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Distr. LIMITED ID/WG.11/6 14 May 1968 ENGLISH ONLY

United Nations Industrial Development Organization

First Meeting of Expert Consulting Group on the Aluminium Industry Vienna, 10 - 17 November 1967

TECHNICAL ASPECTS OF ALUNINA PRODUCTION

by

C. Popov

1/ The views and opinions expressed in this paper are those of the author and do not necessarily reflect the views of the Secretariat of UNIDO.

id.68-040

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This paper was first presented at a mosting of experts consulting on the aluminium industry in Vienna at UNIDO beadquarters, 10-17 November 1967.

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Introduction

1. Since 1956 aluminium production has been second only to that of cast iron and steel, in terms of quantity. In 1963, world aluminium output was 6,739,000 tons, 39 per cent higher than in 1959. By 1975 the United States alone is planning to attain an output of 6.6 million tons of aluminium per year.

2. This rapid rise in aluminium production, which has been surpassed in rate of growth only by the production of lithium, gives reasonable grounds for doubting whether future supplies of row materials will be sufficient. A number of countries that lack beauite, the basic row material, are already engaged in research and are making preliminary plans for the production of aluminium from materials other than bauxite.

3. The known reserves of bauxite in the world (outside the Union of Soviet Socialist Republics) amount to some five billion.^{1/} tons, and this figure is not likely to grow much larger, because the deposits on the surface of the earth's orust have already been thoroughly explored. It is unlikely that deep-lying deposits will be found, considering the conditions under which bauxite is formed. Considering also the way bauxite extraction and processing are developing, the world resources of this commodity are not wast and will be used up in the next few decades. At that time the aluminium industry will have to be adapted to other types of raw materials.

4. Before considering the various types of raw materials other than bauxite, it should be noted that bauxites themselves are not entirely homogeneous materials. They can be divided into two basic groups: residual and sedimentary. Residual bauxites are formed by the process of weathering into laterite of various rocks, including basalt, and they are found at the place of their original formation, covering their matrices. Sedimentary or secondary bauxite deposits are formed by leaching or transfer from their matrices, where, however, they are also found in lenses. A total of 42 chemical elements and some 100 minerals have been found in bauxites. Bauxite composition varies not only from deposit to deposit but also within a single deposit.

1/ Throughout this document the term "billion" signifies one thousand, million.

5. It may be noted, at this point, that matrices of volcanic origin can be considered generally as nature's almost inexhaustible storerooms of riches. They are, as it were, the lower storey of the earth's crust, the upper storey of which is the granite layer, as it is called in geophysical terminology. The total aluminium content of the earth's crust is 7.5 per cent by weight considerably more than the iron content. Thus, while natural forces break down the rocks of the earth's surface with the passage of the various geological epochs, liberating the raw materials needed by man together with many impurities, contemporary science and technology will very soon devise incomparably faster methods, undoubtedly giving a higher degree of purity, for the separation of alumina from basalt and similar rocks.

6. The chemical composition of basalt is, on the average, as follows: 40 to 50 per cent silicon, approximately 15 per cent aluminium, 7 per cent calcium, 6 per cent iron, 5 per cent magnesium and 2 per cent titanium. In addition, basalt contains a certain amount of alkali metals and other chemical elements, including some that are rare and procious. According to research carried out on basalts in India, their aluminium content decreases in proportion to the depth of the deposit.

7. Compared with the richer ores, the number of component elements in basalt may seem small, but basalts contain a number of useful ingredients, the comprehensive extraction of which - if cheap energy and technology are available is an economically sound project. As an illustration, it may be noted that the mechanical processing of granite, basalt and other hard rocks into building, monumental and industrial components by sawing, which was expensive and slow but was until very recently the only method known, can now be replaced by the cheap blowtorch method. Moreover, the harder the rock the more effective the new method is.

8. In his day, the world-renowned D.I. Mendeleev put forward the view that solar energy is most effectively used on the earth for the accumulation of marsh plants and peat, which can renew fuel resources in a very short time without human intervention, and if this process were developed, it could be speeded up even further. It must therefore be assumed that if there is a specific meed for this, modern science will be able to accomplish the artificial "dismantling" ID/NO.11/6 Page 6

of rooks to separate out purer alumina than that liberated by nature over tremendously long periods.

9. Basalt is already being used as a raw material for melting and casting into a number of useful articles which used to be cast from other materials and metals. Evaporation smelting by means of plasma burners makes it easy to break down even basalt into its components, and by introducing the necessary chemical elements into the process it is also possible to obtain new materials in this way.

10. Because of its chemical composition, basalt is the most suitable raw material for the production of pyrocerams, glass-like crystalline materials which are a recent investion potentially as important as the discovery of how to make steel was in its time. Pyrocerams are very similar to aluminium in density (2.6 to 2.65 and 2.7), but they surpass it in strength, resistance to atmospheric corrosion and in value.

11. It has recently been discovered that aluminium is present in 268 minerals, but the only ones that may be considered as ones for the production of alumina or other aluminium compounds are nepheline and the rocks that contain it, alunites, clays and knolins, high aluminium-content silicates and, more rarely, aluminium pulphates such as pickeringite and alunogen. The chemical composition of these minerals is given in table 1.

I. METHODS FOR THE PRODUCTION OF ALIMINA FROM RAW MATERIALS OTHER THAN BAUXITE

12. There are several hundred known methods of producing alumina. This large number of methods is due to the variety of aluminous materials and to the amphoteric properties of aluminium oxide. The best known methods for producing alumina are usually divided into three groups: (a) electro-thermal, (b) acid and (o) alkaline. Naturally this classification is applicable chiefly to the processing of bauxites, which are at present the prodominant raw materials in the alumine industry.

