EFFECTS OF SOLIDIFICATION VARIABLES ON THE STRUCTURE OF ALUMINUM BASE INGOTS

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Cambridge, Massachusetts

Prepared Under Contract No.
DA-19-020-ORD-5706(A)
(June 1965 - June 1966)

For

Department of Army, Materiel Command, MUCOM,
Frankford Arsenal, Philadelphia, Pa. 19137
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ABSTRACT

This work comprises a theoretical and experimental study of macrosegregation in solidification of ingots and castings. A general expression is given for macrosegregation, which is viewed as resulting from mass flow into, or out of, local regions of the ingot during solidification. The mass flow considered in detail results from flow of solute rich liquid to feed solidification and thermal contraction.

Detailed solutions are given for unidirectional heat flow (but where fluid flow is not necessarily unidirectional). It is shown that "inverse segregation" and at least some types of macroscopic "banding" can be quantitatively understood in terms of the analysis. Mold-metal interface resistance is shown to strongly influence macrosegregation, and variations in heat transfer rate during solidification are shown to result in high or low solute bands. Changes in solidification cross-section also result in macrosegregation.

Solutions of the macrosegregation equations are next examined for cases where heat flow is multidirectional. It is shown that centerline segregation can result from the same basic mechanism examined above; i.e., flow of liquid to feed solidification contractions. Other types of segregation considered include (1) the "negative cone" of segregation often found at the base of ingots and (2) high solute regions found under some risers (hot tops). These types of segregation also result in whole or in part from the same mechanism.

A subsequent section considers briefly some types of segregation which may result primarily from thermal contraction or other types of mass flow of solid within the solidifying ingot. This mass flow of solid can accentuate centerline segregation, cause "hot tearing" types of segregation, and result in segregation from settling or floating of solid in early stages of solidification.

Numerical examples of macrosegregation are given for selected limiting cases; calculations are based on simplified forms of the general macrosegregation equation to permit analytical solution. Results of laboratory experiments which qualitatively and quantitatively confirm the analysis are presented. Results are compared with simple analytical solutions, and with more precise results of numerical analysis solutions based on actual thermal data.

The experiments reported are on aluminum-4.5 per cent copper alloy, and include experiments on the following mold designs: (1) simple unidirectional, (2) unidirectional with restricted fluid flow, (3) unidirectional with variable heat extraction rate to illustrate banding, (4) unidirectional with reduced cross sections, (5) edge-risered unidirectional to illustrate centerline segregation, (6) ingot with bi-directional heat flow to illustrate centerline segregation.
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This report summarizes the fourth year of research on study of effects of solidification variables on casting and ingot structure. All experimental work has been on aluminum and magnesium base alloys, but portions of the work are equally applicable to ferrous and other alloys.

Work during the first two years of the research concentrated primarily on microsegregation. Analyses were developed for prediction of effects of alloy and solidification variables on microsegregation; these have shown excellent agreement with experiment\(^1-4\). Work during the second year also considered theoretically and experimentally the problem of pore formation in solidification\(^2,5\). Finally, in the second year, consideration was also given to dendrite structure, and effects of solidification variables on this structure\(^2,4,6\).

In the third year of the research, detailed study was made of elimination of microsegregation by solution treatment\(^7\). This work was undertaken because previous studies had shown that mechanical properties of both cast and wrought aluminum alloys are strongly dependent on degree of homogeneity obtained in the solution treatment\(^8-10\). The previous studies also showed it is extremely difficult to obtain a high degree of homogeneity in most such alloys. A detailed understanding of the kinetics of solutionizing aluminum alloys was therefore felt to be of considerable engineering importance.

Note: References for this Preface will be found on the following page. References for the rest of the report will be found at the end of the report.
In the third year of the research, work was also conducted on the outer "chill zone" found in ingot solidification, and it was shown that the classical theory of "copious nucleation" in the region is inadequate to explain experimental results. Dendrite structure at the chill surface was shown to be different from that in the bulk metal in some rather surprising ways.

Finally, a series of experiments were conducted from which it was concluded that dendrite arm spacing in usual casting and ingot solidification was determined by coarsening ("ripening"). A simple analysis, based on coarsening kinetics, was found to be of engineering applicability to predict effects of alloy analysis and other solidification variables on dendrite arm spacing.

During the period for which this report is written, research has been continued on effects of coarsening kinetics on dendrite structure, and on structure in rapidly solidified alloys. In addition, limited work has been conducted on solidification in composites, and on effects of melt agitation on structure. Finally, a significant effort has been initiated to study mechanical properties of ultra-high strength wrought aluminum-base alloys, produced using solidification techniques to assure complete homogeneity of the original ingot. These studies are to be summarized in a later report; this report will deal with work that has comprised the major effort during this contract year – i.e., macrosegregation in casting and ingot solidification.
This work has provided, for the first time, a basis for understanding a wide variety of apparently different types of macrosegregation. The basic analysis presented provides quantitative as well as qualitative understanding of effects such as inverse segregation, centerline segregation, and banding. The results should be of considerable engineering usefulness in understanding and controlling macrosegregation in ferrous, as well as non-ferrous alloys.

Theoretical portions of this work employ a model for solute redistribution in dendritic solidification discussed in detail by Brody and Flemings, and Bower, Brody, and Flemings. Concepts employed in considering the interdendritic fluid flow are those described by Piwonka and Flemings. Theoretical work of Scheil and Kirkaldy and Youdelis on inverse segregation has served as an important foundation for this study. Finally, special acknowledgement is made of stimulating discussions with Professor John Cahn of Massachusetts Institute of Technology.
REFERENCES TO PREFACE


I. INTRODUCTION

In the works of Vannoccio Biringuccio published in 1540, one of the first references to macrosegregation is noted. He refers to the problem of exudation and inverse segregation encountered in the manufacture of bronze gun barrels. From that day to this, the problem of macrosegregation has been studied by many researchers. The usual approach has been to isolate one type of macrosegregation (i.e. centerline segregation, negative cone of segregation, exudation, inverse segregation, banding, etc.) with the exclusion of all other types. The result has been a multiplicity of theories, each one designed to explain a specific type of macrosegregation. This section will consider many of these macrosegregation effects, present some of the pertinent literature and discuss theories pertaining to each effect. In a subsequent section it will be shown that many of the different types of macrosegregation are not due to unrelated phenomena but can be understood simply in terms of the direction and velocity of solute rich liquid which flows to feed solidification and thermal contractions.

A. **Inverse Segregation**

One of the most thoroughly investigated macrosegregation effects is inverse segregation. In the late 1920's and early 1930's many theories were proposed to explain the mechanism of inverse segregation. Brick and Phillip have reviewed these proposals and rejected most of the suggested mechanisms as being either speculative or unconfirmed by
experimental observations. They did accept, as most observers do
today, that inverse segregation is due to solidification and thermal
volume contraction with resulting flowback of solute enriched
liquid, increasing local average composition above $C_0$ (average melt
composition).

