-	8.3	13	11	12	-
1979-10-26	24H00	FRI	-	0.2	8.0	12	11	12	-
1979-10-27	2H00	SAT	-	0.0	-	-	-	-	-
1979-10-27	4H00	SAT	-	0.0	8.8	7.0	8.6	7.3	-
1979-10-27	6H00	SAT	-	0.6	9.7	14	11	13	-
1979-10-27	8H00	SAT	-	0.7	9.4	13	12	13	-
1979-10-27	10H00	SAT	-	0.6	8.7	13	12	12	-
NO. RESULTS			0	45	44	46	44	44	35
MEAN			-	1.1	9.8	13	12	13	14
STD. DEVIATION			-	1.3	1.9	1.6	1.5	1.6	1.9

TOTAL PHOSPHORUS AS P

DATE	TIME	DAY	G1B6	G1Z12	G1Z22	G1Z32	G1Z42	G1Z52	G1E2
1979-10-21	12H00	SUN	3.7	-	-	-	-	-	4.3
1979-10-21	14H00	SUN	2.9	-	-	-	-	-	4.4
1979-10-21	16H00	SUN	3.6	-	-	-	-	-	4.4
1979-10-21	18H00	SUN	4.6	-	-	-	-	-	4.4
1979-10-21	20H00	SUN	5.4	-	-	-	-	-	4.4
1979-10-21	22H00	SUN	6.1	-	-	-	-	-	4.5
1979-10-21	24H00	SUN	4.9	-	-	-	-	-	4.4
1979-10-22	2H00	MON	6.1	-	-	-	-	-	4.4
1979-10-22	4H00	MON	6.0	-	-	-	-	-	4.4
1979-10-22	6H00	MON	5.4	-	-	-	-	-	4.4
1979-10-22	8H00	MON	4.6	-	-	-	-	-	4.4
1979-10-22	10H00	MON	3.2	-	-	-	-	-	4.5
1979-10-23	12H00	TUE	5.0	-	-	-	-	-	5.7
1979-10-23	14H00	TUE	5.5	-	-	-	-	-	5.4
1979-10-23	16H00	TUE	6.0	-	-	-	-	-	5.6
1979-10-23	18H00	TUE	6.7	-	-	-	-	-	5.1
1979-10-23	20H00	TUE	6.0	-	-	-	-	-	5.0
1979-10-23	22H00	TUE	6.9	-	-	-	-	-	5.2
1979-10-23	24H00	TUE	7.5	-	-	-	-	-	5.0
1979-10-24	2H00	WED	7.0	-	-	-	-	-	5.0
1979-10-24	4H00	WED	6.6	-	-	-	-	-	5.1
1979-10-24	6H00	WED	6.0	-	-	-	-	-	5.2
1979-10-24	8H00	WED	4.9	-	-	-	-	-	5.2
1979-10-24	10H00	WED	4.1	-	-	-	-	-	5.0
1979-10-25	12H00	THU	4.0	-	-	-	-	-	3.1
1979-10-25	14H00	THU	4.7	-	-	-	-	-	2.7
1979-10-25	16H00	THU	5.1	-	-	-	-	-	2.3
1979-10-25	18H00	THU	6.2	-	-	-	-	-	2.7
1979-10-25	20H00	THU	7.0	-	-	-	-	-	2.7
1979-10-25	22H00	THU	7.4	-	-	-	-	-	2.8
1979-10-25	24H00	THU	7.1	-	-	-	-	-	2.9
1979-10-26	2H00	FRI	6.9	-	-	-	-	-	2.8
1979-10-26	4H00	FRI	6.5	-	-	-	-	-	2.9
1979-10-26	6H00	FRI	4.7	-	-	-	-	-	2.9
1979-10-26	8H00	FRI	5.9	-	-	-	-	-	2.9
1979-10-26	10H00	FRI	5.2	-	-	-	-	-	3.1
1979-10-26	12H00	FRI	4.7	-	-	-	-	-	2.7
1979-10-26	14H00	FRI	5.9	-	-	-	-	-	2.5
1979-10-26	16H00	FRI	6.4	-	-	-	-	-	1.8
1979-10-26	18H00	FRI	7.4	-	-	-	-	-	1.7
1979-10-26	20H00	FRI	7.9	-	-	-	-	-	1.8
1979-10-26	22H00	FRI	8.1	-	-	-	-	-	2.0
1979-10-26	24H00	FRI	7.8	-	-	-	-	-	2.4
1979-10-27	2H00	SAT	7.4	-	-	-	-	-	3.8
1979-10-27	4H00	SAT	6.8	-	-	-	-	-	2.6
1979-10-27	6H00	SAT	6.2	-	-	-	-	-	2.4
1979-10-27	8H00	SAT	5.7	-	-	-	-	-	2.2
1979-10-27	10H00	SAT	5.4	-	-	-	-	-	2.0
NO. RESULTS			48	0	0	0	0	0	48
MEAN			5.8	-	-	-	-	-	3.7
STD. DEVIATION			1.3	-	-	-	-	-	1.2

ORTHO - PHOSPHATE AS P

DATE	TIME	DAY	G1B6	G1Z12	G1Z22	G1Z32	G1Z42	G1Z52	G1E2
1979-10-21	12H00	SUN	2,2	4,2	3,8	3,4	3,3	2,8	3,7
1979-10-21	14H00	SUN	2,1	3,1	3,2	3,2	3,3	2,8	3,8
1979-10-21	16H00	SUN	1,9	4,8	3,7	3,3	3,1	2,9	3,8
1979-10-21	18H00	SUN	2,3	5,6	3,4	3,8	3,2	3,0	3,8
1979-10-21	20H00	SUN	2,9	5,3	3,9	3,7	3,7	3,1	3,7
1979-10-21	22H00	SUN	3,5	4,8	3,6	4,0	3,6	3,0	3,8
1979-10-21	24H00	SUN	3,0	5,0	3,1	4,0	3,2	3,7	3,8
1979-10-22	2H00	MON	3,3	2,6	2,4	3,1	2,2	2,1	3,8
1979-10-22	4H00	MON	2,9	4,2	4,6	4,3	-	-	3,8
1979-10-22	6H00	MON	3,2	4,3	4,2	4,4	-	-	0,4
1979-10-22	8H00	MON	2,6	3,8	4,2	4,2	3,8	3,9	3,6
1979-10-22	10H00	MON	1,9	3,6	4,2	4,3	3,9	3,9	3,8
1979-10-23	12H00	TUE	3,0	8,2	5,5	5,0	5,1	5,4	5,2
1979-10-23	14H00	TUE	3,1	8,4	5,2	4,5	4,8	4,9	5,1
1979-10-23	16H00	TUE	3,8	10	5,3	4,5	4,6	4,4	4,9
1979-10-23	18H00	TUE	4,2	13	-	4,5	4,3	4,2	4,6
1979-10-23	20H00	TUE	4,8	17	-	4,8	4,1	4,2	4,5
1979-10-23	22H00	TUE	5,1	18	6,9	5,3	4,6	4,3	4,6
1979-10-23	24H00	TUE	4,9	19	7,1	5,4	4,9	4,5	4,5
1979-10-24	2H00	WED	4,3	17	7,7	6,0	5,3	4,7	4,4
1979-10-24	4H00	WED	4,2	18	7,6	5,4	5,3	4,8	4,5
1979-10-24	6H00	WED	3,8	15	6,9	5,4	5,2	4,8	4,6
1979-10-24	8H00	WED	3,6	11	5,7	4,8	4,8	4,4	4,6
1979-10-24	10H00	WED	2,8	7,3	4,9	4,2	4,4	4,3	4,4
1979-10-25	12H00	THU	2,8	7,3	3,4	2,6	3,0	2,6	3,1
1979-10-25	14H00	THU	3,1	10	3,1	2,9	2,5	2,6	2,7
1979-10-25	16H00	THU	3,4	16	3,9	2,5	2,4	2,2	2,3
1979-10-25	18H00	THU	4,6	18	4,9	3,0	2,5	2,2	2,4
1979-10-25	20H00	THU	4,6	19	5,7	3,5	2,8	2,4	2,5
1979-10-25	22H00	THU	4,7	19	6,2	3,9	3,1	2,6	2,6
1979-10-25	24H00	THU	4,6	20	6,5	4,1	3,4	3,2	2,7
1979-10-26	2H00	FRI	4,6	18	7,0	4,5	3,9	3,2	2,8
1979-10-26	4H00	FRI	4,4	17	7,2	4,6	3,9	3,1	2,9
1979-10-26	6H00	FRI	3,1	19	7,4	4,7	4,0	3,3	2,9
1979-10-26	8H00	FRI	4,1	15	6,4	4,4	3,9	3,4	2,9
1979-10-26	10H00	FRI	3,5	11	5,1	3,8	3,8	3,1	3,1
1979-10-26	12H00	FRI	2,8	5,4	4,2	3,2	3,3	2,7	-
1979-10-26	14H00	FRI	3,6	6,2	3,7	2,9	2,8	2,5	-
1979-10-26	16H00	FRI	3,9	8,6	4,3	2,6	2,3	1,8	-
1979-10-26	18H00	FRI	4,5	9,3	5,0	3,8	2,5	1,7	-
1979-10-26	20H00	FRI	4,6	9,5	5,6	3,1	2,4	1,8	-
1979-10-26	22H00	FRI	5,0	10	5,9	3,2	2,5	1,9	-
1979-10-26	24H00	FRI	4,7	9,6	6,1	3,7	2,9	2,2	-
1979-10-27	2H00	SAT	4,5	6,2	3,5	3,4	3,3	3,7	-
1979-10-27	4H00	SAT	3,9	6,3	3,4	3,2	3,1	2,5	-
1979-10-27	6H00	SAT	3,1	8,2	5,5	3,4	3,0	2,3	-
1979-10-27	8H00	SAT	3,2	8,8	4,9	2,7	3,0	2,2	-
1979-10-27	10H00	SAT	2,8	5,3	4,5	2,4	2,4	1,5	-
NO. RESULTS			48	48	46	48	46	46	36
MEAN			3,6	10	5,0	3,9	3,6	3,2	3,6
STD. DEVIATION			0,9	5,5	1,4	0,9	0,9	1,0	1,0