13. In the electro-thermal process, alumina is obtained by molting the ore with a reducing agent in electric furnaces. This method is not widely used, since it can only be economic when very cheap electric power is available. Purthermore, the alumina produced is not always of satisfactory quality. In the acid process, the raw material is treated with a solution of one of the mineral acids, giving a corresponding soluble aluminium salt, while leaving the bulk of the raw material undissolved. (unfortunately, however, salts of iron and titanium frequently dissolve as well and thus impede the process). After extraction, the aluminium salt is broken down and solid aluminium hydroxide is obtained, which, after calcining, yields alumina. This acid process calls for a costly acid-proof plant, and consequently it is soldom adopted nowadays.

ince of sineral	Chemical formula	
Corundum	Al ₂ 03	<u>Alveine content (%</u>) 100
Diaspore, bookmite	2-3 Al ₂ 0 ₃ .H ₂ 0	85
Gibbsite (hydrargillite)	Al ₂ 0 ₃ .3H ₂ 0	65.4
Cyanite, andàlusite, sillimanite	Al ₂ 0 ₃ .5i0 ₂	63.2
Kaolinite	A1203.28102.2H20	39.5
Alunite	(K, Na,)280, 112 (80,)3.441 (OH)3	37-39
Nepheline	(m,K)20.A1203.25102	36+32
Leucite	K20. A1 ,02.2810	23.5
Ancethite	Ca0.A1_03.2810	36.7
Cryclite	Haz-Al.FA	23.5
Orthoolage	K20.A1203.68102	18.4
Albite	Ma 0. 41 0 . 6810	19.3
Beryl	3Be0. A1 203.68102	14.0
Chrysobery1	Be0. A1 203	80, 0
Spodumene	Li20.A1203.48102	28.0
Pollucite	CE20.A1203.48102	14.0
Pickeringite	Mg.A12(804)4.22H20	11.9
Alunogen	A12(804)3.16H20	16.4

Table 1
Principal sinorals containing aluminium

14. Wide use is made of the more economic alkaline process, which may be carried out in a cheep iron plant, and in which recovery of the alkali is a simple matter. In this process, the raw material is treated to produce sodium aluminate, which can be broken down into aluminium hydroxide. The final stage of the process is to calcine to aluminium hydroxide, thus producing alumina.

15. In this process, the oxides of iron, titanium and calcium do not, in practice, dissolve out of the raw material. Only the presence of silica complicates the processing of the raw material, which must therefore be selected with low silica content if it is to be treated by this method.

II. TECHNICAL CUALITY OF ALURINA

16. The quality of alumina, most of which is intended for processing into metallio aluminium, should be approximately as follows:

Impurities							
		Marinu		ble percer	tiane cont	ent of imp	urition
	Quality:	<u>°00</u>	° o	0 <u>1</u>	0 ₂	03	°4
8102		0.06	0.08	0.15	0.25	0.40	2.0
Fe203		0.03	0.03	0.04	0.05	0.6	1.0
Ma ₂ 0 Other		0.5	0.6	0.6	0.6	0.7	0.8
		1.0	1.0	1.0	1.2	1.5	2.0

Table 2 Quality of alumina (according to USER plassification)

17. The presence of electro-positive and electro-negative impurities is not permissible. Similarly, the amounts of moisture and water of hydration should not exceed the permissible maxima. The dimensions of the alumina particles must be within the range of 50 to 80 microns. Only the first three qualities of alumina are used for the production of aluminium.

The technology of membeline processing

18. Mepheline is one of the apatite-nepheline and urtite rocks used at present in the Union of Soviet Socialist Republics for alumina production. These rocks consist of two minerals: apatite $Ca_3(PO_4)_2$. CaF_2 and nepheline $(Na.K)_2O.Al_2O_3$. 2810₂, plus a certain quantity of secondary minerals. The raw material is concentrated by flotation, in the course of which the phosphatic material, apatite and its tailings, is separated from the nepheline concentrate. The most detrimental impurities are iron compounds.

19. It has been found, in practice, that for the full conversion of the alumina contained in nepheline concentrates into sodium and potassium aluminates, there should be two moles of CaO to one mole of silicon oxide in the charge. The optimum temperature for sintering is approximately 1,300°C, and the decomposition of nepheline by the limestone takes place according to the r action:

 $(\mathbf{M}_{\mathbf{n}},\mathbf{K})_{2}^{0}$. \mathbf{Al}_{2}^{0} . $\mathbf{2510}_{2}$ + $4\mathbf{CaCO}_{3}$ = $(\mathbf{M}_{\mathbf{n}},\mathbf{K})_{2}^{0}$. \mathbf{Al}_{2}^{0} . + $2(2\mathbf{Ca}.\mathbf{510}_{2})$ + $4\mathbf{CaCO}_{2}$

Microscopic analysis of sinter cake obtained under optimum conditions shows that it consists of diculcium silicate and of a solid approximately 30 per cent colution of this silicate in a mixture of alkaline aluminates.

Table] Method of sintering (diagram of process)

Chemical composition of some napheling rooks and their concentrates

		Content in per cont							
Rook/ooncentrate	81 0,2	A1203	Po203	R ₂ O	CaO	TIO	NgO	P205	Other mbstances
Nepheline-sympite	44-35	22.45	10.00	3.47	8.33	0.77	1.12	0.5	4.8
Tailings left after concentra- tion of nepheline apatite	42 .5 0 -	22.08	7.02	16.7	5.0	2.1	1.4	1.2	1.1
Concentrate of tailings	43.1	29.2	3.4	18.5	1.7	0.15	0.6	0.2	1.3

20. Where there is deviation from the optimum sintering temperature - for example, when sintering is carried out at a lower temperature - the nepheline cake produced contains, in addition to the convermentioned products, a cortain amount of undercomposed nepheline and intermediate tertiary compounds of the type:

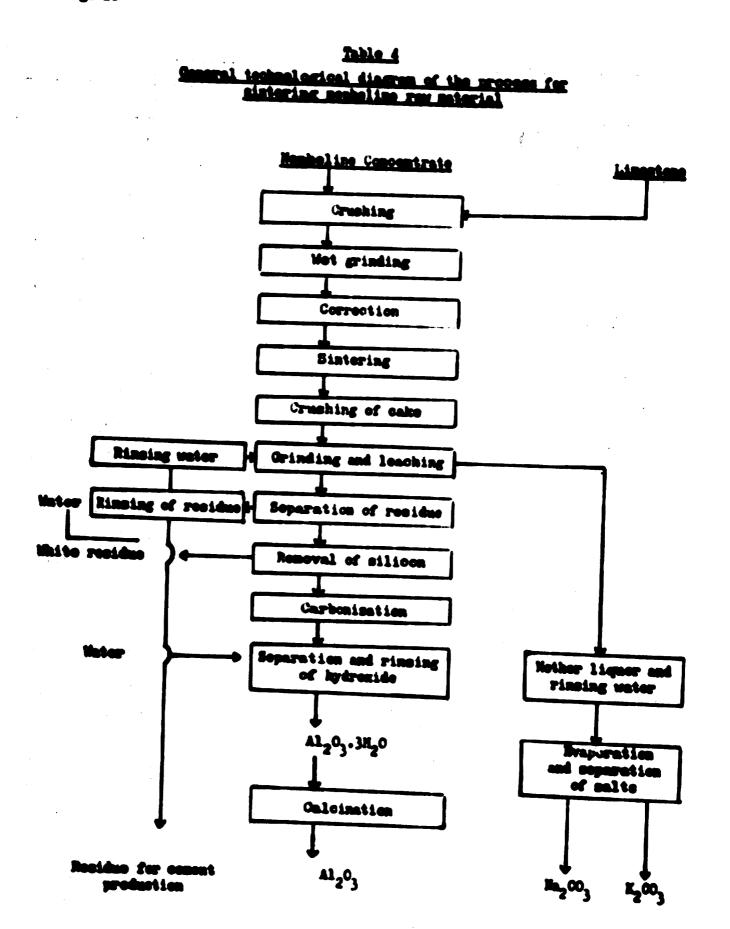
n(Na, K)20.mal203.pero2 and nCaOmal203.p3102

Since these compounds are inscitute in water, they represent a waste of alkali and alumina.

21. Deviation from the optimum composition of the napheline-line (trans composition is equally detrimental. Thus, for example, the inclusion of more limestone in a batch than is called for by the formula for the *Texantica* of dicalcium silicate may give cake from which only a reduced quantity of alumina can be extracted upon leaching. This has to do with the formation during sintering of highly alkaline calcium silicates which, during leaching, generate free line which partially reacts with the aluminate solution and returns a portion of the alumina to the

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insoluble state. On the other hand, an insufficient amount of limestone in the charge leads to incomplete breakdown of the nepheline and to a lower yield of alumina and Ma₂O.

Practical effect of the sintering temperature
Practical effect of the sintering temperature of the monholine charge on the properties of the
cake and the yield of the leaching process

Tebla 5

temperature, °C	Specific gravity	Porosity	Extraction from cake	
			A1203	. Jia_0
1250	2.94	46	79.5	79•3
1270	2.96	31	82.9	80.4
1290	2.99	20	84.6	85.2
1300	3.05	13	86.0	87.1
1310	3.10	7	87.5	86.9
1330-1360	3110	6	89.0	. 90.0

22. The consumption of heat, for a charge humidity level of 26 to 28 per cent and using pulverised coal heating, is 1,300 kilocakries per kilogram of cake. A diagram showing the technological process for treating nepheline is given above.

The new hydrochemical method (14)3

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23. In the new hydrochemical method for breaking down nephelines by means of a solution of caustic sods in the presence of calcium oxide, with heating up to 200-280° in an autoclave, the reaction is as follows:

 $(M_{0},K)_{2}0.Al_{2}0_{3}.28i0_{2} + 2Ca0 + 2NaOHag - Na_{2}0.2Ca0.28i0_{2}.H_{2}0 + 2(Na,K)Al_{2}B.$ 24. This reaction results in the formation of alkaline aluminate, which goes into solution. In establishing the technique for this method, nepheline composed of $Al_{2}0_{3} - 33.3\%$; $Ma_{2}0 - 15.43\%$; $K_{2}0 - 6.47\%$; $8i0_{2} - 42.22\%$; $Fe_{2}0_{3} - 1.2\%$ and Ca0 - 0.9% was treated in 3, 6, 10 and 13 N solutions of NaOH (i.e. containing 94.8; 194.4; 326.9 and 424.5 grams of $Na_{2}0$ per litre, respectively) in the presence of Ca0 for exposure times of 0.2, 5, 10 and 30 minutes at temperatures of 200° and 2800C in an autoolave, the heating rate being 5° per minute. The amount of Ca0 was determined on the basis of the Si0₂ content, taking a molecular ratic of Ca0:Si0₂ = 1.0. The estimated caustic module - molecular ratio, $R_{2}0:Al_{2}0_{3}$, was 12.0.

J Humbers in parontheses refer to the bibliography.

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27.

28.

The breakdown of the nepheline and the formation of new phases ends during the period of heating up to 280° or during the first five to ten minutes of leaching at that temperature. The degree of alumina extraction is 88 to 90 per cent. In the opinion of research workers, this method is also suitable for breaking down other alumo-silicates.

Techniques for the processing of aluminium silicates

The possibility of using clays and kaolins as raw materials for alumina production is being given much study, but the technical conditions under which these materials might be suitable for alumina production have not yet been worked out in detail. The best clays and kaolins for this purpose appear to be those with a composition approaching the the theoretical composition of kaclinite. The best quality materials would therefore be secondary kaclins; the suitability of other deposits for the extraction of alumina would depond on their concentration possibilities.

Kaolins can be converted into alumina by two methods: sintoring and acid treatment. For processing by the acid method, kaolins must have the smallest possible ferric oxide content, at any rate not exceeding 2 to 3 per cent; those with an alumina content of not less than 30 to 32 per cent are considered suitable for processing by the sintering method.

