## John & Kernick FORM P1

REPUBLIC OF SOUTH AFRICA PATENTS ACT, 1978 APPLICATION FOR A PATENT AND ACKNOWLEDGEMENT OF RECEIPT Section 30(1) - Regulation 39

The grant of a Patent is hereby requested by the undermentioned applicant(s) on the present application filed in duplicate

Official application No.

Lodging date

J&K Reference

935072

14th July, 1993

11818 ZA/Mvs

Full Name(s) of applicants:

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Title of Invention

THE PRODUCTION OF HIGH TITANIA SLAG FROM ILMENITE.

The applicant claims priority as set out in the accompanying form P2. The earliest priority claimed is - ZA 92/6007 11th August, 1992. This application is for a Patent of Addition to Patent/Application No. This application is a fresh application in terms of S 37 and based on application no. This application is accompanied by:

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MINTERES

1a A single copy of a provisional specification of

1b Two copies of a complete specification of

2a Informal drawings of sheets

2b Formal drawings of

sheets

3. Publication particulars and abstract (form P8 in duplicate)

4. A copy of Figure

of the drawings for the abstract

5. Assignment of invention (from the inventors) or other evidence of title

6. Certified priority documents (

documents)

7. Translation of priority documents (

documents)

8. Assignment of priority rights

9. A copy of the form P2 and the specification of S.A Patent Application

XX 10. A declaration and power of attorney on form P3

11. Request for ante-dating on form P4

12. Request for classification on form P9

13a Request for delay of acceptance on form P4

13b

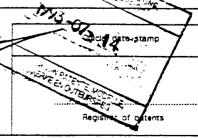
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Address for Service: JOHN & KERNICK, PRETORIA.

Date 14th July, 1993

The duplicate will be returned to the applicant for service as probable lodging but is not valid unless endorsed with official stamp.

01 92/6007



# COMPLETE SPECIFICATION

(Section 30(1) - Regulation 28)

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71 F	ull Name(s) of Applicant(s)			
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72 F	ull name(s) of inventor(s)			
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# THE PRODUCTION OF HIGH TITANIA SLAG FROM ILMENITE

## FIELD OF THE INVENTION

THIS INVENTION relates to the production of a high titania slag from ilmenite and, more particularly, is concerned with a process whereby it may be possible to improve the quality of the product obtained when processing ilmenite, but, possibly more importantly, whereby production costs can be saved at least in areas where the cost of electrical energy is significant.

/BACKGROUND TO

#### BACKGROUND TO THE INVENTION

Natural rutile, owing to its high titanium content and low levels of problem-causing impurities, has traditionally been the preferred feed stock for the production of titanium tetrachloride as an intermediate towards the production of titanium dioxide. Natural rutile is becoming more scarce and, consequently more costly, and the alternative method which uses ilmenite is becoming more favoured.

Ilmenite concentrates have a relatively low titanium content (usually about 50% titanium dioxide compared to about 96% in the case of rutile) and the high impurity content in the balance renders ilmenite generally unsuitable for direct chlorination to titanium tetrachloride as in the case of rutile. In consequence ilmenite has been used as feed stock for the production of pigment by the sulphate process which is becoming increasingly environmentally objectionable.

/The alternative

The alternative to the sulphate route, namely the chloride route, has a major problem associated therewith in that the direct chlorination of ilmenite results in a large quantity of ferric chloride being produced with an associated waste disposal problem. The chloride production of pigment is however preferred over the sulphate route because it requires less processing energy and yields a pigment of superior quality.

general, manufacturers of the intermediate titanium tetrachloride are economically unable to process feed stocks containing less than 80% TiO2 and, in addition, impose stringent limits on some impurities, in particular calcium and magnesium. elements are undesirable The latter two fluidised bed chlorinators because they form high boiling point chlorides which tend to clog both the bed itself and gas ducting leading from the The usual specification imposed reactor. calcium and magnesium is that the sum of the oxides of these two elements should not exceed 1.2% by mass.

