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INTERIM REPORT OF THE
UNIDO FERTILIZER MISSION TO INDONESIA *

(Under SIS ID/OA 220 INDO (8))

From 15 October to 15 November 1968

* This interim report has been prepared for the use of UNIDO and the Government of Indonesia. It has been cleared by them. The views and opinions expressed in this report are those of the author and do not necessarily reflect the views of the secretariat of UNIDO.

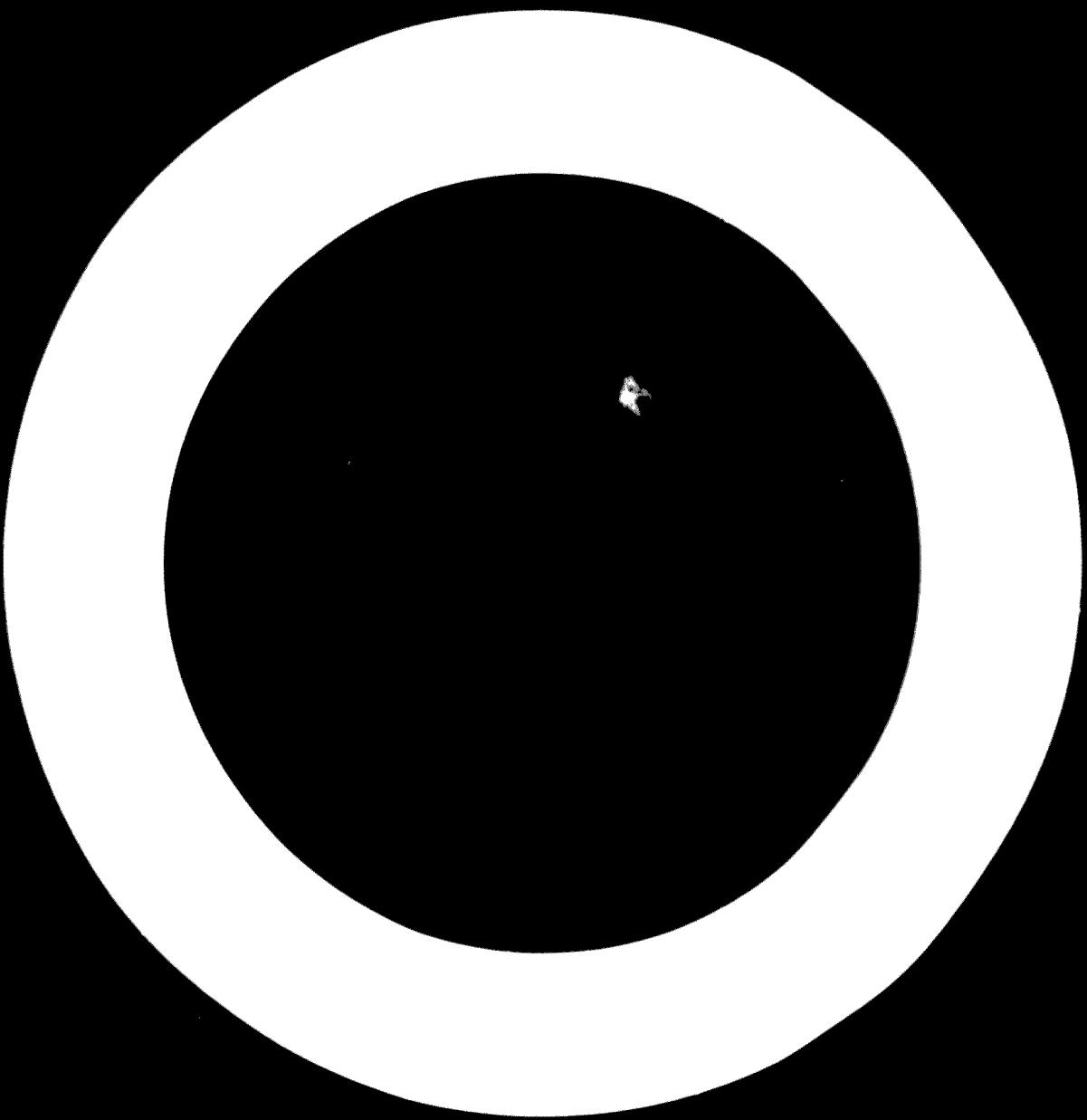


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THE FIFTH MISSION REPORT

INDONESIA

(SRI BUMI TEGAL INDONESIA)

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I. INTRODUCTION

1. During the International Symposium on Industrial Development held in Athens during November-December 1967 contacts were established between the Indonesian Government Delegates and UNIDO. During the discussions an interest was expressed that UNIDO undertake a fertilizer mission to Indonesia to study certain specific problems related to the industry. Subsequently an official request was received through the Resident Representative and the terms of reference were as follows:

The fertilizer mission should study and make recommendations for the Government to take action and also for future UNIDO assistance on

1. the Tjilatjap phosphate plant in south central Java,
2. the Wanaradja sulphur plant in west Java, and
3. the Gresik ammonia plant at Surabaja in east Java.

To undertake the above, UNIDO organized an expert mission under the Special Industrial Services (SIS) consisting of Mr. M.C. Varghese, leader (nitrogen expert), Mr. F.T. Nielsson (expert on phosphates), and Prof. R. Kiyoura (sulphur expert). The first two members of the mission were briefed by the Industries Division of ECAFE on the 18th and 19th October and the visit to Indonesia was carried out between 19 October and 12 November 1968.

2. The mission was greatly assisted by the valuable guidance of the Resident Representative in Djakarta, Mr. Aly Ahmed Gritly, and the Deputy Resident Representative, Mr. Cvido Grisogono. The mission was received by the Director General of Chemical Industries, Col. Eng. Agus Sujono, and the Deputy Director General, Eng. Anwar Ibrahim. After discussing the overall aspects of the mission an itinerary was drawn up and three counterparts were assigned:

Ir. Nico Kansil (leader and nitrogen expert)
Ir. Supratignjo (Tjilatjap plant manager and phosphate expert)
Ir. Sjafri Saarin (sulphur expert)

* The mission is extremely thankful to the above officials for assisting and guiding them in carrying out the mission successfully. A detailed list of the officials with whom the mission had discussions is given in Appendix 4.
.....

II. BACKGROUND INFORMATION ON THE FERTILIZER INDUSTRY
IN INDONESIA

A. The requirements of fertilizers in Indonesia (NPK) have been estimated by different authorities as follows:

1. Mr. J. Mahan and Mr. J. Maes, both FAO experts, have estimated that requirements of fertilizer in Indonesia in the form of basic elements are as follows:

N	749,941 tons
P ₂ O ₅	710,742 tons
K ₂ O	318,467 tons

It should be noted that these are only requirements.

2. Prof. Go Ban Hong of the Department of Agriculture, estimates as follows:

For 1970 - N	225,000 tons
P ₂ O ₅	183,000 tons
K ₂ O	58,000 tons

For 1975 - N	480,000 tons
P ₂ O ₅	457,000 tons
K ₂ O	230,000 tons

3. The Department of Agriculture, going by trends as existing in 1963 and 1964 and disregarding the reduction in consumption in 1965 and 1966, estimates the following consumption of nitrogen, expressed in metric tons:

	1968/'69	1969/'70	1970/'71	1971/'72
Wet Season				
Dry Season	1969	1970	1971	1972
Paddy	101,200	108,100	124,300	140,400
Non-Paddy *	36,800	39,100	41,400	46,000
Estates	<u>40,340</u>	<u>44,890</u>	<u>46,330</u>	<u>47,970</u>
Total	178,340	192,090	212,030	234,370
=====	=====	=====	=====	=====

* Exclusive of nitrogen in compound fertilizers which are imported for special reasons.

4. Using a different approach, namely multiplying total area of fertilized paddy land by nitrogen application per hectare, the nitrogen consumption in 1969 and 1970 is expressed in metric tons as follows:

	Wet Season 1968/69	1969/70
Dry Season	1969	1970
Java	124,636	168,983
Sumatra	18,905	27,430
Rest of Indonesia	<u>10,769</u>	<u>16,420</u>
Total	154,310	212,833
=====	=====	=====

To this must be added consumption on non-paddy land and by the Estates estimated at 90,000 tons per annum.

5. Messrs. John Van der Valk and Associates recently estimated that with the production from the existing Pusri I, namely 48,500 tons per year, and the Petrokemia plant at Gresik now under construction producing 59,500 tons of nitrogen per year by 1973, the above two totalling a figure of 108,000 tons, a possible production by 1973 from the expansion of Pusri (Pusri II) with 600 metric tons of ammonia per day and producing 115,000 tons per year of N, the total availability in 1973 will be 223,000 tons of nitrogen. From the above it can be seen that this indigenous production will not be sufficient to meet the projected nitrogen consumption in 1973 of about 300,000 tons per year. Van der Valk came to a different conclusion than the above.

B. Existing units.

1. There is an ammonia and urea plant in Palembang (Pusri I) producing 100,000 metric tons of urea per year using natural gas as raw material.
2. The Petrokemia at Gresik, which is under construction, is expected to produce by 1970, 45,000 tons of urea per year and 150,000 tons of ammonium sulphate, using indigenous fuel oil and imported sulphur.
3. The phosphate project at Tjilatjap, obtained with bilateral aid from the USSR, is designed to produce 100,000 tons of simple superphosphate per year, using imported phosphate rock and imported or domestic sulphur. This plant is 30% erected but construction has been at a stand still since 1966.

4. The Wanaradja sulphur plant is producing 3 tons of sulphur per day from volcanic sulphur muds.

C. The World Bank and IFC have been involved in the Indonesian fertilizer sector in various ways as follows:

1. In co-operation with USAID, IFC wrote the terms of reference for a study of the feasibility of expanding the Pusri fertilizer plant in Sumatra. USAID has engaged a consultant, John Van der Valk and Associates, who are presently in Indonesia completing this study.
2. IFC has also completed a study on natural gas reserves in Indonesia and furnished a report by IFC consultants, J.C. Sproule and Associates Ltd., to the Government of Indonesia.
3. At the request of the Government of Indonesia, the World Bank Group has undertaken to do a National Fertilizer Study for Indonesia covering all aspects of this sector of the Indonesian economy.

The present UNIDO fertilizer mission therefore did not visit Pusri but concentrated on the Tjilatjap phosphate project, the Wanaradja plant, and the Gresik ammonia plant.

III. TENTATIVE CONCLUSIONS AND RECOMMENDATIONS

1. Since in the Tjilatjap phosphate project 92% of the equipment has been delivered and 30% of the plant has been erected, and a total of about US\$ 8 million has been invested, the mission very strongly recommends the Government of Indonesia to take immediate action to complete this project either by continuation of the bilateral assistance or, failing this, through other assistance. This plant can use the sulphur from Wanaradja when it is developed and can use imported rock phosphate to start with. This plant can produce triple superphosphate at a later stage utilizing imported phosphoric acid, which is being supplied in ocean going ships to India from Mexico. The plant can also use excess ammonia from Gresik to make ammoniated phosphate and produce a $N-P_2O_5-K_2O$
of 5- 20- 0

2. We recommend assistance from UNIDO in the form of an expert on phosphatic fertilizer production for one year to assist this project in its speedy completion and start-up. A draft Job Description is attached (Appendix 10).

3. We recommend that UNDP and UNIDC assist the Indonesian Government in putting up a pilot plant at Wanarodja to develop a process to economically recover sulphur.

..... Ultimately sulphur needed both by the Tjilatjap and Gresik plants (50,000 tons per annum) should be produced in a commercial plant based on the pilot plant results. The pilot plant will use the sulphur muds from the Telaga Bodas crater. A draft Special Fund request is attached to our report. This pilot project will establish the most economic and suitable process for the commercial unit. (App. 11)

4. For the Gresik fertilizer project, in order to assist in suitably disposing of the 6,000 tons of excess ammonia per year, which is most vital to the economic running of this plant, we suggest a Special Fund project for the direct application of ammonia for paddy and sugar can estates close to the Gresik plant. A draft request for this project is attached to our report. (App. 12)

5. The Gresik fertilizer plant has facility for making their product mix flexible, namely, they can produce within certain limits more urea and less ammonium sulphate, or less urea and more ammonium sulphate. The requirements of the market have still not been fully assessed. We recommend that UNIDO assist this project with a marketing expert to assist the plant with carrying out a market survey and to assist in determining the proper product mix. A draft Job Description is attached to our report. (App. 14)

IV. TJILATJAP PHOSPHATE PLANT, SOUTH CENTRAL JAVA

The plant is located 3.5 km east of the city of Tjilatjap and 100 metres southeast of the village of Karangtalung. It was sold by Technopromexport, Moscow under a bilateral agreement between the Government of the USSR and the Government of Indonesia. Part of the contract involved a geological survey by Russian experts of the phosphate and sulphur deposits of Indonesia which was carried out in 1961-1963. Construction of the plant was begun in 1961 and terminated in 1965 with the cessation of Russian aid agreements. Ninety two per cent of the equipment was delivered and thirty per cent of the plant was erected when the project was stopped. Currently there are 420 persons on the plant payroll with 191 employed and the remainder on uncertain leave. The repair shop is operational as a small industry utilizing the foundry and the carpenter shop for industrial contracts.

The initial contracts were for US\$ 9,380,500 of Russian Letters of Credit (L/C) and US\$ 2,117,000 of Czechoslovakian L/C for a total of US\$ 11,497,500. Part of the Russian L/C have not been used. None of the Czechoslovakian L/C have been opened, so the current L/C investment in the plant is US\$ 8,115,000. The figure is necessarily approximate due to variations in the exchange rate of the local currency, the Rupiah.

The housing for expatriate personnel, some 40 dwellings, is complete, some of it now being occupied by local project personnel.

The equipment material storage and repair maintenance shop are essentially complete.

A 2 x 6000 Kw powerplant is part of the project. One boiler is essentially complete, the other is about 30 % finished. The plant power requirements will be 3500 Kw during operation. The power system is tied into the Tjilatjap system which uses 1000 Kw. At present the power for the project and for the housing development is supplied by portable diesel generators.

The sulphuric acid plant is about 60 % complete with buildings, sulphur melter, blowers, converter shell, SO₃ cooler and acid coolers in place.

The phosphate operation is the least advanced with only the steel structure for the building in place.

Except for storage tank sections, equipment is either erected or in crates under cover. The erected equipment is painted and in excellent shape.

Surprisingly, the erected equipment appears to have been put up yesterday instead of three years ago. The project manager is trying his best to keep erected equipment in good shape, e.g. he keeps the piping filled with rain water to minimize corrosion. How long this newly-erected look will continue is questionable in view of the climatic conditions in the area.

One of the questions that was explored at great length was that "farmers do not want to use single superphosphate and prefer triple superphosphate".

Investigation reveals that the single superphosphate in 1962 was powdered material that apparently set up into hard lumps while in transit overseas and kept getting progressively harder in storage in Indonesia. It is understandable that such a product would be deemed unsatisfactory. On the other hand, the triple superphosphate being imported is granular and remains free flowing over a long period of time.

One of the most important points regarding the Tjilitjip phosphate plant is that it has a granulation section in that the product will be granular not fine fl wis. For this reason, we believe that the farmer will accept granular single superphosphate in readily as granular triple superphosphate, especially if the price of the product is favourable.

.....
In Appendices 1 & 2 we show price factors and costs for the manufacture of granular single superphosphate. These cost indicate that with imported triple phosphate and imported sulphur, the production cost will be US\$ 151.0 per ton f_bo_g, if the raw materials are trans-shipped via Singapore at US\$ 145.0 per ton if the raw materials are shipped directly to Tjilitjip. This compares with a C.I.F. imported price in 1962 for powdered single superphosphate of US\$ 175.0 per ton f_bo_g, and the 1967 C.I.F. imported price for granular triple superphosphate of US\$ 174.0 per ton f_bo_g.

The harbour at Tjilitjip is excellent, but the removal of 30 wrecked ships in the harbour so that farmers can difficultly and large ships cannot enter. Clearing of the harbour is underway and should be complete in 1970. The sulphur sulphuric acid plant cannot be finished before 1970. Therefore, about the time the plant is finished, the harbour will be cleared and direct shipments could arrive. As a result, we believe a production cost of US\$ 138.0 per ton of f_bo_g is a realistic cost for a plant in terms for granular single superphosphate at Tjilitjip.

A problem with new plants in Indonesia is that they must be self-sufficient according to Government standards. This means they must provide housing for workers and supervisors, have their own source of electric power, their own sewerage facilities and their own source of water. As a result, the total plant cost is fantastic as compared to a battery limits plant in a developed country.

.....
It appears as if a final cost of US\$ 11,656,000 may be a realistic figure for the total plant investment. One and one-half million Dollars can be assigned to the sulphuric acid plant because this is a self contained facility. Sulphuric acid costs on this basis are shown in Appendix 5. However, we do not know how to allocate the probable revenues from the power plant or the probable income from the housing facilities. Therefore, the remaining US\$ 10,156,000 of capital has been assigned to the superphosphate operation. Such a figure puts a tremendous depreciation burden on the cost of production because normally a granular superphosphate plant could be erected for well under one million Dollars, in which case the production cost of granular single superphosphate under Indonesian conditions would be US\$ 127.0 per ton of f_bo_g.

Also, if the superphosphate operation were isolated, a 30 year depreciation period could be assumed. However, when one considers the large number of permanent concrete buildings, sewage and water treatment facilities, a 40 year depreciation period for the project, exclusive of the acid plant, appears reasonable.

The important thing to remember regarding the Tjilatjap phosphate operation is that because it has a granulation section, infinite variations in plant food manufacture are possible.

Once the plant is finished, an additional investment of about US\$ 30,000 makes possible the production of granular ammoniated superphosphate of a 5-20-0 grade. The addition of urea to the superphosphate feed plus ammoniation could result in an 18-12-0 grade of granular mixed fertilizer, or in any other desired grade with an additional investment of about US\$ 10,000.

Another important factor is that the plant equipment can be used to produce a wide variety of phosphates ranging from 20 % P₂O₅ to 46 % P₂O₅, all in granular form, by using sulphuric acid, phosphoric acid, or a combination of the two acids as described in Appendix 8.

The trade journals have had articles indicating that the Government of India has signed a contract with a Mexican producer to obtain wet-process phosphoric acid, C.I.F. India, at US\$ 125.0 per ton of P₂O₅. We can assume that a ship could discharge most of its acid cargo in India and discharge the remainder at Tjilatjap for an additional US\$ 5.0 per ton P₂O₅, or a cost at the plant of US\$ 130.0 per ton P₂O₅.

To utilize imported phosphoric acid would require about US\$ 50,000 additional investment for storage tanks, piping and meters. With US\$ 130.0 per ton P₂O₅ acid and directly shipped phosphate rock, triple superphosphate could be produced at Tjilatjap for a cost of US\$ 130.0 per ton of P₂O₅ (App. 7). This compares with the 1967 import C.I.F. price of US\$ 172.0 per ton P₂O₅.

Therefore, we recommend that the phosphate plant at Tjilatjap be finished as quickly as possible, even at the expense of a delay in the Pusri II expansion.

Estimates indicate that the capital needed to finish the project will be:

For construction: US\$ 2,117,000
For equipment: US\$ 1,423,385

TOTAL US\$ 3,540,385 at a D.P. rate of Rp 360 =
===== ===== US\$ 1

We recommend also that the UNDP provide the services of one phosphate fertilizer expert for one year under the Technical Assistance programme to:

1. Assist the Government in working out details for ammoniating phosphate using excess ammonia available from Gresik.
2. Study the feasibility of impregnating phosphoric acid to make double or triple superphosphate.
3. Advise and assist in the speedy completion of the project and its economic start-up.
Project Data Sheets and Job Descriptions are attached
.....
for Government approval. (App. 9 & 10)

Although phosphate deposits are widely scattered throughout Java, the deposits are small and not sufficient to sustain an acidulation operation for superphosphates (e.g. in Kragan Grobongan there are 12,000 tons of phosphate scattered among 14 districts). The few known large deposits have been exploited so that they are no longer economic.

V. WANARADJA SULPHUR PLANT - WEST JAVA

The situation is different regarding sulphur.

The Russian survey team, mentioned previously, consisted of 29 persons including 11 geologists. Although their data indicate about 1.5 million metric tons of sulphur reserves throughout 10 districts in Indonesia, the deposits of real interest are at Telaga Bodas, Kawah Putih and Telaga Terus. Data regarding these deposits are shown in the following tabulation:

<u>Location</u>	<u>Distance from Wanaradja, km</u>	<u>Type of Deposit</u>	<u>Economic Sulphur reserves, metric tons S</u>
Telaga Bodas	14	Muddy clay	846,771
Kawah Putih	137	Muddy clay	132,008
Telaga Terus	132	Lump ore	49,463

A factory used to exist at Kawah Putih and 133,400 tons of sulphur were shipped from there in the period 1925 - 1941. In 1944 the sulphur factory was destroyed. The reserves were deemed insufficient for rebuilding the factory.

There is a small factory at Wanaradja built initially to process Telaga Bodas clay. It used to produce 24 tons per day of sulphur before the Japanese invasion, but after the war, partial rehabilitation has resulted in a factory that is producing 3 tons of sulphur per day.

Telaga Bodas is the crater of an inactive volcano. The crater has a lake containing 2,016,000 cubic metres of a muddy clay-sulphur mixture ranging in sulphur content from a low of 15.1% to a high of 70%. Out of a total sulphur content of 18,000 tons, it is estimated that there are 847,000 tons of economic reserves with an average sulphur content of 28% S or better. The economic reserves include about 617,000 tons with an average sulphur content of 55% or better.

When the factory operated before the war, the clay was transported as a slurry through 14 km of pipe from the lake to the factory. Part of the pipe line was rubber, part ceramic, and part an open wooden flume. Now only the ceramic underground portion is in existence and could become part of a new pipe line system.

Now the mud is brought to the factory from the crater by trucks. At the factory the mud is washed and treated in two stages in steam autoclaves. The first stage encourages the growth of the sulphur particles which are found in the original ore as minus 325 mesh particles.

In the second stage the clay is steamed long enough to melt the sulphur and separate it from the clay particles which also are of a minus 325 mesh size. The melted sulphur is cast into blocks and shipped out in bags.

Only about 40% of the sulphur in the clay is being recovered.

At the same time the factory is processing some lump ores (65% - 70% S) from deposits at Lapandajan (90 km) and Tangkuban (150 km) in a single stage melting autoclave step. However, the latter two deposits have a total reserve of only 16,000 tons of sulphur.

It has been suggested that the Telaga Bodas clay could be burned directly at the site in a plant to yield sulphuric acid. However, the water content of the clay, and the fine particle size of the sulphur interspersed within the fine particles of clay make a combustion process infeasible. At the same time, a recovery of only 40% of the sulphur from the clay is an economic tragedy which Indonesia can ill afford.

It is recommended that the UNDP provide a Special Fund project for a technical pilot plant. This pilot plant would study the best means of upgrading the ore through flotation or other means, and of recovering the sulphur from the ore using various types of autoclaves. It is believed that a process can be developed which will recover over 30% of the sulphur present in the ore. If we assume that 50% of the economic reserves can be removed from the crater, then the overall sulphur recovery from the crater would be .90 x .60 = or 72% of 847,000 or over 600,000 tons of sulphur of 99+ % purity.

The phosphate plant at Tjilatjap will require about 43 tons per day of sulphur, while the Gresik plant will require about 110 tons per day of sulphur, for a total requirement of 50,000 tons per annum. At this rate, the Telaga Bodas deposit would last about twelve years, or long enough to justify the 5 million Dollar investment required for a 160 tons of sulphur per day operation.

If we assume the lower price for sulphur of US\$ 54.60 per ton C.I.F. Indonesia, and subtract US\$ 5.0 per ton for amortization and interest charges, then we have a foreign exchange saving of US\$ 45.60 per ton or US\$ 2,280,000 saving in foreign exchange per year with a new sulphur operation at Wanaradja.

However, before anyone will invest in such an operation, there must be engineering data available for designing a large scale plant. The present operation with its 40% recovery system cannot supply such data.

For that reason it is necessary to build the pilot plant mentioned earlier.

A Government request for such a project is attached. (App.11)

VI. GRESIK AMMONIA PLANT, (SURABAJA), EAST JAVA

The Petrokemia plant is about 30% complete and 95% of the equipment is at the site. About 1700 workers are busily engaged in bringing the project to completion. It is expected that the plant will be commissioned early in 1969.

The plant is designed to produce 200 tons per day, ammonia (70,000 tons per annum) of which 6,000 tons are for sale with the balance to be converted into urea and ammonium sulphate.

The plant is based on the partial oxidation of fuel oil and so has a liquid air plant in connection. In addition there is a 350 ton per day sulphuric acid plant on the site.

A 200 ton per day plant using partial oxidation of fuel oil is not a low cost producer at below design capacity, so every effort must be realized to maintain full capacity, and to find a market for the excess ammonia.

The tentative product mix will be:

Ammonia	6,000 tons per annum
Urea	45,000 " " "
Ammonium Sulphate	150,000 " " "

Our recommendation regarding Gresik is that it be brought to completion as rapidly as possible under the original design conditions to yield the tentative product mix listed above, using imported sulphur initially for the production of sulphuric acid.

We suggest two alternatives for ready disposal of the 6,000 tons per annum of excess capacity.

1. The plant has acquired 140 high pressure cylinders capable of holding about one ton of liquid ammonia at atmospheric temperatures. These vessels could be used to transport 20 tons of ammonia per day to the phosphate plant at Tjilatjap for producing ammoniated superphosphate of about a 5-20-0 grade.
2. The excess ammonia could be used as a direct application fertilizer on estate crops and for irrigated paddy. Since this would be a novel use of ammonia in Indonesia (55% of the ammonia in the United States is applied directly to the soil), some preliminary field testing is needed. We recommend that the UNDP provide a Special Fund project to determine the feasibility of direct application of ammonia under the agricultural conditions prevailing in Indonesia. A Government request for such a project is attached. (App. 12)
.....

Although the tentative product mix shows 45,000 tons urea and 150,000 tons ammonium sulphate per annum, the proportions can be varied as dictated by market conditions. However, once the warehouses are full of a certain product mix, it is too late to determine whether the farmers' preference is for the same ratio of products as that found in the warehouses. Therefore it is necessary that a market study be made immediately to determine just what ratio of urea and ammonium sulphate is needed to satisfy consumer demand so the plant can operate at

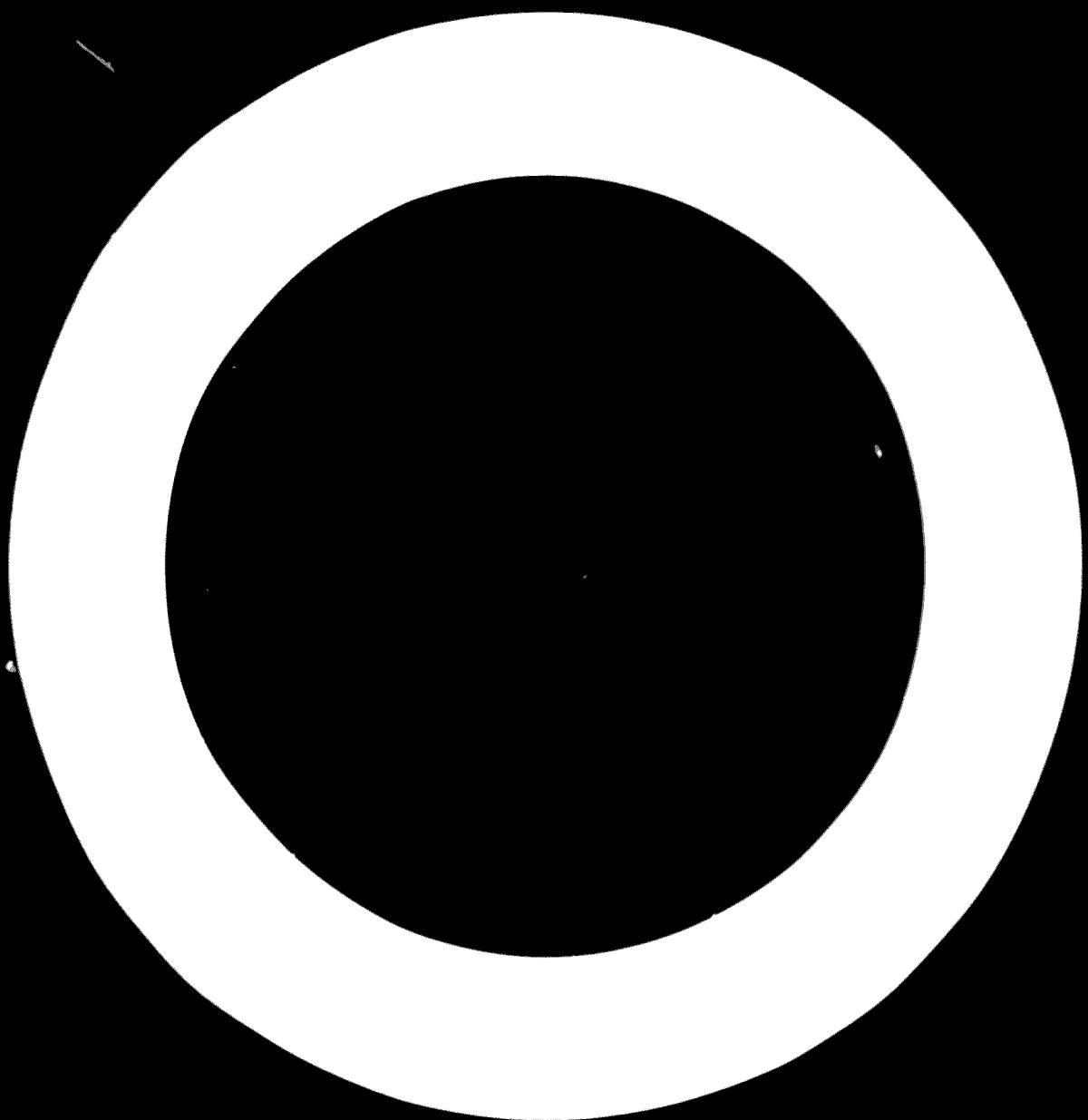
full capacity. We recommend that the UNDP provide a marketing expert under the SIS programme to assist plant management in determining the most profitable product mix for operations.