APPENDIX A5

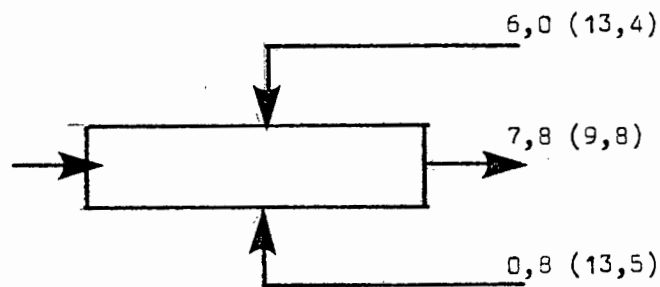
OXYGEN CONSUMPTION RATE

Date	Time of Day h	Oxygen Consumption Rate mg/l/h			Date	Time of Day h	Oxygen Consumption Rate mg/l/h		
		Position in Aeration basin					Position in Aeration basin		
		Start	Middle	End			Start	Middle	End
21/22 October 1979	1200	10,3	9,8	8,4	23/24 October 1979	1200	11,9	10,2	7,0
	1400	9,8	8,8	8,0		1400	13,9	9,2	7,7
	1600	18,9	9,7	7,2		1600	16,4	12,1	7,7
	1800	29,5	14,1	7,7		1800	21,3	13,5	7,7
	2000	20,5	14,1	7,2		2000	23,0	18,1	8,8
	2200	19,7	11,3	6,8		2200	20,5	22,3	8,9
	2400	13,9	10,6	8,3		2400	19,7	19,5	8,3
	0200	10,3	9,2	8,4		0200	20,5	20,3	8,8
	0400	9,8	8,5	8,8		0400	20,5	14,8	8,8
	0600	8,9	8,5	7,7		0600	16,4	10,6	9,3
	0800	8,4	7,8	7,2		0800	8,2	19,7	8,8
1000	7,8	7,1	7,9	1000	10,0	22,6	8,3		
25/26 October 1979	1200	24,6	19,7	5,9	26/27 October 1979	1200	18,0	10,6	9,3
	1400	25,5	21,2	6,2		1400	21,1	11,3	9,3
	1600	29,8	18,3	9,3		1600	29,9	16,9	9,8
	1800	32,3	21,2	9,8		1800	29,5	24,0	9,3
	2000	30,6	28,2	10,3		2000	32,8	31,0	11,6
	2200	42,5	28,2	16,5		2200	31,2	21,2	21,7
	2400	32,3	25,4	11,1		2400	35,9	28,2	21,2
	0200	34,0	28,2	10,3		0200	35,7	29,6	14,0
	0400	34,0	15,6	9,8		0400	35,4	28,3	9,3
	0600	30,6	18,3	9,6		0600	30,6	18,3	9,3
	0800	27,2	12,0	9,0		0800	22,1	12,7	10,3
1000	20,6	10,2	9,3	1000	17,2	11,3	9,8		

APPENDIX A 6.1.

GOUKOPPIE PLANT

CALCULATION OF NITROGEN BALANCE

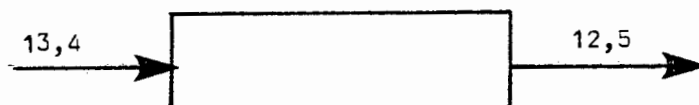
a) Nitrogen loss due to denitrificationPrimary anaerobic/anoxic zones

Figures in brackets depict the nitrate concentration and the other figures depict the flow relative to the incoming flow

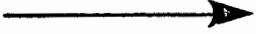
$$\begin{aligned}
 \text{Nitrogen loss in first} &= \text{NO}_3/\text{N}_{\text{in}} - \text{NO}_3/\text{N}_{\text{out}} \\
 \text{anaerobic/anoxic basins} &= 6,0 (13,4) + 0,88 (13,5) - 7,88 (9,8) \\
 &= 80,4 + 11,9 - 77,2 \\
 &= 92,3 - 77,2 \\
 &= 15,1 \text{ mg N/l}
 \end{aligned}$$

Secondary anoxic zone

Nitrate loss in the secondary anoxic zone due to denitrification is depicted below.



In this case the flows entering and leaving the second anoxic zone are the same. The figures depict the nitrate concentration in and out of this zone.

$$\begin{aligned}
 \text{Hence the nitrate nitrogen loss} &= \text{NO}_3/\text{N}_{\text{in}} - \text{NO}_3/\text{N}_{\text{out}} \\
 &= 13,4 - 12,5 \\
 &= 0,9 \text{ mg N/l}
 \end{aligned}$$


b) Total loss of nitrogen

$$\begin{aligned}
 \text{The total loss of nitrogen due to denitrification} &= 15,1 + 0,9 \\
 &= 16,0 \text{ mg N/l}
 \end{aligned}$$

Nitrogen balance

$$\text{TKN influent} = 35 \text{ mg N/l}$$

$$\text{TKN effluent} = 2,0 \text{ mg N/l}$$

$$\text{NO}_3 \text{ in effluent} = 13,5 \text{ mg N/l}$$

$$\text{Nitrogen in waste sludge} = \frac{f_n X_v V}{R_s Q}$$

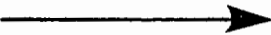
$$f_n = \text{nitrogen fraction of the sludge} = 0,1$$

$$X_v = \text{VSS volatile fraction of MLSS} = 0,72 \text{ MLSS}$$

$$V = \text{volume of process} \quad (l)$$

$$Q = \text{flow to plant} \quad (l/d)$$

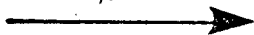
$$R_s = \text{Sludge age} \quad (d)$$

$$\begin{aligned}
 \text{Hence nitrogen loss in waste sludge} &= \frac{0,1 \cdot 0,72 \cdot 1940 \cdot 29,080 \times 10^6}{22 \cdot 30400 \times 10^6} \\
 &= 6,1 \text{ mg/l}
 \end{aligned}$$


Nitrogen loss due to denitrification = 16,0 mg N/l

Total concentration of nitrogen removed = 2,0 + 13,5 + 6,1 + 16,0
= 37,6 mg N/l

Hence nitrogen recovery = $\frac{37,6}{35} \times 100$
= 107 %



APPENDIX A 6.2.