Most of the definitive work which substantiates the volume
contraction and interdendritic flow theory for explaining inverse
segregation was published in the 1940's. Adams$^3$ studied inverse
segregation in aluminum-copper alloys (3 per cent and 7 per cent
copper) and reported extensive experimental results. However, he,
along with Sauerwald$^4$ and Masing$^5$, while attempting quantitative
predictions, derived no analytical expressions. The first analytical
work to predict inverse segregation which could be checked experiment-
ally was proposed by Schell$^6$. His expression predicted maximum
segregation at the chill face as a function of alloy content. Good
order agreement was obtained with experiment. Kirkaldy and Youdelis$^7$
recognized that the basic mechanism that explains inverse segregation
accounts for the details of the segregation distribution in a uni-
directionally solidified ingot. They correlated previously available
data on inverse segregation in aluminum-copper alloys, and developed
expressions in differential form to predict macrosegregation throughout
the length of a unidirectionally solidified ingot; these were solved
by numerical analysis. Later Youdelis and Colton$^8$ extended the work to
a study of inverse segregation in aluminum-zinc ingots. A limiting
case of the macrosegregation expression, given later, is exactly that
of Kirkaldy and Youdelis\textsuperscript{7}; in addition, it is shown that by making certain approximations, a simple analytical solution is possible.

The literature abounds with investigations on inverse segregation. Some of the data are of very early origin and experimental procedures and results are questionable. Extensive reviews of the literature are also available, notably by Vaughn\textsuperscript{9} reviewing work to 1937, by Pell-Walpole\textsuperscript{10} covering work to 1949 and most recently by Vosskuhler\textsuperscript{11} where 65 references are cited, many recent, on inverse segregation. In the theoretical section it will be shown how not only inverse segregation but also the "negative cone" of segregation, banding, centerline segregation, and localized high or low solute regions can be related to the same mechanism which causes inverse segregation; namely flow of solute enriched liquid to feed solidification and thermal contraction. For this section, however, existing theories on these effects will be considered.

B. **Negative Cone of Segregation**

Probably no macrosegregation phenomena is less understood than the cone of negative segregation which often appears at the bottom of large ingots. The usual explanation for this macrosegregation effect is related to initial crystals which form being of lower composition than the average composition of the melt. As the crystals form they are said to settle to the bottom of the ingot (since they are more dense than the liquid) and pile up in a cone of high purity, low solute composition\textsuperscript{12}. Several difficulties exist with the theory:
1. In an ingot, the liquidus temperature is reached by most of the bulk liquid before solidification starts.\textsuperscript{13,14}

2. It can be shown from Stokes' law that the rate of descent of crystals, in liquid metals, compared to solidification times, is too slow to descend the required distance in the available time.\textsuperscript{15}

One of the new ideas on this phenomena is reported by Marburg.\textsuperscript{16} He associates the "negative cone" with accelerated vertical and transverse solidification after the initiation of solidification. In a subsequent section this will be discussed and related to the flow of solute rich liquid to feed solidification contraction.

C. Centerline Segregation

Another macrosegregation effect little understood is centerline segregation. While data are available on the magnitude of the segregation encountered,\textsuperscript{17} theories usually are limited or speculative. Until recently, the generally accepted theory related centerline segregation to a buildup of solute ahead of the interface of the columnar grains growing away from the vertical wall.\textsuperscript{18,19} As the grains grow toward the center, the rejected solute was visualized as being pushed ahead resulting in a solute rich centerline where the growth fronts converge. Recently, however, it has been shown that there can be no significant solute buildup in front of growing tips of dendrites at the growth rates and gradients encountered in usual casting processes.\textsuperscript{20,37,38} In a following section the mechanism for
centerline segregation will also be considered as being related to solute rich fluid flow to feed solidification contraction.

D. Banding

Probably no problem gives the ingot maker or foundryman more problem than banding type macrosegregation. Banding refers to regions where an impurity of solute in the base alloy is localized in a narrow band or strip. In ferrous alloys these bands may be high in inclusion content, but in all alloys are of somewhat different composition than the base alloy (typically .05 to 1.00 per cent, depending on base solute). Banding occurs in large ingots, centrifugal castings and continuously cast billets, and often seems to occur in a random manner. In large steel ingots, Marburg has associated inverted V and V-type banding segregation with changes in the rate of solidification. Takenaka shows that banding type defects in centrifugally cast steel pipe are influenced by vibration in the centrifugal casting system and solidification rate in the melt. This appears to be as close as anyone comes to relating banding to a mechanism similar to that of inverse segregation. Northcott suggests that vibration might cause banding by affecting the boundary layer in front of the growing dendrites. In view of the very small thickness of this layer, however, it seems unlikely that this theory is correct.

Cumberland proposes a theory based on the bands being incipient laps resulting from irregularities in the manner of liquid metal flow within the rotating mold. This theory, however, again requires extensive buildup of solute at the tips of the solidifying interface.
An older theory by Howson\textsuperscript{24} relates banding to a mechanical separation of elements due to centrifugal forces. The theory, however, does not take into account solidification time which can be quite rapid.

Finally, while it is not strictly a banding effect, consideration will be given to the segregation in continuously cast billets and slabs. Wolf\textsuperscript{25} discusses this problem in Al-Zn-Mg-Cu alloys and reports on similar investigations by other researchers. While much data are reported and conclusions drawn as to what affects the segregation in continuous castings, little seems to be available concerning possible mechanisms. Compositions toward the center of continuously cast ingots were found to decrease\textsuperscript{25} which can again be related to solute flow to feed shrinkage.

E. **Solute Enriched Regions**

A type of macrosegregation related to banding is the phenomenon of localized high solute regions usually found in castings and ingots. By localized enrichment is meant any region where solute content can be significantly higher than average base alloy composition. The types of macrosegregation effects usually associated with this phenomena are exudation, hot tearing type defects and solute enrichment under risers. These three phenomena will be considered separately even though they are associated effects, related by the mechanism involved.
1. Exudation.

Exudation is usually found in conjunction with inverse segregation, and the two effects are often confused. Exudation occurs at the mold-metal or chill-metal interface, apparently as a result of metal-mold gap formation. After solidification is started at a wall or chill, contraction of the metal causes an air gap to form. When solute enriched liquid is forced from between dendrites to fill the gap, a layer of enriched solid which can approach eutectic composition is formed. Exudation has been studied and the variables discussed which affect thickness of the layer and its formation. All data indicate, that exudation is related, as are many macrosegregation effects, to the flow of solute enriched interdendritic liquid.

2. Solute Enrichment Under Risers.

One of the effects readily related to solute enriched fluid flow is macrosegregation beneath casting risers and ingot "hot tops". At the later stages of solidification (i.e., when casting or ingot is almost completely solid) some solidification has already occurred in the riser or hot top. Because of the partial solidification in the riser or hot top the liquid has become enriched in solute. When further contraction occurs in the casting or ingot the solute enriched liquid feeds from the riser to the casting. The result is the greatest enrichment in the casting or ingot at the last location to solidify (directly under the hot top or riser). Bishop et al
have reported extensively on this effect in steel castings. It is presently being studied at the MIT casting and solidification section\textsuperscript{31} using aluminum-4.5 per cent copper alloy. It will be seen later in this report that this under-riser segregation is expected to be most severe when the riser is of inadequate size, or inadequately insulated.