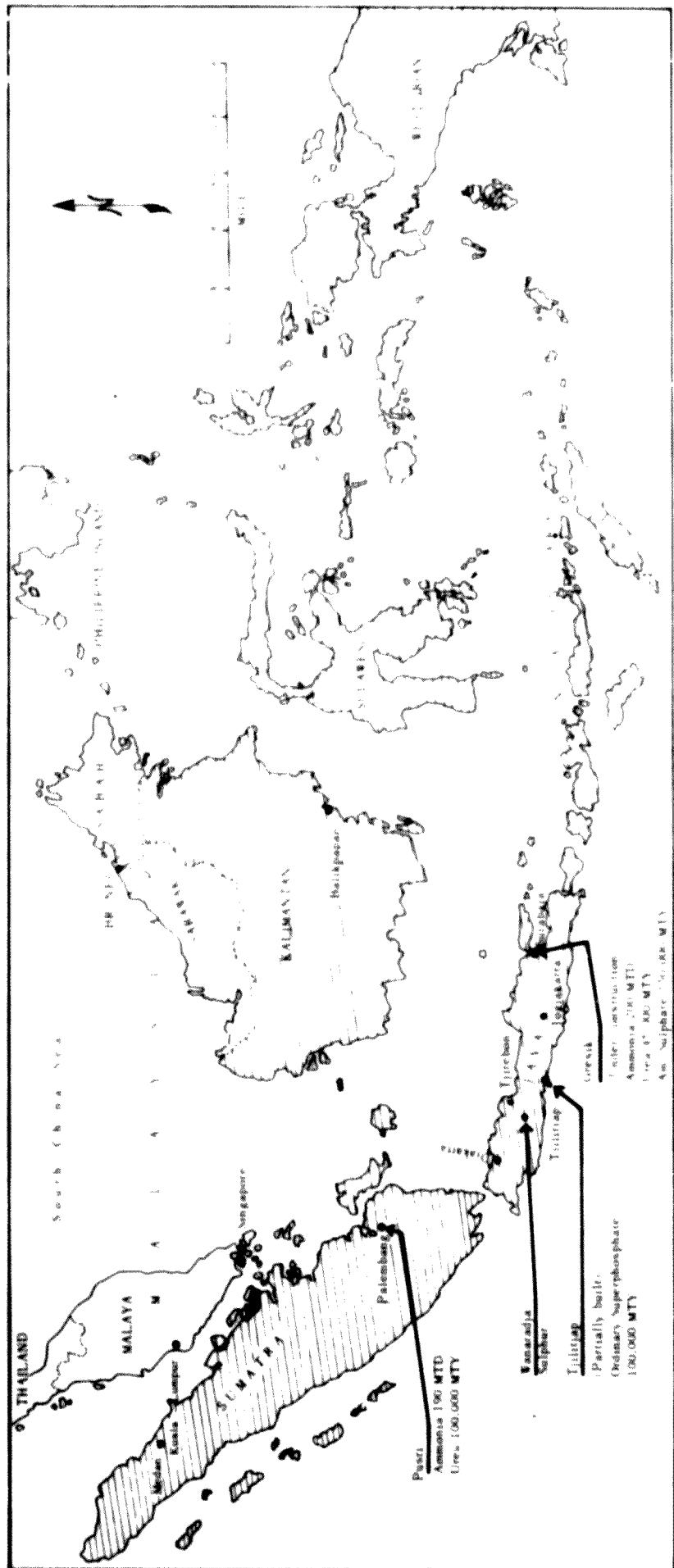
..... A Project Data Sheet (App. 13) and Job Description (App. 14) for the SIS programme is attached for Government approval.

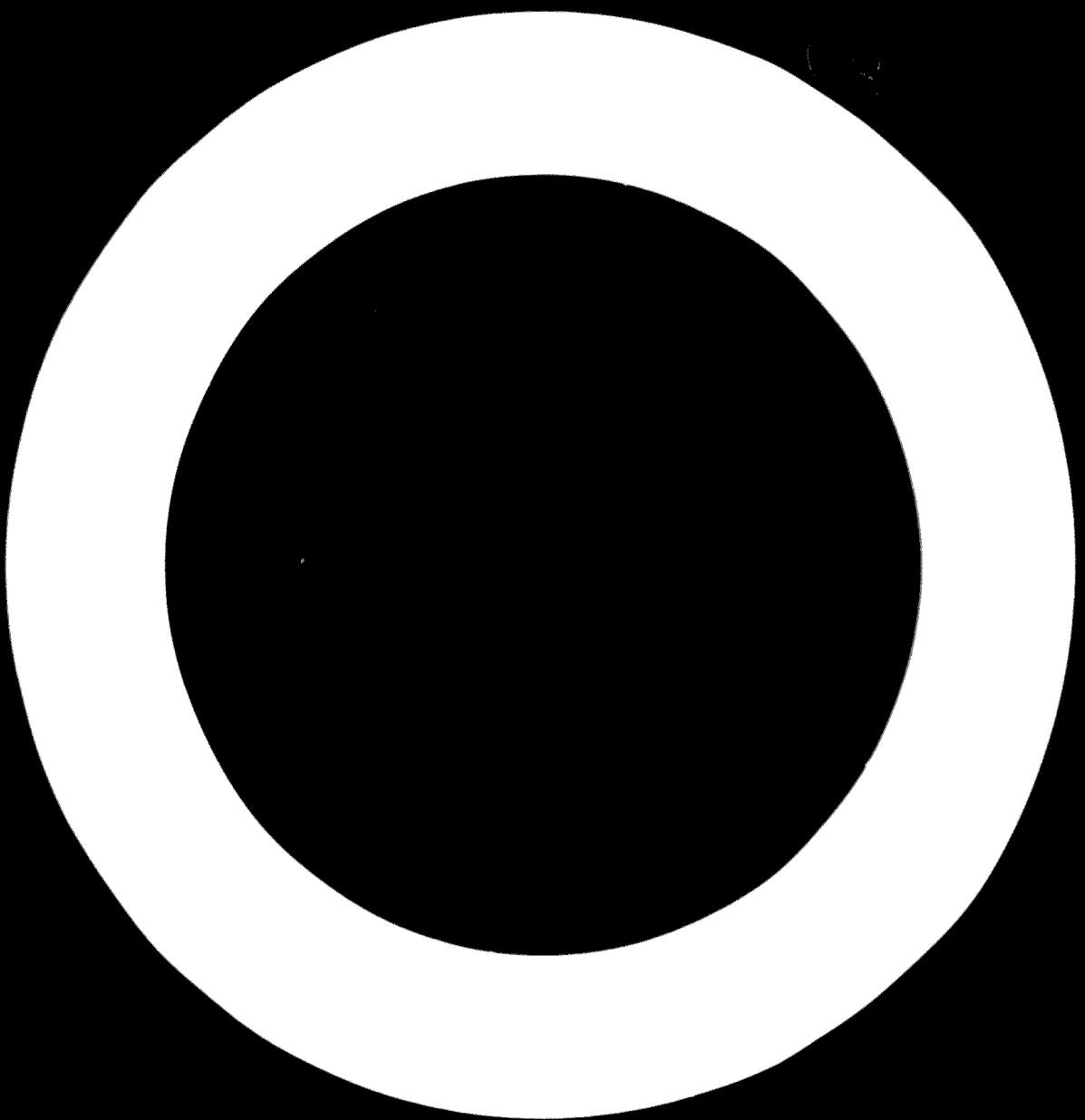
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In conclusion, we repeat our recommendations:

1. Finish the Tjilatjap phosphate plant as planned.
2. Finish the Gresik ammonia, urea, ammonium sulphate plant as planned.
3. Request UNDP assistance for:
 - (a) A technical pilot plant for sulphur recovery at Wanarudja.
 - (b) A pilot project to study the application of anhydrous ammonia directly to the soil, so as to utilize the excess ammonia capacity at Gresik.
 - (c) Technical assistance to aid in the start-up of the plant at Tjilatjap as well as aid in the study of ammoniation, and the manufacture of triple or double superphosphate there.
 - (d) Marketing assistance for determining the best product mix at Gresik.

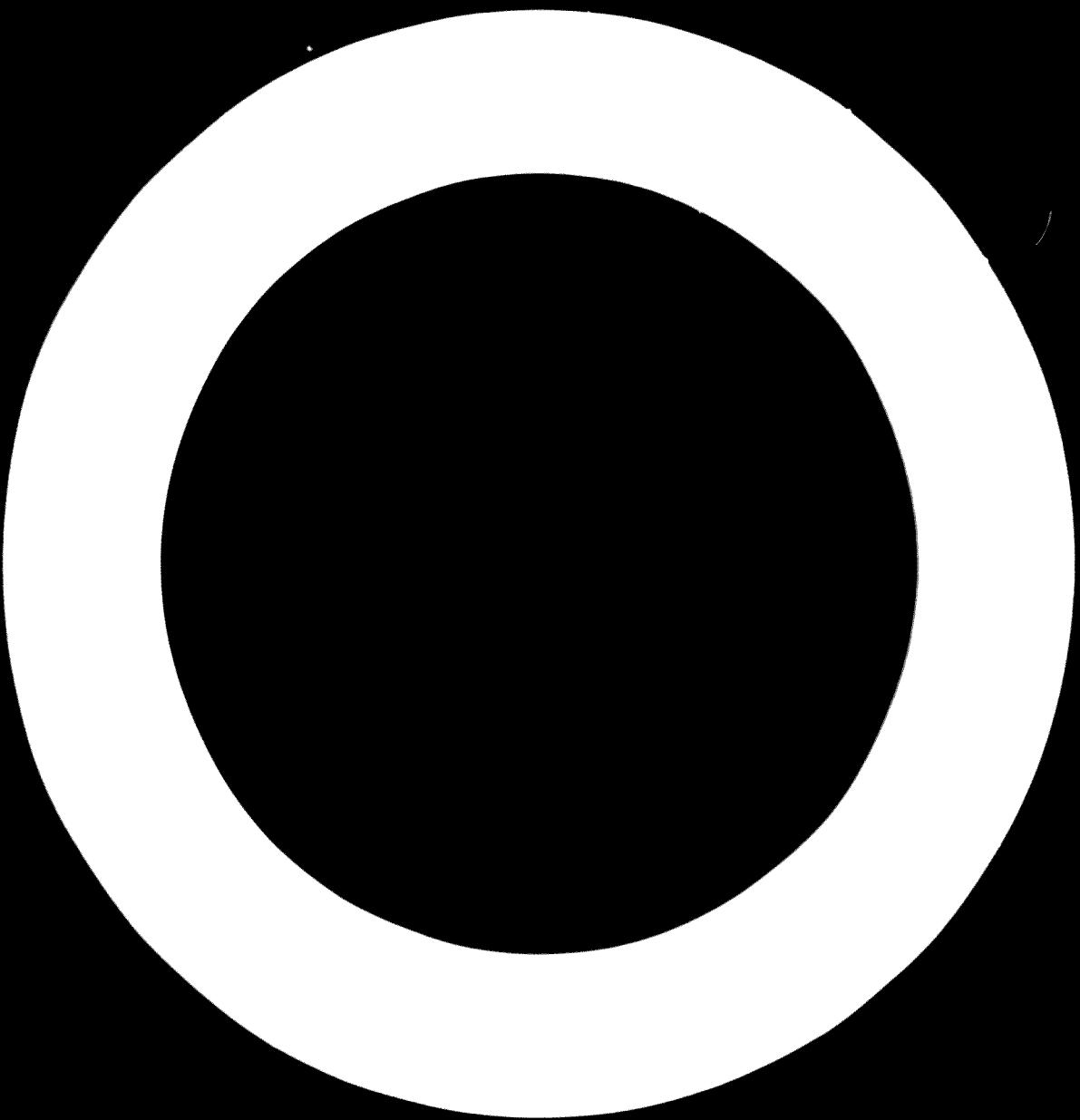






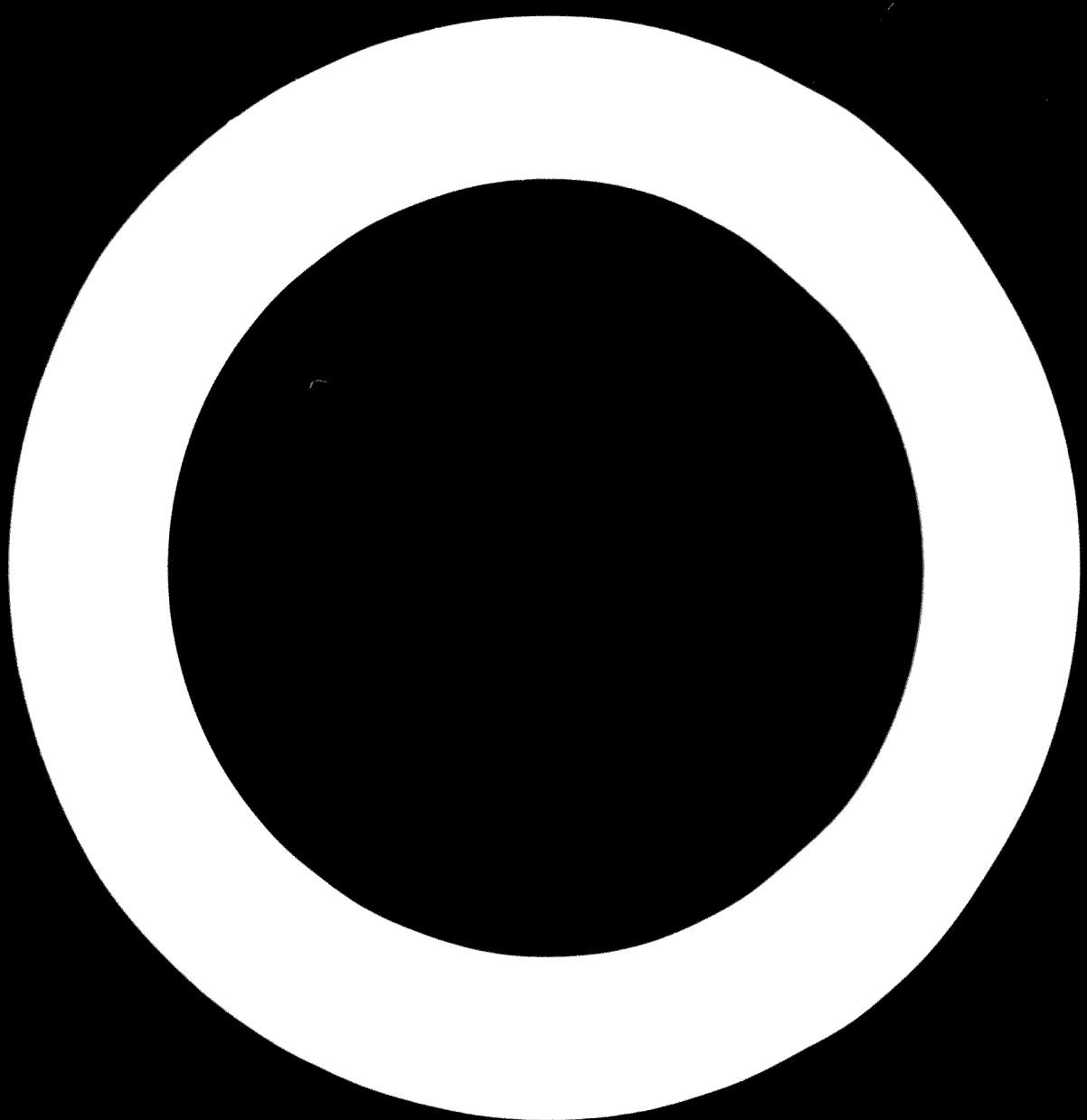
 ITINERARY FOR MR. KIYURU AND MR. VENGERHEIM IN INDONESIA

		Final arrangements for
		Mr. Kiyuru, Mr. Vengerheim and Mr. Djajakarta
27-10-1968	Tuesday	- Final arrangements for the team with project officer and review itinerary of the mission.
28-10-1968	Wednesday	- Counterpart visit to the firm, the General Office of the Ministry of Finance and the Central Bank. - Meeting with the interpretor Dr.
29-10-1968	Thursday	- Counterpart visit to the firm, the General Office of the Ministry of Finance and the Central Bank.
30-10-1968	Friday	- Visit to the - Visit to the - Visit to the firm, the - Sulphur expert Ir. Dr. Kiyuru and interpretor Ir. Syamsudin stay at Wimberly until 3 Nov. 1968
31-10-1968	Saturday	- Visit to the - Visit to the - Visit group leaves for Gresik in the afternoon. - Interpretor expert Ir. Nielssen & counterpart Ir. Supriyatno stay at Tjilik Riwut until 3 Nov. 1968
1-11-1968	Friday	- Gresik
2-11-1968	Saturday	- Gresik - Metallurgist expert Mr. Verheere and project officer Ir. Nico Karsil stay at Gresik until 3 Nov. 1968
4/5-11-1968	Mond./Tuesday	- Preparation of interim report for the Director General in Djakarta.
6-11-1968	Wednesday	- Mr. Verheere leaves Djakarta
12-11-1968	Tuesday	- Mr. Nielssen & Prof. Kiyuru leave Djakarta.



SELECTED DOCUMENTS

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2. Preliminary Report: Study on the Development of Potash Sulphate Fertilizer in Indonesia, Department of Agriculture and Department of Fertilization, Paris, Indonesia, April 1967.
3. The Fertilizer Industry in Indonesia, Development of Raw Material Situraja, ESSO Surinam, Pekanbaru, 4-11 September 1967.
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5. Summary Report: Results of the Indonesian Mission of the Expedition of Superphosphate Raw Materials in 1961-1963, and Recommendations of Further Direction of Works on Sulphur and Phosphates in Indonesia, Ministry of Economic Planning, Djakarta, Indonesia, Ministry of Basic Industry and Mining of the Republic of Indonesia in cooperation with TNO of Soviet Union led by N.J. Marakhin.
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7. Interim Letter Report: Fertilizer Feasibility Study, John Vinkler Volk and Associates, September 1967.
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Appendix

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PERSONS INTERVIEWED IN INDONESIA AND AT THE MEETING

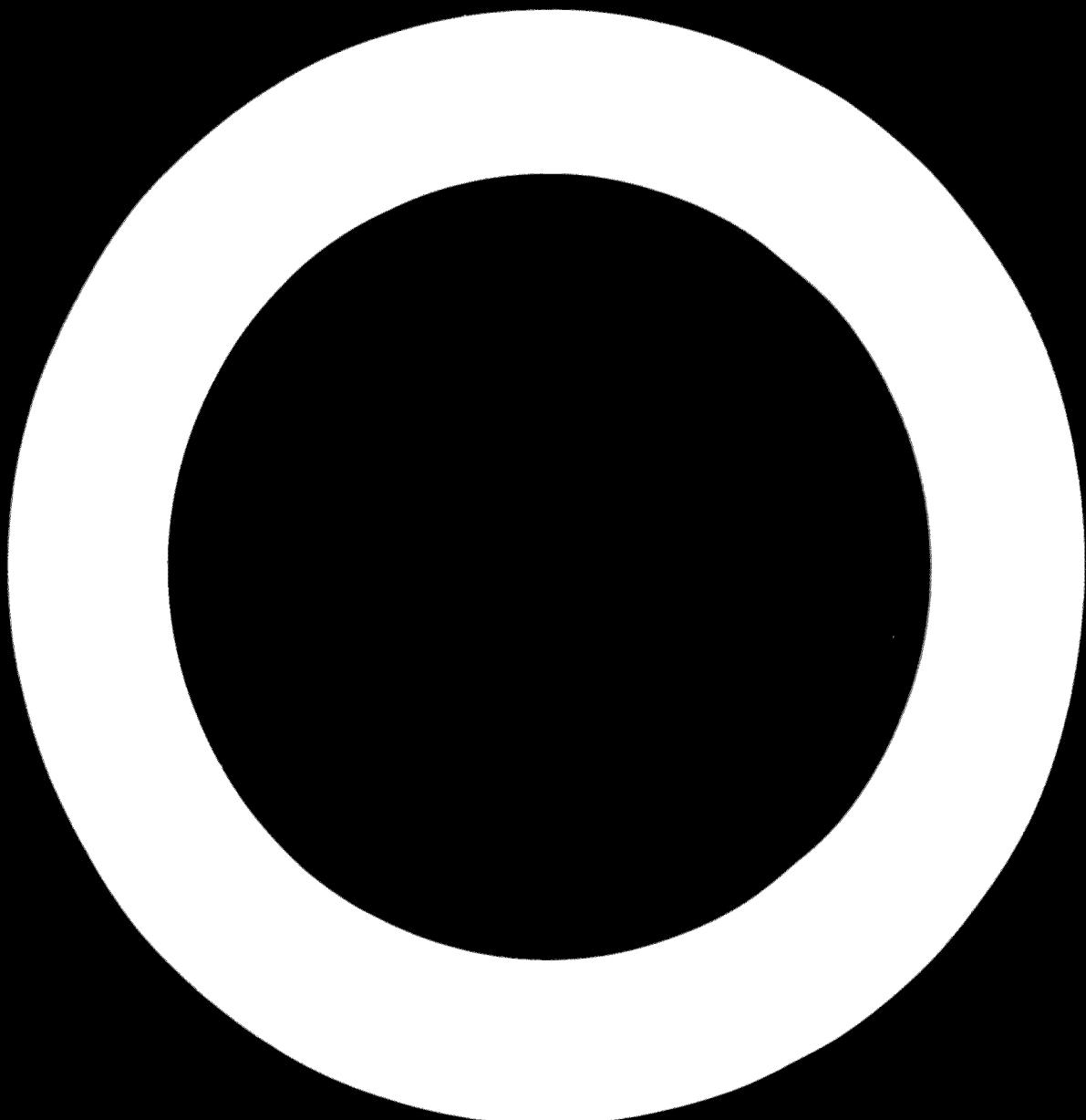
During otherwise stated, the following persons were interviewed, the names:

Mr. A.G. Mennen	Director, Indonesian Institute of Science, Bogor, Indonesia, Bogor, Indonesia, Research Institute for Agricultural and Industrial Development.
Mr. I.Y. Li	Deputy Director, Indonesian Institute of Science, " "
Mr. Fariz	" " "
Mr. Achmed Shamsi	" " "
Mr. Aly Gritly	Bureau of Statistics, Jakarta, Indonesia
Mr. Gvris Griswold	Deputy Director, Bureau of Statistics, Jakarta, Indonesia
Col. Ir. Agus Sugihno	Director General, Ministry of Finance, Indonesia
Ir. Anwar Ibrahim	Deputy Director General, " " "

Mission Counterpart:

1. Ir. Nas Kamill, Leader of Nitrogen Export, Dept. of Mineral Industries.
2. Ir. Sugiritno, Technical Director, Chemical Export.
3. Ir. Syafri Suryati, Chemical Export, Department of Mineral Industries.

Dr. Ing. Haryono Siregar	National Bureau of Planning and Development, Indonesia.
Ir. Abdul Rasyid	Minister of Industry, Republic of Indonesia.
Wagimin Siswoluwih	First Commander, TNI-AU.
En. I. Simardjuntak	Government Representative, Ministry of Finance, Indonesia.
Ir. Suritman	" " " "
Mr. O.C. Woodside	Consultant, " " " "
Alwyn Y. Young	Technical Assistance Expert, UNIDO, Indonesia.
Ir. A.R. Soehadi	Member Foreign Investment Board.
Mr. Rusli Jahja	Director of Marketing, Head of Logistics Division, Department of Agriculture.
Mr. Soedharjo Rawiyo	Acting Director, Directorate General of Agriculture.
Dr. D. Muljadi	Agriculturist, Soil Research Institute, Bogor.
Prof. Dr. Go Ban Hong	Director, Central Research Agriculture Institute, Bogor.
Dr. M. Hartoscedarsa	Director, Research Institute for Estate Crops, Bogor.
Mr. H.L. Beemer	Private Investment & Industry Officer, U.S. Agency for International Development (AID).
Mr. T.W. McCarty	Programme Officer, AID.
Mr. E. Cook	Extension Training Adviser, Agriculture, AID.
Mr. S.M. Sulaiman	International Bank for Reconstruction and Development.
Dr. Ing. M. Harsono	Institute of Industrial Research and Training, Ministry of Industry.
Mr. Soerjo Sediono	Vice Chairman, Office of Foreign Investment Board, and Head, Research Institute for Industrial Development.



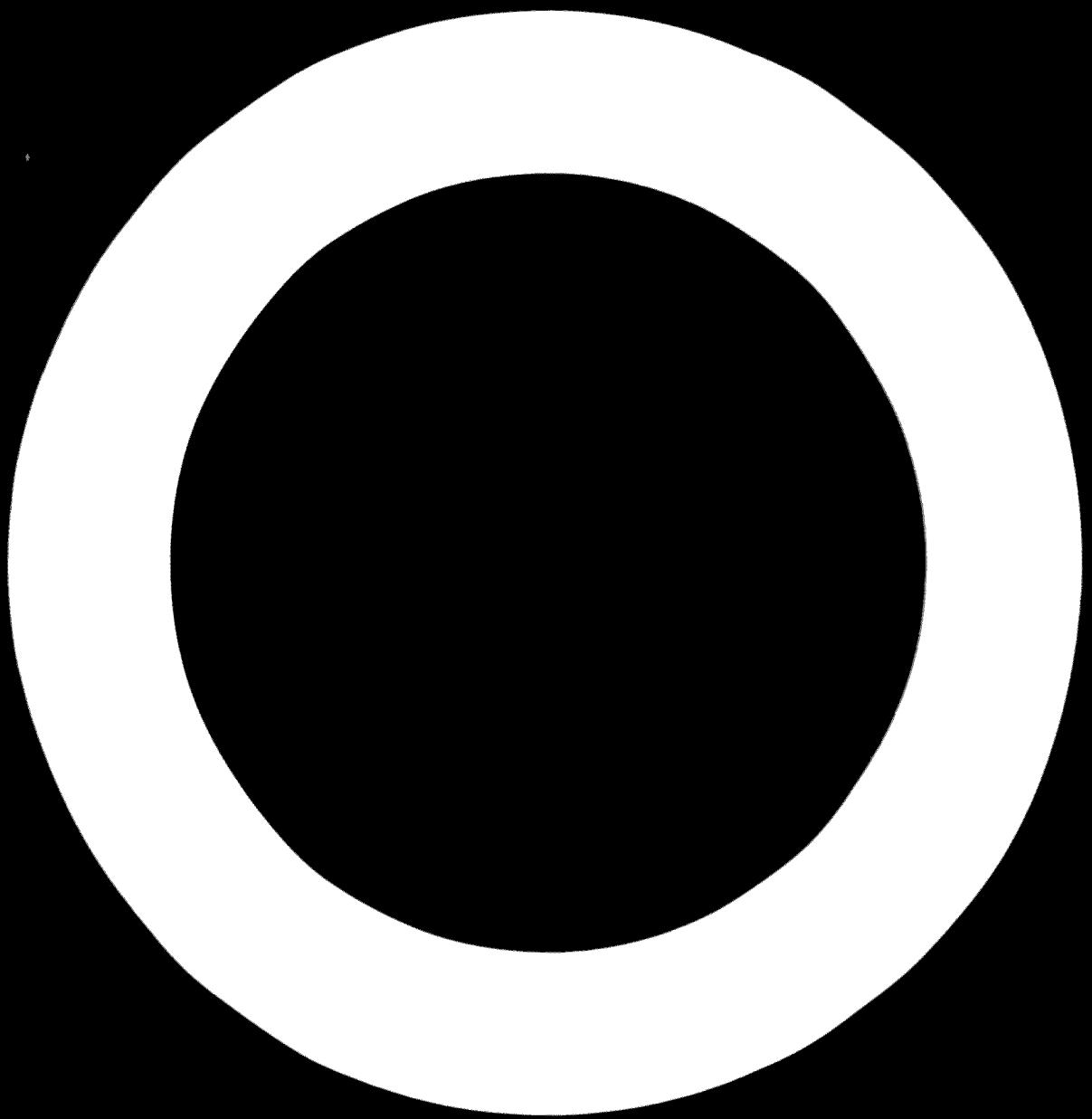
DELIVERY COSTS US\$/metric ton

	<u>via Singapore</u>	<u>Direct</u>
Sulphur F.O.B. Am. riv.	US\$ 15.00	US\$ 14.00
Freight & Insurance	<u>13.0</u>	<u>1.0</u>
C.I.F. Tjilatjap	<u>17.0</u>	<u>1.0</u>
Unloading & Freight	<u>1.0</u>	<u>1.0</u>
Del. plant	<u>1.0</u>	<u>1.0</u>
74/75 % H2SO4 \$ 1.00		
Rock phosphate F.O.B. Tampa	10.15	10.15
Freight & Insurance	<u>13.0</u>	<u>1.0</u>
C.I.F. Tjilatjap	<u>17.0</u>	<u>1.0</u>
Unloading & Freight	<u>1.0</u>	<u>1.0</u>
Del. plant	<u>1.0</u>	<u>1.0</u>
per kg rock	1.02505	1.02505
per kg H ₂ SO ₄	1.0728	1.0728

Assume a US\$ 11,656,000 plant with US\$ 1,500,000 alloted to sulphuric acid at 15 years depreciation, and US\$ 10,156,000 alloted to superphosphate at 40 years depreciation because of the large proportion of building's involved.