GOUKOPPIE PLANT

CALCULATION OF COD BALANCE

One mg/l of nitrate nitrogen is equivalent to 2,85 mg/l of oxygen.

The equivalent oxygen demand for denitrification for the process is denoted by O_{dp}

$$\begin{aligned} \text{Hence } O_{dp} &= \frac{2,85 \text{ (nitrogen loss due to denitrification)} Q}{24 \times V} \\ &= \frac{2,85 (15,1 + 0,9) \cdot 30,4 \times 10^6}{24 \cdot 29,080 \times 10^6} \\ &= 1,99 \text{ mg/l/h} \end{aligned}$$

For every 1 mg N/l of ammonia converted to nitrate 4,57 mg O/l of oxygen are consumed. The nitrification oxygen demand for the process is denoted O_{np} .

$$\begin{aligned} \text{Hence } O_{np} &= \frac{4,57 \text{ (NO}_3 \text{ loss during denitrification + NO}_3 \text{ in effluent)} Q}{24 \times V} \\ &= \frac{4,57 (16,0 + 13,5) \cdot 30,4 \times 10^6}{24 \cdot 29,08 \times 10^6} \\ &= 5,88 \text{ mg/l/h} \end{aligned}$$

The oxygen consumption rate for the two aeration zones is as follows:

$$\text{Primary aeration zone} = O_{t1} = 16,2 \text{ mg/l/h}$$

$$\text{Secondary aeration zone} = O_{t2} = 9,5 \text{ mg/l/h}$$

The total oxygen demand for the process is denoted by O_{tp}

$$\begin{aligned}
 O_{tp} &= \frac{O_{t1}(\text{Vol. primary aeration basin}) + 9,5(\text{Vol. secondary aeration basin})}{\text{Total volume}} \\
 &= \frac{16,2 (14,7 \times 10^6) + 9,5 (2,7 \times 10^6)}{29,08 \times 10^6} \\
 &= \frac{238 + 26}{29,08} \\
 &= 9,1 \text{ mg/l/h} \longrightarrow
 \end{aligned}$$

Equivalent carbon oxygen demand for the process is denoted by O_{cp}

$$\begin{aligned}
 O_{cp} &= O_{tp} + O_{dp} - O_{np} \\
 &= 9,1 + 1,99 - 5,88 \\
 &= 5,21 \text{ mg/l/h} \longrightarrow
 \end{aligned}$$

Carbonaceous material balance

$$\text{Influent COD} = 270 \text{ mg/l} \qquad \text{Effluent COD} = 63 \text{ mg/l}$$

$$\begin{aligned}
 \text{Oxygen demand} &= \frac{O_{cp} \cdot 24 V}{Q} \\
 &= \frac{5,21 \cdot 24 \cdot 29,08 \times 10^6}{30,4 \times 10^6} \\
 &= 119,6 \text{ mg/l}
 \end{aligned}$$

$$\text{COD loss in waste sludge} = \frac{P X_v V}{R_s Q} \qquad \begin{array}{l} P = 1,48 \\ X_v = 0,72 \text{ MLSS mg/l} \end{array}$$

Hence

$$\begin{aligned}
 \text{Loss} &= \frac{1,48 \cdot 0,72 \cdot 1940 \cdot 29,08 \times 10^6}{22 \cdot 30,4 \times 10^6} \\
 &= 89,9 \text{ mg/l} \longrightarrow
 \end{aligned}$$

$$\begin{aligned}\text{Total COD recovered} &= 63 + 119,6 + 89,8 \\ &= 273 \text{ mg/l COD}\end{aligned}$$

$$\% \text{ recovery} = \frac{273}{272} \times 100$$

$$= 99,8$$

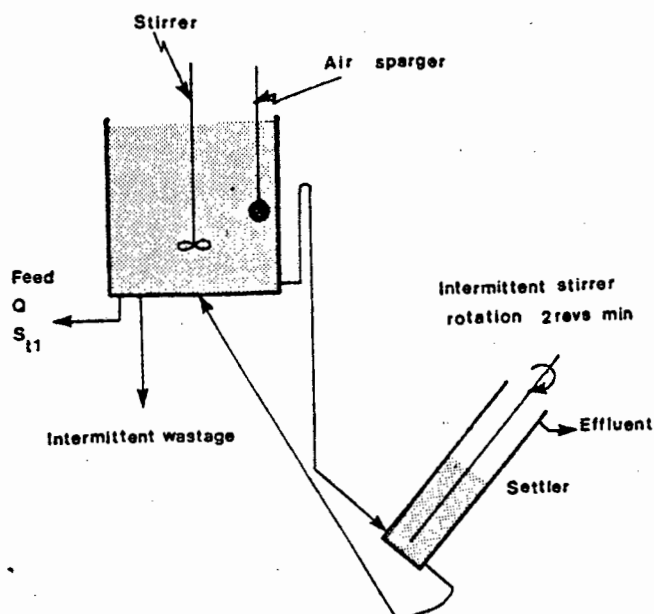


APPENDIX A7

CALCULATION OF READILY BIODEGRADABLE INFLUENT COD FRACTION

1. Reactor configuration

Single reactor (aerobic) with return of settled sludge.

2. Operation

- 1) Each day collect the volume of waste to be fed over 24 hours (Q litres) at the same time each day - place in a feed container which is stirred enough to keep solids in suspension, but not too much or else oxygen is taken up.
- 2) The feed pump must be set to feed the volume Q litres over 34 hours ($\pm 1/2$ hour) It is convenient to start the feed at, say, 9 a.m. Take a sample for COD analysis (S_{ti}).

- 3) Several hours after feed has started (say 2 p.m.) measure the Oxygen Utilisation Rate (OUR) as follows:
 With additional air supply, raise D.O. in reactor to about 7 - 8 mg O/l. With feed still on, switch off air supply and monitor drop in D.O. with time. Call this OUR_Q mg O/l/h
- 4) After first OUR test raise D.O. again to about 7 - 8 mg O/l. Switch off feed pump. After about 3 minutes switch off air supply and measure OUR again. Call this $OUR_{no Q}$ mg O/l/h
- 5) Restart feed. After about 1/2 hour steps 3 and 4 could be repeated to obtain more data if required.