Silicates with a high aluminium oxide content, such as andalusite, cyanite, sillimanite and diaspore-containing clays, may be convorted into alumina by the electro-thormal method. However, they are subject to special requirements regarding their content of such impurities as Fe₂03, Ti02, CaO, MgO, Na2O and K2O, and consequently only their concentrates are processed. In the Union of Soviet Socialist Republics a provisional tochnique has been devised for treating this

Al ₂ 03 not less than	For concentratos of hish-alumine silicates	For concentrates of argillaceous minerals
SiO ₂ not more than	59	38
Pe203 not more than	37.5	47
TiO2 not more than	1.5	0.5
Mag0 + Kg0 not more than	1.2	0.3
CaO + NgO not more than		0,5
Other substances	0.6	0.6
	-	14

29. To obtain alumina from olays, kaolins, coal-ash and other silicates by the sintering method, these materials are sintered with limestone in such a way as to produce spontaneously disintegrating aluno-calcium sinter cake. The amount of limestone in the charge is such as to bring about the formation of the compounds 2CaO.SiO_2 and $5\text{CaO.3Al}_2\text{O}_3$. The charge components are first ground in a wet ball mill to a particle size of approximately 0.088 mm. The wet charge then goes into a rotary furnace where it is sintered at a temperature of 1,375 to 1,425°C. The necessary completeness of interaction between the components of the charge and the requisite property of spontaneous disintegration of the cake are achieved by bringing the charge to the point of enset of fusion and then cooling it for ten to fifteen minutes down to a temperature of 1,300 to 1,325°. Undisintegrated cake is returned to the oven for further sintering, while the disintegrated portion is treated in a solution of soda to obtain a solution of sodium aluminate, which is converted to pure aluminium exide by normal sintering methods.

30. The spontaneous disintogration of the cake during cooling is caused by the reorystallisation of the beta-modification of the dicalcium silicate formed in the sintoring process into a high-volume gamma-modification. In view of the low concentration of Al_2O_3 left in solution after leaching (12-15 grams por litre), the hydrochemical processing of the cake can be so arranged that the operations of silicon removal and evaporation of the residue are entirely eliminated from the technological process.

31. Ferric oxide does not affect the process if the initial material does not contain more than 3 to 5 per cent of it. All that is necessary is to increase the amount of limestone in the charge to a degree that will ensure the formation of disalcium ferrite. Where there is a higher content of ferric oxide, its detrimental effect on the process can be avoided by adding a reducing agent for it to the charge and sintering in such a way as to ensure that the reducing process takes place in the whole charge.

32. A most important innovation in the development of alumina production from various alumo-silicates by the sintering method is the use of sodium sulphate instead of caustic soda in the process. It will be seen from the bibliography (15) that soda and sulphur can be produced jointly with alumina. Study has also been given to the processing of alumites (16), alumina-rich raw materials such as diaspore-boshmite bauxites (17), and clays and kaolins (18). The same researchers have obtained alumina, soda and cement from more acid nepheline symites (23).

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33. For a number of countries which do not have a domostic caustic soda production industry, the use of sodium sulphate - a raw material that is readily available in nature - facilitates and simplifies the problem of producing alumina from almost any material. Even the proliminary conversion of sodium sulphate into sulphite is not really a complicating factor.

34. It is quite possible that even more important data on the processing of sulphates is available at the Government Mines Branch Research Centre in Canada. This centre has a staff of 635 persons, 256 of whom are university graduates, mainly occupied in the extraction of metals from ores, the selective breakdown of mixtures of metal sulphates, and the design of apparatus for these processes (20)

35. A new development (19) in the production of alumina and soda by sintering the raw materials with the use of sodium sulphute is the 80 to 83 per cent extraction of aluminium oxide and the 95 per cent extraction of alkali from sandstones of the following percentage composition: $SiO_2 - 47.48$; $Al_2O_3 - 20.67$; $Fe_2O_3 - 8.36$; $TiO_2 - 0.8$; CaO - 0.72; NgO - 0.95; $(Na_2O + K_2O) - 6.6$; $H_2O - 8.2$ and other substances - 4.48.

The sold method

36. The acid method of separating alumina is being investigated but it has not as yet found favour in industry for the following reasons:

- (a) The corrosive effect of the acids on the materials usually employed for factory equipment. It should be noted, however, that new acidresistant materials such as pyrocerams and plastics should completely eliminate these earlier difficulties;
- (b) The complications presented by any acid method because of the need to purify the resulting solutions from iron salts;
- (c) The fact that the acid method gives a type of alumina that is less suitable for the production of metallic aluminium by modern (electrolytic) techniques.

37. Nevertheless, acid alumina-production mothods may be useful industrially, since they afford greater possibilities, in principle, than alkaline methods for solving the problem of processing high silicon-content materials of the clay and kaolin type, which are available everywhere in large quantities. This is due to the fact that the inertness of silica to acids enables this component of the natural raw material to be taken out of the process in the initial stages without wasting acid reagents. 38. Consequently, the successful development of acid methods should make it possible to widen the range of ores used in the alumina industry by bringing about more extensive industrial use of clays, kaolins and other silicates. This explains the attention that is being paid to acid methods by scientists and technologists, who are striving to oversome the drawbacks mentioned above.

Principal operations in the acid method

39. The principal operations in the acid method are: preliminary treatment of the raw material, breakdown of the raw material by acid, giving a solution of aluminium salts; purifying the aluminium salts of iron compounds; extraction of pure aluminium oxide; and regeneration of the original acid. The preliminary treatment of the raw material comprises crushing, grinding and firing. The purpose of firing is to dehydrate the material to be processed and thus break the chemical bond between its individual elements, thereby increasing the reactivity of the material to any chemical agent, including acids. When kaolin raw materials are broken down by firing, the reaction is as follows:

$$Al_2O_3 \cdot 28iO_2 \cdot 2H_2O = Al_2O_3 + 28iO_2 + 2H_2O$$

The temperature of disintegration of kaolin is between 560 and 580°C. Overheating of kaolin lowers the reactivity of the separated ingredients, and at a temperature of 950 to 960°C they form a stable compound, sillimanite or mullite $(2\Lambda l_2 O_1.8iO_2)$ on which acids have practically no effect whatscever.

