

Feb. 22, 1966

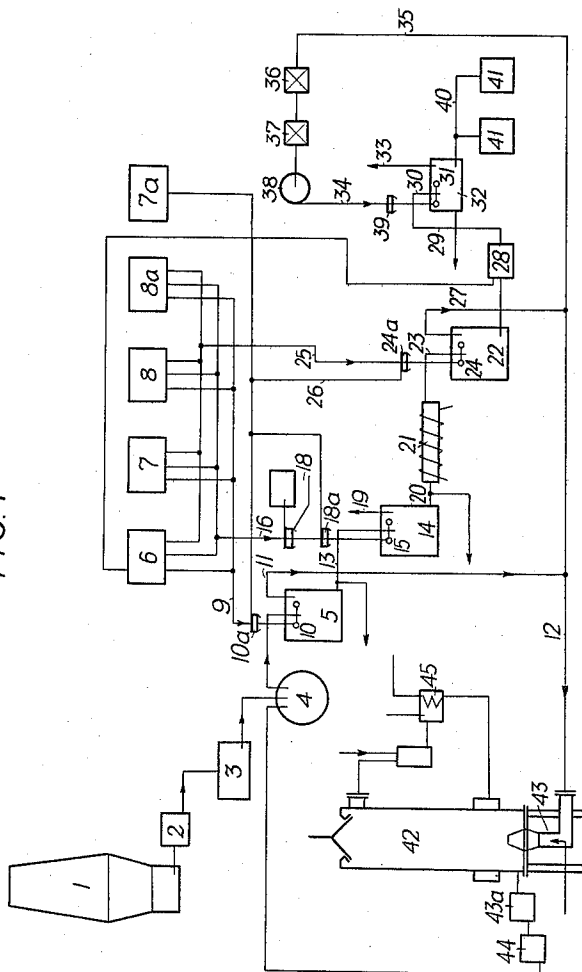
A. MITTERMAYR
PROCESS OF CONTINUOUSLY CONVERTING
MOLTEN CRUDE IRON INTO STEEL

3,236,637

Filed June 25, 1962

2 Sheets-Sheet 1

FIG. 1



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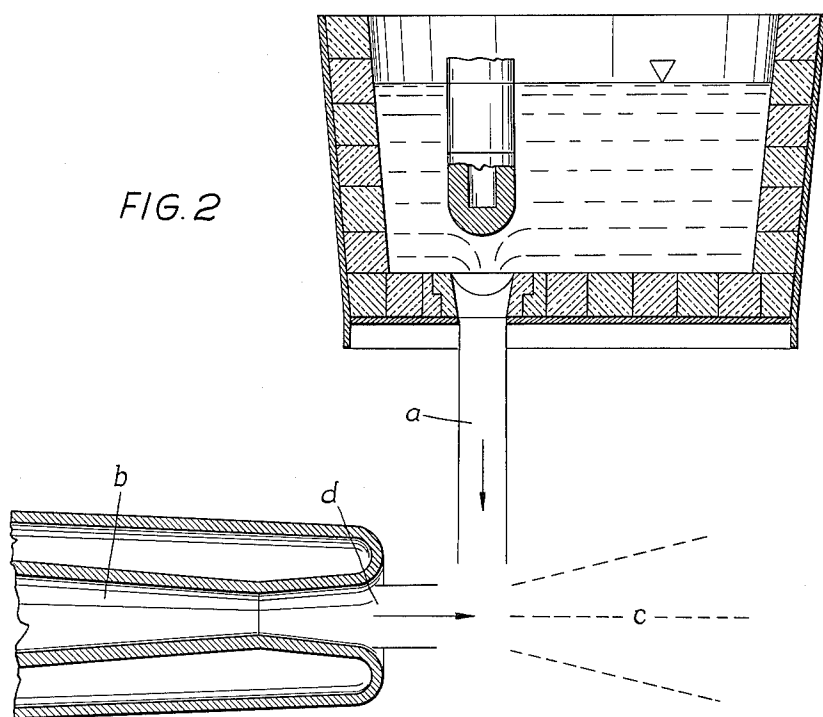
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2 Sheets-Sheet 2



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3,236,637

**PROCESS OF CONTINUOUSLY CONVERTING
MOLTEN CRUDE IRON INTO STEEL**

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Filed June 25, 1962, Ser. No. 204,763

Claims priority, application Austria, June 26, 1961,

A 4,911/61

7 Claims. (Cl. 75—60)

The invention relates to a process of continuously
converting molten crude iron into steel by refining with
a refining agent containing gaseous oxygen in free or
combined form.