3. Calculation

Q = feed rate (l/d)

V = reactor volume (l)

1) $\Delta O = (OUR_Q - OUR_{no Q})$ mg O/l/h

2) Easily biodegradable COD concentration

$$S_{bsi} = \frac{\Delta O \cdot V \cdot 3,33 \cdot 24}{Q}$$

3) Fraction of total COD which is easily biodegradable

$$= S_{bsi} / S_{ti}$$

4) or Fraction of total biodegradable COD which is easily biodegradable

$$f_{bs} = f_{ca} = S_{bsi} / S_{ti} (1 - f_{us} - PF_{up})$$

4. Example calculation for Goudkoppie data

BTE data:

$$\begin{aligned}
 Q &= 50 \text{ l/d} \\
 V &= 7,5 \text{ l} \\
 S_{ti} &= 370 \text{ mg COD/l} \\
 OUR_Q &= 24,1 \text{ mg O/l/h} \\
 OUR_{no Q} &= 20,1 \text{ mg O/l/h}
 \end{aligned}$$

$$\Delta O = 24,1 - 20,1 = 4,0 \text{ mg O/l/h}$$

$$\begin{aligned}
 S_{bsi} &= \frac{4,0 \cdot 7,5 \cdot 3,33 \cdot 24}{50,0} \\
 &= 48 \text{ mg COD/l}
 \end{aligned}$$

Therefore fraction of total COD which is easily biodegradable

$$= 48 / 370 = 0,13 \text{ mg COD/mg COD}$$

3. Advice about loading rate

In order to obtain accurate results it is desirable that OUR is not too small i.e. 4 mg O/l/h. The load on the plant should be such that the OUR_Q is about 35 - 40 mg O/l/h. This can be obtained if the loading rate is about

$$\frac{QS_{ti}}{V} = \frac{36 \cdot 500}{7} \approx 2500 \text{ mg COD/l reactor/d}$$

However, if S_{ti} is well below 500 mg COD/l, then Q may have to be very large to obtain this load. Q must not be so large that the retention time is less than about 0,15 day. i.e. V/Q not less than 0,15 day

APPENDIX A8

NORTHERN WORKS

The raw data obtained during the experimental period

CODE

N3 BE 3	Balancing tank effluent
N3 Z 13	Anaerobic zone
N3 Z 23	Primary anoxic zone
N3 Z 33	Primary aeration zone
N3 Z 43	Secondary anoxic zone
N3 E 3	Effluent

CHEMICAL OXYGEN DEMAND AS O

DATE	TIME	DAY	N3BE3	N3Z13	N3Z23	N3Z33	N3Z43	N3Z53	N3E3
1980- 8-10	7H00	SUN	250	-	-	-	-	-	60
1980- 8-10	9H00	SUN	230	-	-	-	-	-	60
1980- 8-10	11H00	SUN	200	-	-	-	-	-	50
1980- 8-10	13H00	SUN	160	-	-	-	-	-	50
1980- 8-10	15H00	SUN	190	-	-	-	-	-	55
1980- 8-10	17H00	SUN	190	-	-	-	-	-	35
1980- 8-10	19H00	SUN	220	-	-	-	-	-	40
1980- 8-10	21H00	SUN	230	-	-	-	-	-	40
1980- 8-10	23H00	SUN	250	-	-	-	-	-	50
1980- 8-11	1H00	MON	300	-	-	-	-	-	45
1980- 8-11	3H00	MON	310	-	-	-	-	-	50
1980- 8-11	5H00	MON	290	-	-	-	-	-	45
1980- 8-11	7H00	MON	220	-	-	-	-	-	45
1980- 8-11	9H00	MON	210	-	-	-	-	-	60
1980- 8-11	11H00	MON	180	-	-	-	-	-	45
1980- 8-11	13H00	MON	170	-	-	-	-	-	40
1980- 8-11	15H00	MON	230	-	-	-	-	-	40
1980- 8-11	17H00	MON	230	-	-	-	-	-	35
1980- 8-11	19H00	MON	260	-	-	-	-	-	40
1980- 8-11	21H00	MON	290	-	-	-	-	-	40
1980- 8-11	23H00	MON	300	-	-	-	-	-	40
1980- 8-12	1H00	TUE	290	-	-	-	-	-	51
1980- 8-12	3H00	TUE	290	-	-	-	-	-	35
1980- 8-12	5H00	TUE	280	-	-	-	-	-	35
1980- 8-14	7H00	THU	260	-	-	-	-	-	40
1980- 8-14	9H00	THU	230	-	-	-	-	-	45
1980- 8-14	11H00	THU	210	-	-	-	-	-	45
1980- 8-14	13H00	THU	180	-	-	-	-	-	35
1980- 8-14	15H00	THU	200	-	-	-	-	-	40
1980- 8-14	17H00	THU	210	-	-	-	-	-	35
1980- 8-14	19H00	THU	230	-	-	-	-	-	45
1980- 8-14	21H00	THU	240	-	-	-	-	-	66
1980- 8-14	23H00	THU	250	-	-	-	-	-	45
1980- 8-15	1H00	FRI	300	-	-	-	-	-	56
1980- 8-15	3H00	FRI	280	-	-	-	-	-	56
1980- 8-15	5H00	FRI	280	-	-	-	-	-	45
1980- 8-15	7H00	FRI	220	-	-	-	-	-	56
1980- 8-15	9H00	FRI	210	-	-	-	-	-	61
1980- 8-15	11H00	FRI	180	-	-	-	-	-	40
1980- 8-15	13H00	FRI	150	-	-	-	-	-	51
1980- 8-15	15H00	FRI	220	-	-	-	-	-	45
1980- 8-15	17H00	FRI	260	-	-	-	-	-	45
1980- 8-15	19H00	FRI	270	-	-	-	-	-	40
1980- 8-15	21H00	FRI	280	-	-	-	-	-	51
1980- 8-15	23H00	FRI	280	-	-	-	-	-	45
1980- 8-16	1H00	SAT	260	-	-	-	-	-	45
1980- 8-16	3H00	SAT	290	-	-	-	-	-	51
1980- 8-16	5H00	SAT	280	-	-	-	-	-	51
NO. RESULTS			48	0	0	0		0	48
MEAN			240	-	-	-		-	46
STD. DEVIATION			43	-	-	-		-	8

CHEMICAL OXYGEN DEMAND (SOLUBLE) AS O

DATE	TIME	DAY	N3BE3	N3E3	N3Z13	N3Z23	N3Z33	N3Z43	N3Z53
1980- 8-10	7H00	SUN	210	35	80	45	50	35	30
1980- 8-10	9H00	SUN	210	35	70	50	55	35	45
1980- 8-10	11H00	SUN	200	30	70	45	35	30	35
1980- 8-10	13H00	SUN	140	35	65	40	50	35	50
1980- 8-10	15H00	SUN	130	45	70	50	40	70	50
1980- 8-10	17H00	SUN	130	30	80	40	45	40	30
1980- 8-10	19H00	SUN	150	35	75	40	45	35	35
1980- 8-10	21H00	SUN	160	35	85	40	35	35	30
1980- 8-10	23H00	SUN	180	35	65	45	35	40	25
1980- 8-11	1H00	MON	190	45	76	45	40	25	40
1980- 8-11	3H00	MON	230	35	71	51	35	25	40
1980- 8-11	5H00	MON	210	35	66	51	30	30	45
1980- 8-11	7H00	MON	180	40	66	40	30	40	40
1980- 8-11	9H00	MON	170	35	66	40	35	35	40
1980- 8-11	11H00	MON	150	45	56	35	45	25	35
1980- 8-11	13H00	MON	110	40	61	35	30	40	35
1980- 8-11	15H00	MON	140	35	66	76	30	35	40
1980- 8-11	17H00	MON	170	35	66	51	35	35	30
1980- 8-11	19H00	MON	190	30	71	51	30	35	35
1980- 8-11	21H00	MON	220	30	71	40	35	40	40
1980- 8-11	23H00	MON	230	40	66	45	30	45	35
1980- 8-12	1H00	TUE	200	45	85	45	35	30	25
1980- 8-12	3H00	TUE	210	35	85	45	35	30	25
1980- 8-12	5H00	TUE	220	30	75	50	30	35	30
1980- 8-14	7H00	THU	160	40	80	51	60	15	45
1980- 8-14	9H00	THU	160	35	75	45	50	15	40
1980- 8-14	11H00	THU	130	35	70	45	45	14	40
1980- 8-14	13H00	THU	100	35	70	35	45	14	35
1980- 8-14	15H00	THU	130	35	70	35	45	14	35
1980- 8-14	17H00	THU	130	35	60	35	45	14	35
1980- 8-14	19H00	THU	150	35	65	30	35	15	35
1980- 8-14	21H00	THU	150	35	55	45	50	15	35
1980- 8-14	23H00	THU	160	45	80	61	55	16	50
1980- 8-15	1H00	FRI	160	45	80	56	50	45	45
1980- 8-15	3H00	FRI	160	35	85	40	50	55	35
1980- 8-15	5H00	FRI	160	40	80	45	40	50	35
1980- 8-15	7H00	FRI	140	30	90	40	50	35	35
1980- 8-15	9H00	FRI	130	50	65	40	40	30	35
1980- 8-15	11H00	FRI	110	35	55	40	45	45	45
1980- 8-15	13H00	FRI	91	40	55	40	30	40	45
1980- 8-15	15H00	FRI	130	35	55	45	45	45	45
1980- 8-15	17H00	FRI	160	40	75	40	40	50	35
1980- 8-15	19H00	FRI	170	35	96	35	40	30	40
1980- 8-15	21H00	FRI	170	35	71	35	40	40	30
1980- 8-15	23H00	FRI	170	35	66	40	60	35	30
1980- 8-16	1H00	SAT	190	35	91	56	50	40	45
1980- 8-16	3H00	SAT	190	35	81	51	50	35	50
1980- 8-16	5H00	SAT	160	40	81	71	50	40	40
NO. RESULTS			48	48	48	48	48	48	48
MEAN			160	37	72	45	42	34	38
STD. DEVIATION			34	5	10	9	8	12	7