3. Hot Tearing.

The final type of macrosegregation to be considered is "filled hot tears". "Hot tearing" is associated with the solid contraction which occurs when a casting or ingot cools. When the contraction is hindered by the mold, cores, or differences in casting section size, complete rupture of a segment of the casting or ingot may occur. When these fractures occur within the liquid-solid zone, they are termed "hot tears" and when an incipient tear is fed from a surrounding region, the result is a "filled hot tear", generally rich in solute. Rosenberg et al\textsuperscript{32} have studied hot tearing in aluminum-copper alloys and report data which can be accounted for by solute rich flow feeding the "hot tear". While much work has been done on the fundamental mechanism of "hot tearing" in steel\textsuperscript{33,34} and non-ferrous alloys\textsuperscript{35,36} little attention has been given to the resulting solute enriched flow necessary when the "hot tear" is fed.

In summary, five different types of macrosegregation with variations have been discussed, primarily on the basis of generally accepted existing theories. In the next section it will be shown that all these
effects are related to solute enriched fluid flow feeding solidification and thermal contraction, and that they can be quantitatively, as well as qualitatively, understood on this basis.
II. THEORETICAL DEVELOPMENT

In two recent papers\textsuperscript{37,38}, the problem of microsegregation in dendritic solidification was considered. It was concluded that for usual casting and ingot solidification, microsegregation in binary alloys can accurately be predicted by the "non-equilibrium freezing equation" provided there is no diffusion in the solid and no macro-segregation. Up to the limit of solubility of the primary solid phase this is written:

\[(1 - k)C_L df_L = -f_L dC_L\]  

where: \(C_L\) = liquid composition within the "volume element" (a small volume the order of size of the dendrite arm spacing)

\(f_L\) = weight fraction liquid in the volume element

\(k\) = equilibrium partition ratio

For the special case of constant partition ratio, equation (1) is:

\[f_L = \left( \frac{C_L}{C_0} \right) \]  

or, since \(C_S = kC_0\) and \(f_S = (1 - f_L)\)

\[C_S = kC_0(1 - f_S)\]  

(3)
where: $C^* S$ = solid composition at the liquid-solid interface  
$f_S$ = weight fraction solid

Equation (3), for the case of no solid diffusion, describes solid composition as a function of weight fraction solid after, as well as during, solidification. It therefore completely describes "severity of microsegregation", although not its geometry$^{37,38}$.

Here consideration is first given to how the foregoing, and particularly equation (1), is modified by mass flow in or out of the volume element during solidification. The net change in composition of the element at the end of solidification is then calculated (i.e., "macrosegregation").

PART I: GENERAL MACROSEGREGATION EXPRESSIONS

A. Local Solute Redistribution

Consider a small volume element at coordinates, $x, y, z$ within a solidifying ingot. These coordinates are to be such that no solid material enters or leaves the volume element during solidification. Hence the location $x, y, z$ may move with respect to the mold wall as a result of thermal contraction or other solid movement, but refers to the same local region of a given dendrite. Solute enters or leaves the element only by liquid flow to feed shrinkage. Mass flow by diffusion is neglected; such mass flow has been shown to be negligible in the absence of convection sufficiently vigorous to penetrate interdendritic regions. Local perturbations of the macroscopic composition gradient are assumed negligible so $C_L$ is uniform
within the volume element. This model is identical to that described earlier\textsuperscript{37,38}, except that solidification contraction and thermal contraction are considered. The change in total solute in the element per unit time during solidification is then:

\[
\frac{d}{dt} (\bar{\rho}\bar{C}) = -\bar{v}\rho_L \bar{g}_L \bar{C}_L \vec{v}
\]  

(a) (b)

where:

- $\bar{\rho}$ = local average density (of liquid and solid), gm/cm\(^3\)
- $\bar{C}$ = local average composition of liquid and solid, gm/gm
- $\rho_L$ = liquid density, gm/cm\(^3\)
- $\vec{v}$ = local velocity of liquid relative to solid (resulting from solidification contraction), cm/sec
- $\bar{g}_L$ = volume fraction liquid

The two terms are:

(a) Total local rate of change of solute during solidification
(b) Local rate of change of solute resulting from liquid flow relative to the solid.

The change in total mass in the element per unit time is:

\[
\frac{d\bar{\rho}}{dt} = -\bar{v}\rho_L \bar{g}_L \vec{v}
\]  

(5)

and this change is the sum of volume changes in the liquid and solid during solidification:
\[ \frac{d\rho}{dt} = \frac{d}{dt} \left( \rho_S g_S + \rho_L g_L \right) \]  

(6)

where: \( \rho_S \) = solid density, gms/cm\(^3\)

\( g_S \) = volume fraction solid

In the absence of pore formation:

\[ g_S + g_L = 1 \]  

(7)

The term on the right of equation (4) may be rewritten:

\[-\nabla \cdot \rho_L g_L C_L \frac{\rho_L}{C_L} = - C_L \nabla \cdot \rho_L g_L \frac{\rho_L}{C_L} - \rho_L g_L \nabla \cdot \frac{C_L}{C_L} \]  

(8)

Substituting equations (3) - (8) in (4):

\[ \frac{d}{dt} \left( \tilde{C}_S \rho_S g_S + C_L \rho_L g_L \right) = C_L \frac{d}{dt} \left( \rho_S g_S + \rho_L g_L \right) - \rho_L g_L \nabla \cdot \frac{C_L}{C_L} \]  

(9)

Assuming the partition ratio applies at the interface, no diffusion in the solid, and constant solid density, \( \rho_S \):

\[ d(\tilde{C}_S \rho_S g_S) = k_{CL} \rho_S dg_S \]  

(10)

Combining (9) and (10):

\[ \rho_S C_L (1 - k) \frac{dg_L}{dt} = -\rho_L g_L \left( \frac{dC_L}{dt} \right) - \rho_L g_L \nabla \cdot \frac{C_L}{C_L} \]  

(11)
or:

\[
\frac{dC_L}{dt} = - \left( \frac{1 - k}{1 - \beta} \right) \frac{C_L}{g_L} \frac{dg_L}{dt} - \vec{v} \nabla C_L
\]  

(12)

where \( \beta \) is the "solidification shrinkage":

\[
\beta = \frac{\rho_S - \rho_L}{\rho_S}
\]

Equation (12) is the basic local solute redistribution equation used hereafter to calculate macrosegregation. It is written for the general case of three dimensional heat and fluid flow, for (a) constant solid (but not liquid) density during solidification, and (b) negligible net solute change from diffusion.