Continuous refining processes are known, in which the
reactants crude iron, refining agent and slag-forming sub-
stances are continuously introduced into a reaction vessel
and the steel produced is continuously withdrawn. A
known proposal of this kind calls for passing the molten
crude iron in contact with the refining agent through a
horizontal drum-type converter, which is divided by
baffles into a plurality of cylindrical sections. It was
an object of this proposal to divide the process into
stages, in which different reactions (desiliconizing, de-
carburing, dephosphorizing) take place.

According to another proposal of this kind the refining
process is carried out in a plurality of separate reaction
zones, including one or more prerefining reaction zones
and a decarburizing zone, in which zones different tem-
peratures are maintained and into each of which zones
oxygen is blown with lances.

The known processes described have not been de-
veloped to a production scale, as far as is known, be-
cause they lack an essential requirement of continuous
processing, namely, a high rate of reaction within a small
space. In fact, the measures proposed, including the
blowing of the refining agent by means of lances into
relatively voluminous converters and the maintenance of
liquid metal baths in the converters are based on the
intermittent processes and are only suitable for processes
carried out in successive heats.

It is an object of the present invention to eliminate
the disadvantages and difficulties described. The inven-
tion provides a continuous process for converting liquid
crude iron into steel, in which process the desired high
rate of reaction in small reaction chambers is actually
achieved.

The process according to the invention is carried out
in a plurality of separate reaction zones, including at
least one prerefining reaction zone and one decarburizing
reaction zone, in which zones different temperatures are
maintained, and is characterized in that the crude iron
is atomized at its entrance into at least one of the re-
action zones by means of the refining agent. The fact
that the reaction is carried out according to the inven-
tion in a finely divided, e.g., atomized state, ensures the
large contact surface required, which enables a maximum
rate of reaction and, consequently, a minimum refining
time in each zone, while optimum conditions regarding
temperature and oxygen supply are maintained. As a
result, small reaction chambers are sufficient and there
is no need for extensive conveying and auxiliary installa-
tions.

In the process according to the invention, the reaction
zones are completely separated, i.e., closed reaction cham-
bers are used, which do not communicate with each
other. As a result, different conditions regarding tem-
perature and pressure can be maintained in the diverse
chambers; any mutual influence is avoided, a formation

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of smoke and dust is prevented and combustible waste
gases from one chamber can be rendered utilizable. The
reaction zones or chambers are suitably arranged one
behind the other in cascade so that the crude iron can
be transferred by gravity.

As has been mentioned hereinbefore, it is an object
of the invention to divide the refining process into in-
dividual stages and to provide large reaction surfaces to
achieve a maximum reaction. Regarding affinity to oxy-
gen, the elements accompanying the crude iron do not
have a uniform behavior. The consideration of the equi-
librium relations of the oxidation reactions of each of
the elements which accompany crude iron, particularly
Si, Mn, P, Al, C, shows that in the temperature range
in which steel is made the behavior of carbon differs
basically from that of the other elements because the
affinity of carbon to oxygen increases whereas the af-
finity of all other elements decreases with an increase
in temperature. In the temperature range of the liquid
iron manganese has the highest affinity to oxygen; this
is followed by silicon and aluminium.

In the process according to the invention, these rela-
tions are taken into account by the maintenance of
specific temperatures in the individual zones and/or by
the maintenance of the optimum oxygen supply. If the
accompanying elements contained in the crude iron—
besides carbon—comprise only manganese and silicon,
which can be removed without addition of slag-forming
substances, a single prerefining reaction zone is sufficient.
When phosphorus is also present, an additional prere-
fining reaction zone is provided; in this case silicon
and/or manganese will be removed in the first reaction
zone without addition of slag-forming substances and
phosphorus will be removed in the second reaction zone,
to which lime, particularly lime dust, is added. The
prerefining and the dephosphorizing reaction zones are
preferably held at a temperature of 1250–1350° C. where-
as the decarburizing zone is held at a higher temperature,
of 1550–1800° C. In the prerefining reaction zone or
zones a refining agent having a relatively small oxygen
content, such as air or a mixture of oxygen and CO₂
or H₂O is suitably employed. In the decarburizing zone,
best results are obtained with oxygen-enriched air or
technically pure oxygen. Solid oxides, particularly ox-
ides of iron, may be added to the gaseous refining agent,
as will be described more fully hereinafter. The in-
dividual reaction zones are suitably held at the prede-
termined temperature by a heating which is independent
of the refining process.