Sulphuric acid - 36,000 tons per year, 100 %, US\$ 1,500,000 capital

	<u>Sulphur via Singapore</u>	<u>Sulphur Direct</u>
Sulphur 0.333xUS\$ 50.00	US\$ 19.95	0.333x\$55.60 \$ 18.50
Utilities (steam to power plant offsets power)	0.00	0.00
Repair labour 0.3 % of 1.5 m. capital	0.13	0.13
Repair material 4 % of 1.5 m. capital	1.67	1.67
Operating labour	0.08	0.08
Laboratory & Supplies	0.06	0.06
Depreciation, 6.7 % of capital	2.89	2.89
Overhead 40 % of labour	<u>0.01</u>	<u>0.01</u>
Production cost/metric ton	US\$ 24.79	US\$ 23.34
Production cost/kg 100 % H ₂ SO ₄	0.02479	0.02334



GRANULAR SINGLE SUPERPHOSPHATE; 20% CITRATE SOLUBLE P_2O_5 , 10% WATER SOLUBLE P_2O_5 - 100,000 tons/year, US\$ 10,156,000 capital

Acidulation: 1.79 kg H_2SO_4 /kg Cit. S. 1. P_2O_5
2.06 kg Rock/kg Cit. S. 1. P_2O_5

	R or % Sulphur via Sulphuric acid	R or % Sulphur direct
H_2SO_4 1.79x0.02479x20	= US\$ 8.66	= 8.38
Rock 2.06x0.02505x20	= 14.36	= 12.4
Utilities	= 0.15	= 0.1
Repair labour	= 0.11	= 0.11
Repair material	= 0.53	= 0.53
Operating labour	= 0.24	= 0.24
Laboratory & Supplies	= 0.06	= 0.06
Depreciation, 2.5% of capital =	2.54	= 2.54
Overhead, 40% of labour =	0.14	= 0.14
Bags, 25 kg =	<u>3.00</u>	<u>3.00</u>
<u>Production Cost</u> US\$ 30.49		US\$ 27.45
<u>per metric ton F.O.B. plant</u>	=====	=====
<u>Cost per ton P_2O_5</u> US\$ 152.00		US\$ 137.00
If capital were US\$ 1,000,000 production cost per ton P_2O_5 , at 5% depreciation, would be: US\$ 142.00		US\$ 127.00
1962 import price CIF for <u>powdered single super</u> was US. 35.00 per ton or P_2O_5		US\$ 175.00 per ton P_2O_5

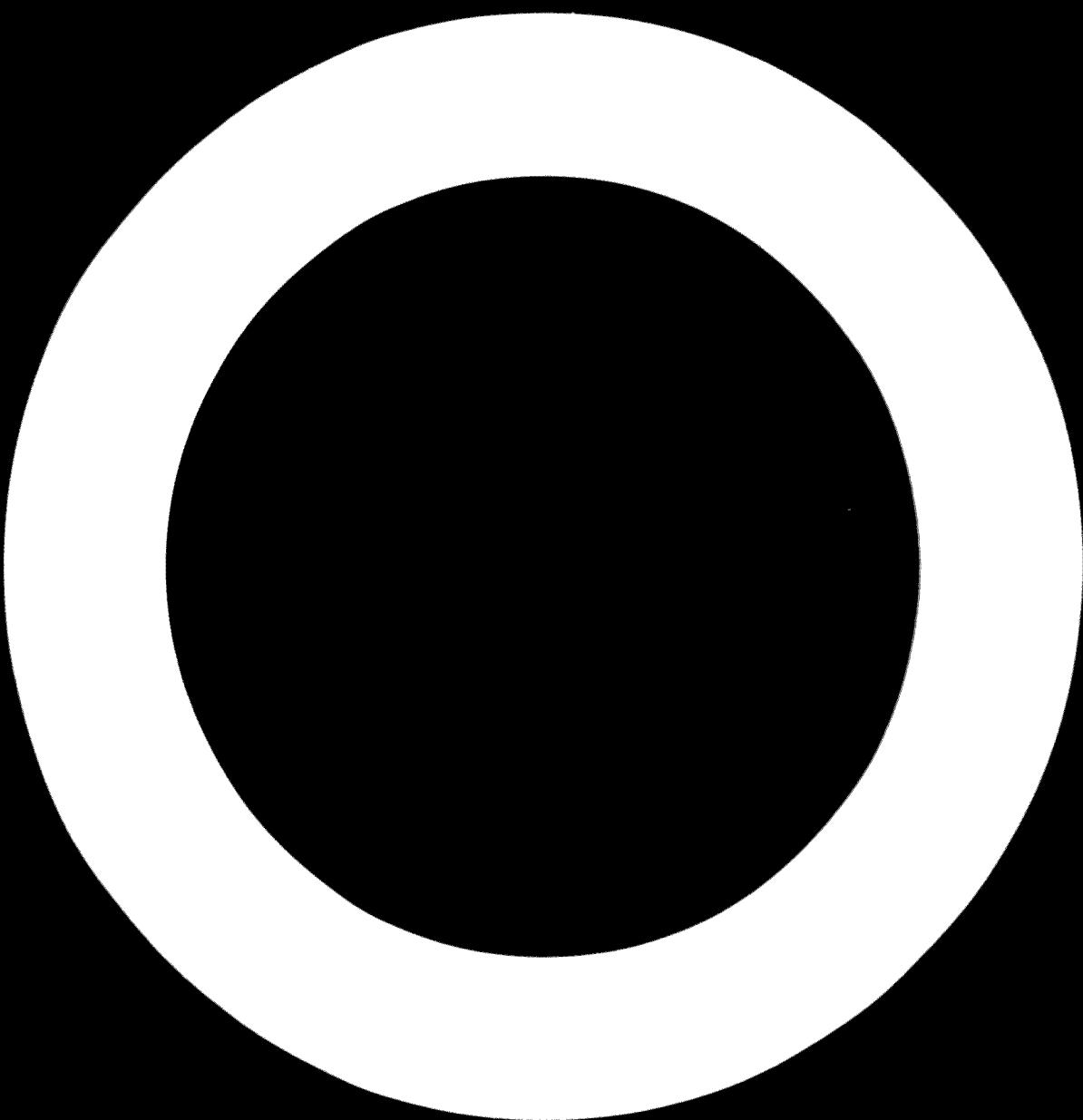


TABLE 10. HOHOLIKA

In view of minimum construction cost per metric ton F.O.B. India at US \$15.00 per metric ton F.O.B., we can assume that partial costs will be unchanged at Trilateral F.O.B. International USC 10 per cent higher than import F.O.B. 10% surcharge triple.

Formulations: 2.70 cu. m. brick, 1 ft. 3 in. x 1 ft.

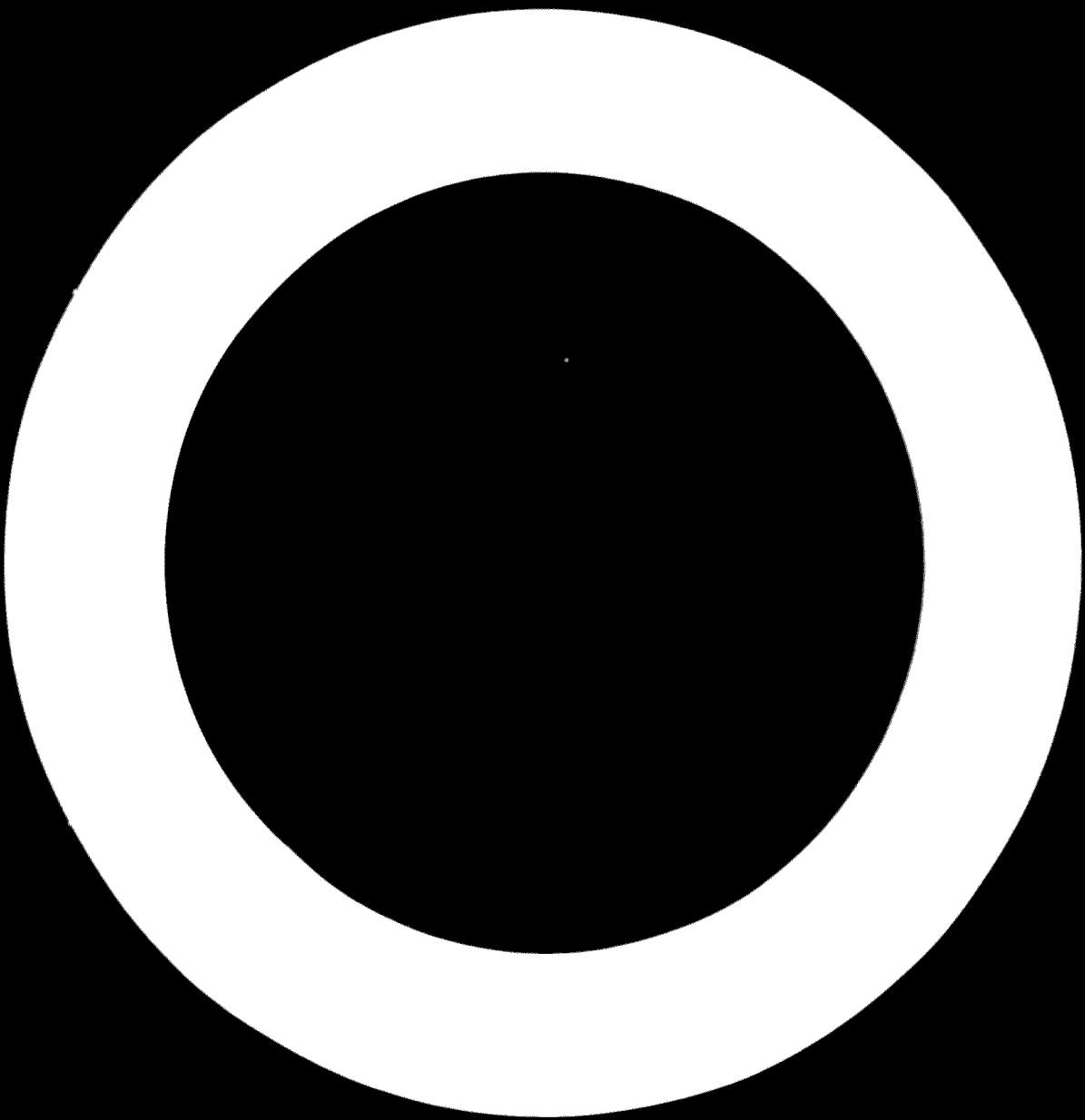
• 30 cu. m. brick, 1 ft. 3 in. x 1 ft.

Cost of soil, 2.70 cu. m.	=	US\$ 1.00
Cost of rock, direct, from mine, .30 cu. m. ²	=	<u>US\$ 1.18</u>
Raw material cost per metric ton F.O.B.	=	US\$ 1.18
Raw material cost per t n product, 28.51 cu. m.	=	US\$ 34.00
Manufacturing cost, incl. taxes, from App. I	=	<u>US\$ 5.77</u>
Production cost F.O.B. plant per ton	=	US\$ 54.77
Cost per ton F.O.B.	=	<u>US\$ 54.77</u>

1967 import price for triple super,

C.I.F., was USC 79.20 per metric ton 46.41 cu. m. product

or USC 172.00 per ton
F.O.B.



Pilot Plant Data for Producing High Grade Superphosphates

IAEC—March 1963

Because the production of phosphoric acid by the wet process may be increased significantly to permit the recovery of uranium salts as a by-product from phosphate rock, there is considerable interest in the feasibility of using wet-process acid for the production of enriched and concentrated superphosphates in existing ordinary superphosphate equipment. This paper describes small scale and pilot plant studies of the acidulation of phosphate rock with mixtures of sulfuric and wet-process phosphoric acid and with wet-process acid alone.

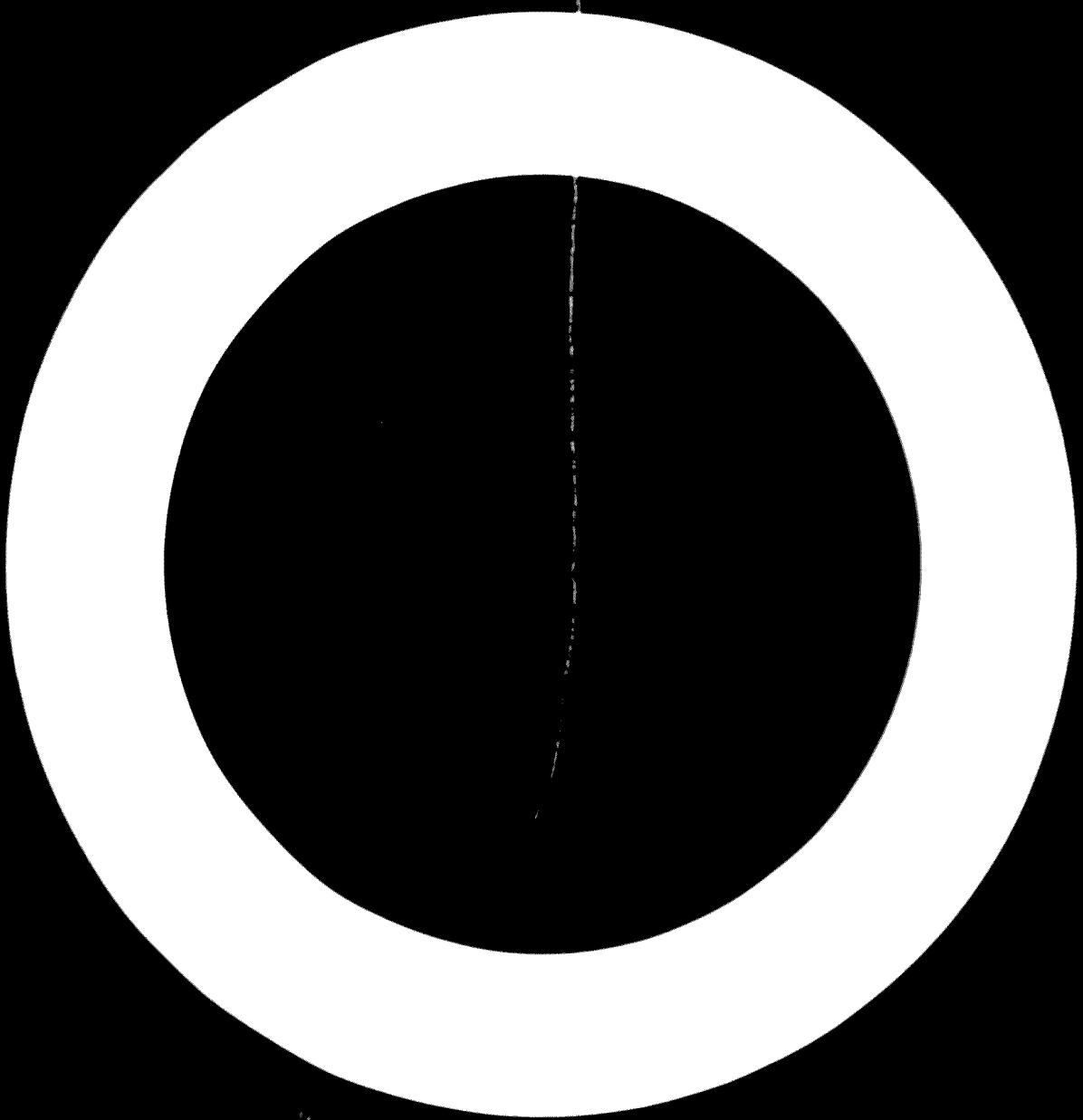
Information was obtained in the pilot plant on the mixing, denning, and curing characteristics of enriched and concentrated superphosphates, and operating conditions that should be satisfactory for use of mixed acid in several types of ordinary superphosphate equipment were identified. The phosphorus pentoxide contents of the enriched and concentrated superphosphates were as available as those in ordinary superphosphate, and the products were as amenable to handling in bags and to ammoniation as is ordinary superphosphate.

The superphosphates produced with mixed acid or phosphoric acid are of high grade and can be used in the production of high analysis fertilizers for which there is a growing demand.

ENRICHED AND CONCENTRATED SUPERPHOSPHATES**L. D. Yoree and F. T. Shulman, E. J. Fox and R. M. Magness,**

Tennessee Valley Authority, Wilson Dam, Ala., U. S. Department of Agriculture, Beltsville, Md.

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PILOT PLANTS



Enriched and Concentrated Superphosphates

L. D. YATES AND F. T. NIELSSON

Tennessee Valley Authority, Wilson Dam, Ala.

AND

E. J. FOX AND R. M. MAGNESS

U. S. Department of Agriculture, Beltsville, Md.

CONCENTRATED superphosphate is the familiar product of acidulation of phosphate rock with phosphoric acid, usually known as double, triple, or triple-superphosphate. Enriched superphosphate may be a mixture of ordinary and concentrated superphosphates or the product of acidulation of phosphate rock with a mixture of phosphoric and sulfuric acids. Because of the increasing demand for high-analysis fertilizers, enriched and concentrated superphosphates appear to be increasingly attractive to producers of ordinary superphosphate. This fact was evidenced by the large attendance of industry representatives at demonstration tests of the production of enriched and concentrated superphosphates in ordinary superphosphate equipment held at Hatfieldburg, Miss., last spring (1). There has been interest on the part of superphosphate manufacturers, but largely academic, because phosphoric acid has not been generally available at prices attractive to them. This picture, however, may be subject to change in the near future as the Atomic Energy Commission has become interested in increasing the production of wet-process phosphoric acid in order to recover uranium salts as a by-product from phosphate rock (2). Consequently, the AEC is encouraging the expansion of present facilities and construction of new ones to make wet-process phosphoric acid. Some of this increased production of phosphoric acid may be available for use in existing ordinary superphosphate plants to produce enriched and concentrated superphosphates.

In 1926 a patent on the production of enriched superphosphate was issued to Larson (10), the equipment and details of the process were not specified. The results of laboratory experiments with mixtures of phosphoric and sulfuric acids are summarized by Bridger (8). A recent paper by Fox and Hill (4) presented a theoretical analysis of the problems that would be encountered in the production of enriched superphosphate. However, only very limited information is available on the techniques of production of enriched and concentrated superphosphates in equipment used in the manufacture of ordinary superphosphate (11). Therefore, the Atomic Energy Commission suggested that consideration be given to procuring the technical information required to carry out these operations effectively. To obtain this information, the U. S. Department of Agriculture and the Tennessee Valley Authority undertook a joint experimental program. USDA carried out small-scale tests, and TVA did the pilot plant work. It is recognized that the equipment in ordinary superphosphate plants varies considerably (7). Some plants have batch-mechanical dens, some have box-type dens (batch),

and others have continuous processes. Since these plants have different processes, but present-day pilot-plant-size types of plants are used. Because of the variation in equipment, present drivers to all the questions that might arise in the manufacture of enriched and concentrated superphosphates is evident, and a great deal of the information in this experimental program is broad in scope, and some experimenting would have to be done in the manufacture of any plant.

The objective of the work reported here was to reduce to a minimum the volume of plant-scale research necessary to any such transition. The small-scale tests were designed to supply information on rock-phosphate conversion to available forms under standard conditions, and to reduce these results to the physical condition of the charge, time of setting, and reaction temperatures and to relate batch temperature and consistency to the concentration, composition, and temperature of the acidulant. The pilot plant experiments were intended primarily to determine the mixing, dening, and curing characteristics of full-size batches of suitable mixed and superphosphates in order to provide data for guidance in the selection of appropriate conditions for their manufacture. The possibility of one type of continuous den in mixed-acid treatment may be inferred from the published results of recent factory-scale experiments (12). Because enriched and concentrated superphosphates may be expected to be used mainly in mixed fertilizers, the behavior of the pilot plant products on ammoniation was also studied. Since some of the more concentrated superphosphates are likely to be used for direct application to the soil, tests were made to determine their caking properties in bag storage and their drilling characteristics.

$$\text{Acidulation} = \frac{\text{Mole Ratio of } \text{P}_2\text{O}_5 + \text{SO}_3}{\text{CaO}}$$

Florida land-pebble phosphate was used in all the experiments. The compositions and screen analyses of the rocks are shown in Table I. Virgin 93% sulfuric acid was used in all tests. The phosphoric acids that were used had been made by the wet process; their compositions are given in Table II.

In the rock-acid reactions, 1 mole of sulfur dioxide or phosphorus pentoxide is equivalent to 1 mole of calcium oxide. The acidulating values of the phosphoric acids were determined by titration with sodium hydroxide to the methyl orange end point, which titrates the first hydrogen ion of pure phosphoric acid. In

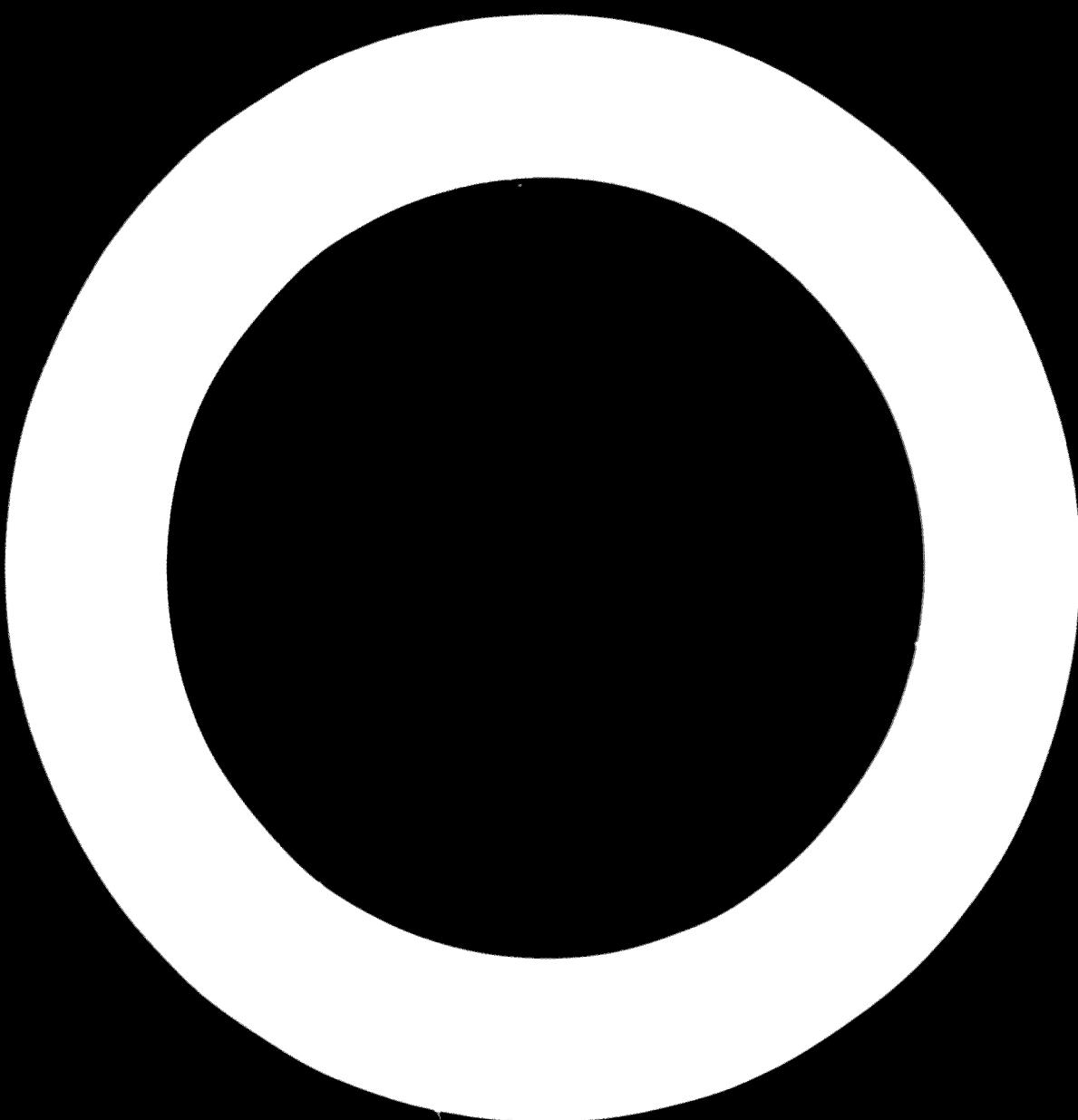


Table I. Compositions and Screen Analyses of Phosphates

Scale of Work	Composition, %*						Ignition loss	Moisture	
	P ₂ O ₅	CaO	F	Fe ₂ O ₃	Al ₂ O ₃	CO ₂			
Small scale	32.2	47.3	3.37	1.28	1.32	2.9	1.2	4.6	0.7
Pilot plant	34.7	49.5	3.6	1.3	1.3	2.8	0.3	4.6	0.6
Screen Analysis									
	Wet Cumulative % through Indicated Sieve Size U.S.								
Small scale	60	100	150	200	250	300			
Pilot plant	65.4	87.5	78.2	57.1	48.7	34.7			

* Air-dry basis.

preparing mixed acid, sulfuric acid was replaced with an equivalent amount of wet-process phosphoric acid on the basis of 2 moles of titratable phosphoric acid per mole of sulfuric acid. Since the wet-process phosphoric acids contained appreciable amounts of iron and aluminum phosphates as well as other impurities, the total phosphorus pentoxide in the acid in some cases exceeded the phosphorus pentoxide equivalent of the titratable acid by an appreciable amount (Table II).

Table II. Compositions of Phosphoric Acids

	Composition, %*						
	H ₃ PO ₄ †	P ₂ O ₅	CaO	F	Fe ₂ O ₃	Al ₂ O ₃	
Small scale test	74.3	58.6	0.23	0.6	1.2	0.8	0.7
Pilot plant tests							
A*	65.4	50.6	0.3	0.5	1.1	0.9	0.7
B*	75.3	57.6	0.1	0.5	1.1	0.7	0.9
C*	74.3	53.1	0.3	1.3	2.5	1.6	2.8

* Acidity determined by titration with standard NaOH to methyl orange end point.

† Determined by Karl Fischer method as described by Zerban and Rettler.

‡ This procedure was tested for phosphoric acid.

§ Used in tests of production of enriched superphosphate.

¶ Prepared by concentrating acid A; used in tests of production of concentrated superphosphate.

* Used in tests of production of concentrated superphosphate.

In the small scale tests the amount of acid used was equal to 57.5 pounds of sulfuric acid per 100 pounds of rock. The degree of acidulation was 0.985, expressed according to Equation 1.

$$\text{Acidulation} = \frac{\text{Moles P}_2\text{O}_5 + \text{Moles H}_3\text{PO}_4 \text{ equivalent to H}_2\text{PO}_4}{\text{from rock}} + \frac{\text{Moles SO}_3 \text{ from sulfurous acid}}{\text{Moles CaO from rock}}$$

This equation also was used for expressing acidulation in the pilot plant work. For the purpose of simplification the ratio shown on the right side of Equation 1 is referred to in the text

and in the tables as the $\frac{\text{P}_2\text{O}_5 + \text{SO}_3}{\text{CaO}}$ mole ratio.

Acid mixtures are described in terms of percentages of sulfuric acid replaced with phosphoric acid. Since 2 moles of the latter is required to replace 1 mole of sulfuric acid and the molecular weights of these acids are sensibly the same, it is evident that their combined weights may be expressed as

$$W_a = S(1 + z)$$

where

- W_a = combined weight of H₂SO₄ and H₃PO₄
- S = parts of H₂SO₄/100 parts of rock, by weight
- z = fraction of H₂SO₄ replaced with H₃PO₄

Thus, with ordinary superphosphate z = 0 and W_a = S, and with concentrated superphosphate z = 1 and W_a = 2S—that is, the weight of phosphoric acid required is twice the weight of sulfuric acid. Consequently, the weight of the charge per unit weight of rock as well as the percentage of phosphorus pentoxide in the product increased with the percentage replacement.