/In consequence

In consequence of the unsuitability of ilmenite for direct use in the chloride process, processes involving the thermal reduction of ilmenite to produce a titania rich slag as well as a process involving combined prereduction and chemical leaching procedures to form a synthetic rutile have been proposed and used.

Of these the thermal reduction approach yields a product having a lower titanium grade but it does have the advantage of producing iron in a directly recoverable state.

Conventional alternating current open-arc smelting of various types of ilmenite ores is currently being carried out. The feed materials are introduced into a conventional six-in-line open-arc smelter via multiple feedports (typically more than 20 feed ports are used) most of which are situated near the sidewalls of the furnace in order to protect the furnace sidewalls from refractory erosion. This type of feeding renders process control extremely difficult and, as a result, irrespective of sophisticated computerised control

which can be applied to such furnaces, localised regions of overreduced slag are periodically produced which gives rise to a foaming slag. This is regarded as loss of control and corrective measures which need to be taken lower the thermal efficiency and availability of the furnace.

A limited amount of testwork involving smelting of ilmenite in a d.c. transferred-arc plasma furnace at a 50 kW scale has been described by A D Brent on behalf of the applicant in Mintek report no. M304 dated 15th July 1987. This process has not been implemented by industry due to the following disadvantages:

- 1) The process was of a batch nature and therefore the degree of control required for feeding materials on a continuous basis with periodical or continuous withdrawal of on-grade titanium-rich slag and pig-iron from the furnace was not determined.
- 2) The feed materials are introduced via a single port situated near the sidewall of the furnace

and not directly into the molten pool. This mode of operation could result in similar problems as those encountered in conventional a.c. open-arc smelting in addition to difficulties in the maintenance of a protective freeze lining.

of a magnesite-based lining resulted in product contamination due to an inaccurate feed to power balance and therefore the inability to provide a protective freeze lining. A graphite lining cannot be employed in practice because it would be consumed by reaction with the furnace fume and gases. In addition, a graphite lining would result in an unknown reductant input and thus would affect the required ilmenite to reductant feed ratio and render metallurgical control of product grade difficult.

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It is the object of this invention to provide a process for the thermal reduction of ilmenite in

which the disadvantages outlined above are, at least to some extent, obviated.

#### SUMMARY OF THE INVENTION

In accordance with this invention there is provided a process for the reduction of ilmenite to yield a titania rich slag and pig iron, the process comprising feeding ilmenite simultaneously with carbonaceous reductant, in the absence of fluxes, to the central region of the molten bath of a circular furnace of d.c arc furnace type having one or more electrodes situated in the roof acting as cathode and the molten bath acting as anode, and wherein the feeding of the furnace is on a continuous basis; periodically or continuously withdrawing titania rich slag and pig iron from the furnace, and wherein a frozen lining is maintained between the molten bath and refractory lining.

Further features of the invention provide for the freeze lining to be maintained by suitably adjusting the feedrate to power ratio, optionally in combination with water cooling; for the electrodes situated in the roof of the furnace to be graphite electrodes; for there to be only one

such electrode which is located centrally; for the one or more electrodes to be hollow in which case reactants are introduced into the furnace by way of the bore of such hollow electrode or electrodes; for air to be substantially excluded from the interior of the furnace in which case enhancement of the exclusion of air may be achieved by operating the furnace at a slightly positive pressure; and for the furnace to be a plasma d.c. arc furnace.

A still further and important feature of the invention provides for the feed materials to be preheated. Preferably, in such a case, preheating is achieved, at least predominantly, by using the thermal calorific value of the off-gases from the furnace, optionally after cleaning.

Clearly in all cases the feed rate of materials and the energy input into the furnace are adjusted to achieve and maintain desired tapping temperatures of both slag and molten metal as well as the protective freeze lining in order to assure product

/quality and

quality and protect the furnace lining against excessive wear.

Preferably the carbonaceous reductant is employed in quantities in excess of the stoichiometric amount required to reduce all the iron oxides present in the ilmenite concentrate to metallic iron and to insure that oxygen in the off-gases is substantially in the form of carbon monoxide and the temperature of the furnace is controlled to between 1650 and 1750°C.