In view of the above-mentioned principles of the pro-
cess according to the invention, a preferred embodiment
for the conversion of crude iron which contains phos-
phorus in addition to silicon and manganese comprises
the following stages:

I. Oxidation of silicon and manganese at a low tem-
perature and removal of the oxidation products of iron
formed.

II. Oxidation of phosphorus and sulfur in the presence
of slag-forming substances at a low temperature, separa-
tion of the slag from the iron.

III. Heating up to the carbon reaction temperature
and steel melting temperature.

IV. Oxidation of carbon at a high temperature.

V. Deoxidation of steel with gaseous and/or solid
reducing agents.

VI. Killing and alloying.

In the heating-up zone, the crude iron is heated to
the temperature of the decarburizing zone. As dis-
tinguished from the known refining processes the heat
of combustion of silicon, manganese, phosphorus and
iron, which are highly expensive heating means, is not

required for this purpose, the heating being accomplished with an inexpensive heating means, such as gas or electric heat. The oxidation of the crude iron-accompanying elements in the prerefining reaction zones should not result in a substantial temperature rise because this would lower the reaction rate in the prerefining zones. As has been mentioned hereinbefore, prerefining is preferably effected with a slightly oxidizing gas, such as CO_2 , H_2O or air or mixtures thereof. In particular, the refining agent may be selected from the group consisting of air, a mixture of oxygen and CO_2 , H_2O , and a mixture of oxygen and H_2O . Alternatively, a mixture of solid and gaseous reactants may be employed. The most economical operation will be obtained when a mixture of gaseous and solid oxidizers is used for refining because in this case the iron ore (solid oxidizer) is reduced to iron in a direct steelmaking process by the elements silicon, manganese, phosphorus, and carbon. In this phase, the refining process constitutes a combined smelting and refining process.

The process according to the invention and the plant for carrying it out will now be explained more in detail with reference to the drawing. FIG. 1 is a flow sheet, FIG. 2 is a diagrammatic showing of a preferred method of atomizing the metal.

In FIG. 1, 1 is a blast furnace, which can be operated with a minimum coke requirement (stahleisen). The crude iron passes over a slag separator, which is shown as a forehearth 2, to the desulfurizing unit 3, in which the crude iron is to be desulfurized to a high degree because a special desulfurizing step is not provided for in the succeeding refining process.

A mixer 4 precedes the cascade refining plant according to the invention. This mixer serves as a crude iron storage and equalizing container. The mixer is not essential because the crude iron may be supplied directly from the blast furnace if the same supplies crude iron continuously.

The cascade refining process comprises the previously listed stages I to VI; the oxidizing and refining stages are carried out in the closed oxidizing chambers 5, 14 and 22 lined with refractory material.

Stage I is carried out in reaction chamber 5. Gaseous refining agent is blown through the nozzle 10 onto the crude iron entering in a vertical jet and the crude iron is atomized. CO_2 , O_2 , H_2O steam, air from storage containers 6, 7, 8 and 8a or mixtures thereof may be used as refining agents, and iron ore from the storage container 7a may be added. The refining agent is conducted through the conduit 9 to the nozzle 10. In the preceding mixing nozzle 10a the gaseous and solid refining agents are mixed.

Before the refining begins, the reaction chamber 5 is heated to the desired temperature of about 1250°C . Electric heating is preferable to gas heating because the former will not produce waste gases which contaminate the reaction product and will not increase the volume of gas in the reaction space. It is sufficient if iron having low silicon and manganese contents is obtained in stage I. The oxidation of silicon and manganese in stage I is about 70% so that about 0.1–0.3% silicon and manganese remain where steelmaking iron is used as a starting material. A complete oxidation in stage I is not desired because the iron should be protected from oxidation in the following stages. Slag and metal flow through an outlet out of chamber 5. The slag is separated from the metal outside of the reaction chamber, e.g., in a device as is used at the crude iron tap of the blast furnace.