B. Local Solute Redistribution, Planar Isotherms

Assume isotherms are planar and perpendicular to x axis where x is the distance from the mold wall. Liquid composition at the liquid-solid interface is a function only of temperature provided undercooling from solute, radius of curvature, and kinetics effects is negligible. Hence, planar isotherms require planar isoconcentrates and:

\[
\nabla C_L = \vec{v} \left( \frac{\partial C_L}{\partial x} \right)
\]  

(13)

also:

\[
\vec{v} = i v_x + j v_y + k v_z
\]  

(14)
Combining (13) and (14) with (12):

\[ dC_L = - \left( \frac{1 - k}{1 - \beta} \right) \frac{C_L \, dg_L}{g_L} - \left( \frac{\partial C_L}{\partial x} \right) v_x \, dt \]  

(15)

Now \( C_L \) is a function only of temperature. Isotherms and isoconcentrates move with velocity \( \frac{dx}{dt} \) at \( x \). Hence,

\[ \left( \frac{\partial C_L}{\partial x} \right) \frac{dx}{dt} = - \frac{dC_L}{dt} \]  

(16)

and:

\[ \frac{\partial C_L}{\partial x} = - \frac{dC_L}{dx} \]  

(17)

Similarly:

\[ \frac{\partial T}{\partial x} = - \frac{dT}{dx} \]  

(18)

where: \( T = \) temperature

Now, let \( G = \frac{\partial T}{\partial x} = - \frac{dT}{dx} \) (temperature gradient), and \( \varepsilon = \frac{dT}{dt} \) (rate of temperature change), then:

\[ \frac{\varepsilon}{G} = - \frac{dx}{dt} \]  

(19)

Combining (17), (18), and (15):

\[ \frac{dg_L}{g_L} = - \left( \frac{1 - \beta}{1 - k} \right) \left( 1 + \frac{v_x G}{\varepsilon} \right) \frac{dC_L}{C_L} \]  

(20)

Equation (20) is the solute redistribution expression (equation 12) for the case of unidirectional heat flow, but where fluid flow is not necessarily unidirectional, \( k \) and \( \beta \) are, in general, functions of \( C_L \).
G and c (at x, y, z) are also functions only of temperature, and hence of \( C_L \); these quantities are readily measured experimentally, or in principle, calculated from heat flow considerations. Equation (20) becomes identical to that of Kirkaldy and Youdelis\(^7\) when \( v_y = v_z = 0 \). Evaluation of \( v_x \) is discussed below.

C. Evaluation of Flow Velocity Vector, Planar Isotherms

1. Flow Perpendicular to Isotherms.

Consider first the case of flow only perpendicular to planar isotherms. Equation (5) and (6) then become, for locations \( x' \):

\[
\frac{d\rho}{dt} = -\frac{3}{\partial x'} (\rho_L' g_L' v') = \frac{d}{dt} (\rho_S' g_S' + \rho_L' g_L') \tag{21}
\]

where: \( x' \) is distance from the chill face to the location considered.

When the chill face is above the non-equilibrium solidus, \( T_E \),
\( 0 < x' < x \), and when the chill face is below \( T_E \), \( x_E < x' < x \). The primes on other quantities denote that these are at \( x' \). Now assuming constant \( \rho_S \) and considering the case where the chill face \( (x' = 0) \) is above \( T_E \):

\[
\begin{align*}
\rho_L g_L v_x &= \int^{x'}_{x=0} (\rho_L g_L v_x)\ dx' = \int^{x'=x}_{x'=0} (\rho_L' - \rho_S') \frac{dg_L'}{dt} \ dx' + \int^{x'=x}_{x'=0} g_L' \left( \frac{dp_L'}{dt} \right) \ dx' \\
\end{align*}
\]

or

\[
\begin{align*}
\rho_L g_L v_x &= \int^{x'}_{x'=0} (\rho_L' - \rho_S') \left( \frac{dg_L'}{dt} \right)(dx') - \int^{x'=x}_{x'=0} g_L' \left( \frac{dp_L'}{dt} \right) \ dx' \tag{22a}
\end{align*}
\]
when the chill face is below the non-equilibrium solidus, in the case of an alloy containing some eutectic, eutectic of volume fraction \( g_E \) precipitates at the non-equilibrium solidus, \( T_E \). Then:

\[
\begin{align*}
\rho_L g_L \frac{dx'}{dx} &= \int \left( \rho_S' - \rho_L' \right) \left( \frac{dg_L'}{dt} \right) dx' - \int g_L' \left( \frac{dp_L'}{dt} \right) dx' - \\
&= \left( \rho_{SE}' - \rho_{LE}' \right) g_E U_E \\
\end{align*}
\]

(22b)

where: \( \rho_{SE}' \), \( \rho_{LE}' \) = densities of solid and liquid eutectic respectively

\( U_E \) = velocity of eutectic isotherm, \( T_E \)

\( dg_L' \) is a function of \( x' \) as well as macrosegregation at \( x' \).

Hence, to rigorously determine local solute redistribution at \( x \), equation (22) and equation (30) must be solved by a tedious iterative procedure. To permit simpler, analytical, solutions for the segregation, an alternate procedure is employed in much of the following.

\( \rho_S \), \( \rho_L \) and \( k \) are assumed constant, and for velocity calculation, \( g_L \) is assumed to vary linearly with temperature within the liquid-solid zone.


Calculation of the general case of macrosegregation when fluid flow is not perpendicular to isotherms requires detailed knowledge of \( v_x' = f(x', y', z') \). In the special case where fluid flow is not
unidirectional but where $v_x'$ and $g_L'$ are independent of $y'$ and $z'$, an expression similar to equation (22b) is written (for the chill face below $T_E$):

$$A \rho_L g_L v_x = \int (\rho_S' - \rho_L')(\frac{dg_L'}{dt})A' \delta x' - \int g_L' (\frac{d\rho_L'}{dt})A' \delta x' -$$

$$A_{E} (\rho_{SE} - \rho_{LE}) g_{E} U_{E}$$  \hspace{1cm} (23)

where $A$, $A'$ and $A_E$ are cross sectional areas at $x$, $x'$ and $x_E$', respectively.

$U_E$ = velocity of eutectic isotherm

D. Calculation of Macrosegregation

Macrosegregation at $x$, $y$, $z$, is defined as

$$\Delta C = \bar{C}_S - C_0$$  \hspace{1cm} (24)

where: $\bar{C}_S$ = average composition at $x$, $y$, $z$ after solidification

$C_0$ = bulk composition

"Positive segregation" refers to $\bar{C}_S > C_0$, and "negative segregation" for $\bar{C}_S < C_0$. For a binary alloy containing a primary phase $a$ and some eutectic, the segregation after solidification is given by:
\[ \Delta C = \frac{\rho_a g_a \bar{C}_a + (1 - g_a)(\rho_{SE} C_E)}{\rho_a g_a + (1 - g_a)\rho_{SE}} - C_0 \] 

(25)

where: \( \rho_a, \rho_{SE} \) are densities of \( \alpha \) and eutectic, respectively

\( (\rho_a = \rho_S = \text{constant}) \)

\( \bar{C}_a = \) average composition of the entire amount of primary \( \alpha \) phase in the final solid ingot

\( C_E = \) composition of eutectic

The average composition, \( \bar{C}_S \), of any portion of the \( \alpha \) phase, \( g_S \), is given by:

\[ \frac{g_S + \Delta g_S}{\rho_S \bar{C}_S \Delta g_S} = \frac{C_S d g_S}{g_S} \]

(26)

and the mass of solute in the entire primary phase, \( \rho_S g_a \bar{C}_a \) determined by summing equation (26) for all portions of the phase \( \Delta g_S \), which total to \( g_a \). \( C_S \) as a function of \( g_S \) is determined from equation (20) recalling that \( C_S = kC_L \), and \( g_S = 1 - g_L \).

E. Incomplete Feeding

In the foregoing and in most of the following, it is assumed that feeding is complete. However, under many, if not most, solidification conditions, pores are expected to develop in the late stages of solidification as a result of resistance to fluid flow, precipitation of dissolved gas or a combination of both. The preceding analyses are readily adaptable to treatment of macrosegregation under these conditions.
In the simplest case, feeding is assumed to stop completely at a critical fraction, liquid, $g_L^V$, at time $t_v$, location $x_v$ and liquid composition becomes $C_L^V$. Hence flow velocity to feed shrinkage will be less in the case of pore formation, since the lower integration limit on, for example, equation (22b) becomes $x_v$ instead of $x_E$.