The total weight of the acidulant, W_a, is obtained by dividing the acid weight, W_a, by the combined acid concentration. Assuming water as the third component of the acidulant, the acid

concentration and its composition may be defined in terms of per cent water and per cent replacement as given in Tables III through VI.

Phosphate rock conversion to available forms was calculated by difference between the estimated phosphorus pentoxide contributed by the rock and the extra-insoluble phosphorus pentoxide found by analyses of the product. In small scale tests calculations were based on the make-up of the acidulates; complete analyses were not obtained. In pilot plant products the calculations were based on the chemical analyses of the products for CaO, P₂O₅, and SO₃.

Small Scale Experiments Covered Wide Range of Acidulants and Operating Conditions

The change-can type mixer (6) used in the small scale tests was equipped with stainless steel blades and can 11 inches in diameter by 12 inches deep. The off-center stirrer assembly attached to a hood-type cover, revolved at 90 r.p.m. in opposite direction to the can rotated at 45 r.p.m. Evolved gases were exhausted through a 3-inch diameter telescoping vent pipe attached to the cover, which could be raised and lowered by a rack and pinion gear. A drag stick operating against spring tension, with pointer moving across a calibrated scale to measure the consistency of the charge, and a thermocouple well were inserted through and attached to the cover, which also had a 3-inch diameter opening through which the acid was added. The junction of the iron-constantan thermocouple was silver-soldered to a thin stainless steel tip electrically and thermally insulated from the stainless steel well by a stout hard rubber shield. This highly sensitive couple junction was positioned about 2 inches away from the side and 1½ inches above the bottom of the can.

A 12-pound charge of phosphate rock was weighed in the can, the latter placed on the mixer and the cover lowered. The requisite amount of preheated acid was then poured rapidly onto the rock while the mixture was being stirred. Time measurements started when all the acid had been added. Systematic observations on the consistency and temperature of the charge were recorded, and the course of the reaction in terms of acid consumed was followed by the titration of residual free acid in weighed

grab samples, usually taken at 10-minute intervals during the first 30 minutes. In one series of experiments, mixing was continued until the mixture thickened; in another the mixing was limited to 2 minutes or less, depending on the rate of set. Temperature measurements were continued for 30 minutes or until the temperature generated by the heat of reaction reached a maximum and started to decline. The cover of the mixing machine then was raised, and the adhering material was scraped from the mixing blades and other parts of the equipment. The can with its charge then was covered and transferred to an electrically heated oven maintained at 155°, 175°, or 185° F., usually the latter, for curing overnight. About 24 hours later the charge was reweighed and sampled by cutting a core from top to bottom of the charge about midway between its center and circumference. In a few instances the product was too crumbly to permit a solid core being taken, in which case reasonable precautions were observed to secure portions of the sample from various parts of the charge. These samples were ground, screened, well mixed by rolling on paper, and analyzed immediately. Other pertinent details of the procedure are noted in the presentation and discussion of the results given in Table III and shown in Figure 1.

Acid consumption by reaction with the rock during the first 20 minutes in mixtures made with 75, 70, and 65% acid solutions (approximately 25, 30, and 35% water, respectively) is plotted against acidulant composition in Figure 1, in which the curves for initial acid temperatures of 104°, 140°, and 176° F. for each of the three acid concentrations are shown. The acid consumption during the 20- and 30-minute periods did not greatly exceed the

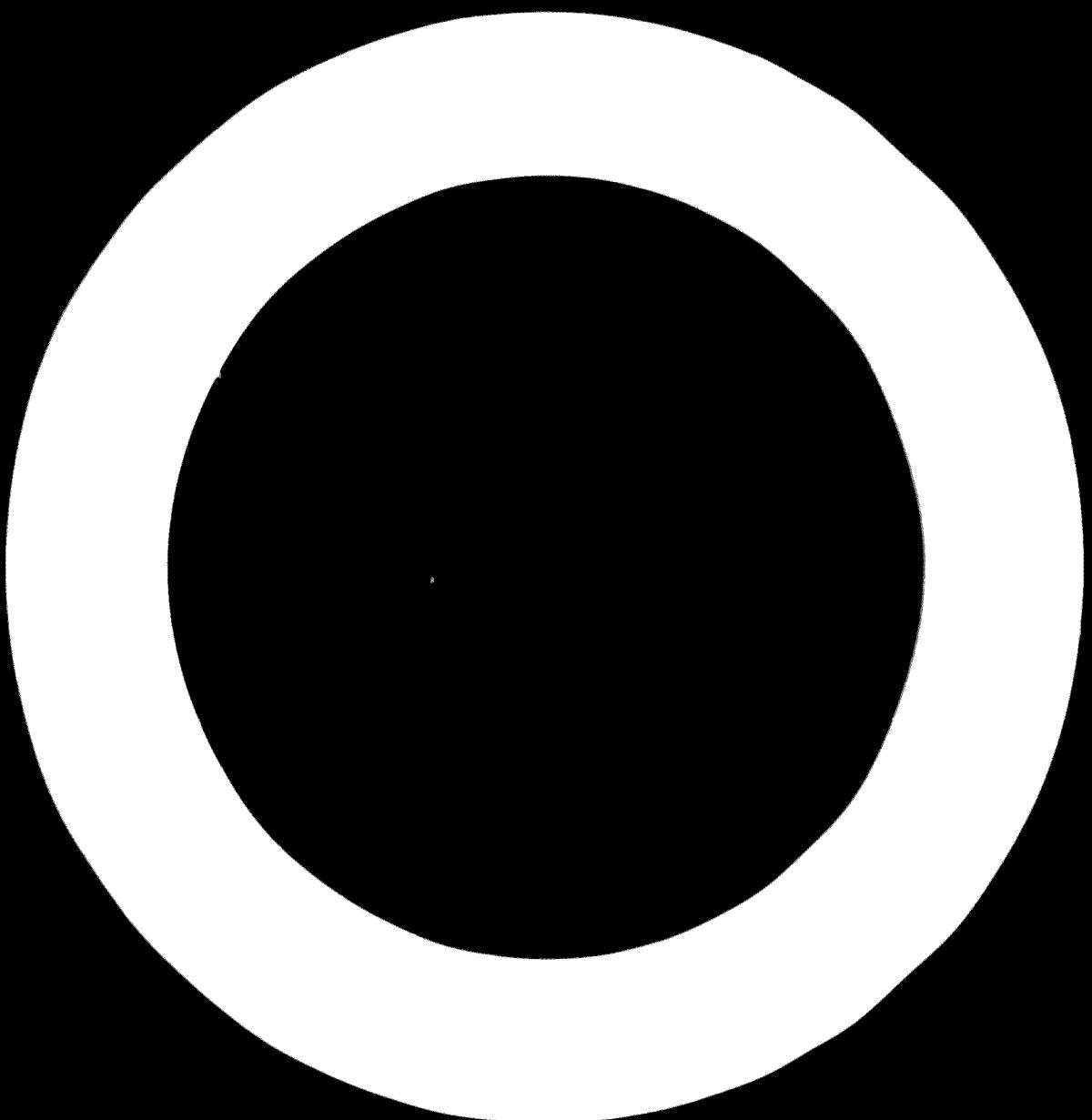


Table III. Character of Acidulants and Quick-Cured Superphosphate Prepared in 20- to 30-Pound Lots

Acidulant	H ₂ O _n	Temp., °F.	Mixing Time, Min.	Batch Temperature, °F. at End of Reaction	Condition of Product at End of 20 hr. Curing	Weight Loss During Mixing	Ergs./hr. of Product	Cure % of Rock Phosphate	Weight Loss During Mixing		Ergs./hr. of Product	Cure % of Rock Phosphate
									24 hr.	24 hr.		
0% Replacement												
25	104	1	233	213	soft	0	18	44	0	8	44	3
	140	2	237	240	243	2	0	44	0	26	44	8
	176	1	237	242	244	2	0	44	0	26	44	8
30	104	1	201	214	220	4	0	44	0	26	44	6
	140	2	214	218	219	3	0	44	0	26	44	8
	176	2	223	226	228	3	0	44	0	26	44	8
33	104	3	149	166	167	10	0	44	0	26	44	6
	140	1	149	166	167	10	0	44	0	26	44	6
	176	1	171	210	214	2	0	44	0	26	44	8
34.3% Replacement												
25	104	2	168	172	203	120	0	18	0	0	18	0
	140	2	166	173	217	127	0	44	0	26	44	0
	176	2	223	228	242	242	0	44	0	26	44	0
36	104	2	141	154	172	208	7	44	0	26	44	6
	140	2	163	168	207	217	8	44	0	26	44	6
	176	2	212	225	227	227	0	44	0	26	44	6
38	104	2	131	138	141	166	40	18	0	18	18	6
	140	2	170	180	194	198	10	44	0	26	44	6
	176	2	191	193	198	198	0	44	0	26	44	6
43.3% Replacement												
25	104	2	180	174	184	212	2	18	0	0	18	0
	140	30	185	188	196	196	3	18	0	0	18	0
	176	2	206	207	223	234	20	44	0	26	44	0
	176	1	210	217	203	197	2	44	0	26	44	0
36	104	2	163	167	174	209	3	18	0	0	18	0
	140	20	160	162	172	207	20	44	0	26	44	0
	176	2	190	192	210	210	20	44	0	26	44	0
	176	10	181	183	190	194	18	44	0	26	44	0
38	104	2	130	136	138	138	30	18	0	0	18	0
	140	4	128	138	138	138	20	44	0	26	44	0
	176	2	187	188	190	196	20	44	0	26	44	0
	176	1	194	187	194	207	2	44	0	26	44	0
48.3% Replacement												
25	104	2	111	116	124	183	40	18	0	0	18	0
	140	2	118	123	127	187	20	44	0	26	44	0
	176	10	185	183	182	186	20	44	0	26	44	0
36	104	2	120	124	126	196	20	18	0	0	18	0
	140	3	144	140	152	183	40	44	0	26	44	0
	176	1	183	183	187	194	20	44	0	26	44	0
38	104	2	107	107	120	138	20	18	0	0	18	0
	140	2	140	147	153	162	20	44	0	26	44	0
	176	1	162	172	181	187	20	44	0	26	44	0
40% Replacement												
25	104	2	106	108	114	180	40	18	0	0	18	0
	140	0.5	118	107	121	186	20	44	0	26	44	0
	176	0.7	121	121	121	186	20	44	0	26	44	0
36	104	0.8	111	111	111	186	20	18	0	0	18	0
	140	0.8	114	114	117	186	20	44	0	26	44	0
	176	1	126	127	126	183	20	44	0	26	44	0
38	104	1.1	94	102	120	169	40	18	0	0	18	0
	140	1	111	124	142	155	20	44	0	26	44	0
	176	1	165	172	180	183	20	44	0	26	44	0

^a Temperature at which highest observed temperature was attained.^b Character of acidulate. A = soft and crumbly; B = damp, sticky, or thick, with no sticky tendency; C = sticky, sticky, with no sticky tendency; D = slow, thin, watery paste; E = lumpy. For example, top 4 to 5 inches in each lot were 100% when measured immediately after mixing.^c Character of product. 1 = dry and easily crumbled; 2 = crumbly, somewhat moist; 3 = damp and hard; 4 = porous and hard; 5 = dense and hard; 6 = porous, mixed as evidenced by the rock in both the product and the raw materials.^d Result calculated from weights of ingredients and product.^e Result calculated from analysis of sample of cured product. Figures in parentheses determine the percentage of acid consumed by starting 20 hr. to 24 hr.^f Oven temperature was 154°F.^g Oven temperature was 175°F.^h Temperature still rising.

consumption observed during the last 10 minutes so that the results for the 20-minute period will serve to illustrate the results for all three.

With sulfuric acid (0% replacement) the effects of varying the acid concentration and the initial acid temperature were relatively small because the heat of reaction in all cases raised the batch temperature to, or near to, the boiling point of the solution where large heat losses by evaporation tended to level off the reaction temperature. With increasing replacement of sulfuric acid the effects of acid concentration and temperature increased, as is indicated by the spread of the curves at 30% replacement. With an initial acid temperature of 101°F., acid consumption at all three concentrations decreased from about 67% at 0% replacement to about 25% at 30% replacement, then increased again at higher replacement to about 40% of the total acid.

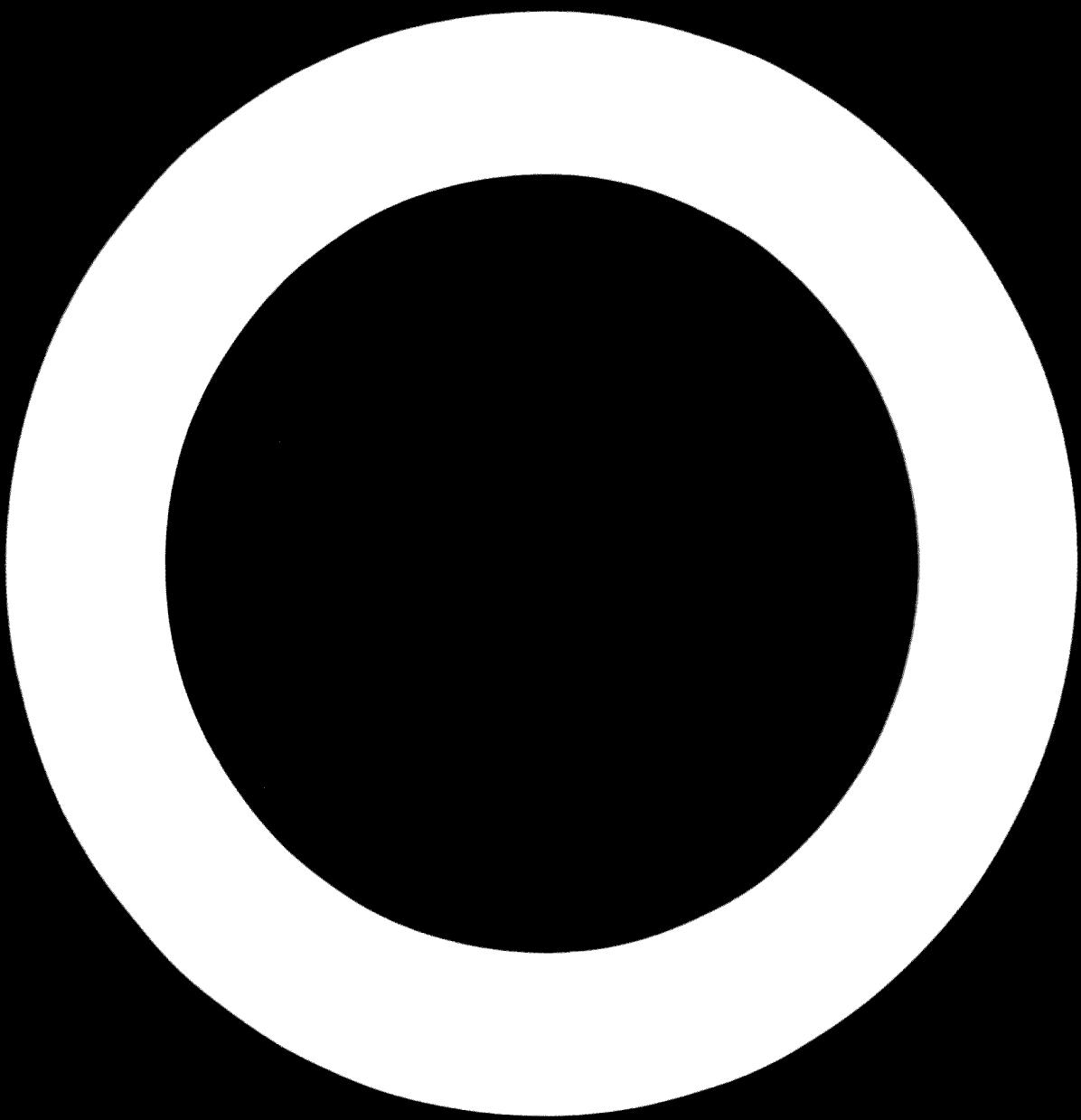
Progressive increases in acid consumption with increasing replacement over the range 0 to 30% were caused by raising the temperature of the acid. Because of the rapid set of mixtures made with 75% acid at 83.3 and 100% replacement, proper mixing was not obtained and acid consumption was not determined.

The chemical behavior of the acidulates as shown in Figure 1 is reflected in the temperature data and physical characteristics pre-

sent in Table III. The increase in acid consumption with increasing replacement of sulfuric with phosphoric acid (Figure 1) is accompanied by a corresponding decrease in batch temperature (columns 4 to 7, Table III) and a prolongation of the period of fluidity as indicated by the mixing time (column 3). Limiting the agitation to 2 minutes or less generally resulted in higher batch temperature due to diminished heat losses occasioned by the agitation of the charge. A corresponding delay in the attainment of the maximal batch temperature also occurred, as is shown by the results for 33.3% replacement. On the other hand, pronounced segregation of the solid and liquid phases was noted in several instances where agitation was stopped before the mixture stiffened.

The physical characteristics of the oven-cured products present a pattern analogous to that of the acidulates at the end of 20 minutes (columns 8 and 9, Table III). High temperature and water content improved the tractability of the products of the mixed acidulants, especially those derived from the sticky pastes at 33.3% replacement.

Rock phosphate conversion to available forms at the end of 24 hours (column 13, Table III) under the stipulated conditions of curing did not show consistent dependence on any of the studied



factors. Such variations as did occur appear to be more closely associated with the lack of uniformity of mixing due to differences in the setting rates than to any of the controlled variables of acid temperature, composition, or concentration. However, when all the tests are considered together, the trend is toward higher conversion with 65% acid at 140° F. With respect to effect of

acid, 60 pounds of sulfuric acid per 100 pounds of the phosphate rock in the production of ordinary superphosphate. However, acidulation was determined from analyses of samples of the superphosphates, varied from 0.93 to 1.01 (equivalent to 57 to 61.5 pounds of sulfuric acid per 100 pounds of phosphate rock). Where possible, comparisons were made in a narrower range of acidulations.

Figure 2 and 3 show the equipment used in the pilot plant tests. Mixing was carried out in a 1-ton Model E Stedman mixer. A sheet-metal jacket containing steam coils was placed around the pan so that it could be preheated to the equilibrium temperature that was obtained when several batches of ordinary superphosphate were made in it consecutively. In this way a single batch of superphosphate could be made without it being cooled unduly by the pan. Thermocouples were provided for measuring the temperature of the metal of the pan and of the superphosphate in it. The acid storage and mixing tanks were equipped with agitators, and the mixing tank and acid feed tank contained coils for steam or water to permit adjustment of acid temperature. Phosphate rock was fed to the mixer through a weigh hopper mounted above the mixer.

The superphosphate fell from the mixer through a 12-inch pipe and into a horizontal wooden cylindrical den (44 inches inside diameter by 37 inches inside length) that could be rotated. The side walls of the den were 4 inches thick and the back end had two walls with an air space between. These walls served to insulate the den effectively. A heavy wooden door at the front of the den could be opened or closed quickly. The temperature of the material in the den was obtained through use of a thermocouple that could be inserted into the iron through the back wall.

The superphosphates were cut from the den with a blade that extended from the center of the den to one of its edges. The blade was attached to a hydraulically operated piston. As the den was rotated, the blade was advanced into the den so that it removed a 1-inch layer of superphosphate. As it was cut from the den, the superphosphate crumbled and fell onto the lower end of an eccentric conveyor, by means of which it was elevated and dumped into heated open-top curing bins. The bins, which were 18½ inches by 18½ inches by 10 feet deep and were constructed of wood and lined with asphalt-impregnated paper, had a capacity of 1000 pounds of superphosphate. The superphosphates were held in the bins at about 130° F. to simulate curing in large piles.

When either ordinary or an enriched superphosphate was being made, the mixing procedure was as follows. The flow of phosphate rock to the pan was started a few seconds before the flow of acid

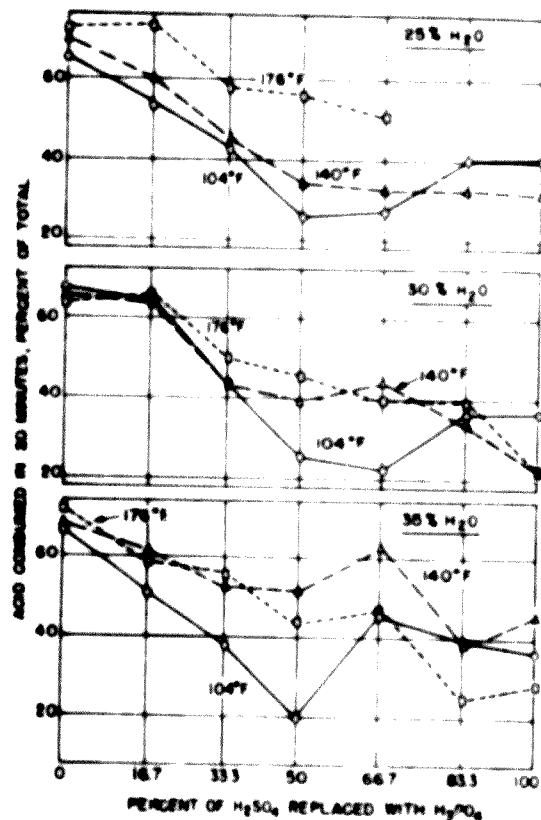


Figure 1. Effect of Replacement on Acid Consumption in 20 Minutes

temperature of the acid, these results agree with those obtained in the pilot plant. However, the concentration of the acid had little effect on conversion of rock phosphorus pentoxide in products that had been cured for 14 days or more.

In Pilot Plant Tests, Acidulation Averaged 0.97 (60 Lb. H₂SO₄/100 Lb. Phosphate Rock)

The pilot plant tests were made with 0, 20, 33, and 100% replacement acid at 110°, 130°, and 150° F.; 20% replacement corresponds to a 1:2 weight ratio of phosphoric to sulfuric acid and 33% replacement to a 1:1 weight ratio. The water contents of the acids were varied from 18 to 32%. The ordinary superphosphates (0% replacement) were made with 70% sulfuric acid at 130° F. and were carried through the tests as standards for comparison. The desired acidulation for all tests was 0.97, which is equivalent to the use

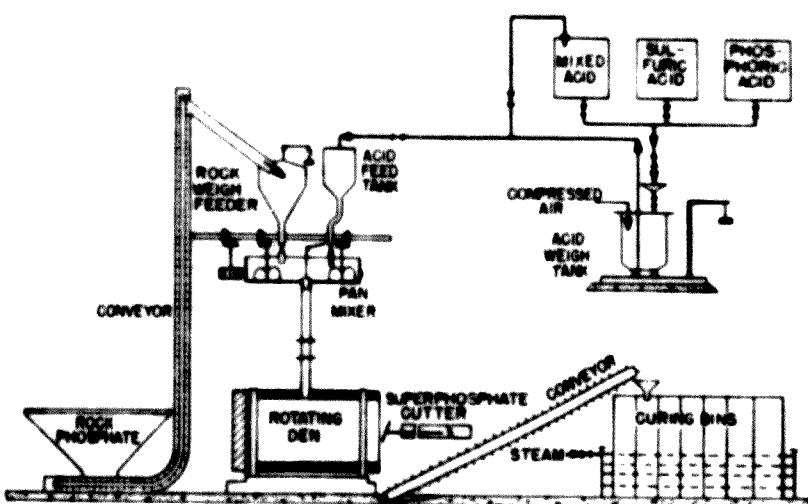
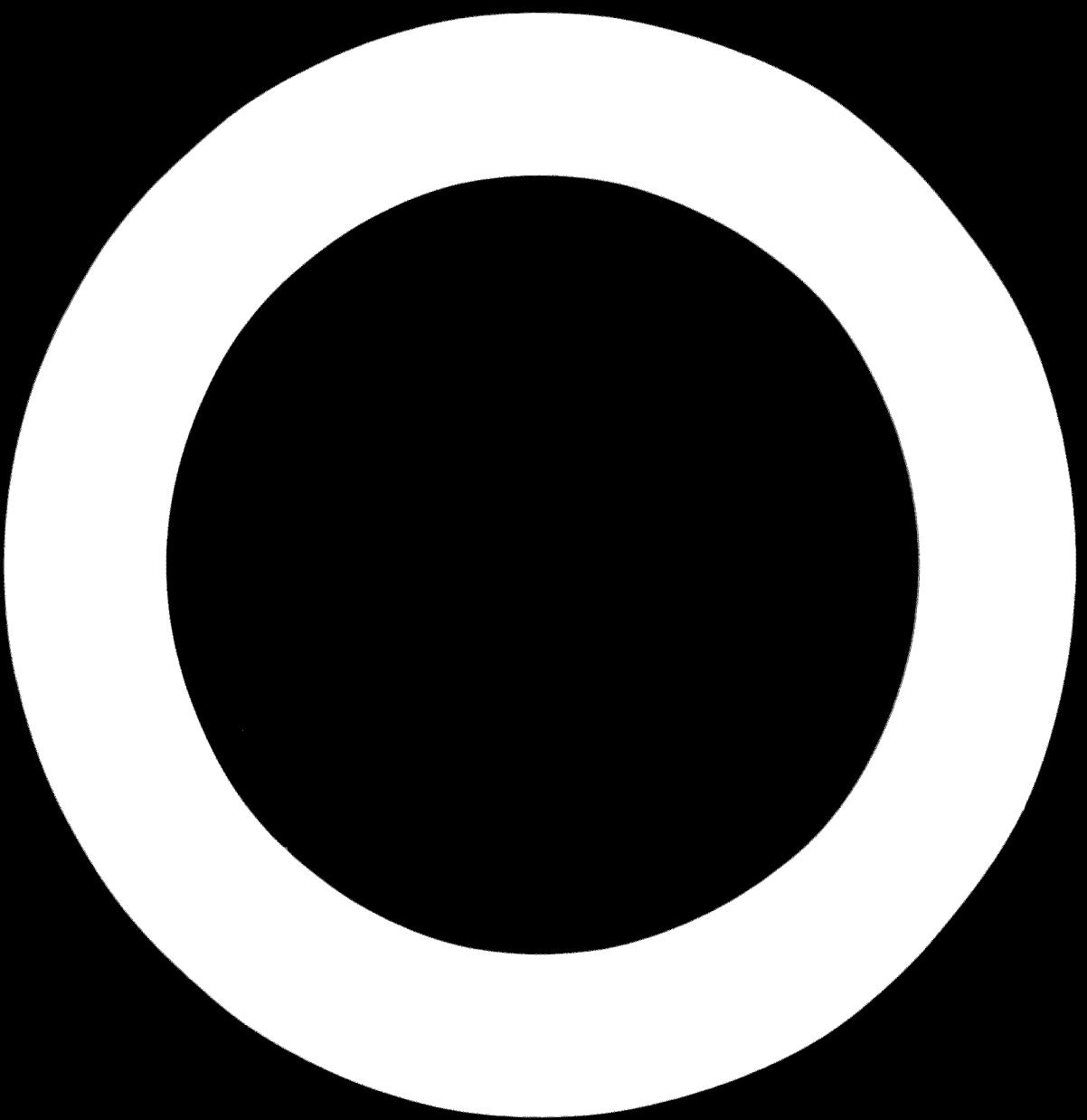


Figure 2. Flow Sheet



mixing was continued for about 20 seconds after maximum fluid had occurred, after which the pan was discharged. Charging, mixing, and discharging required about 2 minutes.

This mixing procedure was not satisfactory for making concentrated superphosphate; when it was used, the concentrated superphosphate in the den contained many lumps of partially reacted phosphate. It was found that by adding all the rock and about half of the acid simultaneously, followed in a few seconds by the addition of the remainder of the acid, a well-mixed anhydrite could be produced. The concentrated superphosphates set more rapidly than those of other types and when type B phosphoric acid (Table II) was used and the water content of the acid was in the lower part of the range investigated, the allowable mixing time was only about 45 seconds. When type C acid, which contained more impurities than type B, was used, the superphosphate remained fluid and could be retained in the pan for about 90 seconds. In making concentrated superphosphate, it appeared that best mixing was obtained when the temperature of the acid was 150° F.

Batch Temperature Decreased with Increase in Replacement

Figure 4 shows time-temperature curves for superphosphates in the mixer and in the den. The temperature of the mixed acid used in the tests was 130° F. except that 150° F. phosphoric acid was used for the concentrated superphosphate. The water contents of the acids were in the range 27.4 to 32.2%. The break in each curve represents the transfer of the superphosphate from the mixer to the den. The first readings for the den temperatures were low because of the time required for the thermocouple to reach the temperature of the superphosphate.

The curves show, as do the results of the small scale tests (Table III), that as the replacement of sulfuric acid with phosphoric acid was increased the maximum temperature attained by the superphosphate decreased and the rate of temperature rise also decreased. These observations are in agreement with predicted temperature response to sulfuric acid replacement (4). The higher acidulant temperature (150° F.) used in the case of 100% replacement is partly responsible for the position of this curve relative to the one for 33.3% replacement.