The CO liberated when 2CO_2 are decomposed into 2CO and O_2 is conducted through the waste gas conduit 11 to the manifold 12. The same applies to some H_2 formed by a decomposition of H_2O .

In stage I, carbon, phosphorus and sulfur are hardly changed and correspond approximately to the values they

have in the crude iron. In conduit 13 the metal is conducted into the reaction chamber 14, which is at 1250°C ., and in which stage II is carried out. In this stage, oxygen from container 7 and ore from container 7a are supplied through the mixing nozzle 18a and the conduit 16. The metal flowing in through the conduit 13 is atomized by means of the nozzle 15. To pass the resulting phosphoric acid into slag, lime dust is used, which is supplied through the mixing nozzle 18 in a mixture with the oxygen. The share of ore must be increased if the reaction temperature is too high when pure oxygen is used.

The P_2O_5 -containing slag is separated from the iron as in the first chamber. The iron flowing out of stage II is almost free of manganese, silicon, phosphorus and sulfur.

In furnace 21, the high-carbon iron is heated from about 1300°C . to 1650°C . (stage III). Heating is effected by an electric or flame heating system. When discharged from the furnace, the iron is at steel melting temperature, which enables also the carbon oxidizing reaction.

Carbon is oxidized in chamber 22 (stage IV) of the refining cascade. The metal enters through nozzle 23 and is atomized by the refining agent supplied through nozzle 24. The oxidizing effect may be controlled by mixing O_2 , Fe_2O_3 , H_2O steam, CO_2 and air. This control of the oxygen supply serves to prevent a combustion of iron. The carbon reaction chamber 22 is preheated to 1650°C . to 1800°C . The CO formed during the reaction is conducted through conduit 27 into the manifold 12. Any existing slag is separated from the iron outside chamber 22, as in connection with stages I and II.

The iron is caused to flow through the several stages I to IV of the cascade as a result of the head. If this is prevented by the structural arrangement, i.e., when the required head is not available for all zones, the metal may be raised between zones with an elevator. This possibility is indicated in the diagrammatic showing of FIG. 1 after zone IV. Behind the carbon reaction stage, the iron flows into a pressure chamber 28, which serves to raise the iron to a level which corresponds to the structural arrangement. In this case the iron is raised with the aid of CO_2 ; this may be effected with a mammoth pump.

Iron passes through conduit 29 and nozzle 30 into the reaction chamber 32, in which the deoxidizing stage V is carried out. The deoxidation is carried out in accordance with the same principle as the oxidizing processes described, i.e., with the metal in atomized condition so that a maximum reaction surface is obtained and the residence time in the deoxidizing chamber corresponds to the residence time in the oxidizing chambers.

By the deoxidation, the oxygen content of the iron should be reduced and the resulting deoxidation product should be completed separated from the iron. A complete separation can be effected mostly reliably when the deoxidation product is gaseous. A gaseous deoxidation product will be obtained where carbon is used as a deoxidizer. This deoxidizer, however, can be applied only in rare cases (only with high-carbon steels). In most cases, other deoxidizers must be used, such as magnesium, aluminium etc. These agents are blown in powdered form together with the atomizing gas. The atomizing gas is a non-oxidizing gas, such as CO. CO is removed through conduit 35 from the CO conduit 12 and is cooled in cooler 36 to purifying temperature, purified in the washer 37 and fed by the pump 38 through the conduit 34. In a mixing nozzle 39 the atomizing gas is mixed with the solid deoxidizer. The metal is atomized with the aid of nozzles 30 and 31. The deoxidizing stage is heated to 1650°C . The steel and the deoxidation slag flow together out of chamber 32 and are separated outside. The resulting waste gas is discharged through the pipe 33.

A simpler deoxidation is enabled by a deoxidizing slag,

which is held in a liquid state in the deoxidizing chamber and into which the steel to be deoxidized is introduced or blown. The device used for this purpose is virtually the same as before, but the slag-forming substances must be blown in together with the atomizing gas.

Alloying stage VI is carried out in chamber 41, into which alloying elements are introduced in solid or liquid condition. Chambers 41 should be heatable for the production of high-alloy steels.