AMMONIA AS N

			N3E3	N3Z13	N3Z23	N3Z33	N3Z43	N3Z53	N3E3
DATE	TIME	DAY							
1980- 8-10	7H00	SUN	24	14	5,4	0,0	0,0	0,0	0,0
1980- 8-10	9H00	SUN	24	14	5,4	0,0	0,0	0,0	0,0
1980- 8-10	11H00	SUN	24	14	5,2	0,0	0,0	0,0	0,0
1980- 8-10	13H00	SUN	24	14	5,2	0,0	0,0	0,0	0,0
1980- 8-10	15H00	SUN	25	14	5,2	0,0	0,0	0,0	0,0
1980- 8-10	17H00	SUN	29	15	7,2	0,0	0,0	0,0	0,0
1980- 8-10	19H00	SUN	29	16	6,4	0,0	0,0	0,0	0,0
1980- 8-10	21H00	SUN	30	15	5,2	0,0	0,0	0,0	0,0
1980- 8-10	23H00	SUN	29	14	6,4	0,0	0,0	0,0	0,0
1980- 8-11	1H00	MON	23	22	12	2,5	3,0	0,0	0,0
1980- 8-11	3H00	MON	24	21	11	2,0	2,5	0,0	0,0
1980- 8-11	5H00	MON	23	21	11	1,5	2,0	0,0	0,0
1980- 8-11	7H00	MON	23	23	12	1,5	2,5	0,0	0,0
1980- 8-11	9H00	MON	21	23	12	1,5	3,5	0,0	0,0
1980- 8-11	11H00	MON	21	21	11	0,0	0,0	0,0	0,0
1980- 8-11	13H00	MON	22	21	11	1,5	0,0	0,0	0,0
1980- 8-11	15H00	MON	29	26	12	2,0	0,0	0,0	0,0
1980- 8-11	17H00	MON	29	25	14	2,0	0,0	0,0	0,0
1980- 8-11	19H00	MON	23	26	14	2,5	2,5	0,0	0,0
1980- 8-11	21H00	MON	27	25	13	3,0	3,5	0,0	0,0
1980- 8-11	23H00	MON	23	23	13	2,5	2,5	0,0	0,0
1980- 8-12	1H00	TUE	24	16	7,0	1,8	1,0	0,0	0,0
1980- 8-12	3H00	TUE	25	15	7,0	1,1	0,0	0,0	0,0
1980- 8-12	5H00	TUE	23	14	6,3	1,1	0,0	0,0	0,0
1980- 8-14	7H00	THU	19	12	4,5	0,0	0,0	0,0	0,0
1980- 8-14	9H00	THU	19	11	4,2	0,0	0,0	0,0	0,0
1980- 8-14	11H00	THU	17	12	3,9	0,0	0,0	0,0	0,0
1980- 8-14	13H00	THU	19	11	3,4	0,0	0,0	0,0	0,0
1980- 8-14	15H00	THU	28	15	4,4	0,0	0,0	0,0	0,0
1980- 8-14	17H00	THU	23	15	4,6	0,0	0,0	0,0	0,0
1980- 8-14	19H00	THU	11	16	4,7	0,0	0,0	0,0	0,0
1980- 8-14	21H00	THU	21	16	5,0	0,0	0,0	0,0	0,0
1980- 8-14	23H00	THU	24	16	5,3	0,0	0,0	0,0	0,0
1980- 8-15	1H00	FRI	22	13	4,9	1,1	0,0	0,3	0,0
1980- 8-15	3H00	FRI	20	13	4,4	0,5	0,0	0,0	0,6
1980- 8-15	5H00	FRI	20	12	4,1	0,0	0,0	0,0	0,7
1980- 8-15	7H00	FRI	20	14	4,6	0,0	0,0	0,2	0,0
1980- 8-15	9H00	FRI	20	12	4,1	0,6	0,0	0,0	0,0
1980- 8-15	11H00	FRI	18	12	3,9	0,7	0,0	0,2	0,0
1980- 8-15	13H00	FRI	20	11	3,8	0,8	0,0	0,0	0,0
1980- 8-15	15H00	FRI	23	15	4,8	0,6	0,0	0,0	0,0
1980- 8-15	17H00	FRI	28	16	5,0	0,6	0,0	0,0	0,0
1980- 8-15	19H00	FRI	24	16	5,3	0,6	0,0	0,0	0,0
1980- 8-15	21H00	FRI	26	13	4,8	0,9	0,0	0,0	0,0
1980- 8-15	23H00	FRI	22	14	5,5	1,1	0,0	0,0	0,0
1980- 8-16	1H00	SAT	24	13	5,6	0,0	0,0	0,0	0,0
1980- 8-16	3H00	SAT	24	14	5,1	0,8	0,0	0,0	0,0
1980- 8-16	5H00	SAT	22	12	4,5	0,7	0,0	0,0	0,0
NO. RESULTS			48	48	48	48	48	48	48
MEAN			24	16	6,8	0,7	0,5	0,0	0,0
STD. DEVIATION			3,9	4,4	3,3	0,9	1,1	0,1	0,1