Further, in evaluation of the solute redistribution expression, equation (20), one set of limits now becomes $g_L^V$ and $C_L^V$; for $g_L^V < g_L$ no further feeding occurs. Final macrosegregation is then simply calculated from summing the solute in the solid and liquid phases at $t_v$ (since after $t_v$ no further solute enters or leaves the volume element, although solidification progresses).

PART II: UNIDIRECTIONAL SOLIDIFICATION

As examples of application of the foregoing to calculations of macrosegregation, the case of "unidirectional" solidification is first considered; i.e., solidification with planar isotherms, unidirectional heat flow. Some limiting cases are first considered, and then segregation versus distance from the chill wall is examined for various conditions of heat and fluid flow.

A. Limiting Cases

As examples of the foregoing, solutions to equation (20) for three limiting cases are considered first here: (a) $v_x G/\varepsilon = 0$, (b) "steady state solidification", and (c) non "steady state", but finite $v_x G/\varepsilon + f(g_L)$. $\beta$ and $\beta_e$ are assumed constant and equal.
1. \( \frac{v_x G}{\varepsilon} = 0 \).

At a chill face, \( v_x = 0 \), and provided \( G/\varepsilon \) is finite, \( v_x G/\varepsilon = 0 \).

For constant, \( k, \beta \) and no solid movement during solidification, equation (20) becomes:

\[
\int \frac{g_L}{g_L} \frac{dC_L}{C_L} = -\left( \frac{1 - \beta}{1 - k} \right) \int \frac{dC_L}{C_L} \quad (27)
\]

and

\[
g_L = \left( \frac{C_L}{C_0} \right)^{1 - \beta} \quad (28)
\]

Here, increasing \( \beta \) results in increasing \( C_L \) at a given \( g_L \) and hence in greater positive macrosegregation.

2. Steady State Solidification.

Consider now unidirectional "steady state" solidification of a long ingot; i.e., planar isotherms move at a constant velocity \( U = -\varepsilon/G \), along the \( x' \) direction in an ingot of constant cross section, and chill face is below the non-equilibrium solidus. For \( \beta = \beta_e = \text{constant} \), equation (22b) reduces to:

\[
v_x = \left( \frac{\beta}{1 - \beta} \right) \frac{\varepsilon}{G} \quad (29)
\]

Substitution of equations (29) in (20) and integrating from the limit \( g_L = 1 \) at \( C_L = C_0 \) to \( g_L, C_0 \) yields:
Equation (30), for the special case of $\beta = \beta_e$ = constant yields, at the end of solidification, precisely the same solute distribution as equation (2). The final composition (weight fraction or volume fraction solute) is exactly that of the original liquid and no macro-segregation results.

For the special cases where $4 f(g_T)$, and where the chill face is below the solidus temperature, equation (20) can be directly integrated:

$$g_L = \left( \frac{C_L}{C_0} \right) - \left( \frac{1}{1 - k} \right) \left( 1 + \frac{v_x G}{\varepsilon} \right)$$

(31)
D. Limiting Cases, Numerical Examples

Consider for numerical example of the foregoing, aluminum-4.5 per cent copper alloy, solidifying unidirectionally from a chill face. The appropriate portion of the phase diagram is shown in Figure 1a; Figure 1b shows schematically the liquid-solid "mushy" zone during solidification. Dendrite tips are at the liquidus temperature, \( T_L \), and liquid exists within the dendrite arms to the eutectic temperature \( T_E \) (Figure 1c). It should be noted that while dendrites are shown schematically, the following analysis is independent of dendrite morphology.

From the phase diagram, \( k = 0.172 \), maximum solid solubility at the eutectic is 5.65 per cent copper, and eutectic composition is 33 per cent copper, and assume \( \beta = \beta_e = 0.055 \), a constant. (\( \beta \) is not, in fact, a constant for this alloy; however, the assumed value of 0.055 results in calculated segregation in good agreement with that observed experimentally, as will be shown later.)

Using the foregoing data, the solute redistribution equations for each of the foregoing limiting cases are plotted in Figure 2; that for the third case is plotted for \( \frac{v_x C}{\epsilon} = \frac{108}{1 - \beta} \). In each case the final average composition at the location in question is given by the area under the curve drawn; this average composition decreases with increasing \( \frac{v_x C}{\epsilon} \).

This final average composition is readily determined graphically or by solution of equation (26). Figure 3 shows the general result
Figure 1a: Aluminum rich portion of aluminum-copper phase diagram.
Figure 1b: Schematic diagram of dendrites in liquid-solid mushy zone during solidification.

Figure 1c: Temperature distribution in unidirectional solidification.
Figure 2: Solute redistribution for three solidification cases
(a) \( \frac{v_x G}{\varepsilon} = 0 \), i.e., chill face solidification,
(b) \( \frac{v_x G}{\varepsilon} = \frac{\beta}{1 - \beta} \), i.e., steady state solidification,
(c) \( \frac{v_x G}{\varepsilon} = \frac{10 \beta}{1 - \beta} \), i.e., 10/1 change in solidification cross section.
Figure 3: Average composition versus $v \frac{C}{\epsilon} (1 - \frac{B}{\rho})$ for aluminum-4.5 per cent copper alloy.
of average composition $\bar{C}_S$ (and segregation, $\Delta C$) plotted against the dimensionless group of variables, $p = \frac{vG}{\epsilon} \frac{1 - \beta}{1 + \beta}$. For the case of the chill face, $p = 0$, for steady state, $p = 1$. In the general case, however, $p$ can have a wide range of positive or negative values with the marked segregation illustrated.

C. Macrosegregation Versus Distance from Chill

1. Description of Model, Constant $k$, $\beta$

We consider first the case of macrosegregation versus distance from the chill, assuming constant $k$ and $\beta$. Assume, as in Figure 4a, liquidus and non-equilibrium solidus (eutectic) isotherms move according to the general relations:

$$x_L = n_L t^b + a_L \quad (32a)$$

$$x_E = n_E t^b + a_E \quad (32b)$$

where $x_L$ and $x_E$ are the distance from the chill to the liquidus and eutectic isotherm respectively, $t$ is time and constants are as defined by Figure 4a. By proper choice of $a_L$ and $a_E$ we can readily use relations derived above for a variety of unidirectional solidification cases including (1) solidification, at constant rate and gradient (Figure 4b), (2) $x_L$ and $x_E$ proportional to $t^{1/2}$ as expected for unidirectional solidification against a chill wall with no mold-metal interface resistance (4c) and (3) $x_L$ and $x_E$ proportional to $t^{1/2}$ plus a constant; expected for unidirectional solidification against a chill wall with mold-metal interface resistance $^{38}$. 
Figure 4(a): Liquidus and eutectic isotherm movement, interface resistance case, $n_L + n_E$. 
Figure 4: Liquidus and eutectic isotherm movement: (b) steady state, n_L = n_E = constant, (c) no interface resistance, a_L = a_E = 0, (d) general case, a_L ≠ a_E, n_L ≠ n_E.
For analytic solution, it is now necessary to assume a distribution of fraction liquid, $g'_{L}$ versus $x'$ in the mushy zone. Reasonable alternatives include:

(1) A distribution as given by the steady state solute redistribution expression, equation (30). Such a distribution would clearly be correct for linear temperature distribution and small macrosegregation. This is sketched in Figure 5a.