In the process according to the invention, scrap cannot be melted in the various stages. Scrap can be melted, however, with the aid of the CO from stages I and IV so that the process according to the invention is not inferior to other methods also in this respect.

The shaft furnace 42 is provided for melting the scrap.

Because the cascade refining process according to the invention is carried out in gastight chambers, there are no or only small gas losses. For this reason the waste gases which contain combustible components may be collected and used for heating purposes so that heat is available for melting the scrap.

The shaft furnace is a combined gas and solid fuel furnace. The CO from stages I and IV is blown in unpurified condition into conduit 12. The bottom nozzle 43 of the shaft furnace 42 consists of a burner, in which the unpurified CO is burned. The coke layer in the shaft furnace reduces any Fe₂O₃-containing dust which may be blown in with the CO so that a separate cleaning of the gas is not required.

The process of the shaft furnace is not a pure melting process but lies between that of a blast furnace and that of a foundry shaft furnace. The iron from the shaft furnace passes over the forehearth 43a and the soda desulfurizer 44 to the hot metal mixer 4 and from there into the cascade refining plant. Depending on its composition, the waste gas from the cupola is either passed into the hot blast stove 45 or supplied to a different use.

As has been mentioned, the heating up in stage III to steel-melting temperature may be electrically effected by means of a flow furnace 21. This heating step can be omitted if the heating up to steel-melting temperature and decarburization are simultaneously effected in stage IV. Stage IV must be held at 1650–1750° C. and the atomization must be effected with the strongly oxidizing heating flame. Owing to the large metal surface available due to the atomization, the radiation effects a quick transfer of heat so that stage III can be eliminated.

A preferred structure of an atomizing nozzle is diagrammatically shown in FIG. 2. The atomizing nozzle has an outlet piece *a* for forming a dropping metal jet and blowing nozzles *b* which extend at right angles or at an oblique angle thereto and have an orifice member *d*. The pressurized gaseous refining agent is blown in through the blowing nozzles. *c* is the turbulence-producing edge. The relative position of *a* and *d* influences the atomizing effect.

Example

Crude iron was refined in a cascade refining plant according to the invention. In separate chambers, a pre-refining reaction zone and a decarburizing reaction zone with an intervening heating-up zone were provided. The pre-refining chamber was held at a temperature of 1250° C. Air under a pressure of 8 kg./sq.cm. gauge was used as a refining agent in the pre-refining zone. The atomizing nozzle was 12 mm. in diameter. The outlet member for forming the dropping crude iron jet was 14 mm. in diameter. Complete atomization was achieved under these conditions.

The decarburizing chamber was held at a temperature of 1550° C. Pure oxygen was used as a refining agent. The atomizing conditions were the same as in the pre-refining zone (nozzle diameter 12 mm., diameter of iron jet 14 mm., pressure 8 kg./sq.cm. gauge).

The following test results were obtained:

EXPERIMENT NO. 1

| 5 Sampling after seconds | First Oxidizing Stage (Air) | | | | | | Second Oxidizing Stage (O ₂) | | |
|-----------------------------------|-----------------------------|---------------------|---------------------|--------------------|---------------------|---------------------|------------------------------------------|---------------------|---------------------|
| | C, per- cent | Si, per- cent | Mn, per- cent | C, per- cent | Si, per- cent | Mn, per- cent | C, per- cent | Si, per- cent | Mn, per- cent |
| | Before oxidation | | | After oxidation | | | After the oxidation | | |
| 10 30----- | 4.10 | 0.63 | 1.74 | 3.70 | 0.12 | 0.36 | 0.92 | 0.10 | 0.26 |
| 45----- | ----- | ----- | ----- | 3.71 | 0.13 | 0.35 | 0.98 | 0.08 | 0.26 |
| 60----- | ----- | ----- | ----- | 3.72 | 0.11 | 0.34 | 0.95 | 0.12 | 0.25 |
| 75----- | ----- | ----- | ----- | 3.68 | 0.12 | 0.35 | 0.95 | 0.12 | 0.25 |
| 15 90----- | ----- | ----- | ----- | 3.71 | 0.13 | 0.35 | 0.96 | 0.11 | 0.25 |