In order that the invention may be more fully understood, one example thereof will now be described in greater detail.

# DETAILED DESCRIPTION OF AN EXAMPLE OF THE INVENTION

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In this example tests were conducted in a 3,2 MVA furnace manufactured by applicant. The furnace was of a known d.c. plasma arc furnace type employing a single hollow graphite electrode located centrally above the furnace bath. A direct current power supply was employed in which the molten bath

/formed, in

formed, in use, the anode, while the graphite electrode formed the cathode.

The furnace which had an outside diameter of 1780mm and a refractory lining thickness of only 140mm, was lined with a refractory material wherein the MgO content was approximately 96%. The hearth was lined with a similar material to a thickness of 800mm and a number of mild steel rods were used to make the d.c. (anode) electrical connection from the molten bath through the hearth refractory to Spray water-cooling was cable. anode incorporated on the furnace shell sidewalls in order to assist in maintaining a protective slag freeze-lining. The molten bath in the furnace was heated to the operating temperature of between 1650°C and 1750°C with an initial metal charge.

The feed materials consisted of ilmenite ore concentrate and anthracite having the compositions detailed in Table 1. The feed materials were passed through the single hollow graphite electrode, situated centrally in the furnace and the liquid products were tapped intermittently

during a continuous campaign lasting 11 days at a 500 kW gross energy input. The power and feedrate were balanced in order to maintain a slag tapping temperature of between 1650 and 1750°C. A power density of 0,2 MW/m<sup>2</sup> based on shell diameter, proved suitable in order to maintain a protective freeze lining thickness of approximately 0,15m.

Results of the smelting tests showing metal, slag and dust compositions are given in Table 2.

Table 1:

Chem will analysis of the feed material (mass %).

Compenent	Ilmen <b>ite</b>	Anthracite
TiO <sub>2</sub>	48,17	0,11
Fe <sub>2</sub> O <sub>3</sub>	51,50	1,18
MgO	0,38	0,10
CaO	<0,05	0,13
Al <sub>2</sub> 0 <sub>3</sub>	0,21	2,25
sio <sub>2</sub>	1,02	3,75
MnO	0,78	-
V <sub>2</sub> O <sub>5</sub>	0,10	-
Cr <sub>2</sub> O <sub>3</sub>	0,05	- <b>-</b>
P <sub>2</sub> O <sub>5</sub>		0,03
Na <sub>2</sub> O		0,03
K <sub>2</sub> O		0,21
Fixed Carbon		83,5
Volatiles		6,5
Moisture		2,0

All metal oxide components were analysed for the total metallic content and expressed as the oxide.

Table 2 : Smelting of ilmenite ore concentrate

Test Series	Ilmenite	Anthracite kg		Metal kg	Slag kg	Dust kg
A B C C C C C C C C C C C C C C C C C C	2556 2000 1004 3499 1999 3492 5256	325 273 140 492 289 494 737		513 287 707 1031 978 1086	1615 1163 1691 753 1733 2406	78 114 52 196 83 218
Metal Analyses	Titanium S	Silicon	Manganese	Carbon	Sulphur	Phosphorus
K H O O H F O	<0,005 <0,007 0,020 0,019 0,019 0,011	<0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 <0,05 	0,008 0,026 0,025 0,018 0,033 0,027	1,50 1,83 1,90 1,69 1,83 1,52	0,10 0,12 0,11 0,11 0,11 0,11	0,032 0,026 0,026 0,026 0,023 0,024 0,025

Slag Analyses	TiO2	FеО %	Мдо	CaO \$	A1203	SiO <sub>2</sub>	MnO *
<b>ನ</b> ದ೧೦೫೯೧	74,6 84,5 90,3 84,1 87,8 85,5	18,7 9,6 5,2 10,8 6,8 9,6	0,83 1,13 0,96 0,81 1,13 0,81	0,14 0,18 0,16 0,17 0,23 0,20 0,19	1,35 2,30 2,08 1,74 2,49 1,95	1,96 2,30 2,18 2,52 2,40 2,19	1,20 1,17 1,18 1,19 1,26 1,21 1,26
Dust Analyses	Ti02	Fe203	Мдо	Ca %	A1203	sio <sub>2</sub>	Mno
K B O O B F D	27,7 36,7 45,8 44,6 44,2 48,0	41,3 39,8 33,3 27,3 27,3 32,6	5,51 2,12 1,41 1,72 2,69 1,31	0,92 <0,32 <0,20 <0,24 <0,24 <0,21 <0,20	2,05 1,74 1,61 1,68 1,63 1,75	9,85 10,04 9,63 9,80 9,06 9,15	2,95 3,79 3,79 3,62 3,86