40. The raw material is disintegrated by means of sulphuric, nitric, hydrochloric or sulphurous acid. This produces a solution of aluminium and iron salts. The silicon in the raw material remains in the sediment. Sulphuric acid has long been used to break down raw materials containing alumina for the production of aluminium sulphate.

41. Nitric, hydrochloric and sulphurous acids have the advantage of being comparatively easy to regenerate. Sulphurous acid is also a by-product of a number of metallurgical plants, but to achieve the necessary high concentration of SO₂, the breakdown process must be carried out under a pressure of several atmospheres in autoclaves.

42. Another method is to break down the raw material in a furnace by heating it together with ammonium sulphate:

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 $Al_2O_3.25iO_2.2H_2O + 4(NH_4)_2.SO_4 = 25iO_2 + (NH_4)_2.Al_2(SO_4)_3 + 6NH_3 + 5H_2O$ The ammonia extracted is used in one of the later stages of the process to obtain ammonium sulphate.

43. The elimination of iron compounds can be achieved by a number of methods. One such method, based in the differential solubility of certain aluminium and corresponding iron compounds, enables the aluminium salts to be precipitated out, leaving the iron salts in solution.

44. Thus, for example, in the sulphuric acid method it is possible to make use of the property processed by aluminium sulphate of forming a double salt with ammonium sulphate. This salt, which is known as ammonium alum, $Al_2(80_4)_3$ $(NH_4)_2SO_4.24H_2O_7$, is easily soluble in hot water but almost insoluble in cold. In view of the fact that ferric sulphate also forms a similar salt with ammonium sulphate, while the lower (ferrous) salt does not interact with ammonium sulphate but remains in solution when it is present, the production of ammonium alum should be preceded by the reduction of the iron compound in solution to the lower (ferrous) form. Either a solution of ammonium bisulphite (NH_4HSO_3) or else sulphur dioxide can be used as a roducing agent, the reaction being as follows:

$$Fo_2(SO_4)_3 + SO_2 + 2H_2O = 2FoSO_4 + 2H_2SO_4$$

45. Potash alum, $K_2SO_4 \cdot Al_2(SO_4)_3 \cdot 24H_2O_4$, reacts in the same way as ammonium alum to hot and cold water.

46. As another example it may be noted that, to purify a solution of aluminium ohloride, the solution may be saturated with gaseous HCe, which precipitates aluminium chloride hexahydrate, AlCl₃.6H₂O, while the iron compounds remain almost entirely in solution. Another method worthy of attention is based on the use of certain surface-active substances, such as manganese dioxide (MnO₂), which are capable of absorbing (adsorbing) the iron compounds on their surface and thus taking them out of solution. This last method, which is connected with the new and rapidly developing branch of science known as physico-chemical mechanics, should be especially borne in mind both for the above-mentioned use and for the investigation of other likely possibilities in the technology of enriching the raw and partially processed materials for aluminium production. Fure aluminium oxide is extracted by calcining such of the compounds containing it as are decomposible by heat. As well as aluminium oxide, calcination produces the corresponding acidio oxides and, in the case of aluminium chloride (which is calcined in the presence of steam), gaseous hydrogen chloride also.

47. When potash alum is fired with aluminium oxide, potassium sulphate is also obtained. The fired product is therefore washed, the resulting solution evaporated, and the potassium sulphate extracted from it. Part of this potassium sulphate can be disposed of commercially (as, for example, when processing alunite raw material), while part is used as a feedback product in the earlier stages of the process.

48. Although most aluminium salts break down completely at comparatively low temperatures (200-400°C) and only aluminium sulphate requires higher temperatures (770° and over), in practice, in the case of purely acid methods, the calcination of aluminium salts must end with the normal temperature for the calcination of alumina, which is in the region of 1200°C. This is necessary in order to fully remove the acid residue from the aluminium salts and to ensure that the aluminium oxide is not hygroscopic. The firing operations are carried out in rotary furnaces, in multi-hearth furnaces with mechanical super heating from hearth to hearth, or in a fluidized bed furnace.

49. The breakdown of the ammonium alum can also be done chemically by processing with ammonia according to the formula:

 $A1_2(80_4)_3 \cdot (NH_4)_2 SO_4 \cdot 24H_2 O + 6NH_4 OH = 2A1(OH)_3 + 4(NH_4)_2 SO_4 + 24H_2 O$

50. The residue of the aluminium hydroxide is separated from the solution and washed and fired as usual. The ammonium sulphate, on the other hand, is a feedback product and is used, as was mentioned above, in the earlier stages of the process, although in some methods the ammonium sulphate is also a commercial product, sold for use as fertilizer.

51. The firing of aluminium salts which crystallize with the liberation of a large amount of water, such as potash alum, presents clear technical difficulties, owing to the ease with which such salts, even at low temperatures, mult in their own water of crystallisation. In such cases, in addition to firing in a fluidised bed furnace, recourse may be had to a special procedure based on the observation that when the alum solution is heated in autoclaves to a temperature of about 200°C, the alum breaks down according to the reaction:

 $3K_2SO_4 \cdot Al_2(SO_4)_3 \cdot 24H_2O = K_2SO_4 \cdot 3Al_2O_3 \cdot 4SO_3 \cdot 9H_2O + 2K_2SO_4 + 5H_2SO_4 + 58H_2O$ The so-called basic alum which is thus precipitated is finally dehydrated by the usual method, and the above-mentioned complications do not arise.