EXPERIMENT NO. 2

| | | | | | | | | | |
|---------------|-------|-------|-------|------|------|------|------|------|------|
| 20 30----- | 4.30 | 0.95 | 1.50 | 4.10 | 0.24 | 0.38 | 0.86 | 0.12 | 0.21 |
| 45----- | ----- | ----- | ----- | 4.00 | 0.23 | 0.34 | 0.84 | 0.10 | 0.26 |
| 60----- | ----- | ----- | ----- | 3.90 | 0.24 | 0.31 | 0.83 | 0.10 | 0.20 |
| 75----- | ----- | ----- | ----- | 4.10 | 0.20 | 0.31 | 0.85 | 0.11 | 0.22 |
| 90----- | ----- | ----- | ----- | 4.00 | 0.18 | 0.31 | 0.85 | 0.10 | 0.21 |

EXPERIMENT NO. 3

| | | | | | | | | | |
|---------------|-------|-------|-------|------|------|------|------|------|------|
| 25 30----- | 4.00 | 0.80 | 1.00 | 3.93 | 0.21 | 0.33 | 0.83 | 0.08 | 0.26 |
| 45----- | ----- | ----- | ----- | 3.98 | 0.21 | 0.33 | 0.80 | 0.10 | 0.26 |
| 60----- | ----- | ----- | ----- | 3.97 | 0.18 | 0.31 | 0.81 | 0.11 | 0.24 |
| 75----- | ----- | ----- | ----- | 4.00 | 0.14 | 0.35 | 0.80 | 0.06 | 0.25 |
| 90----- | ----- | ----- | ----- | 3.98 | 0.16 | 0.33 | 0.82 | 0.10 | 0.25 |

Whereas in the preferred embodiment of the invention the crude iron is reacted in atomized or sprayed condition in each reaction zone, i.e., in the pre-refining reaction zone or zones and in the decarburizing reaction zone, the inversion in an atomized or sprayed state is not effected in vent in an atomized or sprayed state is not effected in all stages but only in part thereof.

What I claim is:

1. A process of continuously converting molten crude iron containing more than 0.1% silicon and more than 0.3% manganese and also containing phosphorus, sulfur, and carbon into steel, comprising the steps of placing the iron in a first zone, maintaining the iron in the first zone at a temperature of about 1250° C. to about 1350° C., forming a dropping jet of molten iron in the first zone, blowing a slightly-oxidizing gas into the jet in the first zone to atomize the iron in the first zone and simultaneously facilitate oxidation of at least a part of the silicon and manganese to form oxidation products and reduce the silicon content of the iron to about 0.1% and the manganese content of the iron to about 0.3%, removing the oxidation products, placing the iron in a second zone separate from the first zone, maintaining the iron in the second zone at a temperature of about 1250° C. to about 1350° C., forming a dropping jet of molten iron in the second zone, blowing oxygen and a slag-forming substance into the jet in the second zone to atomize the iron in the second zone and simultaneously facilitate oxidation of at least a part of the phosphorus and sulfur to form slag and render the iron substantially free of phosphorus and sulfur, removing the slag, placing the iron in a third zone separate from the first and second zones, bringing the iron in the third zone to a temperature of about 1300° C. to about 1650° C., placing the iron in a fourth zone separate from the first through third zones, maintaining the iron in the fourth zone at a temperature of about 1550° C. to about 1800° C., forming a dropping jet of molten iron in the fourth zone, blowing an iron-combustion-preventing mixture into the jet in the fourth zone to atomize the iron in the fourth zone and simultaneously facilitate oxidation of at least a part of the carbon to form decarburized iron and reaction products, removing the reaction products, placing the iron in a fifth zone separate from the first through fourth zones, maintaining the iron in the fifth zone at a temperature of about 1600° C. to about 1700°

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C., forming a dropping jet of molten iron in the fifth zone, blowing a deoxidizing slag suspended in an inert gas into the jet in the fifth zone to atomize the iron and simultaneously facilitate deoxidation of the iron to form deoxidized iron and deoxidation products, removing the deoxidation products, placing the iron in a sixth zone separate from the first through fifth zones, and forming an alloy in the sixth zone, the alloy comprising the de-