Other tests showed that when replacement was held constant decreasing the water content of the acid increased the maximum temperature.

Minimum Time in Den and Rate of Hardening Established Denning Characteristics

The study of denning characteristics involved the determination of (1) minimum denning time, which was defined as the length of time that the superphosphate had to be retained in the den to avoid slumping and to ensure satisfactory disintegration into free-flowing crumbs, and (2) the rate of hardening of the superphosphate after minimum denning time had been reached. Consideration was given to the various types and sizes of dens in use and to the length of time that ordinary superphosphate usually is retained in these dens. In continuous dens of the Broadfield or Brackett type, all the superphosphate retains in the den for the same relatively short periods. In batch dens the retention time of each succeeding mixer batch decreases and the



Figure 3. Pilot Plant

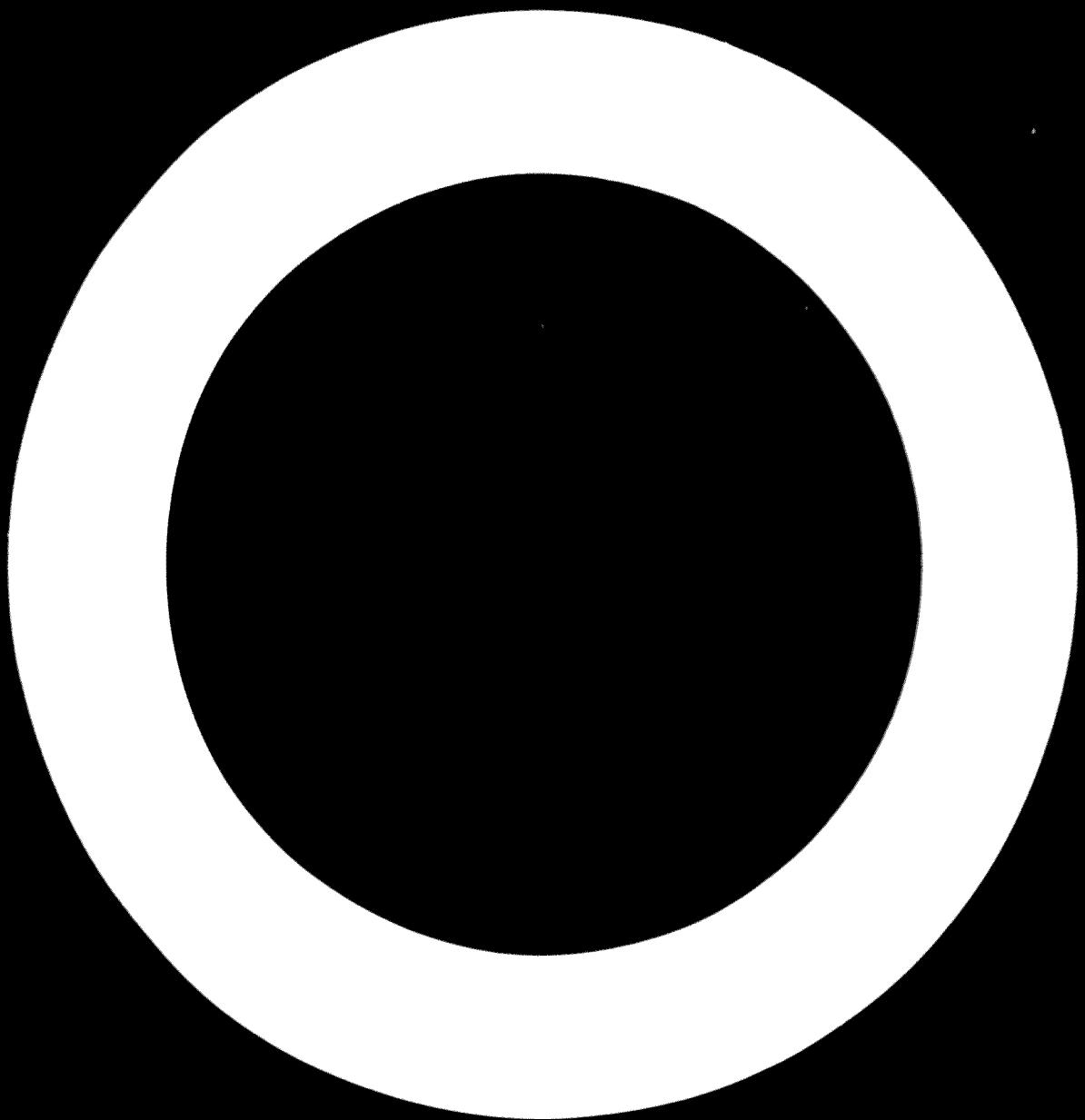


Table IV. Denning Characteristics of Superphosphates

Acidulant	Acidulation ^a P _H = 0.85 (g)	Molar Ratio in Superphosphate	First cut		Second cut		Third cut		Fourth cut	
			Water content %	Temp. °F	Minimum denning time min.	Cutter motor load ^b amp	Time min.	Cutter motor load ^b amp	Time min.	Cutter motor load ^b amp
0% Replacement										
30	130	0.94	28	0.1	43	0.2	73	0.3	—	—
30	130	0.94	18	0.2	33	0.1	63	0.3	—	—
30	130	0.94	30	0.2	210	0.2	490	0.4	—	—
30	130	0.95	28	0.2	43	0.2	73	0.4	—	—
30	130	0.96	18	0.1	33	0.2	63	0.3	—	—
30	130	0.97	18	0.1	33	0.2	63	0.3	—	—
30	130	0.97	14	0.1	29	0.2	74	0.3	—	—
30	130	0.98	23	0.1	34	0	74	0.2	—	—
20% Replacement^c										
19.5	110	0.96	28	0.1	43	0.3	73	0.6	—	—
19.5	130	0.96	23	0.2	33	0.2	73	0.6	—	—
19.5	130	0.97	28	0.2	43	0.3	63	0.6	—	—
22.5	130	1.01	33	0.1	48	0.1	106	0.3	—	—
25.5	110	0.98	43	0.1	58	0.2	106	0.2	—	—
25.5	130	0.98	43	0.1	58	0.2	84	0.2	—	—
25.5	130	0.99	18	0.2	53	0.2	84	0.2	—	—
30.5	130	0.99	13	0.1	28	0.2	53	0.2	—	—
30.5	110	0.95	29	0.1	44	0.2	74	0.2	—	—
30.5	130	0.96	30	0.1	210	0.2	74	0.2	—	—
30.5	130	0.96	18	0.1	53	0.2	84	0.2	—	—
31.5	130	0.95	24	0.2	39	0.2	63	0.3	—	—
31.5	130	0.96	28	0.1	43	0.2	74	0.2	—	—
31.5	130	0.98	30	0.2	53	0.2	74	0.2	—	—
31.5	130	1.00	24	0	49	0.1	74	0.2	—	—
32.5	130	1.01	33	0	49	0.1	74	0.2	—	—
33% Replacement^c										
19.5	110	0.96	37	0.1	50	0	86	0.3	—	—
21.0	110	0.98	50	0.1	54	0.1	84	0.3	—	—
21.0	130	0.98	54	0.1	50	0	86	0.4	—	—
21.0	130	0.99	57	0.2	50	0.1	74	0.3	—	—
21.0	130	0.99	50	0.2	50	0.1	84	0.3	—	—
21.0	130	0.99	24	0.2	50	0.1	84	0.3	—	—
22.5	110	0.98	28	0.2	48	0.1	74	0.3	—	—
22.5	130	0.98	30	0.1	53	0.2	84	0.3	—	—
22.5	130	0.94	46	0.1	61	0.2	84	0.3	—	—
22.5	130	0.95	36	0.2	41	0.2	84	0.3	—	—
27.5	130	0.97	50	0.1	270	0.2	260	0	—	—
29.5	130	0.93	118	0.3	133	0.2	163	0.1	—	—
30.5	130	0.97	107	0.1	300	0.2	340	0.3	1020	0.3
30.5	130	0.99	46	0.1	180	0.2	340	0.3	1140	0.3
32.5	130	0.97	139	0.2	340	0.1	260	0.1	—	—
100% Replacement										
18.5	130	0.98	25	0.2	55	0.3	115	0.4	1320	1.6
19.5	130	1.00	16	0.1	81	0	84	0.1	1080	0.3
20.5	130	0.93	9	0.2	74	0.1	84	0.1	—	—
22.5	130	0.99	52	0.1	62	0.2	74	0.2	—	—
25.5	130	0.98	6	0.2	21	0.2	53	0.2	—	—
27.5	130	0.98	51	0.2	61	0.2	74	0.2	—	—

^a Calculated from analysis of sample received from den.
^b Difference in current with cutter turning in superphosphate and in air.
^c Prepared using type A phosphoric acid, Table II.

^d Started cutting fragments greater than 2.5 mm.

^e Prepared using type B phosphoric acid, Table II.

^f Prepared using type C phosphoric acid, Table II.

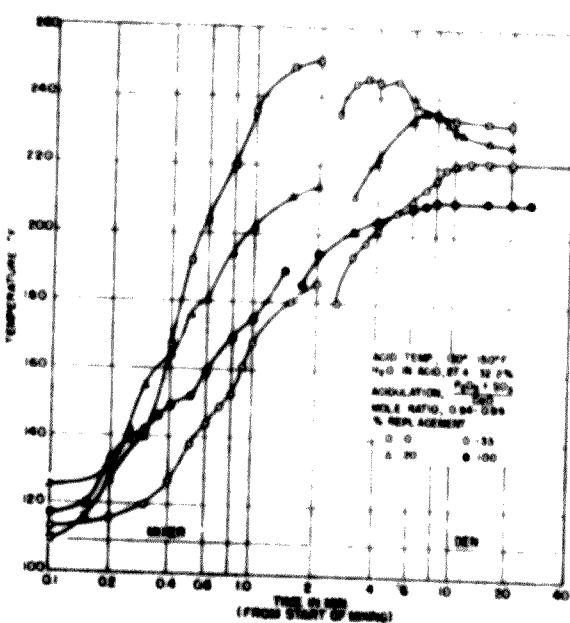
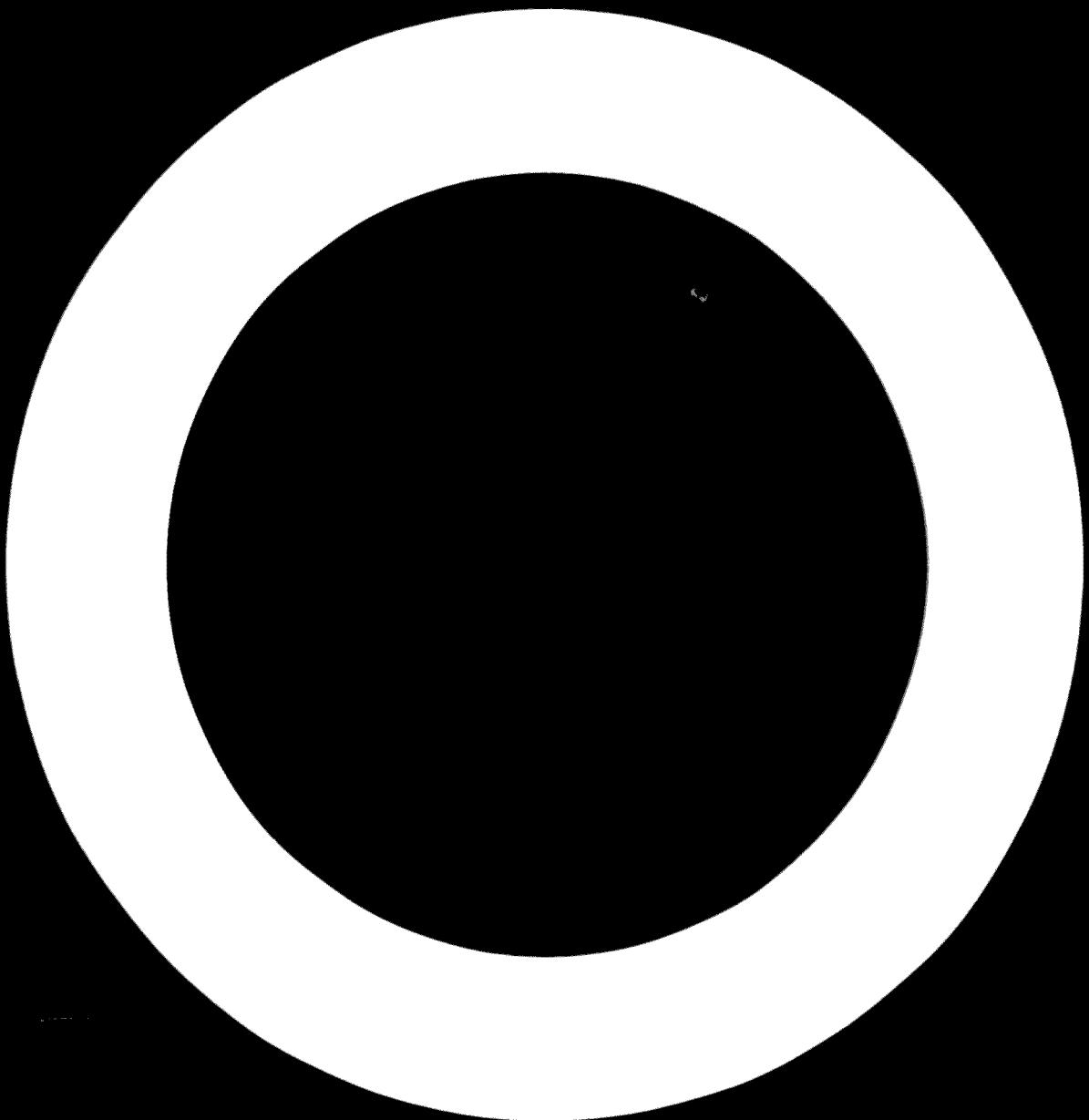


Figure 4. Effects of Proportion of Phosphoric Acid on Mixer and Den Temperatures

minimum denning time as determined in this work would apply most closely to the last batch to enter the den. It is recognized that mixing of batches in a den takes place to some extent, but the effect of that factor was not determined.

Some of the superphosphates became extremely hard a short time after they had reached a condition suitable for densification, and it appeared doubtful that they could be removed satisfactorily from the den by use of conventional excavating equipment. It was considered that a superphosphate might be removed from a den of a given type satisfactorily if it were not appreciably harder than ordinary superphosphate at the normal time of excavation for the den.

The minimum denning time and the rate of hardening were determined through use of the double-wing cutter shown in Figure 5. This cutter was rotated at 35 rpm by a small electric motor which was connected to the cutter through a speed reducer. It could be attached to the large cutter mentioned previously and advanced by the hydraulic cylinder. Through use of the double-wing cutter, a hole 10 inches in diameter and about 12 inches deep could be bored into the superphosphate in the den. The rate at which this cutter was advanced was such that each blade continuously shaved off a layer of superphosphate $\frac{1}{8}$ inch thick. The current drawn by the cutter motor while a cut was being made was recorded. The difference in current required to turn the cutter in the air and to cut the superphosphate was taken as a measure of the hardness of the superphosphate.



In testing a batch of superphosphate, the door of the den was kept closed except when cutting was in progress. The earliest time at which the superphosphate would not stamp and a satisfactory cut could be made with the double-wing cutter was called the minimum denning time. Other cuts were made later to determine the rate of hardening.

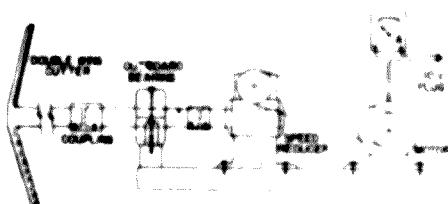


Figure 5. Double-Wing Cutter Assembly

dine the rate of hardening, the time between cuts varied with the type of superphosphate being tested. Results of the tests are shown in Table IV.

For ordinary superphosphate, the minimum denning time varied from 18 to 30 minutes and averaged 22 minutes. Figure 6 shows a smoothed curve for hardness, as measured by cutter motor load, versus denning time for ordinary superphosphate made at acidulations in the range 0.94 to 0.98. Increasing the acidulation in this range had no uniform effect on minimum denning time or on rate of hardening.

Results Indicate Satisfactory Operation for Enriched and Concentrated Superphosphates in Most Standard Dens

20% Replacement. For enriched superphosphate made with acid in which replacement was 20% (Table IV), the minimum denning time did not vary uniformly with acidulation or with the temperature or water content of the acid; it ranged from 11 to 43 minutes and averaged 31 minutes, which is not very different from the average for ordinary superphosphate. The rate at which the superphosphate hardened in the den was not appreciably affected by acid temperature. Figure 7 shows the effect of time in the den on hardness for enriched superphosphates made with acids with different water contents. The curve for ordinary superphosphate is shown for comparison. Smoothed curves are shown. As the water content of the acid increased, the rate of hardening in the den decreased. When the water content of the acid was less than about 25%, the enriched superphosphate hardened more rapidly than ordinary superphosphate. When the acid contained more than about 25% water, the rate of hardening was less rapid than that of ordinary superphosphate.

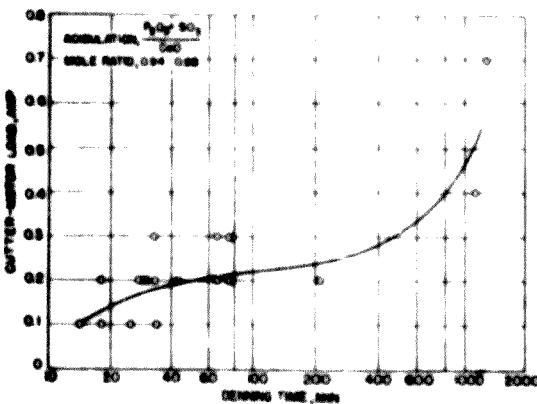


Figure 6. Effect of Time in Den on Hardness of Ordinary Superphosphate 20% Replacement

The data indicate that for a continuous den in which the superphosphate is retained for only about 30 minutes the water content of the acid could be as low as 19%. Acid of this water content probably would not be satisfactory for use with a batch-mechanical den in which the denning time is 1.5 to 2 hours because at the end of such a long period the superphosphate would be appreciably harder than ordinary superphosphate for which the excavating equipment is designed. With such a batch-mechanical den the data indicate the use of acid containing 22 to 28% water. Similarly, for use with box dens in which the superphosphate would be retained for even longer periods of time, acid containing 25 to 40% water is indicated.

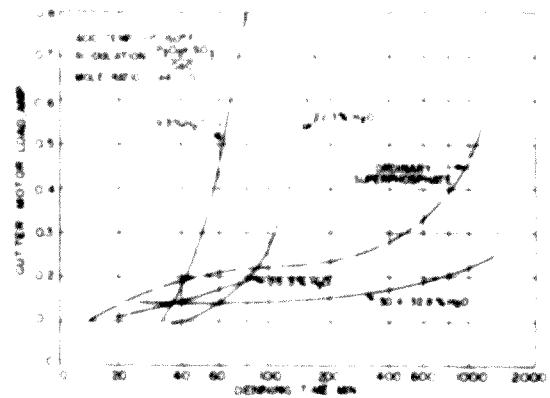


Figure 7. Effect of Time in Den on Hardness of 20% Replacement Superphosphate

33% Replacement. For enriched superphosphate in which replacement was 33%, the minimum denning time increased with water content of the acid. Figure 8. With acids containing from 21 to 22% water, minimum denning times were longer when the acid temperature was 110°F than when it was 130° or 150°F, and an increase in acid temperature increased the rate of hardening. Figure 9 shows the effect that time in the den exerts on hardness of the enriched superphosphates made with acids of several water contents at 130°F. When made with acids containing 19 to 22% water, the superphosphates hardened more rapidly than ordinary superphosphate. When made with acids containing 30 to 32% water, the superphosphates hardened at about the same or at a lower rate than ordinary superphosphate. In a test with acid containing 27.6% water the enriched superphosphate was no harder than ordinary superphosphate after 270 minutes in the den.

In the production of enriched superphosphate with acid in which replacement is 33% and containing 19 to 20% water probably would be satisfactory for use with a continuous den, and one containing 30% water probably would be satisfactory for use with a box den in which the superphosphate is retained overnight. It appears that acid containing about 22% water would be best for use with a batch-mechanical den, but the period between the minimum denning time and the time at which the superphosphate becomes appreciably harder than ordinary superphosphate would be short, and shorter than usual denning cycles probably would have to be used. A small change in water content of the acid in this range made a big difference in the denning characteristics of the superphosphate. Therefore, close control of water content would be necessary.

No tests were made of the effect of purity of phosphoric acid on the denning characteristics of enriched superphosphates.

Concentrated Superphosphate. The data for concentrated superphosphate (Table IV) show that when type B phosphoric acid was used the minimum denning time was shorter than when the more impure acid, type C, was used.

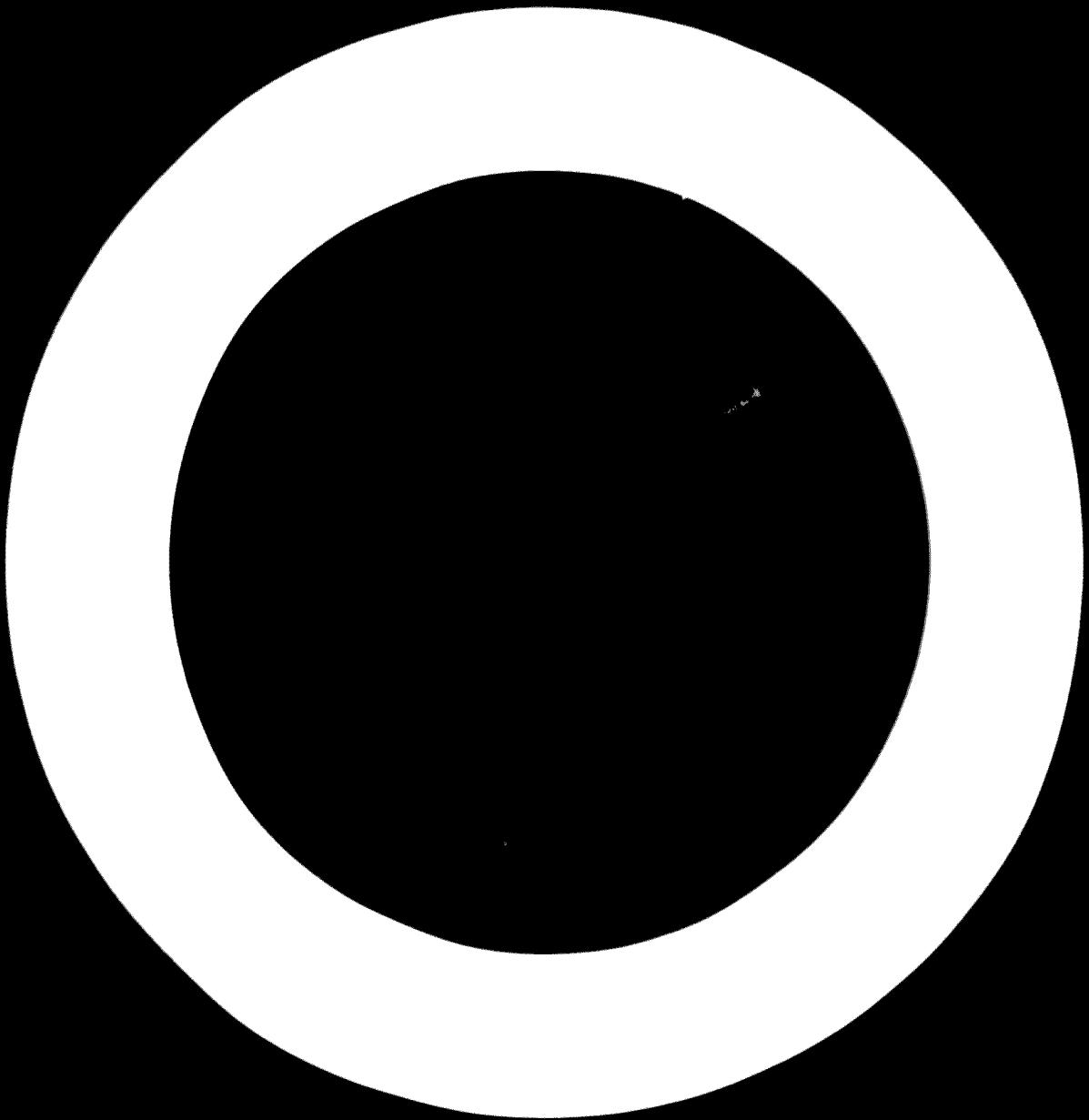


Table V. Composition of Cured Superphosphates

Acidulation P ₂ O ₅ + SO ₃ Mole Ratio	Water Content of Acid, %	Days in Storage	Composition, %				
			Total	Avail. able	Water soluble	Fine solid	H ₂ O
0% Replacement							
0.97	30	14	20.8	19.7	17.7	3.3	7.2
0.97	30	30	20.7	20.7	18.7	3.7	5.6
20% Replacement*							
1.01	22.3	14	20.2	20.7	20.0	8	4.0
0.99	21.4	14	27.8	27.2	23.9	4.3	8.0
0.99	21.4	30	28.2	27.9	25.8	3.3	5.6
33% Replacement*							
0.99	21	14	34.6	33.8	30.0	1.8	3.4
0.97	32.2	14	37.3	31.7	28.1	4.8	3.8
100% Replacement*							
0.98	18.3	14	59.1	48.1	43.7	7.2	4.0
0.98	27.4	14	47.8	47.1	40.8	3.7	4.0
0.97	27.4	30	48.1	47.8	41.8	4.3	3.7

* Samples from curing time.

† Determined by drying at 100°C. for 2 hours in air containing 10% ammonia.

‡ Prepared using type A phosphoric acid (Table II).

* Prepared using type C phosphoric acid (Table II).

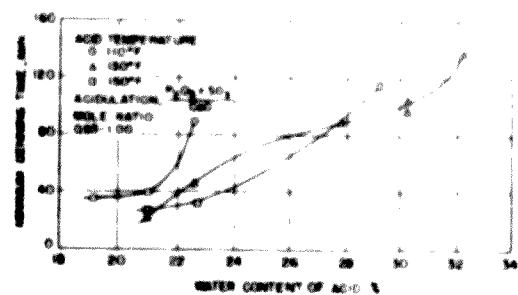


Figure 8. Effect of Water Content of Acid on Minimum Denning Time of 33% Replacement Superphosphate

However, for both types of acids and in the entire range of water contents studied, 18 to 32%, the minimum denning time did not exceed 32 minutes, which was about the same as for ordinary superphosphate. The solid curve of Figure 10 correlates denning time and hardness of concentrated superphosphates that were made with acid of type C containing 18.1 to 27.4% water and for one made with acid of type B containing 25.2%. This curve practically coincides with that for ordinary superphosphate; this indicates that any of these concentrated superphosphates probably could be handled in any kind of ordinary superphos-

phate den. The dotted curve is for concentrated superphosphate made using acid of type B containing 20.1% water. This superphosphate hardened more quickly than ordinary superphosphate. It appears that increasing the purity of the acid increases rate of hardening.

For the range of acid purity that might be encountered in commercial practice, the data on curing and denning characteristics indicate the use of acid containing from 25 to 27% water in the production of concentrated superphosphate in equipment commonly used in producing ordinary superphosphate.

Cured Products Had High P₂O₅ Availability

Only products made with acid of type C were prepared for curing because analyses of samples made with acid of type B and quick-cured at 100°C. had indicated that concentrations of rock phosphate pentoxide were about equal when the temperature of the acid was 100 or 150°F., and that a difference was found when the acid was 100°F. The same general results were noted in the small-scale test (Table III). Samples of the concentrated superphosphates that had been made with acid of type C and were better mixed in the past were cured in the oven.