52. The sulphuric acid obtained by the above reaction is a feedback product and is used for breaking down new batches of minerals. The acid is regenerated, in the case of nitric, sulphurous and hydrochloric acids, by simple absorption by water of the volatile products of calcination, or in the case of sulphuric acid by more complex special chemical processing. Sometimes, as may be seen from the above formula for the breakdown of potash alum in an autoclave, one product of the decomposition of aluminium salts is the original acid, which for further use only requires boiling down and regenerating.

53. The Canadian aluminium industry is already producing alumina from clays and shales, and five technological plans for this process have been published (21). The raw material is crushed and ground, leached with sulphuric acid, and the silica, titanium and iron compounds, and other impurities are filtered off, after which the aluminium sulphate solution is cooled, causing the aluminium sulphate to crystallize out. The aluminium sulphate is then centrifuged out of the mother liquor, which is returned to the raw material acid leaching stage. The journal article reformed to also reviews the economics of the process.

54. An account of what appears to be the same process is published in another source (22), where reference is made to a method developed in 1961 for the chemical processing of clay and shale into high-quality alumina by the Olin Mathieson method. In the method described, the raw material is leached with sulphuric acid, the iron and the aluminium sulphate are separated by crystallisation and the aluminium sulphate thus obtained is broken down thermally.

55. In the method developed by the Anaconda firm, the raw material is treated with hydrochloric acid, the resultant aluminium chloride and iron are concentrated by evaporation, the AlCl₃ is broken down thermally, and the iron is removed by heating with NaOH or by sintering with sodium carbonate (soda).

56. It is possible that tomorrow's alumina and aluminium technology will also make use of the very interesting French studies of acid soils (24), in which appreciable quantities of free aluminium, not forming part of the orystal lattice of the argillaceous minerals of the soil, were observed. It is conjectured that the aluminium is present in the form of an exchange-ion. Similar studies of soils in Poland (25), in which 2,800 samples taken from an area of 4,500 hectares in the Oder valley were analyzed, showed free aluminium in 667 samples in a quantity of more than 5 mg per 100 g of soil, the conclusion being that at least 20 per cent of the soil contained exchange-ion aluminium.

57. In Poland, too, aluminium sulphate is produced from clay with a low aluminium oxide content by treating the clay with sulphuric acid and calcining in rotary kilns. Relatively pure aluminium sulphate is leached out of the cake thus obtained (26). Research into the production of aluminium sulphate from clays is also being carried out in Egypt (27).

The production of alumina from alunite

58. Recent research into the technology of alunite processing has led to the introduction of reduction heating as a separate step from the firing operation and has confirmed the expediency of treating calcined alunite by the combined ammonium-alkaline method.

59. In this new process, the alunite is ground, after which it is fired and reduction heating takes place in fluidized bed furnaces. The reaction that takes place is as follows:

 $(\text{Na K})_2 \text{SO}_4 \cdot \text{Al}_2(\text{SO}_4)_3 \cdot 4\text{Al}(\text{OH})_3 = (\text{Na K})_2 \text{SO}_4 \cdot \text{Al}_2(\text{SO})_4 + 2\text{Al}_2\text{O}_3 + 6\text{H}_2\text{O};$ (Na K)_2 \text{SO}_4 \cdot \text{Al}_2(\text{SO}_4)_3 + 2\text{Al}_2\text{O}_3 + 3\text{CO} = (\text{Na K})_2 \text{SO}_4 + 3\text{Al}_2\text{O}_3 + 3\text{SO}_2 + 3\text{CO}_2.

60. A theoretical basis for the selection of the optimum firing temperature is provided by the curve for the heating of alunite. This curve shows that the thermal stoppage of dehydration occurs in the temperature range 500 to 520°C, which is also the optimum temperature for firing. At lower temperatures, dehydration is only partial, while higher temperatures lossen the reactivity of the gamma-oxide of aluminium present in dehydrated alunite. For these reasons, the reduction of alunite is carried out at temperatures not exceeding 540°C.

61. As a result of the separate firing and reduction of alunite, sulphur dioxide is obtained in a concentrated state (up to 75 per cent by volume) and can comveniently be used for the production of sulphuric acid. Because the alunite is processed at low temperatures, the alumina obtained dissolves readily in caustic soda solutions of relatively low concentration (110-120 g per litre) at

temperatures not requiring the use of autoclaves. The aluminate solution obtained through leaching, which contains approximately 100 g of Al_2O_3 per litre, has a caustic module of 1.8-1.9 and an SO₃ content of 30 to 40 g per litre, is separated from the mud and converted into alumina by means of the apparatus and technology normally employed in the Bayer process.

62. The only distinctive feature of the method just described is that the aluminste solution obtained, the silicon modulus of which does not exceed 130, must go through an operation for the removal of silicon. This is done at a temperature of 103 to 105°C, by mixing the aluminate solution with seed sodium alumosilicate obtained in preceding silicon-removal operations.

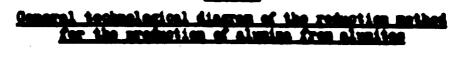
63. The recirculated solution is evaporated, thus separating out the potassium and sodium sulphates. Since, in the reduction firing process, up to about 15 per cent of the aluminium sulphate contained in the alumite remains undecomposed, an equivalent quantity of caustic alkali is converted in the leaching process into an alkali sulphate, while some more of the caustic alkali originally introduced into the process is used up in the formation of alkali alumosilicats in the leaching and silicon removal processes. To compensate for these lesses, part of the sulphate separated out in the evaporation of the recycled solution is converted into caustic alkali in a secondary branch of the process by sintering it with aluminium hydroxide in the presence of a reducing agent. The remainder of the potassium sulphate/sodium sulphate mixture, which is available for commercial disposal, is convorted into the more valuable potassium sulphate, for which purpose it is treated with potassium chloride solution.