2. A process of continuously converting crude iron into steel comprising the steps of

- (a) heating said crude iron to bring it to a molten state,
- (b) effecting introduction of said molten crude iron into a prerefining reaction zone in which a given reaction involving separation from said molten crude iron of at least one element selected from the group consisting of silicon and manganese is adapted to take place,
- (c) concurrently with said introduction of said molten crude iron into said prerefining reaction zone, maintaining said prerefining reaction zone at a temperature of 1250° C. to 1350° C.,
- (d) also concurrently with said introduction of said molten crude iron into said prerefining reaction zone, effecting introduction into said prerefining reaction zone of a first gaseous refining agent having a given oxygen content in such a manner as to atomize said molten crude iron,
- (e) effecting withdrawal of said molten crude iron from said prerefining reaction zone,
- (f) effecting introduction of said molten crude iron into a decarburizing zone separate from said prerefining reaction zone,
- (g) concurrently with said introduction of said molten crude iron into said decarburizing zone, maintaining said decarburizing zone at a temperature of 1550° C. to 1800° C., and
- (h) also concurrently with said introduction of said molten crude iron into said decarburizing zone, effecting introduction into said decarburizing zone of a second gaseous refining agent having an oxygen content higher than that of said first gaseous refining agent.

3. In the process of claim 2, the step of

- (i) selecting said first gaseous refining agent from the group consisting of air, a mixture of oxygen and carbon dioxide, steam, and a mixture of oxygen and steam.

4. In the process of claim 2, the steps of

- (i) selecting oxygen as said second gaseous refining agent and
- (j) effecting introduction of said oxygen into said decarburizing zone in such a manner as to atomize said molten crude iron.

5. In the process of claim 4, the step of

- (k) passing said molten crude iron through a heating zone separate from said prerefining reaction zone and said decarburizing zone subsequently to said withdrawal of said molten crude iron from said prerefining reaction zone and prior to said introduction of said molten crude iron into said decarburizing zone.

6. In the process of claim 2, the steps of

- (i) selecting air as said first gaseous refining agent,
- (j) selecting oxygen as said second gaseous refining agent,
- (k) effecting introduction of said oxygen into said decarburizing zone in such a manner as to atomize said molten crude iron,

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- (l) passing said molten crude iron through a heating zone separate from said prerefining reaction zone and said decarburizing zone subsequently to said withdrawal of said molten crude iron from said prerefining reaction zone and prior to said introduction of said molten crude iron into said decarburizing zone,

- (m) effecting introduction of said molten crude iron into a deoxidizing zone separate from said prerefining reaction zone, said decarburizing zone, and said heating zone subsequently to said withdrawal of said molten crude iron from said decarburizing zone,

- (n) concurrently with said introduction of said molten crude iron into said deoxidizing zone, maintaining said deoxidizing zone at a temperature of 1600° C. to 1700° C., and

- (o) also concurrently with said introduction of said molten crude iron into said deoxidizing zone, effecting introduction into said deoxidizing zone of an inert gas containing an oxidizable metal suspended therein in such a manner as to atomize said molten crude iron.

7. In the process of claim 2, the steps of

- (i) selecting air as said first gaseous refining agent,
- (j) selecting oxygen as said second gaseous refining agent,
- (k) effecting introduction of said oxygen into said decarburizing zone in such a manner as to atomize said molten crude iron,
- (l) effecting introduction of said molten crude iron into a dephosphorizing zone separate from said prerefining reaction zone and said decarburizing zone subsequently to said withdrawal of said molten crude iron from said prerefining reaction zone and prior to said introduction of said molten crude iron into said decarburizing zone,
- (m) concurrently with said introduction of said molten crude iron into said dephosphorizing zone, maintaining said dephosphorizing zone at a temperature of 1250° C. to 1350° C.,
- (n) also concurrently with said introduction of said molten crude iron into said dephosphorizing zone, effecting introduction into said dephosphorizing zone of oxygen having lime dust suspended therein in such a manner as to atomize said molten crude iron,
- (o) effecting withdrawal of said molten crude iron from said dephosphorizing zone, and
- (p) passing said molten crude iron through a heating zone separate from said prerefining reaction zone, said decarburizing zone, and said dephosphorizing zone subsequently to said withdrawal of said molten crude iron from said dephosphorizing zone and prior to said introduction of said molten crude iron into said decarburizing zone.

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