Figure 11 shows conversion of phosphorus pentoxide availability in superphosphate obtained from the curing time after storage for various periods of time. These superphosphates were made with acids in which the replacement ratios were 20, 21,

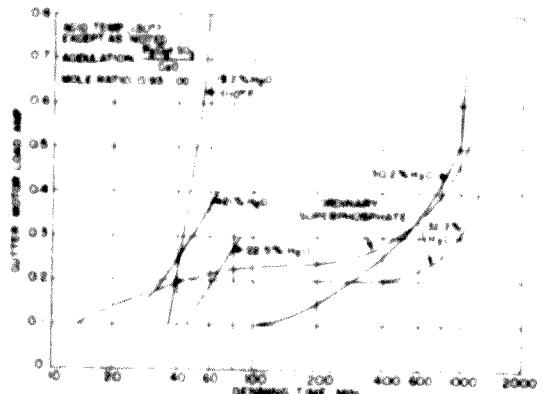


Figure 9. Effect of Time in Den on Hardness of 33% Replacement Superphosphate

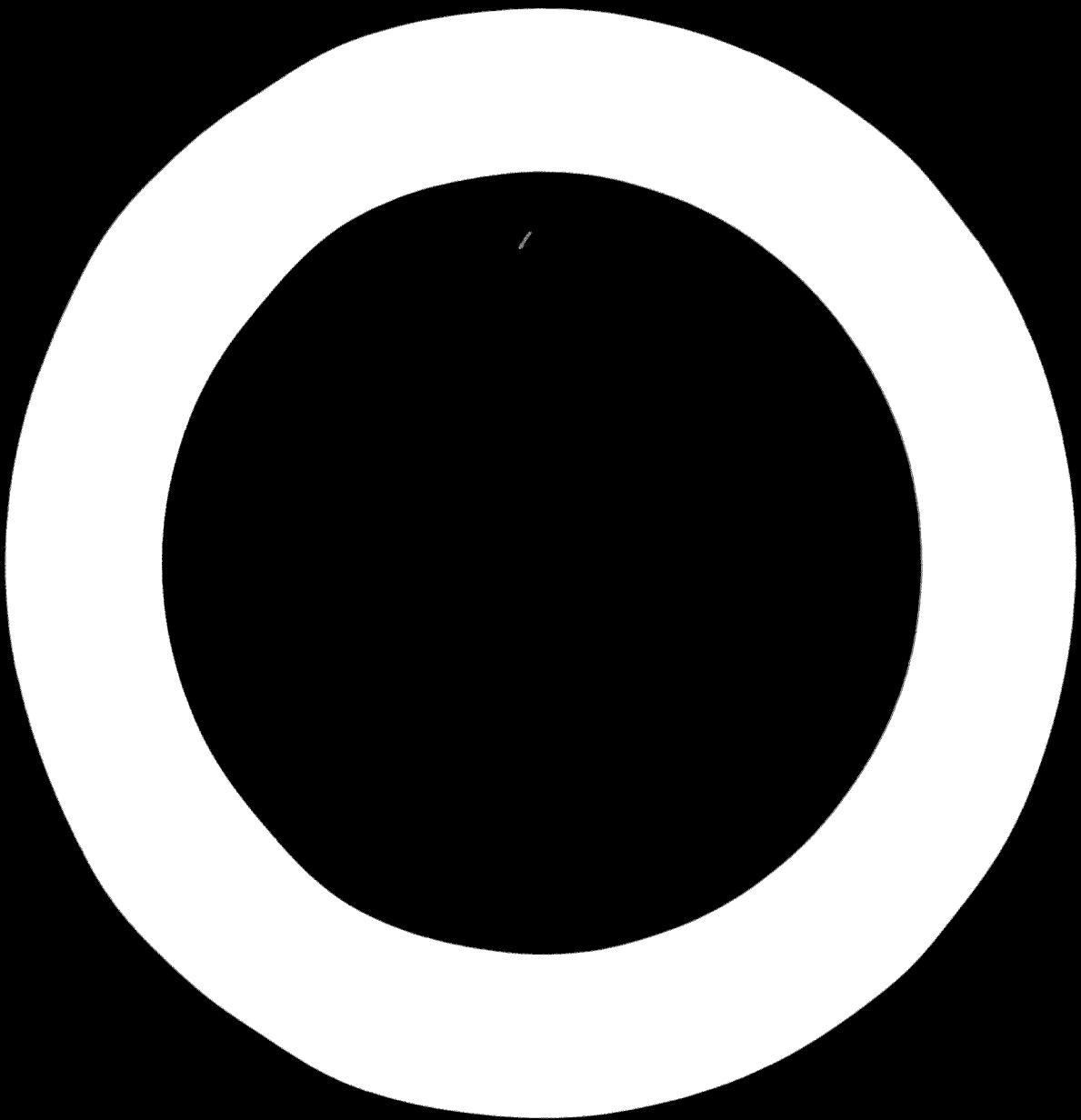
Table VI. Results of Ammoniation of Superphosphates

Water content of acid, %	Acidulation P ₂ O ₅ + SO ₃ Mole Ratio	Curing days	Description of Sample	Ammoniation Lb. NH ₃ /Tun Dolys in Feed	Total	P ₂ O ₅ analysis, %	Ammoniated Superphosphate Characteristics			P ₂ O ₅ availability
							Available	Water soluble	Nitrogen as NH ₃ , No.	
0% Replacement										
30	0.98	14	Feed Product	2.72	20.8	20.1	18.0	12.4	1.1	98.1
20% Replacement*										
22	1.01	14	Feed Product	2.68	30.1	30.3	25.1	15.0	1.0	98.7
31.1	0.98	14	Feed Product	2.68	27.8	27.2	15.7	10.0	1.0	97.8
32.3	0.97	14	Feed Product	2.90	26.7	26.3	24.3	14.0	1.0	98.5
33% Replacement*										
21	0.99	14	Feed Product	3.06	35.0	34.3	30.5	21.0	1.0	98.0
32.3	0.97	14	Feed Product	3.06	29.9	29.0	15.0	10.0	1.0	97.3
32.3	0.97	30	Feed Product	3.23	31.7	28.1	23.2	16.0	1.0	98.1
100% Replacement*										
18	0.98	21	Feed Product	2.53	48.4	47.8	31.4	23.2	1.2	98.8
27.4	0.97	30	Feed Product	2.68	42.4	40.8	23.2	16.0	1.0	98.1
27.4	0.97	30	Feed Product	2.68	48.1	47.8	41.3	22.1	1.0	98.1
27.4	0.97	30	Feed Product	2.68	39.4	38.3	22.1	8.3	0.4	97.2

* Prepared using type A phosphoric acid (Table II).

† Product indicated 18% replacement.

‡ Prepared using type C phosphoric acid (Table II).



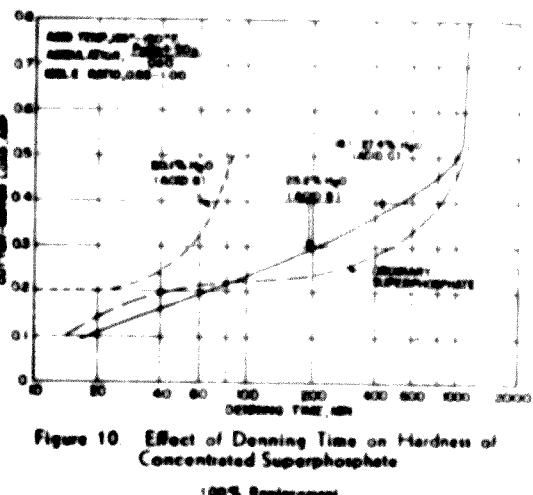


Figure 10. Effect of Denning Time on Hardness of Concentrated Superphosphate
100% Replacement

and 100% and the water contents were 10, 11, 12, 21, and 27.4% respectively. Type A phosphoric acid was used in making the enriched superphosphates and type C acid was used in the concentrated superphosphate.

After 14 or 30 days of curing, there was little difference in conversions of rock phosphorus pentoxide or availability of phosphorus pentoxide in the superphosphates. Conversion of rock phosphorus pentoxide in the concentrated superphosphate increased most rapidly with time of curing. Figure 12 shows the effects of water content of the acid on conversion of enriched and concentrated superphosphates. Conversion in the enriched superphosphate was increased about 17% when the water content of the acid was increased from about 20 to 30%. Conversions in concentrated superphosphates made with acids containing 18 or 27% water and cured 14 to 30 days were about equal. Somewhat lower conversions were obtained in the concentrated superphosphate made with acid that contained 21.1% water, but it appears that conversion in this material might have equaled that in the other concentrated superphosphates if curing had been continued for a longer time. After 21 days of curing, phosphorus pentoxide availability was 97.1% in this superphosphate which was higher than that in the ordinary superphosphate that had been made at the same acidulation and cured for a like period.

Analysis of bin-cured superphosphates. Table V shows that the grade is affected by replacement, water content of acid, and curing time. The effects of replacement and water content are so pronounced that close control of acid composition would be required in commercial operation. It was noted that conversion and grade had increased slightly after disintegration and handling of the superphosphates in the pilot plant for ammoniation or bagging. However, such increases probably would not occur in a large plant because there would be less opportunity for the product to dry.

The data on fluorine evolution during mixing and curing are too few for presentation on a quantitative basis. However, they indicate that the proportion of fluorine evolved from the rock and acid decreased as replacement increased.

Curing in the bins was found to be comparable to curing in large piles. Nomographs published by Sheld, Wright, and Souchelli (11) show that ordinary superphosphate made under the condition of acid concentration and temperature, phosphate composition, and acidulation used as standard in the present work should contain 20.5% total phosphorus pentoxide and about 19.9% available phosphorus pentoxide after being cured in a plant pile for 30 days. The averages of total and available phosphorus pentoxide contents of three batches of the standard ordinary superphosphate cured for 30 days in the bins were 20.7 and 20.1%, respectively (Table V).

The data indicate that obtaining high phosphorus pentoxide availability presents no problem with any of these superphosphates. Using acids with low water content in the range studied, the phosphorus pentoxide availability in all the superphosphates would be about the same after they had been cured. However, to produce a superphosphate of maximum grade and to ensure that a mixed fertilizer containing this superphosphate would be in the best condition, the water content of the acid should be as low as possible. The data indicate that the minimum water content that could be used agrees with the limiting characteristics of the superphosphate and the type of fertilizer used.

Products Showed No Caking after Storage for 6 Months

Superphosphates that had been cured in 30 days were bagged in five-ply paper bags having the surface untreated poly and stored in piles twelve bags high. The superphosphates showed no caking after 6 months of storage. The enriched and concentrated superphosphates (filled with 100% rock phosphate) superphosphate through a John Ring No. 300 wire mesh fertilizer belt after storage.

Ammoniation of the Superphosphates Resulted in Only Small Losses in P.O. Availability

Some of the cured superphosphates were ammoniated to determine how they would respond to this treatment in the preparation of mixed fertilizers. Although in the manufacture of mixed fertilizers, superphosphate usually is mixed with other fertilizer ingredients before being ammoniated, the superphosphates prepared in the present work were ammoniated by themselves.

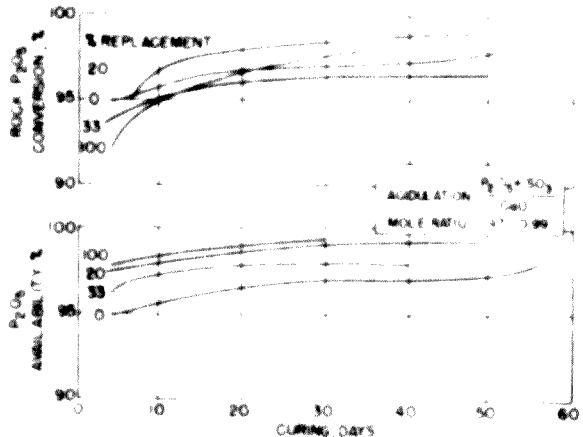


Figure 11. Effects of Proportion of Phosphoric Acid and of Curing on P.O. Availability and Conversion

A 1-ton drum-type mixer was used in the ammoniation of 1000-pound batches of the cured superphosphates with ammoniating solution prepared in the laboratory to have the composition of Barrett No. III nitrogen solution (55.5% NH_4NO_3 , 26% NH_3 , and 18.5% H_2O). The average degree of ammoniation was 2.8 pounds of neutralizing ammonia per unit of available phosphorus pentoxide in the unammoniated superphosphate, which probably is lower than that used by some manufacturers but is in the upper part of the range recommended in the manual "Barrett Standard Nitrogen Solutions" (1). The average rate of addition of solution to the mixer was such that 26 pounds of neutralizing ammonia was added per minute. The superphosphates that were used had been made at acidulations of from 0.95 to 1.01 and had been cured for 14 to 30 days before being ammoniated.

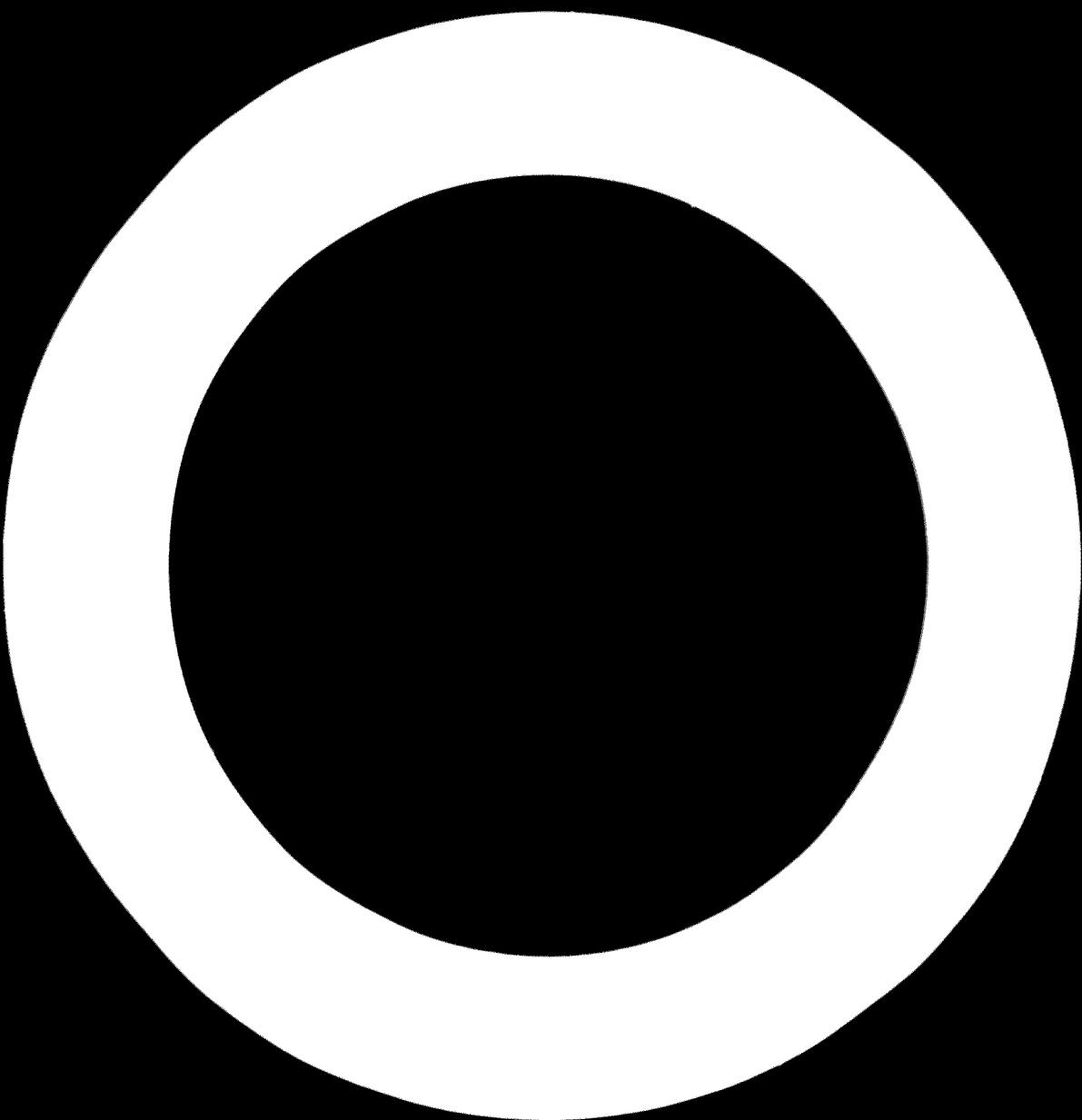


Table VI shows data and results from the ammoniation tests. Losses of ammonia were practically nil. In tests with ordinary superphosphate the average loss of phosphorus pentoxide availability was 2.8%. In the ammoniation of concentrated superphosphate that had been made with acid that contained 18 or 27% water, loss of phosphorus pentoxide availability was about

2.8%. In the ammoniation of the enriched superphosphates, loss of availability increased from about 1% to 4 to 5% as the water content of the acid used in preparing the superphosphate was increased from about 20 to 30%. Hecht et al. (6) and Jones and Rohner (8) have shown that increasing the moisture content of superphosphate or of mixtures containing superphosphate increased the loss of phosphorus pentoxide availability incurred on ammoniation.

None of the losses of availability is considered serious. However, as shown by Keenan (9), the losses would be much smaller if the superphosphate were ammoniated in mixture with other materials or if the ammoniated superphosphate were mixed with other materials. Dry mixing the ammoniated 33% replacement superphosphate made with acid containing 32.3% water (Table VI) with enough ammonium sulfate and muriate of potash to produce a fertilizer of 11-11-11 grade increased phosphorus pentoxide availability from 92.9 to 95.7%, which, in effect, reduced the loss of availability from 5.2 to 2.4%. The behavior of the experimental superphosphates on being ammoniated appeared satisfactory.

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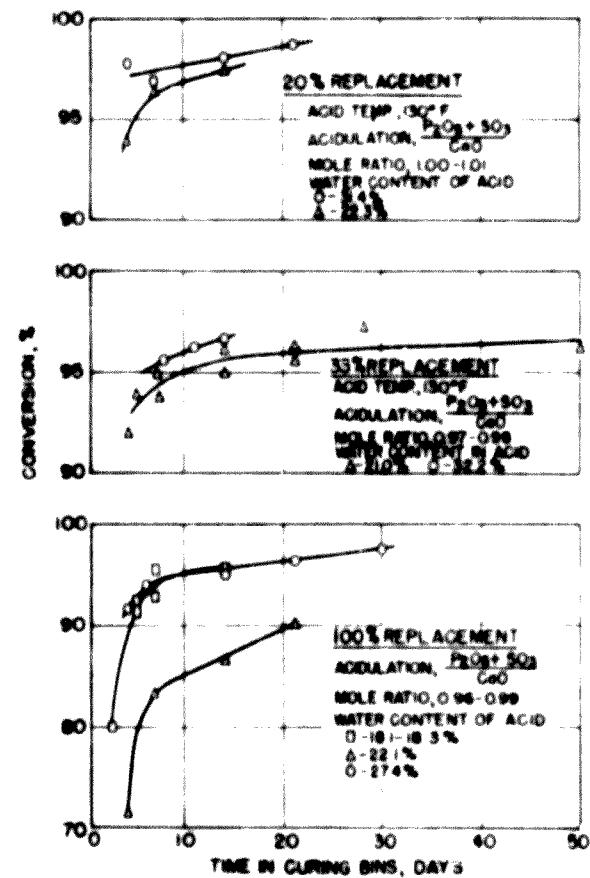
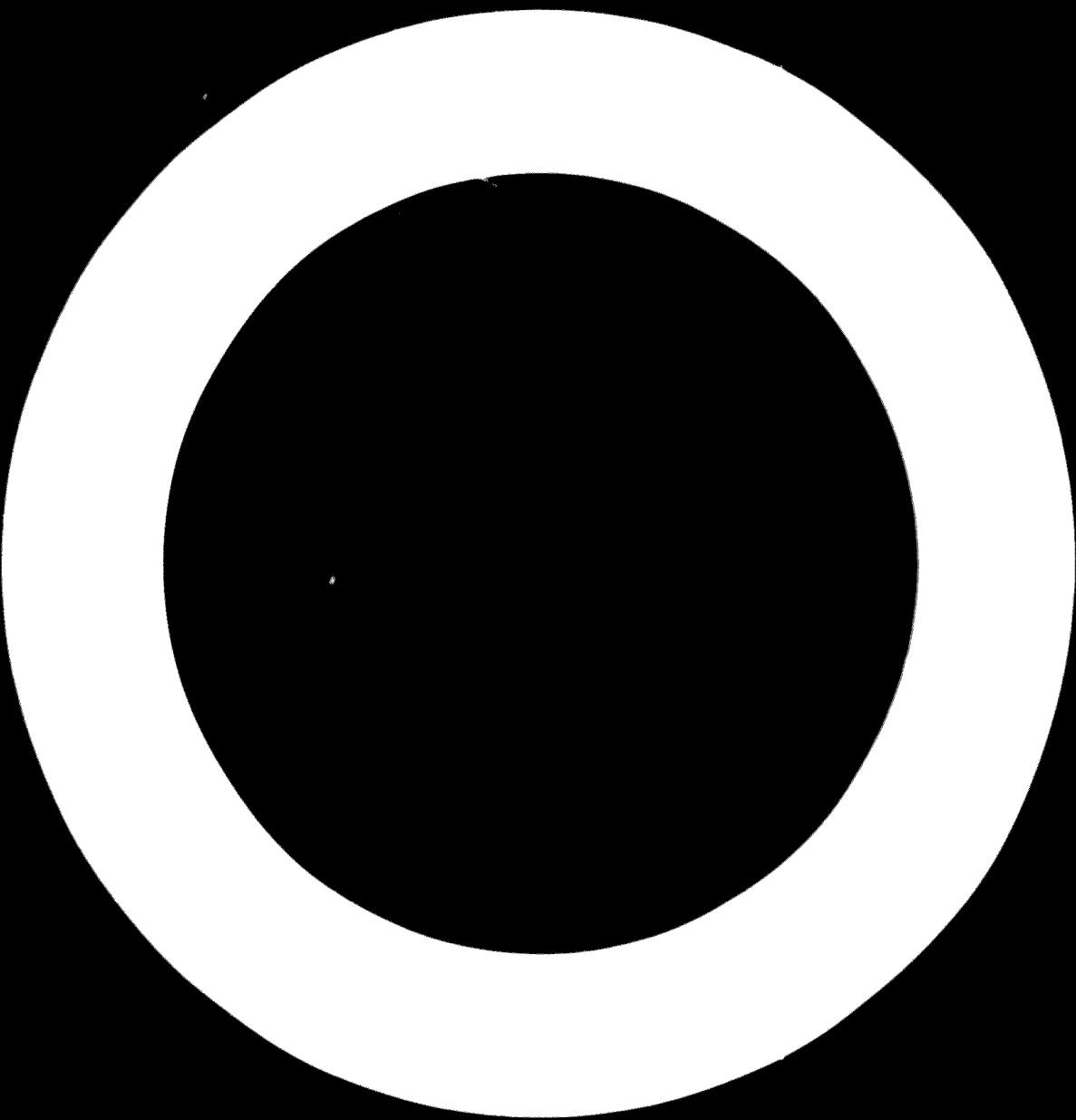


Figure 12. Effect of Curing Time and Water Content of Acid on Conversion of Rock P₂O₅.

Received for review October 2, 1952

Accepted January 27, 1953



JOB DESCRIPTION

Post Title: A Senior Chemical Engineer - Chemical Fertilizer Project Expert.

Duration: One year.

Date required: January 1970.

Duties: The expert will be assigned to work at the phosphate plant in Brazil.

He will be given the plant project manual and the final completion stages of the plant to review assistance during the start-up of the sulphuric acid and superphosphate fertilizer plant.

He will also do feasibility study of the production of phosphoric acid and the production of single and triple superphosphate.

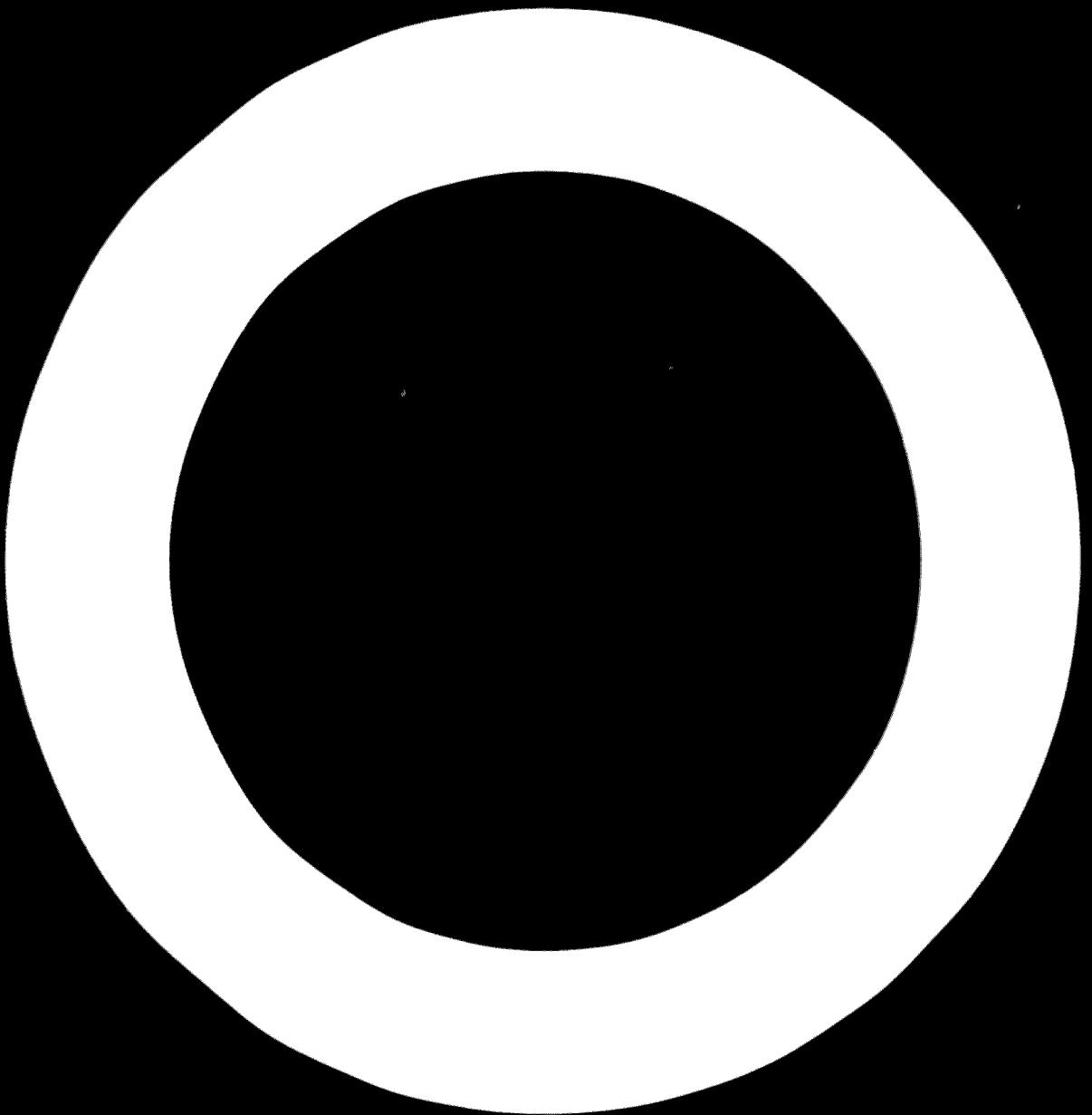
In addition, he will assist the government in working out the details of using excess current from the Brazilian plant in producing ammonia superphosphate and other complex fertilizers at the plant.

Qualifications: University Degree in Chemical Engineering, or Mechanical Engineering, and at least 5 years experience in the production of superphosphate or ammonia superphosphate fertilizers.

Language: English.

Background Information: The phosphate plant capability was started in 1961 under bilateral agreements and construction was stopped in 1965 when the agreements were terminated. 92% of the equipment is on hand and the erection of the plant is about 30% complete. It is expected that construction will be renewed in 1969 and that the plant will be finished in 1970.

The plant is designed to produce 100,000 tons per year of granular single superphosphate. The plant consists of 120 ton per day sulphuric acid plant, phosphate rock grinding facilities, superphosphate mixing and storage facilities



Appendix I cont'

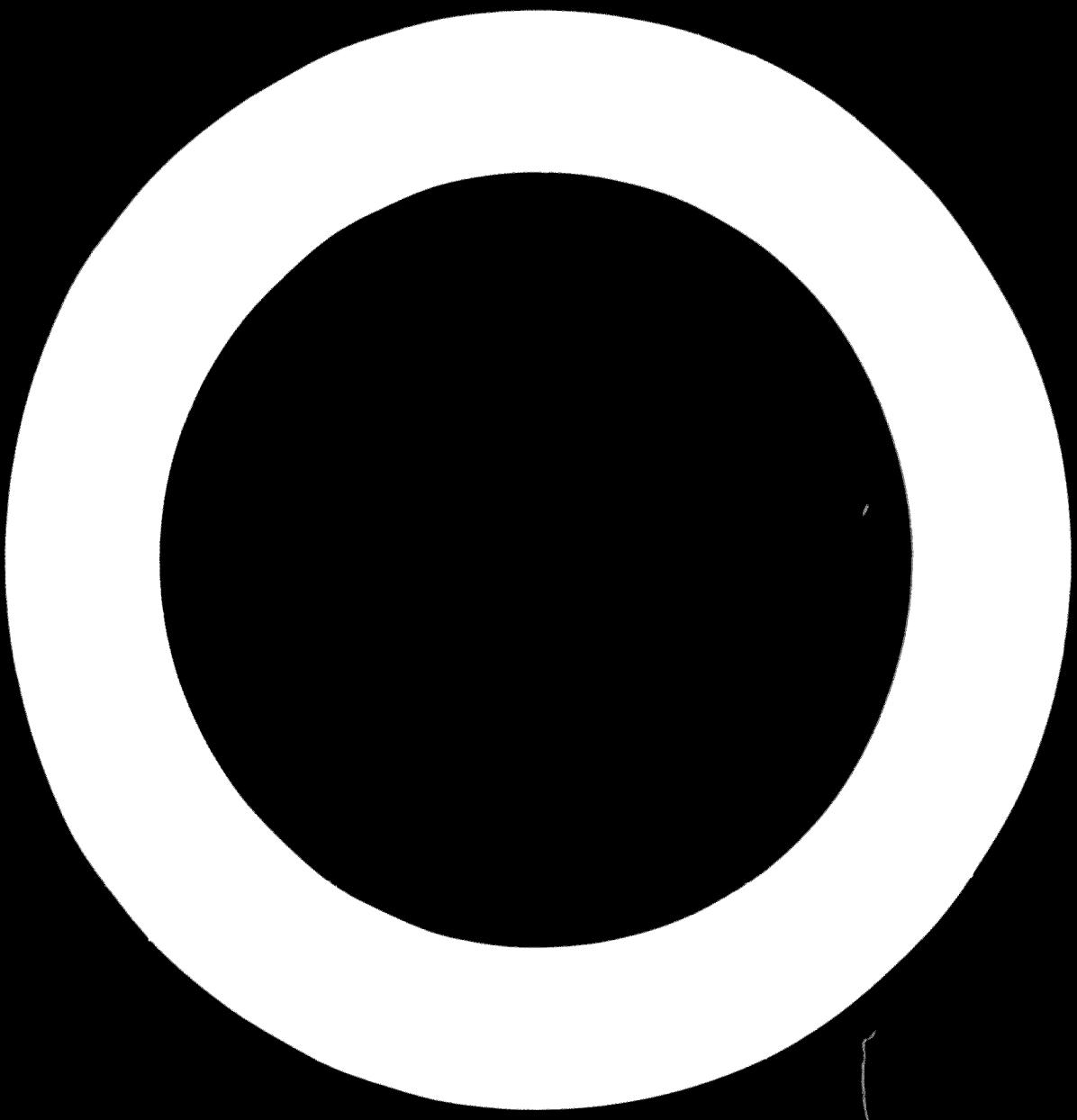
1.