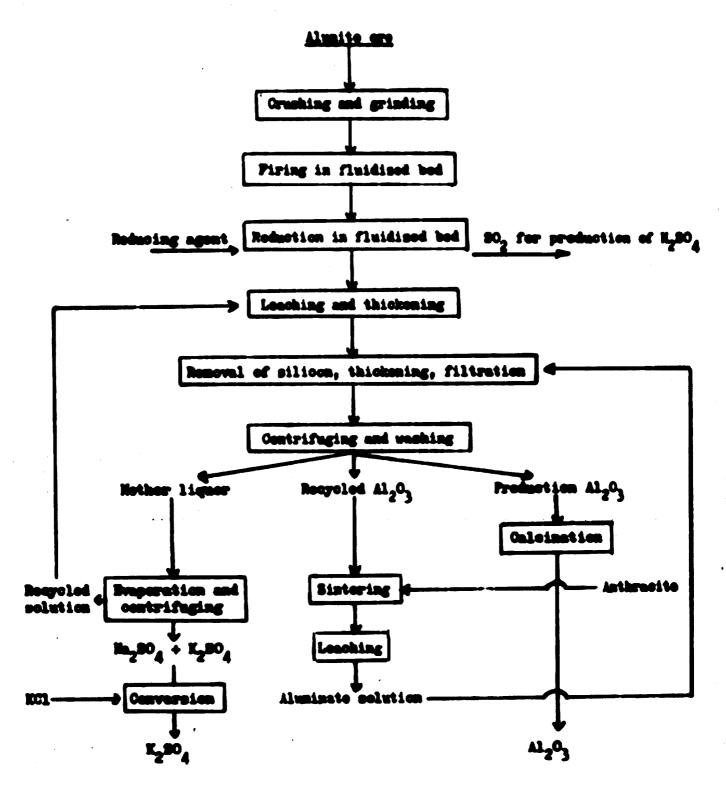
64. In the processing of alunite rock containing about 50 per cent alunite (the rost being quarts), 6.66 tons of ore, 710 m³ of producer gas (1450 kcal/m³), and around 7 tons of steam are needed to produce 1 ton of aluminium oxide. The byproducts obtained are sufficient sulphur dioxide for the production of 1.26 tons of sulphuric acid monohydrate and 0.235 tons of potassium sulphate, plus 0.14 tons of KCl. A flowsheet of the reduction method of producing alumina from alu mate is given in table 6. (2) (3)

65. A now type of alumina-containing raw material is natural sulphates of aluminium, which occur in two forms: in combination with magnesium $MgAl_2(SO_4)_4$. 22H₂O (pickeringite) and as alunogon $Al_2(SO_4)_3$.16H₂O. In the deposits referred

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Table 6





to below, these materials are accompanied by opsomite, MgSO4.7H2O, certain amounts of sulphates of iron and nickel, and by limonite.

66. The deposits of these materials which have been explored in the Argentine province of San Juan near the town of Calingasta contain reserves of approximately twolve million tons. A deposit of pickeringite is also known to exist in Salta province but its extent has not been explored. The deposits of pickeringite in the neighbouring republic of Chile are believed to be even larger.

67. The average chemical composition of the sulphates in the deposits that have so far been explored, according to the findings of geologist Natalia Rossi, are as follows: $Al_2(SO_4)_3 - 20.75\%$; $MgSO_4 - 10.66\%$; $Fo_2(SO_4)_3 - 6.74\%$; $PoSO_4 - 3.12\%$; remainder: water. The average content of anhydrous SO_3 is 12 per cent, while the ratio of useful axides Al_2O_3 : MgO = 4.5. In the opinion of several Argentine geologists, more extensive exploration of the mountainous regions bordering upon the areas referred to might well reveal further deposits of the sulphates mentioned above.

68. The theoretical composition of natural pickeringite is as follows: $Al_2(SO_4)_3 = 39.8\%$; NgSO₄ = 13.9%; and H₂O = 46.2%, giving, in the anhydrous state, an aluminium sulphate content of 74.1 per cent and a magnesium sulphate content of 25.9 per cent. This would correspond to an aluminium oxide content of about 12 per cent in the natural state and 22 per cent in the anhydrous state.

Technical considerations on the method of obtaining alumina from pickeringite 69. Up to the present time, practical investigations on the treatment of pickeringite have either not been carried out or else their results have not been published. It is only possible to speak of the methods of treatment of pickeringite in a general manner. It is well known that pickeringite, alumogen and other sulphates found in rock are easily soluble in water. It seems reasonable therefore to assume that they could be freed of possible accompanying impurities that are not soluble in water by centrifuging an aqueous solution.

70. After treating the solution thus obtained with annonia by the method desoribed above, annonium alum compounds of aluminium are obtained which are soluble with difficulty in cold water; these too are separated by contrifuging. It is well known that megnosium sulphate does not form difficult isoluble compounds with annonia, and the same applies to the other ingredients, except for ferric iron which must therefore be transformed into the lower ferrous form as a preliminary step.

71. The natural minoral boussingaulite $(NH_4)_2 Mg(SO_4)_2 \cdot 6H_2 O$ is soluble in water, while the other natural minoral morenomite NiSO₄ $\cdot 7H_2 O$ and the form of morenomite in combination with magnesium (pyromelin) are readily soluble in water. In addition to the methods considered above for separating the sulphate components of pickeringite, these components can also be separated by using differential heat treatment and breakdown in an exidising mediums

	$\underline{\mathrm{Al}_{2}(\mathbf{s0}_{4})_{3}}$	MeBO4
Heat of formation, koal/mole	820.98	305.5
Temperature of decomposition in an oxidising medium	770°C	1100-1200°C

72. There may also be other mothods of solving the problems of breaking down the sulphate components of pickeringite which have been worked out in practice but not yet published. The most reasonable thing to do, in this connexion, would be to turn first of all to the work of the Canadian Government Mines Bureau Research Centre, which specializes in the breakdown of motal sulphates.

73. The practical solution of this question must undoubtedly be preceded by a fully laboratory investigation of the dotails of the technological process and a test of the process in a half-scale pilot plant; this would be extremely useful for countries possessing deposits of pickeringite, as it would enable them to use the extremely valuable components making up this material.