(2) Linear $g'_{L}$ versus $x'$, assuming $g_{E}$ constant and given by equation (30) ($g_{E} = .0902$ for aluminum-4.5 per cent copper), Figure 5b.

(3) Linear $g'_{L}$ versus $x'$, assuming $g_{L} = 0$ at $x_{E}$, Figure 5c.

An analytic solution for segregation based on condition (1) above would be quite complicated. A solution based on (2) is much simpler, and is given in part, in Appendix B for locations $x > x_{c}$ where $x_{c}$ is defined in Figure 4a as the location of the liquidus when the chill face reaches the eutectic temperature. For small $g_{E}$, condition (2) reduces to condition (3) which is by far the simplest for analytical solutions. The equation for (3) is written:

$$g'_{L} = \frac{x' - x_{E}}{x_{L} - x_{E}}$$

(33)

Numerical examples in Appendix B show that for aluminum-4.5 per cent copper alloy, values of $\frac{v_{G}}{C}$ calculated from (3) vary by only a few per cent from those calculated from (2). Resulting calculated values of segregation are similar (within .02 per cent). Condition (3) is, therefore, used for the following.
Figure 5: Distribution of fraction liquid in the liquid-solid region assuming (a) steady state solute redistribution, (b) linear $g_L$ versus $x'$ assuming $g_E$ as given by steady state equation, (c) linear $g_L$ versus $x'$ assuming $g_E = 0$. 
2. Macrosegregation, $x > x_c$. 

Through manipulation of equations (22b), (32), and (33), an expression for $\frac{v_x G}{\epsilon}$ is readily obtainable (Appendix B) and is written:

$$\frac{v_x G}{\epsilon} = \frac{1}{2} \left( \frac{\beta}{1 - \beta} \right) \left( \frac{g_L n_q + 2}{g_L n_q + 1} \right) \tag{34}$$

where $n_q = \frac{n_L - n_E}{n_E}$

Combining equation (34) with (20) and integrating from $g_L = 1$ at $C_L = C_0$ to $g_L$, $C_L$ yields the "solute redistribution expression" comparable to equations (28), (30) and (31):

$$C_L = C_0 g_L \left( \frac{(1 - a)(1 - k)}{a} \right) \tag{35}$$

and macrosegregation is readily determined graphically or analytically from (35) and (24 - 26).

3. Macrosegregation at $x < x_c$. 

For locations $x < x_c$, a somewhat more complicated expression for $\frac{v_x G}{\epsilon}$ results than that of equation (34). This expression, derived in Appendix B is:

$$\frac{v_x G}{\epsilon} = \frac{x}{2} \frac{\beta}{1 - \beta} \left[ \frac{2(x - x_E) + g_L n_q (x - 2x_E)}{(x - x_E)^2 (q_L n_q + 1)} \right] \tag{36a}$$

where

$$x_E = \frac{x + g_L a_F (n_q - 1) + a_L g_L}{g_L n_q + 1}$$
And $C_L$ can now be determined as a function of $g_L$ by combining equations (36) and (20) and integrating between the limits $g_L = 1$ at $C_L = C_0$ and $g_L, C_L$ (Appendix B):

\[
C_L = C_0 \left( \frac{N + Mg_L}{N + M} \right) \frac{1 - k}{M}
\]

where

\[
N = \frac{8x (2P - xn_q)}{2p^2}
\]

\[
M = \frac{x80n_q + 2(1 - \beta)p^2}{2p^2}
\]

\[
P = n_q x - a_n q + a_E - a_L
\]

\[
Q = n_q x - 2a_n q + 2a_E - 2a_L
\]

It should be noted that $N, M, P$ and $Q$ are constants at any distance $0 < x' < x_c$.

Analytical solutions for solute redistribution between $x = 0$ to $x = x_L$ are obtained by solving equation (37) up to $g_L = g_L^c$ where $g_L^c$ is the volume fraction liquid at location $x_c$ when $x_E = 0$. After $g_L^c$ is reached equation (35) is used by integrating between the limits $g_L = g_L^c$ at $C_L = C_L^c$ and $C_L, g_L$

\[
C_L = C_L^c \left( \frac{g_L^c - (1 - k) g_L n a + 1}{g_L^c q + (1 - a)(1 - k) g_L q a + 1} \right)
\]

Consider macrosegregation in the last region to solidify of a unidirectional ingot of length L. Final composition in this region will, in general, be less than \( C_0 \) (for positive \( \beta \), \( k < 1 \)) as a result of draining away of solute-rich feed metal to feed shrinkage.

When the dendrite tips reach the ingot end \( (x_L = L) \), at time \( t = t_v \) subsequent solidification drains liquid back from the tips, voids result since no further supply of feed metal exists. The location \( x_v \), of the end of the liquid at time \( t_v \) is:

\[
L - x_v = -\int_{t^0}^{t_v} v_x \, dt
\]

From which, by appropriate substitution, equation (39) can be solved to determine liquid fraction, \( g_L^v \) at \( x_v \). This liquid fraction will all be drained from the volume element at \( x_v \), leaving a void of volume \( 1 - g_L^v \). Finally, local solute redistribution in the ingot at \( x_v \) is the same for \( g_L^v > g_L^v \) as in cases previously considered, but final composition is lower because of the voids present.

For the special case of \( n_q = 0 \), the critical value, \( g_L^v \) at \( x_v \), needed to calculate macrosegregation (from Appendix C):

\[
g_L^v = 1 - \frac{L - x_v}{(\beta)(x_L - x_E)}
\]

Some numerical examples of the foregoing are considered here, for unidirectionally solidified aluminum-4.5 per cent copper using the same value for constants as in previous examples.

(a) Constant Thermal Gradient.

Figure 6 shows composition versus distance from the chill for constant thermal gradient \( (n_q = 0) \), assuming a total length of mushy zone at steady state of 0.2 \( L \) where \( L \) is ingot length. Note the initial transient (where alloy composition is above \( C \)) is 0.2\( L \) in length. At the chill face, the maximum composition calculated earlier is reached (4.92 per cent copper). The final transient, resulting from depletion of feed metal, is shown on an expanded scale. In later graphs this zone is omitted for clarity.

(b) Thermal Gradient Decreasing with Time.

Figure 7 shows composition versus distance curves for several cases of varying \( n_q \). Curve 1 shows segregation for \( a_L \) and \( a_E = 0 \), \( n_q = 0.52 \). (These are approximate values to be expected for aluminum-4.5 per cent copper alloy solidifying against a metal chill, no interface resistance as sketched in Figure 4c.) Note that overall composition of the ingot is perceptibly higher than \( C_0 \) (approximately 4.53 as compared with \( C_0 = 4.50 \)) and that no solute rich "initial transient" zone exists.
Figure 6: Composition versus distance from chill wall for constant thermal gradient \( n_q = 0 \). Length of mushy zone at steady state is 0.2L.
Figure 7: Composition versus distance from chill wall for varying thermal conditions: (1) $a_L = a_E = 0$, $n_q = .52$,
(2) $a_L = 0$, $a_E = -2$, $n_q = .52$, (3) $a_L = a_E = 0$, $n_q = \infty$,
(4) $a_L = 0$, $a_E = -\infty$, $n_q = .52$. 
Curve (2) is identical to curve 1 except that reasonable values are employed for $a_E$ and $a_L$ for the case of moderate mold-metal interface resistance to heat transfer ($a_L = 0, a_E = -2$). As in the constant thermal gradient case surface composition again rises to the maximum chill face composition ($4.92$ per cent copper). Composition of the bulk ingot is unchanged ($4.53$ per cent copper).