Background Information

The plant will be located in the power plant, water in New Mexico part of the project. The plant will start up around May 1970 and run about.

The plant has enough equipment so that right additional investment will result in the construction of superphosphate, the manufacture of triple superphosphate, the production of triple superphosphate.

The ammonia plant at Irenex will be constructed this year. It uses ammonia to produce fertilizer or by-product sulphur. This excess ammonia will be utilized during the chemical operations of the Irenex plant. At 1 C,000 t/h per year superphosphate fertilizer will be output at 1,000 t/h. There will be no by-products.



THE STATE OF THE ARTINTRODUCTION

Officially, Indonesia's foreign trade policy is to import what it needs and export what it can. In practice, however, the country has been unable to meet its foreign exchange requirements.

FOREIGN TRADE

An joint representation by the
INTP, Sinar Indah

Proposed government participation
Contribution

Contribution to war
Local operating costs

Proposed Government
Co-operation

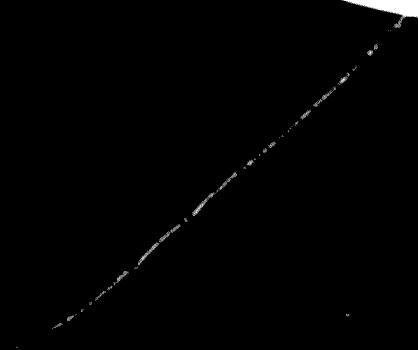
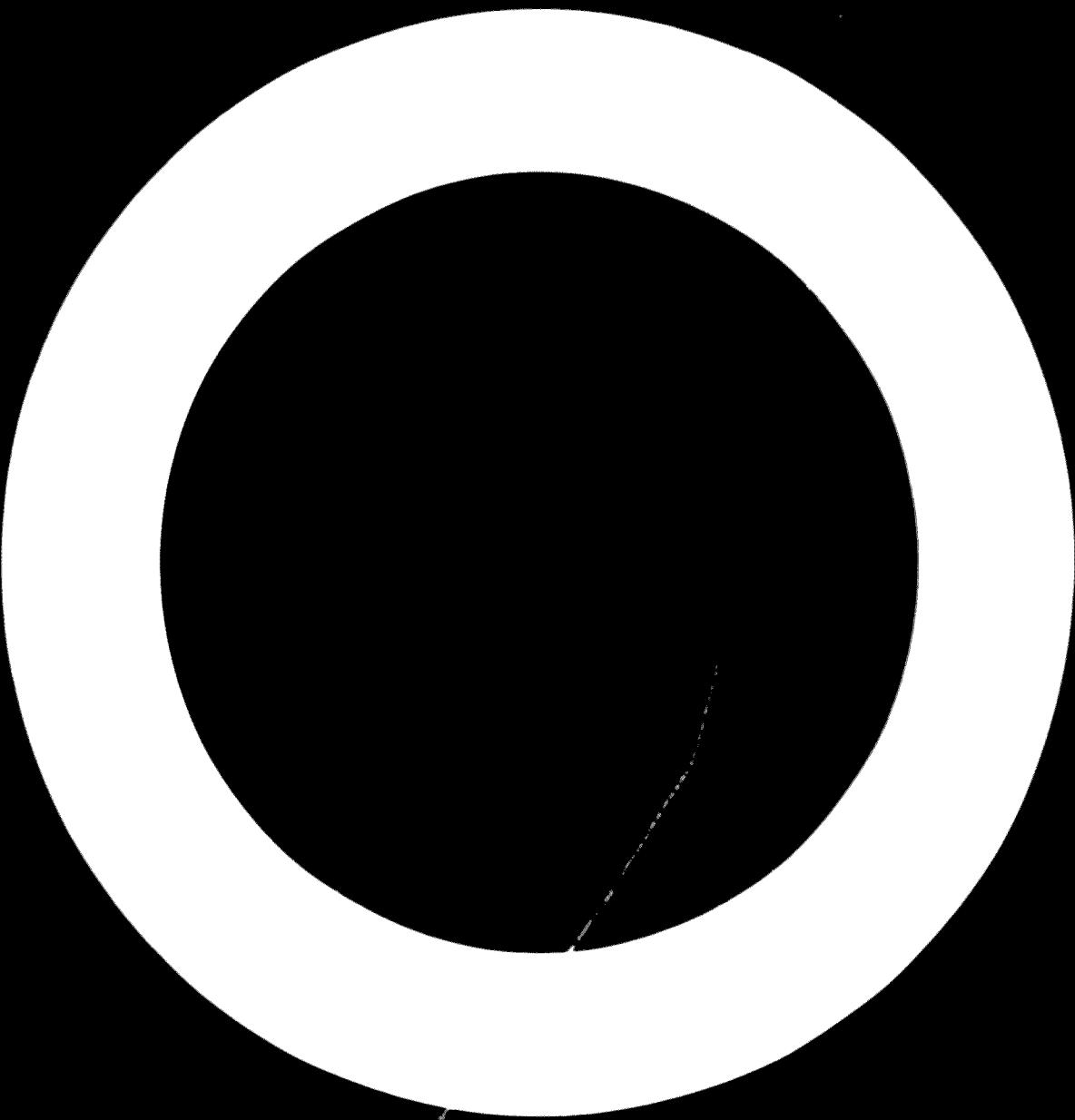
Proposed contribution

Department of Finance
Industry

Participation in Executive Agency
Indonesia Industrial
Development Organization

I. Background

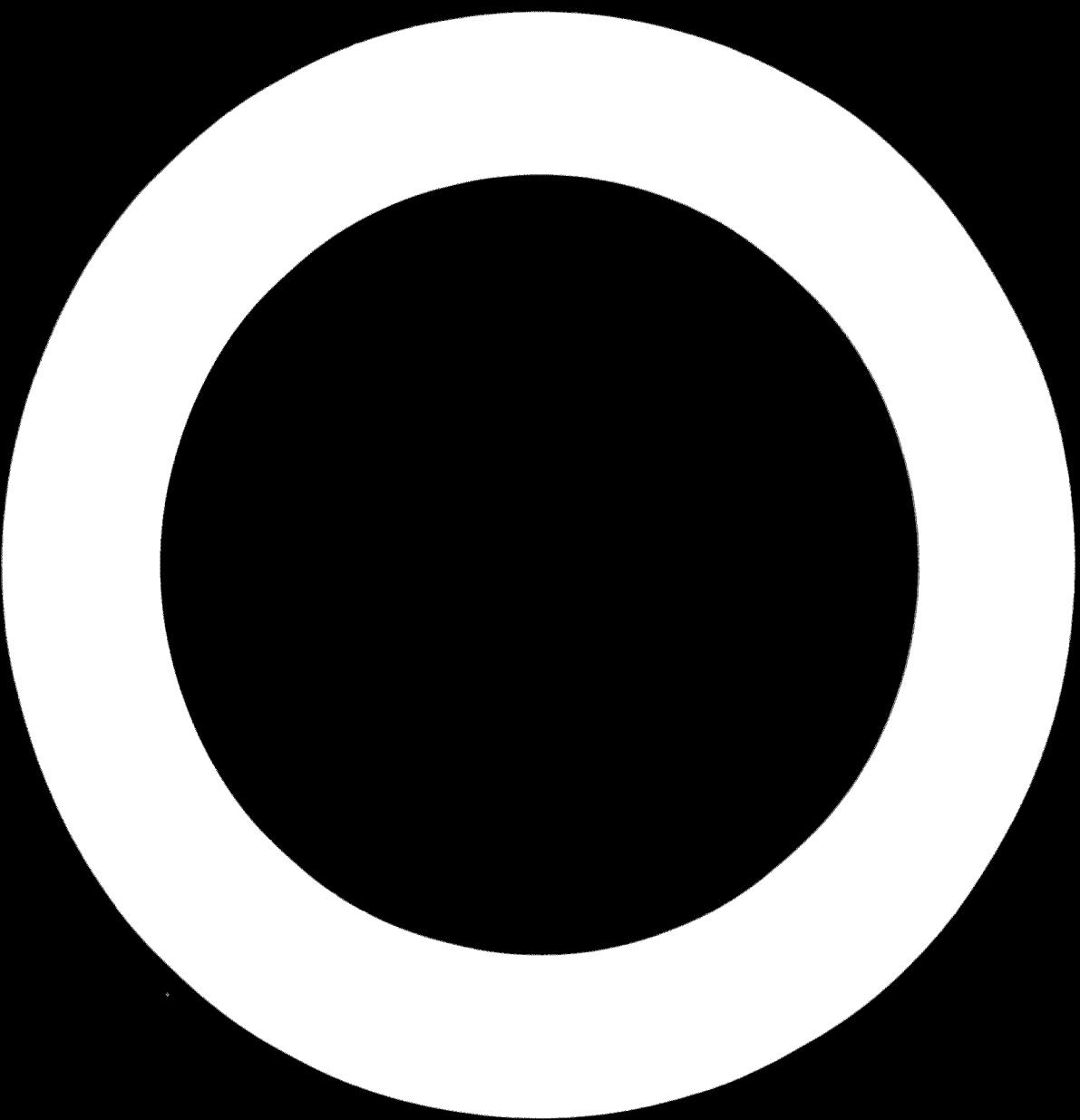
1. The Government of Indonesia is attempting to increase agricultural input industries. Indonesia's purchase of large quantities of rice (up to 1.7 million tonnes in the world market) and consequent heavy expenditure of scarce foreign exchange underlines the importance of increasing domestic rice and other food production. The central role of cash crops in Indonesia's exports highlights the need for higher yields and expanded output of these crops. To achieve this will require, among other things, the use of much greater quantities of fertilizers.
2. An ammonium sulphate facility being built at Gresik and a superphosphate plant under construction at Tjilatjap will go far in saving foreign exchange being used to import fertilizers. However, these facilities will start up using imported sulphur. The development of an indigenous sulphur industry will save US\$ 2,250,000 yearly, in foreign exchange if a 160 ton per day facility to supply the needs of the Gresik and Tjilatjap plants is installed at Wanaradja to process the sulphur muls from the Telaga Bodas crater. This crater has economic reserves of over 846,000 tons of sulphur at a distance of 14 km from an existing plant at Wanaradja.



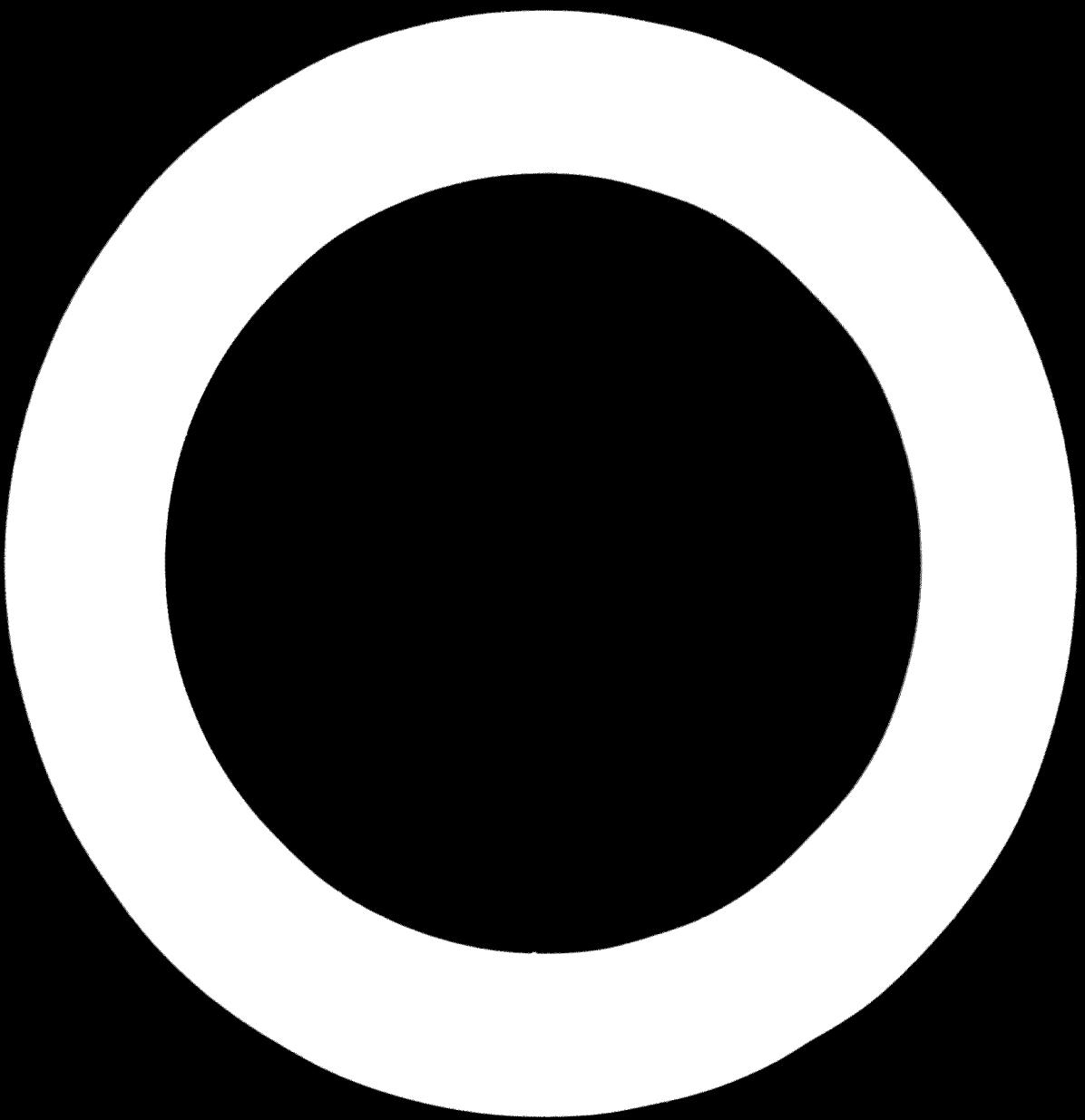
3. However, the existing plants will manufacture tonnes of sulphur in a lower grade form which will be free from sulphur from the feed clay, and the project must therefore build a sulphurizing plant per se facility.
4. A technical pilot plant is needed to obtain experience from laboratory plant. The amount of laboratory work requested, the assistance of the UNDP Special Fund in establishing this pilot plant.
5. The project has been evaluated with the aid of a consultant from the United Nations Industrial Development Organization.

II. The Project

6. The purpose of the project, the duration of which is two years, is to develop technical recommendations for the design of a large-scale facility which will produce 100 tonnes per day of 99% sulphur. The sulphur will supply sulphuric acid facilities being built in fertilizer plants at Iresik and Tjilatjap. The pilot plant will have a capacity of 1 tonne of sulphur per 24 hours day.
7. The pilot plant is expected to provide the following:
 - (a) Means for concentrating the sulphur in the feed clay from a range of 20% - 65% S to a uniform feed containing about 75% sulphur. This will involve studies using classifying, flotation and thickening equipment.
 - (b) Means for separating the sulphur from the clay particles in the feed. This will involve studies with different kinds of float clavie arrangements.
 - (c) Means for removing organic impurities from the melted sulphur. This will involve studies with various filtration aids.
 - (d) Technical problems involved in direct combustion of the feed clay.
 - (e) Means for transforming the melted sulphur into a solid suitable for shipment. This will involve a study of casting techniques.



8. The UNDP (Special Fund) is requested to provide:
 - the services of an internationally recruited expert Project Manager for a total period of two man-years,
 - the services of an internationally recruited expert with relevant experience for a total period of two man-years,
 - pilot plant equipment,
 - certain office equipment,
 - costs of other necessary imported equipment,
 - expenses for travel of experts outside the country.
9. The Government proposes to provide:
 - modernization of the guest house at Gharut for experts living quarters,
 - the services and salaries of national counterparts totalling 7½ man-years,
 - secretaries, chauffeurs and maintenance staff,
 - skilled and unskilled labour for the operation of the pilot plant,
 - necessary civil engineering works connected with the erection of the pilot plant,
 - for the haulage of all imported equipment from the point of delivery to the site and to provide skilled and unskilled labour and locally available auxiliary construction material required in the erection of the process equipment furnished by the United Nations Development Programme,
 - use of the existing facilities at the Wanarega sulphur plant where possible,
 - meeting such operating costs for the running of the pilot plant as can be provided from local sources.



Budget Estimate

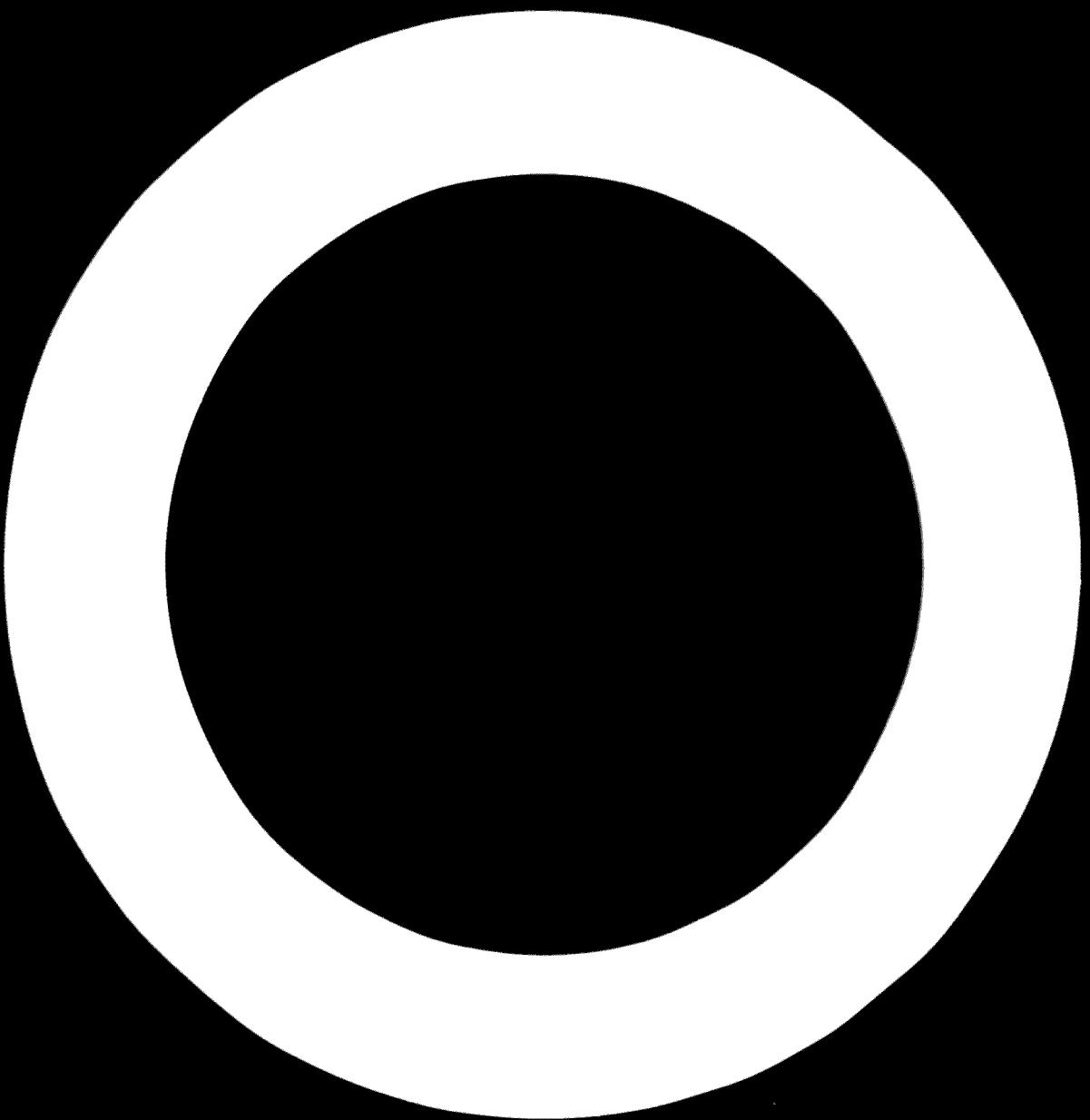
III. Financial Data

A. Contributions from International Organizations

	US\$	£	Rs.
(a) Experts	Sub-tot	Sub-tot	Sub-tot
Project Leader	21	1	100
Grants-in-aid			
Equipment	26	2	120
			120,000
(b) Equipment			
Imported office equipment	10,000	1,000	50,000
Joe:			
Trucks (2)	10,000	1,000	50,000
Furniture			
1. Classifier	1,000	100	500
2. Flotation cell	1,000	100	500
3. Thinner	1,000	100	500
4. Autoclave	1,000	100	500
5. Filter unit compressor	1,000	100	500
6. Pump unit, 110V	1,000	100	500
7. Casting materials	1,000	100	500
8. Instrumentation and laboratory equipment	20,000	2,000	10,000
9. Chemical reagents	1,000	100	500
	Sub-tot		120,000

(c) Miscellaneous

Documentation	US\$ 5,000	£ 500
Travel expenses outside the country	5,000	
Contingency	<u>30,000</u>	
	Sub-tot	US\$ 40,000
TOTAL amount requested from the UNDP (Special Fund)	US\$ 310,000	



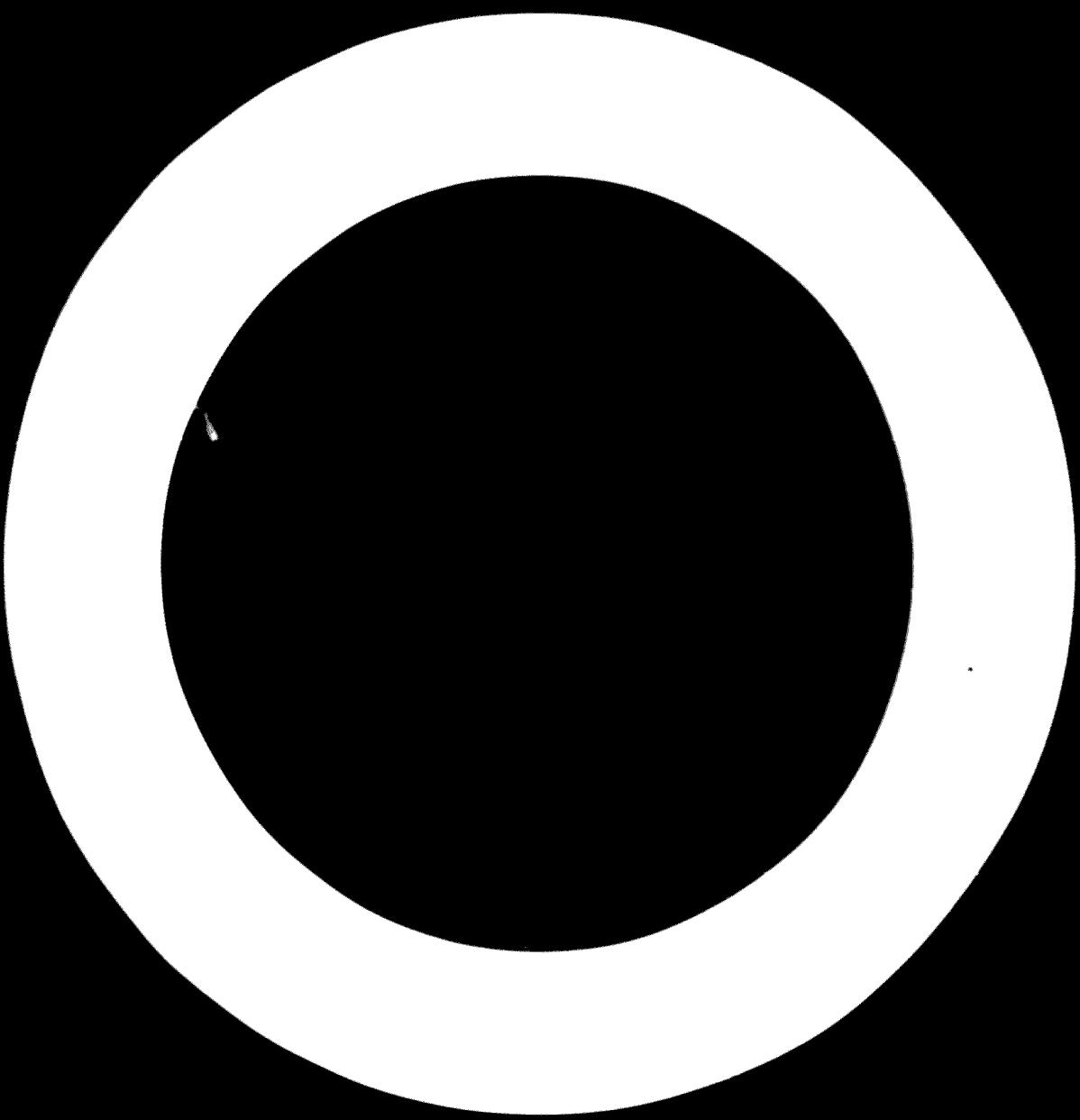
Air Force Budget

B. Project and Government Counter-Part Contributions

<u>Counter-Part Staff</u>	<u>Total</u>	<u>Total</u>	<u>Y</u>	<u>Y</u>
	<u>US\$</u>	<u>US\$</u>	<u>%</u>	<u>%</u>
Project Manager	40	40	10	10
Mining Engineer	10	10	2.5	2.5
Chemical Engineer	10	10	2.5	2.5
Chemist	10	10	2.5	2.5
	<u>40</u>	<u>40</u>	<u>10</u>	<u>10</u>
Other Salaries	100	100	25	25
Rental of office space & furniture	100	100	25	25
Office supplies	10	10	2.5	2.5
Civil engineer	10	10	2.5	2.5
Operating expense	100	100	25	25
Transportation	10	10	2.5	2.5
Domestic travel	10	10	2.5	2.5
Maintenance of guest house	10	10	2.5	2.5
	<u>100</u>	<u>100</u>	<u>25</u>	<u>25</u>
			<u>US\$ 215.00</u>	
	<u>TOTAL</u>	<u>US\$ 215.00</u>		
	<u>NUMBER</u>	<u>NUMBER</u>		

C. Local Operating Costs

The Government proposes to pay an amount estimated at US\$ 16,500 as a cash contribution towards the local operating costs of the project.



CHINESE NATIONALITY IN INDONESIAChinese Fertilizer

This report from the Chinese Nationality in Indonesia, the Chinese Government, and the Chinese Fertilizer Company, will be submitted to the Ministry of Agriculture.

Introduction

As far as we can ascertain, the Chinese Government has been instrumental in the development of agriculture in Indonesia.

The following are the main contributions:

Centralization of water control systems.

Implementation of government

agricultural enterprises.

Participation in Executive Agency (Bina Darat) in agricultural development organizations.

I. Background

1. Chinese Government's contribution to Indonesian agriculture. Indonesia is second largest variety of climate in the world. Out of the country's total of 115 million acres, only 35 million are in use, 55 million being fallow. Thus, the second most important crop is maize at 23% in acreage. Indonesia's food production occurs principally in Java, where almost three-fourths of the total population of 110 million live. Expansion of agricultural production in Java can be attained only by higher yield per unit of land area, since virtually all the land suitable for crops is under cultivation. Fertilizer is one of the most important inputs to increase yields.