NITRATE AS N

DATE	TIME	DAY	N3BE3	N3Z13	N3Z23	N3Z33	N3Z43	N3Z53	N3E3
1980-8-10	7H00	SUN	-	5.0	9.7	16	17	13	15
1980-8-10	9H00	SUN	-	3.7	9.7	16	15	15	15
1980-8-10	11H00	SUN	-	3.7	9.7	15	15	15	15
1980-8-10	13H00	SUN	-	3.8	9.8	15	15	15	15
1980-8-10	15H00	SUN	-	4.0	9.8	15	15	15	15
1980-8-10	17H00	SUN	-	4.4	11	15	15	15	15
1980-8-10	19H00	SUN	-	4.3	9.6	16	16	15	15
1980-8-10	21H00	SUN	-	4.0	9.8	16	16	15	15
1980-8-10	23H00	SUN	-	5.0	12	-	14	15	17
1980-8-11	1H00	MON	-	4.7	11	17	17	13	19
1980-8-11	3H00	MON	-	4.7	10	17	16	17	19
1980-8-11	5H00	MON	-	4.0	10	16	16	16	19
1980-8-11	7H00	MON	-	5.1	12	13	18	13	18
1980-8-11	9H00	MON	-	4.6	12	19	18	13	18
1980-8-11	11H00	MON	-	5.5	12	13	17	13	18
1980-8-11	13H00	MON	-	5.1	12	13	17	17	18
1980-8-11	15H00	MON	-	4.8	12	13	17	17	13
1980-8-11	17H00	MON	-	5.5	12	19	17	13	18
1980-8-11	19H00	MON	-	4.9	13	19	17	13	18
1980-8-11	21H00	MON	-	4.8	13	19	13	19	13
1980-8-11	23H00	MON	-	5.5	12	13	13	19	13
1980-8-12	1H00	TUE	-	2.0	3.6	15	14	15	15
1980-8-12	3H00	TUE	-	2.2	3.3	15	14	15	14
1980-8-12	5H00	TUE	-	2.2	3.4	15	14	15	14
1980-8-14	7H00	THU	-	3.2	9.4	16	15	13	16
1980-8-14	9H00	THU	-	2.9	3.6	15	15	16	15
1980-8-14	11H00	THU	-	3.2	9.1	15	14	15	15
1980-8-14	13H00	THU	-	3.4	3.8	15	14	15	15
1980-8-14	15H00	THU	-	3.6	9.5	15	14	13	15
1980-8-14	17H00	THU	-	3.5	9.7	16	14	14	15
1980-8-14	19H00	THU	-	3.3	9.2	16	15	15	15
1980-8-14	21H00	THU	-	3.2	3.6	16	15	14	15
1980-8-14	23H00	THU	-	4.1	11	13	16	17	16
1980-8-15	1H00	FRI	-	4.5	11	13	17	13	18
1980-8-15	3H00	FRI	-	4.3	11	17	16	17	17
1980-8-15	5H00	FRI	-	4.2	11	17	16	17	13
1980-8-15	7H00	FRI	-	4.1	11	17	17	17	17
1980-8-15	9H00	FRI	-	4.6	11	17	16	17	17
1980-8-15	11H00	FRI	-	4.9	12	17	17	17	17
1980-8-15	13H00	FRI	-	5.4	11	16	16	17	17
1980-8-15	15H00	FRI	-	5.0	11	16	16	17	17
1980-8-15	17H00	FRI	-	5.0	11	17	16	17	17
1980-8-15	19H00	FRI	-	3.2	12	13	16	17	17
1980-8-15	21H00	FRI	-	4.9	11	13	16	17	17
1980-8-15	23H00	FRI	-	5.3	12	13	17	13	13
1980-8-16	1H00	SAT	-	4.6	10	17	16	17	16
1980-8-16	3H00	SAT	-	4.3	10	17	16	17	17
1980-8-16	5H00	SAT	-	4.1	11	17	17	17	17
NO. RESULTS			0	43	43	47	43	43	43
MEAN			-	4.3	10	17	16	16	17
STD. DEVIATION			-	0.9	1.1	1.3	1.2	1.5	1.4

TOTAL PHOSPHORUS AS P

DATE	TIME	DAY	N3BE3	N3Z13	N3Z23	N3Z33	N3Z43	N3Z53	N3E3
1980- 8-10	7H00	SUN	9.0	-	-	-	-	-	7.4
1980- 8-10	9H00	SUN	9.2	-	-	-	-	-	7.3
1980- 8-10	11H00	SUN	10	-	-	-	-	-	7.3
1980- 8-10	13H00	SUN	9.0	-	-	-	-	-	7.2
1980- 8-10	15H00	SUN	8.4	-	-	-	-	-	7.1
1980- 8-10	17H00	SUN	8.0	-	-	-	-	-	7.4
1980- 8-10	19H00	SUN	8.8	-	-	-	-	-	7.8
1980- 8-10	21H00	SUN	9.6	-	-	-	-	-	9.0
1980- 8-10	23H00	SUN	9.8	-	-	-	-	-	8.0
1980- 8-11	1H00	MON	15	-	-	-	-	-	7.1
1980- 8-11	3H00	MON	15	-	-	-	-	-	7.2
1980- 8-11	5H00	MON	15	-	-	-	-	-	7.3
1980- 8-11	7H00	MON	9.0	-	-	-	-	-	7.2
1980- 8-11	9H00	MON	8.4	-	-	-	-	-	7.3
1980- 8-11	11H00	MON	7.6	-	-	-	-	-	7.4
1980- 8-11	13H00	MON	7.4	-	-	-	-	-	7.3
1980- 8-11	15H00	MON	9.6	-	-	-	-	-	7.3
1980- 8-11	17H00	MON	11	-	-	-	-	-	7.0
1980- 8-11	19H00	MON	13	-	-	-	-	-	6.8
1980- 8-11	21H00	MON	14	-	-	-	-	-	6.9
1980- 8-11	23H00	MON	15	-	-	-	-	-	7.2
1980- 8-12	1H00	TUE	11	-	-	-	-	-	7.7
1980- 8-12	3H00	TUE	9.6	-	-	-	-	-	8.0
1980- 8-12	5H00	TUE	9.2	-	-	-	-	-	7.6
1980- 8-14	7H00	THU	8.0	-	-	-	-	-	8.4
1980- 8-14	9H00	THU	7.6	-	-	-	-	-	8.2
1980- 8-14	11H00	THU	6.6	-	-	-	-	-	8.2
1980- 8-14	13H00	THU	6.2	-	-	-	-	-	8.4
1980- 8-14	15H00	THU	7.2	-	-	-	-	-	8.0
1980- 8-14	17H00	THU	8.0	-	-	-	-	-	7.3
1980- 8-14	19H00	THU	8.8	-	-	-	-	-	7.6
1980- 8-14	21H00	THU	9.6	-	-	-	-	-	7.7
1980- 8-14	23H00	THU	10	-	-	-	-	-	7.4
1980- 8-15	1H00	FRI	12	-	-	-	-	-	7.6
1980- 8-15	3H00	FRI	10	-	-	-	-	-	8.6
1980- 8-15	5H00	FRI	10	-	-	-	-	-	8.7
1980- 8-15	7H00	FRI	7.8	-	-	-	-	-	8.5
1980- 8-15	9H00	FRI	7.0	-	-	-	-	-	8.7
1980- 8-15	11H00	FRI	6.2	-	-	-	-	-	7.5
1980- 8-15	13H00	FRI	6.4	-	-	-	-	-	7.2
1980- 8-15	15H00	FRI	8.0	-	-	-	-	-	7.3
1980- 8-15	17H00	FRI	10	-	-	-	-	-	7.1
1980- 8-15	19H00	FRI	11	-	-	-	-	-	7.0
1980- 8-15	21H00	FRI	12	-	-	-	-	-	7.0
1980- 8-15	23H00	FRI	12	-	-	-	-	-	7.3
1980- 8-16	1H00	SAT	11	-	-	-	-	-	7.1
1980- 8-16	3H00	SAT	11	-	-	-	-	-	8.1
1980- 8-16	5H00	SAT	9.4	-	-	-	-	-	7.8
NO. RESULTS			48	0	0	0	0	0	48
MEAN			9.7	-	-	-	-	-	7.6
STD. DEVIATION			2.4	-	-	-	-	-	0.5