III. GENERAL CONCLUSIONS

74. The solution to the problem of providing raw materials for the future production of alumina must be sought basically in the comprehensive processing of the great rock matrices which are the sources of the present raw materials for alumina production; namely, bauxites and laterites.

75. Naturally, an effective method must first of all be worked out for the prooessing of laterites and mountain rock deposits that have begun the process of weathering and which contain either high concentrations of aluminium oxide or large contents of other components suitable for comprehensive extraction.

76. The separation of the alumina and other useful components is facilitated by finely dispersing the ores and the iron-reducing agent, using the most active type of peat and coal of vegetable origin (the consumption of these is only half that of coke and they are quickly restored by nature), and bringing the reduced iron into a finely-dispersed state in which it is easily removed by electro-magnetic separation.

77. The quickest and most effective solution of this whole problem can only be reached under the auspices and through the resources and efforts of the Netal- - lurgical Industries Section of the United Nations.

78. The processing of aluminium silicates into alumina becomes industrially feasible when there is the right combination of economic factors such as the availability of cheap raw materials, cheap power and cheap ohemical reagents (e.g. sodium sulphate), the generation of SO_2 as a by-product, or the production of surplus chlorine, when aluminium silicates can be treated by chlorination (silicones and similar compounds).

79. The utilization of sulphate raw material such as pickeringite and alunite may in the near future satisfy the local needs of countries that have such raw muterial at small expense.

80. For the future production of alumina, it would be best to concentrate on the processes described in references (7) and (31).

BIBLIOGRAPHY

- 1. Aluminium, Bulletin of the US Bureau of Mines, 1965 (630).
- 2. LABUTIN, G. (1965) <u>Alunity</u> [Alunites].
- 3. BELYAEV, A. and N. GREVER (1963) <u>Osnovy metallurgii</u> [Fundamentals of Netallurgy], vol. 3.
- 4. LAINER, A. (1961) Proisvodstvo glinosema [Alumina Production].
- 5. KUZNETSOV, S. (1956) Proizvodstvo glinozema [Alumina Production].
- 6. MAIZEL, V. (1950) Proisvodstvo glinozema [Alumina Production].
- 7. SHCHERBAKOV, D. (1957) Klady v bazalte [Treasures in Basalt], Tekh. Mol.
- 8. BETERHTIN, A. (1950) Mineralogiva [Mineralogy].
- 9. <u>Metody isuchenivs osadochnykh porod</u> [Methods of Studying Sedimentary Rocks], Geological Institute of the Acadomy of Sciences of the USSR, vols. 1 and 2, 1957.
- 10. <u>Motallurgiva. obogashchenie i ogneuporv</u> [Metallurgy, Boneficiation and Refractory Materials], Academy of Sciences of the Kazakhstan SSR, vol. 3, 1960.
- 11. RETERTIN, A. <u>et al.</u> (1946) <u>Kurs mestoroshdenii poleznykh iskopaenykh</u> [The Pattern of Deposits of Usoful Ninerals].
- 12. SEVRYUKOV, N., B. KUZMIN and E. CHELINTSEV (1954) Obshchava metallurgiya [Goneral Motallurgy].
- 13. SAZHIN, V. and V. DENISEVICH (1966) Zhur, Prikl. Khimii (12), 2617-2622.
- 14. PONCMAREV, V. (1946) "Sovmostnoe poluchenie glinozema, sody i sery" [Combined Extraction of Alumina, Sodium Carbonate and Sulphur], <u>Vostnik AN Nasekh. SER</u> (11).
- 15. PONOMAREV, V. and B. ERMEKOVA (1957) "Vyshchelachivania glinosema is alunitovykh porod rastvorami sernistogo natriya" [Leaching of Alumina from Alunitorich Rock with Sodium Sulphide Solutions], <u>Izv. AN Kazakh. SSR</u> (<u>Metallurgical</u> <u>Series</u>) no. 5 (16).
- 16. PONCMAREV, V. and M. ERDENBAEVA (1953) "The Interaction of Aluminium Hydroxide and Silice with Solutions of Sodium Sulphide at High Temperature", <u>Isvestiva</u> <u>Akad. Neuk Kasakhakoi SSR</u> (Chemical Series) no. 7.
- 17. PONCMAREV, V. and Kh. NURMAGAMBETOV (1955) "The Treatment of Clays by the Sulphide Method", <u>Trudy Kasakhakogo Gorno-Metallurgicheskogo Instituta.</u> <u>Alma-Ata</u> no. 10.

- 18. The Chemistry and Technology of Alumina, Akad. Nauk Kasakhskoi SSR, Alma-Ata, 1961.
- 19. "Chemical Aspects of the Work at the Mines Branch", Chem. Canada 12(84, 86, 88), (1960).
- 20. <u>Canad. Ches. Process</u>. <u>46</u>(4), 33-34 (1962).
- 21. Mining Preinsoring 14(2), 74-75 (1962).
- 22. Vostnik Akad, Neuk Kasakhakoi SSR 1962 (4), 23-24.
- 23. Ann. Inst. Net. Tech. Agron A 11 (6), 621-649 (1960).
- 24. Room. Glebosnewcze 1961 (10), 760-761.
- 25. Preen. Chem 42(11), 649-650 (1963).
- 26. Sprechaal Koremik, Glas, Email, Silik. 97(16), 449-450 (1964).
- 27. Snravochnik khimii (kratkii) [Short Handbook of Chemistry], 1954.
- 28. Kratkava khimicheskava entsiklopediva [Short Chemical Encyclopaedia], 1961-1964.
- 29. CHERNEAH, A. (1965) Khimicheskos obogashchenie rud [The Chemical Enrichment of Ores].
- 30. POZIN, M. (1961) Tekhnologiva mineral'nykh solei [The Technology of Mineral Salts].
- 31. CLEAVER, G.H. (1954) "Now Notallurgical Process May Unlock Refractory Ore", Ingineering and Mining Journal 155(7), 98-99.