It is possible to envision two other simple thermal cases. In one, excessively steep thermal gradients are maintained so $n_q \rightarrow 0$, $a_E \rightarrow 0$, $a_L \rightarrow 0$; here composition is maintained at exactly $C_0$ throughout. In a second, $a_E$ and $a_L$ remain zero so the chill face is brought instantaneously to the eutectic temperature $T_E$. Thereafter the chill face is held at $T_E$ while the liquidus isotherm moves with a velocity proportional to $t^b$ (i.e., $n_q \rightarrow \infty$). In this case, equation (24) becomes:

$$\frac{v_x G}{\epsilon} = \frac{1}{2} \left( \frac{\beta}{1 - \beta} \right)$$

and the solute redistribution equation is written as:

$$C_L = C_0 (g_L)^{\frac{1 - k}{a}} \quad \text{where} \quad a = 1 - \frac{\beta}{2}$$

Macrosegregation calculated from the solute redistribution expression (42) is shown as curve 3, Figure 7.

Finally, for the case of solidification under extremely shallow gradients ($a_E = \infty$), the limit of segregation calculated using equations
(36a) and (36b) is precisely the same as that at the chill face (Figure 4, curve 4). This is most easily seen by inspection of equation (20). At shallow gradient, and finite cooling rate, the term $\frac{vGx}{\varepsilon}$ approaches zero, as at a chill face.

(c) Variable Thermal Gradients.

Consider unidirectional solidification with movement of isotherms as sketched in Figure 8. Resulting composition is shown in Figure 9. Region A is the initial transient, and final composition is similar to that shown in Figure 6. In Regions B and F, $n_q = 0$ and final composition is $C_0$. Region D is drawn with $n_q = 2.5$ with the resulting increase in composition (see Figure 10). Regions C and E are transient zones between B–D and D–F respectively. The compositional variations in these two transient regions are depicted schematically.

Macrosegregation in regions B, D, and F is readily calculated from the solute redistribution equation (35) shown as Figure 10. Figure 9 shows the final macrosegregation in the ingot. Note the positive segregation (solute rich "band") at region C resulting from decreasing thermal gradient. Similarly an increasing thermal gradient will result in negative segregation (solute poor "band").

(d) Variable Ingot Cross Section.

Consider an ingot of design shown in Figure 11 solidifying unidirectionally upwards with constant thermal gradient for liquid–solid region is of length 0.2L for $x_E > 0$. The ingot is reduced, on an area basis 10/1 at 0.5L.
Figure 8: Isotherm movement for a unidirectionally solidified ingot.
Figure 9: Composition versus distance from the chill wall for ingot with isotherm movement as shown in Figure 8.
Figure 11: Sketch of reduced cross section ingot.

CHILL
Final solid composition, Figure 12, is as in the comparable constant cross section ingot (Figure 6) up to \( x/L = 0.5 \). At \( x/L = 0.5 \), \( v_x \), and hence \( \frac{v_x}{c} \) increases discontinuously by a factor of 10 (see equation 23). Hence, solid composition at \( x/L = 0.5 \) is given directly by Figure 3 for \( \frac{v_x}{c} \frac{1 - \beta}{\beta} = 10 \) (\( \bar{c}_S = 2.3 \) per cent copper).

For the transient zone, \( 0.5 < x/L < 0.7 \) a region exists where local solute redistribution is expressed, from equation (20), for \( g_L > g_L^c \) where \( \frac{v_x}{c} \frac{1 - \beta}{\beta} = 10 \) by:

\[
g_L = \left( \frac{c_L}{c_0} \right) - \frac{1 + 9\beta}{1 - k} \tag{43a}
\]

For \( g_L < g_L^c, \ c_L > c_L^c, \frac{v_x}{c} \frac{1 - \beta}{\beta} = 1 \) and equation (20) in the integrated form becomes

\[
g_L = g_L^c \left( \frac{c_L}{c_L^c} \right) - \frac{1}{1 - k} \tag{43b}
\]

For \( 0.7 < x/L < 1.0 \) final composition in the reduced cross section ingot, Figure 12, is again identical to that of the comparable constant cross section case, Figure 6, where \( \frac{v_x}{c} (1 - \beta) = 1 \).

A second type of ingot of variable cross section is one in which cross section varies continuously. Consider, for example, the prism of Figure 13 in which

\[
A' = \left( 1 - \frac{x'}{L} \right) A_0 \tag{44}
\]
Figure 12: Composition versus distance from chill for ingot shown in Figure 11.
Figure 13: Sketch of prism ingot.
where:  \( A' \) = cross sectional area at \( x' \)
\( A_0 \) = cross section area at chill face

Again assuming constant temperature gradient, \( \frac{v_x G}{\epsilon} \) is calculated from equation (23) which, for the assumptions used herein reduces to, for \( x > x_c \) (Appendix D)

\[
\frac{v_x G}{\epsilon} = \frac{\beta}{1 - \beta} \left( \frac{1}{L - x} \right) \left[ \frac{2L(x_L - x_E) + 2x_a T + a_T^2 g_L}{2(x_L - x_E)} \right]
\]  

(45)

where \( a_T = a_L - a_E \)

Local solute distribution and macrosegregation are then calculated as before by integrating (20) after substituting (45) and then solving (26).

For \( x < x_c \) equation (45) becomes

\[
\frac{v_x G}{\epsilon} = \frac{\beta}{1 - \beta} \left( \frac{1}{L - x} \right) \left[ -\frac{x(2L - x)}{2R_L} \right]
\]

(45b)

and macrosegregation, is determined by a procedure similar to that for the ingot of constant cross section (equations 36 - 38).

For \( g_L > g_C^C \), \( C_L < C_C^C \) equation (45b) is substituted in (20) and integrated to solve for solute redistribution. When \( g_L < g_C^C \), \( C_L > C_C^C \) equation (45a) is used in (20) and integrated between appropriate limits. The results are plotted as Figure 14.
Figure 14: Composition versus distance from chill wall for prism ingot.
(e) Variable $k$, $\beta$.

To permit analytic solutions, the foregoing calculations have been made assuming constant $k$ and constant $\beta$. In aluminum-4.5 per cent copper these factors are, however, not constant. Figure 15 shows $k$ versus $C_L$; Figure 16 summarizes best available data on $\rho_S$ and $\rho_L$ versus $C_L$; Figure 17 shows $\beta$ versus $C_L$, calculated from the data of Figure 15. Note that in this alloy, $\beta$ is negative during much of solidification (negative $\beta$ denotes expansion on solidification).