ORTHO - PHOSPHATE AS P

DATE	TIME	DAY	N3BE3	N3Z13	N3Z23	N3Z33	N3Z43	N3Z53	N3E3
1980- 8-10	7H00	SUN	6.4	7.0	6.8	6.6	6.4	6.0	6.2
1980- 8-10	9H00	SUN	6.0	6.6	6.6	6.4	7.0	6.2	6.2
1980- 8-10	11H00	SUN	5.2	6.6	4.6	6.4	6.8	6.2	6.3
1980- 8-10	13H00	SUN	4.6	3.0	6.4	6.4	6.4	6.0	6.3
1980- 8-10	15H00	SUN	4.8	5.8	6.0	6.2	6.0	5.8	6.3
1980- 8-10	17H00	SUN	5.2	5.8	6.0	5.8	6.0	5.8	6.4
1980- 8-10	19H00	SUN	5.6	6.4	5.6	6.0	6.2	5.6	6.3
1980- 8-10	21H00	SUN	6.2	6.8	5.8	5.8	6.2	5.6	6.4
1980- 8-10	23H00	SUN	6.6	6.6	6.4	5.8	6.2	5.6	5.9
1980- 8-11	1H00	MON	9.0	8.3	7.5	7.1	7.1	7.0	6.3
1980- 8-11	3H00	MON	9.4	8.6	8.0	7.6	7.7	7.5	6.3
1980- 8-11	5H00	MON	9.4	8.5	8.0	7.9	7.9	7.7	6.2
1980- 8-11	7H00	MON	6.2	6.5	6.4	6.4	6.5	6.4	6.1
1980- 8-11	9H00	MON	5.8	6.3	6.2	6.3	6.3	6.4	6.1
1980- 8-11	11H00	MON	5.2	5.7	6.0	6.1	6.4	6.2	6.3
1980- 8-11	13H00	MON	4.6	5.4	5.7	5.9	6.3	7.8	6.2
1980- 8-11	15H00	MON	6.0	5.9	5.7	5.8	6.0	6.1	6.2
1980- 8-11	17H00	MON	7.6	6.3	6.1	5.8	6.0	6.0	6.2
1980- 8-11	19H00	MON	8.4	7.0	6.4	6.2	6.1	6.0	6.1
1980- 8-11	21H00	MON	8.8	7.3	6.8	6.6	6.4	6.4	6.2
1980- 8-11	23H00	MON	9.4	7.9	7.3	6.8	6.7	6.7	6.2
1980- 8-12	1H00	TUE	8.2	8.1	7.6	7.3	7.4	7.4	7.2
1980- 8-12	3H00	TUE	7.4	7.9	7.6	7.3	7.6	7.5	7.3
1980- 8-12	5H00	TUE	6.8	7.7	7.6	7.5	7.7	7.6	7.4
1980- 8-14	7H00	THU	6.0	7.5	7.1	7.1	7.0	7.1	7.0
1980- 8-14	9H00	THU	5.4	7.3	7.0	7.1	7.0	7.1	7.0
1980- 8-14	11H00	THU	4.6	6.6	6.7	6.9	6.9	7.0	7.1
1980- 8-14	13H00	THU	3.8	5.8	6.4	6.6	6.7	6.8	6.9
1980- 8-14	15H00	THU	4.8	6.2	6.2	6.3	6.5	6.8	6.9
1980- 8-14	17H00	THU	5.8	6.8	6.4	6.2	6.3	6.5	6.7
1980- 8-14	19H00	THU	4.8	7.3	6.5	6.2	6.3	6.3	6.6
1980- 8-14	21H00	THU	5.0	7.6	6.7	6.3	6.4	6.4	6.6
1980- 8-14	23H00	THU	7.6	7.3	6.6	6.4	6.3	6.2	6.3
1980- 8-15	1H00	FRI	3.0	7.8	6.9	6.8	6.5	6.4	6.3
1980- 8-15	3H00	FRI	7.2	7.9	7.3	7.0	6.7	6.8	6.8
1980- 8-15	5H00	FRI	7.4	7.6	7.1	6.9	6.9	7.0	6.9
1980- 8-15	7H00	FRI	6.0	7.0	6.7	6.6	6.6	6.6	6.5
1980- 8-15	9H00	FRI	5.0	6.4	6.6	6.5	6.5	6.5	6.5
1980- 8-15	11H00	FRI	4.0	6.5	6.4	6.6	6.4	6.4	6.6
1980- 8-15	13H00	FRI	4.0	5.9	6.3	6.5	6.5	6.5	6.5
1980- 8-15	15H00	FRI	5.6	5.9	6.1	6.2	6.1	6.2	6.5
1980- 8-15	17H00	FRI	6.8	6.8	6.1	6.2	6.1	6.0	6.4
1980- 8-15	19H00	FRI	7.8	6.9	6.4	6.2	6.1	6.0	6.4
1980- 8-15	21H00	FRI	8.2	7.8	6.8	6.6	6.4	6.3	6.4
1980- 8-15	23H00	FRI	8.8	7.5	6.8	6.6	6.4	6.3	6.4
1980- 8-16	1H00	SAT	8.0	7.5	7.0	6.6	6.7	6.7	6.4
1980- 8-16	3H00	SAT	7.8	7.7	6.9	6.5	6.6	6.6	6.9
1980- 8-16	5H00	SAT	7.2	7.1	7.0	6.9	6.7	6.8	6.7
NO. RESULTS			48	48	48	48	48	48	48
MEAN			6.5	6.9	6.6	6.5	6.6	6.5	6.5
STD. DEVIATION			1.6	1.0	0.7	0.5	0.5	0.6	0.3

KJELDAHL NITROGEN AS N

			N3BE3	N3Z13	N3Z23	N3Z33	N3Z43	N3Z53	N3E3
DATE	TIME	DAY							
1980- 8-10	7H00	SUN	36	-	-	-	-	-	2,6
1980- 8-10	9H00	SUN	34	-	-	-	-	-	2,8
1980- 8-10	11H00	SUN	34	-	-	-	-	-	2,3
1980- 8-10	13H00	SUN	33	-	-	-	-	-	2,6
1980- 8-10	15H00	SUN	38	-	-	-	-	-	1,9
1980- 8-10	17H00	SUN	43	-	-	-	-	-	1,6
1980- 8-10	19H00	SUN	42	-	-	-	-	-	1,3
1980- 8-10	21H00	SUN	45	-	-	-	-	-	1,6
1980- 8-10	23H00	SUN	42	-	-	-	-	-	2,3
1980- 8-11	1H00	MON	38	-	-	-	-	-	1,3
1980- 8-11	3H00	MON	38	-	-	-	-	-	1,4
1980- 8-11	5H00	MON	38	-	-	-	-	-	1,3
1980- 8-11	7H00	MON	36	-	-	-	-	-	1,4
1980- 8-11	9H00	MON	34	-	-	-	-	-	1,4
1980- 8-11	11H00	MON	32	-	-	-	-	-	1,4
1980- 8-11	13H00	MON	32	-	-	-	-	-	1,3
1980- 8-11	15H00	MON	44	-	-	-	-	-	1,5
1980- 8-11	17H00	MON	42	-	-	-	-	-	1,1
1980- 8-11	19H00	MON	42	-	-	-	-	-	1,6
1980- 8-11	21H00	MON	40	-	-	-	-	-	1,3
1980- 8-11	23H00	MON	38	-	-	-	-	-	1,3
1980- 8-12	1H00	TUE	37	-	-	-	-	-	2,2
1980- 8-12	3H00	TUE	33	-	-	-	-	-	1,9
1980- 8-12	5H00	TUE	32	-	-	-	-	-	1,5
1980- 8-14	7H00	THU	34	-	-	-	-	-	3,5
1980- 8-14	9H00	THU	36	-	-	-	-	-	3,2
1980- 8-14	11H00	THU	34	-	-	-	-	-	2,5
1980- 8-14	13H00	THU	38	-	-	-	-	-	2,2
1980- 8-14	15H00	THU	48	-	-	-	-	-	2,8
1980- 8-14	17H00	THU	44	-	-	-	-	-	2,4
1980- 8-14	19H00	THU	46	-	-	-	-	-	2,4
1980- 8-14	21H00	THU	44	-	-	-	-	-	2,6
1980- 8-14	23H00	THU	44	-	-	-	-	-	2,5
1980- 8-15	1H00	FRI	40	-	-	-	-	-	1,6
1980- 8-15	3H00	FRI	34	-	-	-	-	-	3,0
1980- 8-15	5H00	FRI	38	-	-	-	-	-	2,8
1980- 8-15	7H00	FRI	38	-	-	-	-	-	3,0
1980- 8-15	9H00	FRI	38	-	-	-	-	-	2,6
1980- 8-15	11H00	FRI	32	-	-	-	-	-	2,9
1980- 8-15	13H00	FRI	36	-	-	-	-	-	1,7
1980- 8-15	15H00	FRI	46	-	-	-	-	-	2,6
1980- 8-15	17H00	FRI	50	-	-	-	-	-	2,1
1980- 8-15	19H00	FRI	44	-	-	-	-	-	3,0
1980- 8-15	21H00	FRI	40	-	-	-	-	-	2,2
1980- 8-15	23H00	FRI	38	-	-	-	-	-	2,8
1980- 8-16	1H00	SAT	38	-	-	-	-	-	1,7
1980- 8-16	3H00	SAT	42	-	-	-	-	-	2,8
1980- 8-16	5H00	SAT	34	-	-	-	-	-	3,3
NO. RESULTS			48	0	0	0	0	0	48
MEAN			39	-	-	-	-	-	2,1
STD. DEVIATION			4,6	-	-	-	-	-	0,7