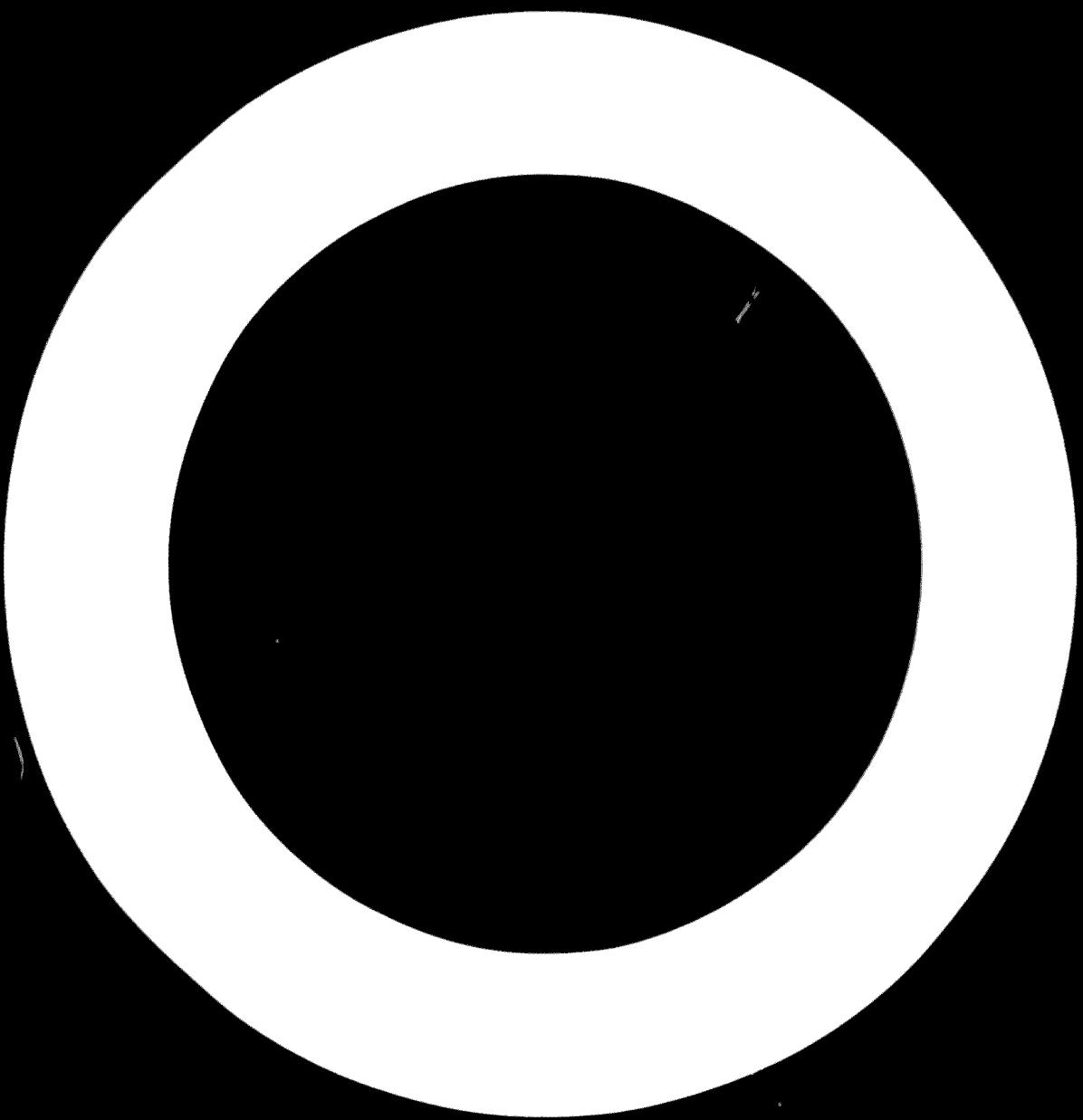
2. A government-owned enterprise at Palembang, Sumatra manufactures 106,000 tons of urea per annum.

3. Another government-owned enterprise at Gresik (Surabaya), manufacturing 70,000 tons ammonia per annum, will come on stream in early 1970 with the following tentative product mix:

Ammonia, 6,000 tons per annum

Urea, 45,000 " " "

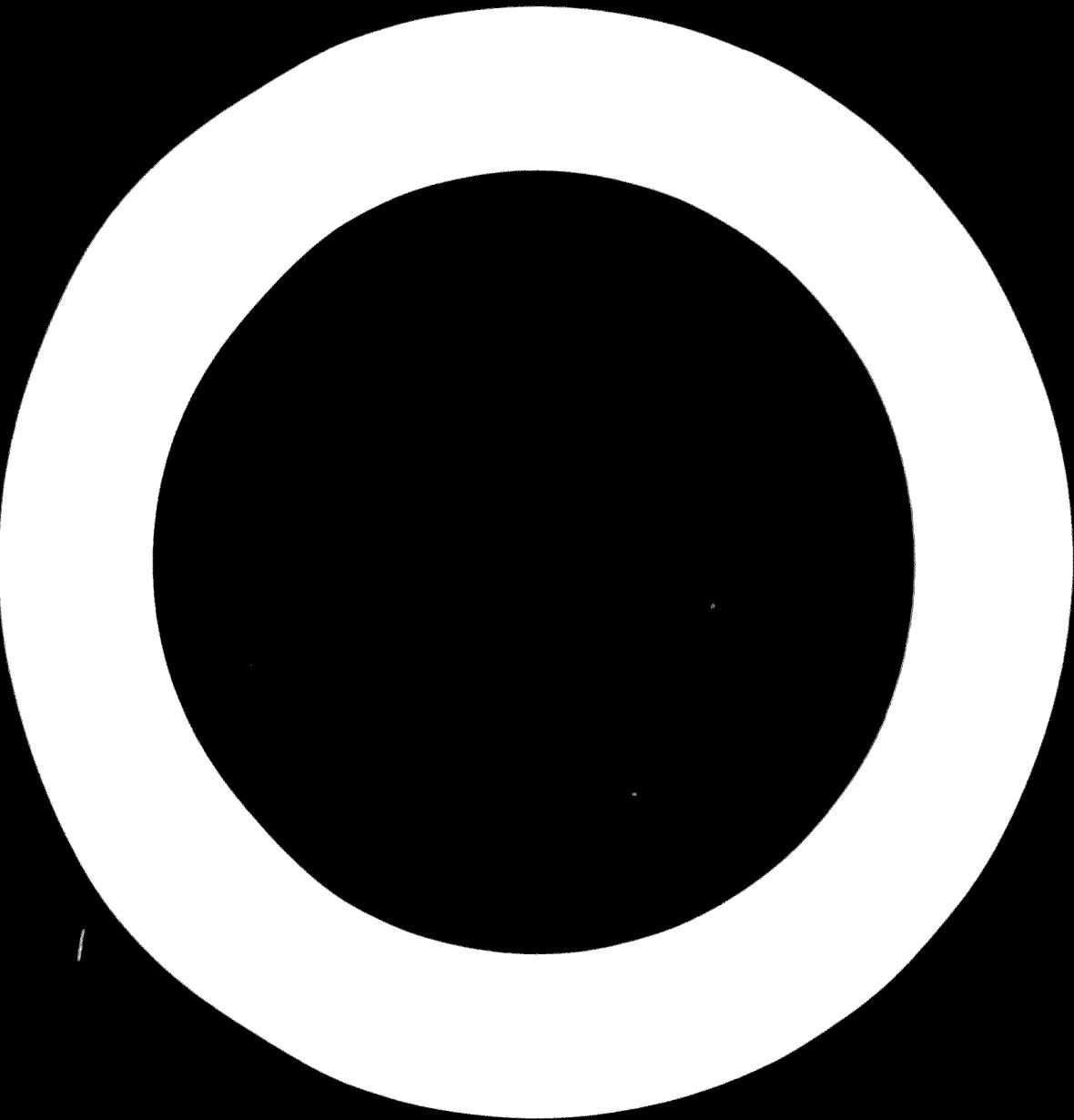
Ammonium Sulphate, 150,000 " " "



6. The ZK technology can be used in the field application of fertilizer. This project will not be limited to production of liquid manure, as every effort will be realized to maintain capacity and to find opportunities to extend services.
7. One method of using the ZK technology will be to make a direct application of nitrogen fertilizer to the soil. It may be with an amount of fertilizer which is equivalent to United States no-till injection to the soil, thus will be a novel method in Indonesia. The possibilities of low cost nitrogen fertilizer are dependent on the direct application of fertilizer to the ground. The government is requesting assistance in the establishment of a pilot project for this purpose.

II. THE PROJECT

8. The purpose of the project, the duration of which is three years, is to determine:
 1. The feasibility of pilot projects applying the ZK nutrient injection process in agricultural fertilizer.
 2. The economic feasibility of the ZK nutrient injection process compared to conventional fertilizer.
 3. The most feasible and economic ways of transplanting the project to the formal fertilizer industry.
9. The UNDP (United Fund) is responsible to provide:
 - the services of an internationally recruited expert as Project Manager for a total period of three one-years,
 - the services of an internationally recruited expert with agricultural engineering experience for a total period of two one-years,
 - process equipment for carrying out only from ammonia into aquaculture,
 - application tanks to inject ammonia into the soil as well as fuel storage tank, and tank transport vehicles,
 - cost of other necessary imported equipment,
 - certain office equipment,
 - expenses for travel of experts outside the country.



d. The Government (r) must provide:

- the services of sufficient qualified water supply technicians;
- secretaries, chauffeurs and interpreters,
- auxiliary unskilled labour for the equipment,
- necessary civil engineering services to assist the creation of field installations,
- for the salaries following the payment of the amount of delivery to the contractor of private skilled and unskilled labour and all other auxiliary construction materials required in the creation of the equipment furnished by the United Nations Development Programmes,
- a team back up for the completion of the remainder of the project as and when provided from time to time.

III. Financial Data**A. Contribution of the United Nations Development Program**

(a) <u>Experts</u>	Total	Total	Per Month		
	Min-Month	US\$	I	II	III
Project Manager	9	US\$ 1,000	12	12	
Agricultural Engineer	24	US\$ 300	—	12	12
	33				

(b) Equipment

Imported office equipment	US\$ 15,000
Jeep	5,000
Tractor	5,000
Trucks (2)	20,000
Application tanks	20,000
Ammunition converter (for gun)	20,000
Tank transport wagon	3,000
Field storage tanks	10,000
Storage parts	5,000
Maintenance equipment	10,000

Sub-Total US\$ 120,000

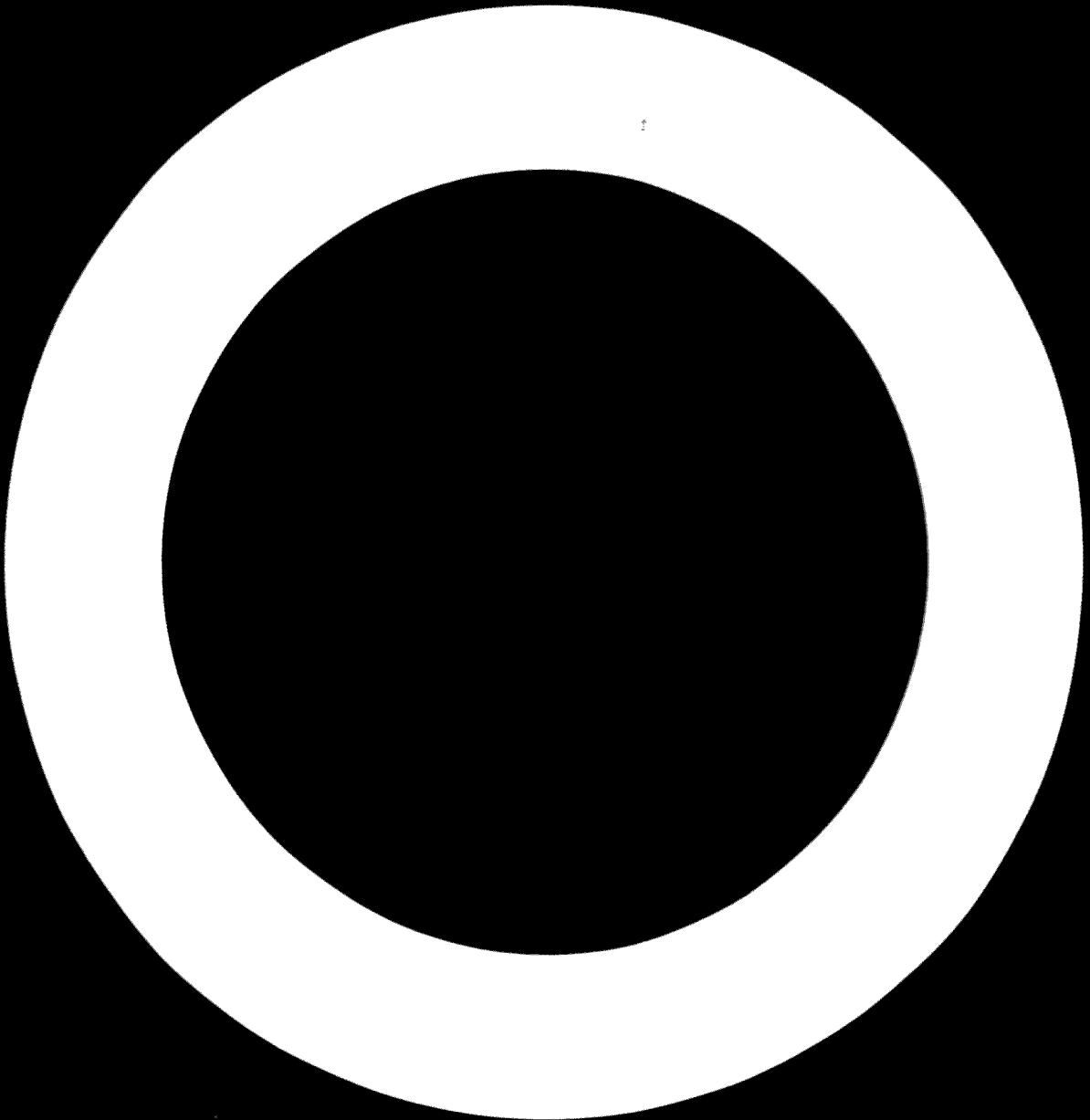
(c) Miscellaneous

Documentation	US\$ 5,000
Travel outside the country	5,000
Contingency	30,000

Sub-Total US\$ 40,000

TOTAL amount requested from
UNDP (Special Fund)

US\$ 320,000



B. Projected Government Counterpart Contribution

Counterpart item	Total US\$ equivalent	Capital US\$	Op. US\$	Op. US\$
Project Management	—	—	1,000	1,000
Agricultural Engineer	—	—	1,000	1,000
Archivist	—	—	1,000	1,000
Chemist	—	—	1,000	1,000
Other services	—	—	1,000	1,000
Rental of fire equipment	—	—	—	—
Office supplies	—	—	1,000	1,000
Civil engineering	—	—	1,000	1,000
Operating expenses	—	—	1,000	1,000
Transportation	—	—	1,000	1,000
Domestic travel	—	—	1,000	1,000
Farm rental	—	—	1,000	1,000

US\$ 172,000

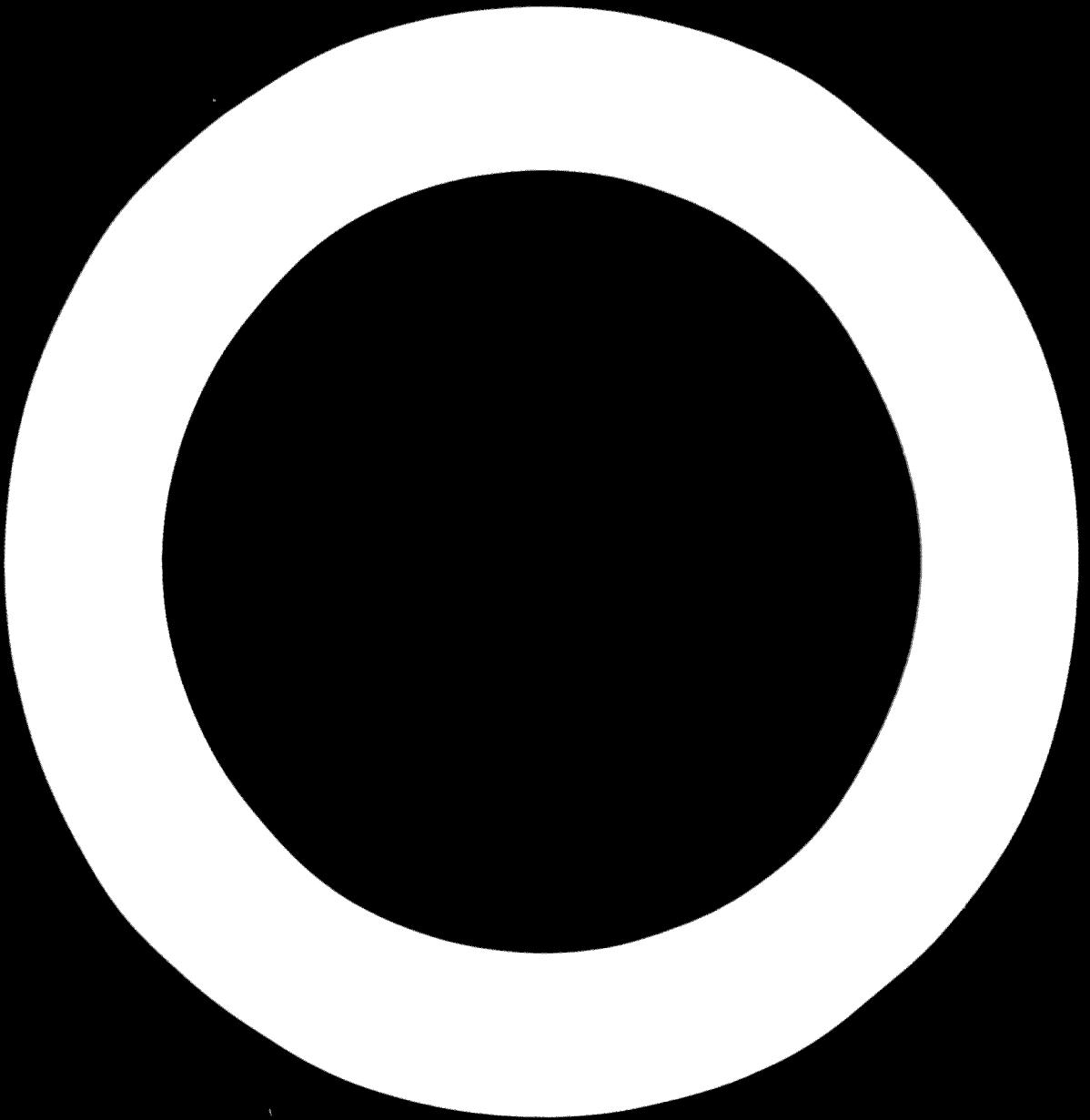
TOTAL

US\$ 212,000

including interest @ 5% p.a.

C. Local operating costs

The Government proposes to pay an amount estimated at US\$ 21,000 as a cash contribution towards the local operating costs of the project.



UNITED NATIONS INDUSTRIAL DEVELOPMENT ORGANIZATIONUnited Nations Development ProgramSpecial Industrial ServicesMarketing Assistance

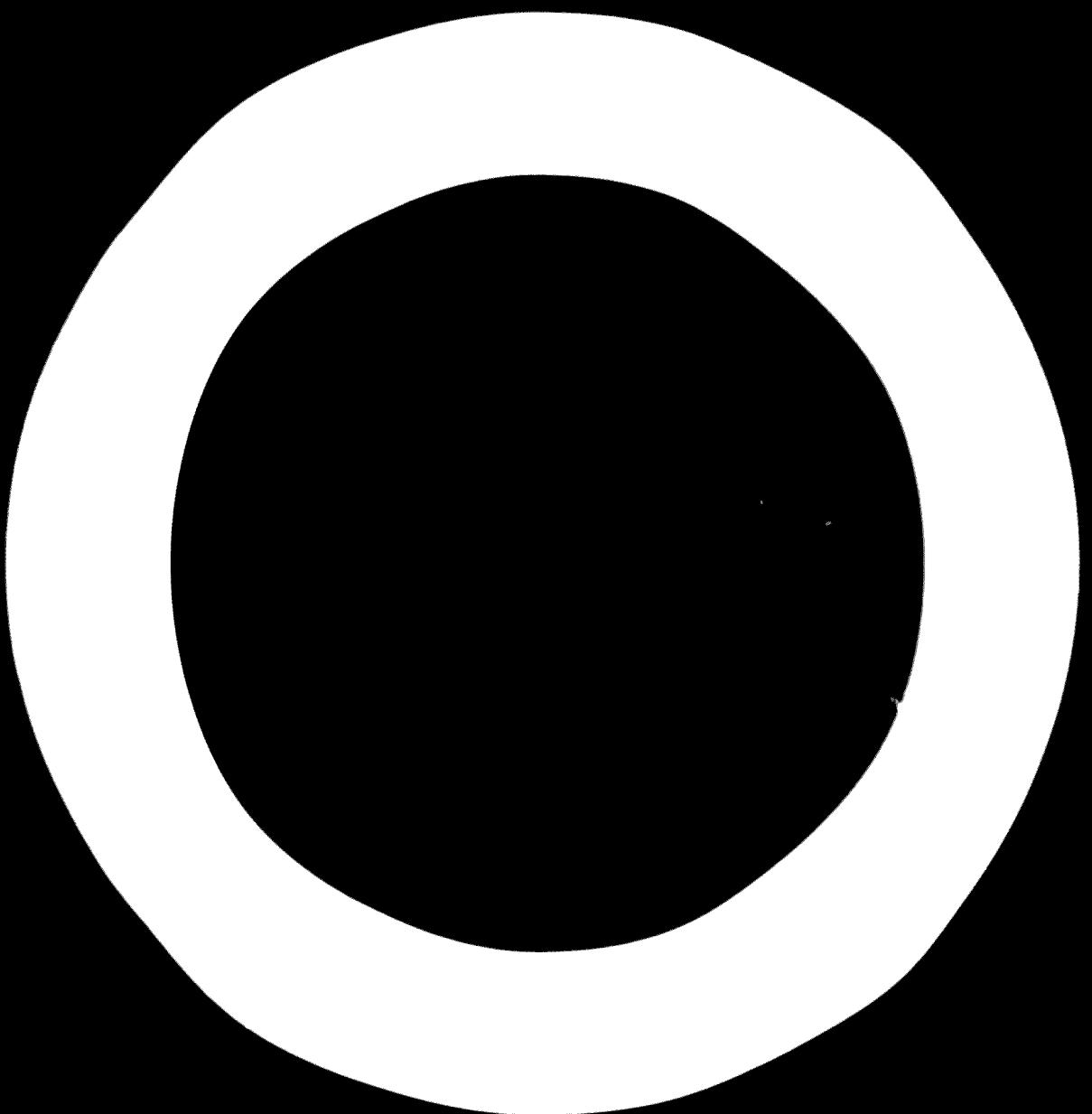
1. Reference No.: SIS/IND -
- Country: Indonesia
- Project Title: Short-term Technical Assistance to the Gresik Phosphate and Potash Fertilizer Project, Indonesia.
- Date Formal Request received:
- Government Department submitting Request: Department of Chemical Industry, through the Resident Representative
- Specific Government Agency concerned with the Project: Department of Chemical Industry
2. Description of the Project:
A marketing expert will be assigned to the Ammonium Nitrate-Ammonium Sulphate plant at Gresik to advise on the proportions of urea and ammonium sulphate that should be manufactured.
3. Summary of major grounds for Request:
This request is based on a UNID fertilizers team's mission that visited Indonesia in Oct.-Nov. 1968. The Government of Indonesia is keenly interested in maintaining the output of the Gresik plant at its maximum.
4. Relationship with other Technical Assistance Projects: Nil
5. Project Components, Duration and Estimated Costs:
- | <u>Field of Activity</u> | <u>Duration</u> | <u>Costs</u> |
|--------------------------|-----------------|--------------|
| A Marketing Expert | 6 months | |
6. Request approved:

For UNDO

Date

FOR UNDP

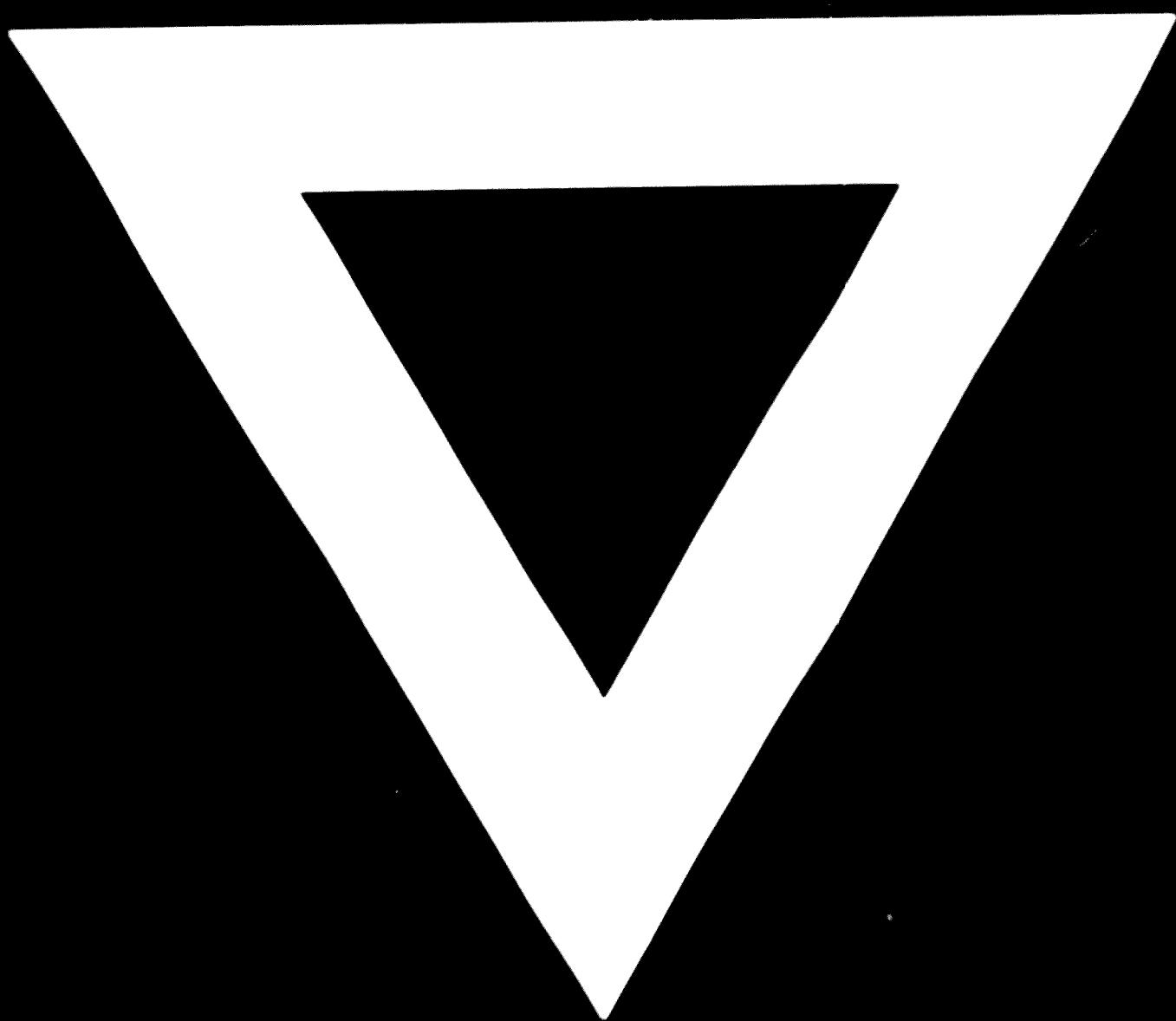
Date



CF Unit 101

<u>Job Title:</u>	Miner, Ammonium Fertilizer Plant
<u>Duration:</u>	Six months.
<u>Date required:</u>	As soon as possible.
<u>Duties:</u>	The worker will be called in the agricultural industry to help in the mining plant, mainly to mine, transport, and mix with the firm extension service, the fertilizer proportions for mining sulphur. He will also take charge of the services and call the miners for the future. He will be determined to control the use of fertilizers to the soil system in the various warehouses in the area. He will be assigned to experiment in mixing up a very fine mix that will be distributed to the products of the plant.
<u>Qualifications:</u>	University graduation with a lot of experience and knowledge with the practice of fertilizing.
<u>Language:</u>	English.
<u>Background Information:</u>	The mining plant has a capacity of 1000 tons of fertilizers per day. It is designed to produce 70,000 tons per annum from which 50,000 tons are exported and the balance will be converted into ammonium sulphate. Tentative plans are for production of 50,000 tons per year of urea and 150,000 tons per year of ammonium sulphate. However, these figures obviously cover a fairly wide range, depending on market conditions. However, once the warehouses are full for a certain product mix, it is time to determine if the farmers preference is for the same ratio of products as found in the warehouses.
	The plant at Gresik can only be an economical producer if the ammonia plant is operated at maximum capacity throughout the year.





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