All metal oxide components were analysed for the total metallic content and expressed as the oxide.

From the above it will be seen that acceptable grades of titanium dioxide are produced in most instances and, accordingly, in consequence of the effectiveness of control on a furnace of the nature described above, variables can be controlled to maintain the calcium and magnesium contents of the titanium dioxide at acceptable limits. Also the titanium dioxide content itself can be maintained above the minimum of 80% required for the chloride process and can be increased to 90% if required.

Pre-heating of the feed material has not yet been tested but theoretical calculations indicate that approximately a 30% saving on electrical energy input may be expected if the calorific value of the off-gases is employed for pre-heating the feed materials. It is envisaged that preheated feed materials, or even hot pre-reduced feed materials will not provide any difficulty with respect to maintaining the desired high temperature in feeding through a single small number of or in contra-distinction arrangements to the

/conventional six-in-line

conventional six-in-line furnaces where numerous inlets are provided.

It is envisaged that an extremely effective and economic process is provided by this invention the use of which will result in the simplified employment of ilmenite as a source of high titania slag for use as a feedstock to the chloride process for titanium tetrachloride production and thence titanium dioxide pigments.

#### CLAIMS

- A process for the reduction of ilmenite to 1. yield a titania rich slag and pig iron, the process comprising feeding ilmenite simultaneously with carbonaceous reductant, in the absence of fluxes, to the central region of the molten bath of a circular furnace of d.c arc furnace type having one or more electrodes situated in the roof acting as cathode and the molten bath acting as anode, and wherein the feeding of the furnace is carried out on а continuous basis: periodically or continuously withdrawing titania rich slag and pig iron from the furnace, and wherein a frozen lining is maintained between the molten bath and refractory lining.
- 2. A process as claimed in claim 1 in which the electrodes situated in the roof of the furnace are graphite electrodes.

- 3. A process as claimed in either of claims 1 or 2 in which there is only one electrode in the roof of the furnace.
- 4. A process as claimed in any one of claims 1 to 3 in which at least one electrode is hollow and reactants are introduced into the furnace by way of the bore of such hollow electrode or electrodes.
- 5. A process as claimed in claim 4 in which the hollow electrode is made of graphite.
- 6. A process as claimed in any one of the preceding claims in which air is substantially excluded from the interior of the furnace.
- 7. A process as claimed in claim 6 in which the furnace is operated at a slight positive pressure to achieve said exclusion of air.
- 8. A process as claimed in any one of the preceding claims in which the furnace is a plasma d.c. arc furnace.

- 9. A process as claimed in any one of the preceding claims in which the feed materials are preheated prior to introduction to the furnace bath.
- 10. A process as claimed in claim 9 in which preheating is achieved using the off-gases of the furnace, optionally after cleaning.
- 11. A process as claimed in any one of the preceding claims in which the carbonaceous reductant is used in quantities in excess of the stoichiometric amount required to reduce all the iron oxides to metallic iron.
- 12. A process as claimed in any one of the preceding claims in which the temperature of the furnace is controlled to between 1650 °C and 1750 °C.
- 13. A process as claimed in any one of the preceding claims in which the frozen lining is maintained by suitably adjusting the feedrate to power ratio.
- 14. A process as claimed in claim 13 in which the furnace is, in addition, watercooled to maintain said frozen lining.
- 15. A process substantially as herein described and exemplified.

DATED THIS 14TH DAY OF JULY 1993

JOHN & KERNICK

FOR THE APPLICANT