OXYGEN CONSUMPTION RATE

NORTHERN WORKS

Date	Time of Day h	Oxygen Consumption Rate mg/l/h			Date	Time of Day h	Oxygen Consumption Rate mg/l/h		
		Position in Aeration Basin					Position in Aeration Basin		
		Start	Middle	End			Start	Middle	End
10/8/80	0700	34,2	22,4	9,6	14/8/80	0700	32,4	29,8	8,7
	0900	34,2	22,4	8,7		0900	34,2	29,8	6,3
	1100	34,2	34,0	7,8		1100	32,4	27,2	4,8
	1300	33,4	28,1	7,8		1300	32,4	23,8	6,3
	1500	32,4	22,1	6,9		1500	36,0	26,4	5,4
	1700	34,2	26,4	7,5		1700	36,0	29,8	6,0
	1900	34,2	28,1	7,5		1900	36,0	28,9	10,2
	2100	36,0	30,6	10,5		2100	32,4	30,6	14,7
11/8/80	2300	36,0	26,4	13,2	15/8/80	2300	32,4	29,8	13,8
	0100	36,0	29,8	-		0100	34,2	29,8	12,6
	0300	36,0	31,5	7,5		0300	32,4	28,9	11,7
	0500	36,0	32,3	10,5		0500	34,2	27,2	12,3
	0700	36,7	31,5	9,0		0700	34,2	28,1	7,2
	0900	34,0	28,9	8,7		0900	34,2	26,4	6,0
	1100	33,4	27,2	7,8		1100	37,5	23,0	5,1
	1300	34,0	23,8	7,2		1300	37,5	27,2	5,1
12/8/80	1500	31,6	23,0	6,3	16/8/80	1500	40,5	26,4	4,8
	1700	28,2	21,3	7,5		1700	39,0	28,9	7,5
	1900	34,0	31,5	13,5		1900	39,0	-	10,8
	2100	35,2	32,3	18,6		2100	39,0	30,6	12,9
	2300	36,0	26,4	13,2		2300	39,0	30,6	11,7
	0100	36,0	29,8	-		0100	36,0	28,9	13,2
	0300	36,0	31,5	7,5		0300	36,0	28,1	15,6
	0500	36,0	32,3	10,5		0500	36,0	28,9	9,6

APPENDIX A 9

NORTHERN WORKS

NITROGEN BALANCE

TKN influent	(mg N/l)	39
TKN effluent	(mg N/l)	2,1
Nitrate nitrogen effluent	(mg N/l)	17
Total nitrogen in waste sludge	(mg N/l)	2,9
Nitrogen loss due to denitrification	(mg N/l)	16,8
Total nitrogen recovered	(mg N/l)	38,8
% recovery		99

COD BALANCE

Influent COD	(mg/l)	240
Effluent COD	(mg/l)	46
Oxygen demand	(mg/l)	128
COD in waste sludge	(mg/l)	43
Total COD recovered	(mg/l)	217
% Recovery		90

COD concentrations; raw sewage N3I, settled sewage N3TE,
balancing tank effluent N3BE2

				N 3 I	N 3 T E	N 3 B E 2
1980-7-6	2H00	SUN		600	370	420
1980-7-6	4H00	SUN		520	360	430
1980-7-6	6H00	SUN		430	320	410
1980-7-6	8H00	SUN		350	310	390
1980-7-6	10H00	SUN		350	340	380
1980-7-6	12H00	SUN		410	350	370
1980-7-6	14H00	SUN		740	270	380
1980-7-6	16H00	SUN		790	290	330
1980-7-6	18H00	SUN		740	310	310
1980-7-6	20H00	SUN		350	330	420
1980-7-6	22H00	SUN		770	330	360
1980-7-6	24H00	SUN		630	430	350
1980-7-7	2H00	MON		510	360	390
1980-7-7	4H00	MON		510	340	320
1980-7-7	6H00	MON		430	310	330
1980-7-7	8H00	MON		270	300	310
1980-7-7	10H00	MON		310	380	240
1980-7-7	12H00	MON		300	250	280
1980-7-7	14H00	MON		440	320	290
1980-7-7	16H00	MON		770	360	350
1980-7-7	18H00	MON		610	410	390
1980-7-7	20H00	MON		450	430	360
1980-7-7	22H00	MON		640	360	350
1980-7-7	24H00	MON		680	420	360
1980-7-8	2H00	TUE		600	410	340
1980-7-8	4H00	TUE		510	370	350
1980-7-8	6H00	TUE		430	320	360
1980-7-8	8H00	TUE		340	360	350
1980-7-8	10H00	TUE		320	300	410
1980-7-8	12H00	TUE		670	310	400
1980-7-8	14H00	TUE		730	390	310
1980-7-8	16H00	TUE		750	420	340
1980-7-8	18H00	TUE		710	430	340
1980-7-8	20H00	TUE		830	470	170
1980-7-8	22H00	TUE		690	470	390
1980-7-8	24H00	TUE		670	480	330
1980-7-9	2H00	WED		680	450	240
1980-7-9	4H00	WED		650	420	420
1980-7-9	6H00	WED		420	440	420
1980-7-9	8H00	WED		290	430	400
1980-7-9	10H00	WED		290	360	360
1980-7-9	12H00	WED		790	330	390
1980-7-9	14H00	WED		590	310	340
1980-7-9	16H00	WED		590	330	400
1980-7-9	18H00	WED		730	410	390
1980-7-9	20H00	WED		770	420	420
1980-7-9	22H00	WED		750	420	410
1980-7-9	24H00	WED		770	460	-
1980-7-10	2H00	THU		790	450	410
1980-7-10	4H00	THU		630	470	410
1980-7-10	6H00	THU		620	450	390
1980-7-10	8H00	THU		300	290	380
1980-7-10	10H00	THU		250	230	370
1980-7-10	12H00	THU		650	380	330
1980-7-10	14H00	THU		700	340	300
1980-7-10	16H00	THU		760	370	310
1980-7-10	18H00	THU		730	370	320
1980-7-10	20H00	THU		700	390	340
1980-7-10	22H00	THU		660	440	350
1980-7-10	24H00	THU		660	370	370
1980-7-11	2H00	FRI		670	360	390
1980-7-11	4H00	FRI		670	390	370
1980-7-11	6H00	FRI		430	360	360
1980-7-11	8H00	FRI		750	330	330
1980-7-11	10H00	FRI		830	360	310
1980-7-11	12H00	FRI		840	280	310
1980-7-11	14H00	FRI		790	350	330
1980-7-11	16H00	FRI		760	370	370
1980-7-11	18H00	FRI		760	410	370
1980-7-11	20H00	FRI		760	400	360
1980-7-11	22H00	FRI		680	440	380
1980-7-11	24H00	FRI		680	450	380
1980-7-12	2H00	SAT		700	440	340
1980-7-12	4H00	SAT		720	420	350
1980-7-12	6H00	SAT		440	360	340
1980-7-12	8H00	SAT		370	400	290
1980-7-12	10H00	SAT		350	350	300
1980-7-12	12H00	SAT		330	800	280
1980-7-12	14H00	SAT		330	790	310
1980-7-12	16H00	SAT		350	780	350
1980-7-12	18H00	SAT		350	760	370
1980-7-12	20H00	SAT		370	790	360
1980-7-12	22H00	SAT		370	630	400
1980-7-12	24H00	SAT		370	700	410
NO RESULTS				84	84	83
MEAN				580	400	350
STD. DEVIATION				180	120	50

19 JUL 1983