However, a net flow of feed metal, negative $v_\chi$, still results throughout steady state solidification, because of thermal contraction of the liquid (i.e., the last term on the right of equation (12) is sufficiently large (negative) that $v_\chi$ is negative even though the first term on the right is positive).

Figure 18 shows local solute redistribution at the chill face for aluminum-4.5 per cent copper, determined using the data of Figures 15 - 17 by graphical integration of equation (20). For comparison with the case of constant $\beta$ (and hence constant $\rho_S$, $\rho_L$), the data are plotted as weight fraction solid, $f_S$, where:

$$f_S = \frac{\rho_S g_L}{\rho_S (1 - g_E) - \rho_S g_E}$$

(44)

where: $\rho_S$ and $\rho_E$ = densities of the $\alpha$ phase and eutectic respectively (at just below $T_E$)

$g_E$ = volume fraction eutectic
Figure 15: Equilibrium partition ratio versus liquid composition for aluminum rich, aluminum-copper alloy.41
Figure 16: Liquid density ($\rho_L$) for aluminum rich aluminum-copper alloys and solid density ($\rho_s$) for an aluminum-4.5 per cent copper alloy.
Figure 17: Solidification shrinkage for aluminum-4.5 per cent copper alloy.
Figure 18: Steady state solute redistribution for two cases: (a) constant, $k$, $\beta$ direct calculation of weight = volume fraction solid, and (b) variable $k$, $\beta$ - calculation of volume fraction, from density data Figure 16 conversion to weight fraction.
Graphical integration of curve (b), Figure 18, gives a final chill face composition of 5.03 per cent. This curve, based on best available data for alloy constants, yields a final value of chill face composition close to that calculated by Scheil using slightly different constants. The curve (a) in Figure 18 was calculated with constant \( k (k = 0.172) \), and constant \( \beta (\beta = 0.055) \), with the value of \( \beta \) chosen so that final chill surface composition is that calculated by Scheil (4.93). Note the similarity of curves (a) and (b), making reasonable the approximation used throughout much of this work, of constant \( \beta \).

PART III: NON-UNIDIRECTIONAL HEAT FLOW

A. Introduction

The foregoing has been concerned only with cases where isotherms could be assumed planar, and heat flow unidirectional from the metal to a single flat mold wall. In this section we examine macrosegregation in castings or ingots in which heat flow is not unidirectional. Types of segregation include (1) centerline segregation, (2) negative "cone" of segregation, (3) positive segregation beneath risers (hot tops), and (4) others. It is assumed in most of the following, as before, that feeding is complete (no voids form during solidification). This however, is expected to be an increasingly poor assumption as thermal gradients decrease.

Figure 19 shows schematically solidification of an alloy in a metal mold. In Figure 19a, a fully liquid region remains at the ingot
center and in Figure 19b, at a later stage of solidification, this fully liquid zone has disappeared. It will be shown below that some centerline segregation will, in general, result from flow to feed shrinkage during and after the time of solidification sketched in Figure 19b. Also it will be shown that flow to feed shrinkage throughout most of solidification must result in negative segregation near the ingot base (negative "cone" of segregation).

Figure 20 is a sketch of a plate-type sand casting which may be made with, or without, a metal chill at the end of the plate opposite the riser. Figure 21 shows schematically the solidification of aluminum-4.5 per cent copper alloy in a 3/4-inch thick plate of this geometry. Temperature differences throughout most of the plate during solidification are small compared with the freezing range of the alloy, and so solidification proceeds in "mushy" manner as shown in Figure 21a. When a chill is placed at the end opposite the riser, longitudinal thermal gradients are introduced in the metal and solidification proceeds directionally towards the riser as sketched in Figure 21b. The two different types of solidification result in two distinctly different patterns of macrosegregation as discussed below.

As a final example of macrosegregation in non-unidirectional solidification, the case of segregation under a riser ("hot top") of insufficient size is considered. As example we will examine a cube, risered with a tall riser of square cross section, which is smaller than that of the cube, Figure 22. The small riser cools through the
Figure 19a: Sketch of ingot solidification in a metal mold showing liquid, liquid plus solid, and fully solid regions.
Figure 19b: Sketch of ingot solidification in a metal mold during late stages of solidification, after liquidus isotherms have met at the casting centerline.
Figure 21: Schematic diagram of equiaxed solidification of sand cast plate. Left side, unchilled plate; right side end-chilled plate. Plate 3/4" thick: (a) 1 minute after casting, (b) 4 minutes after casting, (c) 7 minutes after casting.
Figure 22: Top risered cube casting.
solidification range somewhat more rapidly, than the casting and therefore contains somewhat more solid during most of solidifica-
tion than does the casting. It will be shown below that this results in positive macrosegregation under the riser.

B. Centerline Segregation

1. No Contraction Strain.

Consider solidification from two flat parallel faces, so isotherms are planar and meet at the centerline, Figure 19. Before the centerline reaches the liquidus temperature, as in Figure 19a, liquid flow is expected to occur in the y vertical direction only in the region of no solid (region between the dendrite tips and the centerline) within the liquid-solid region, \( v_y = v_z = 0 \), and \( v_x \) is as in steady state unidirectional solidification.

When the liquidus isotherm reaches the centerline, Figure 19b, flow within the liquid-solid region must occur in the y direction. Three possible flow paths, with resulting segregation, will now be considered. In each, for simplicity, assumptions employed will include constant \( \beta, k, G, r \), and linear distribution of solid in the liquid-solid zone. Alloy used as example is again aluminum-4.5 per cent copper; \( n_q \) is assumed equal to 0 and at steady state the thickness of the mushy zone is \( 0.2L \), where \( L \) is half thickness of this ingot.

(a) \( v_y = 0 \) within the liquid-solid zone except at \( x' = L \). Here, all flow in the y direction is in a layer of vanishingly small thickness at the centerline. Hence, \( v_x \) at all locations up to \( x' = L \) is exactly
as in steady state solidification, and $G/c$ is constant. Final local solute content is as in steady state solidification and no macrosegregation results at the centerline, curve (1), Figure 23.

(b) $v_y = 0$ at $x_L < L; v_x = 0$ after $x_L = L$. Here, the instant the liquidus isotherms reach the centerline, flow in the $x$ direction is assumed to cease. If at this instant the fraction liquid at $x'$ is $g^c_L$, then it is clear that for $g_L > g^c_L$, solute redistribution is as in steady state solidification, and for $g_L < g^c_L$, solute redistribution is at a chill face. Hence, local solute redistribution is calculated from:

\[
g_L = \frac{c_L}{c_0} - \frac{1}{1 - k}\quad \text{for } g_L > g^c_L \quad (45a)
\]

\[
g_L = g^c_L \left( \frac{c_L}{c_C} \right) - \frac{1 - g}{1 - k}\quad \text{for } g_L < g^c_L \quad (45b)
\]

and segregation is calculated as before, and shown as curve (2), Figure 23.

(c) $v_y = 0$ at $x_L < L; v_y = f(g_L)$ for $x_L > L$. While cases (1) and (2) represent limits on centerline segregation which can occur in the examples chosen, $v_y$, and hence $v_x$ and segregation can generally be expected to lie between the extremes chosen. For example when $v_y = f(g_L)$ solute content in the final transient region will at all locations lie between case (1) and case (2). At the beginning of the transient region composition will be $C_0$ while at the centerline it is the same composition as case (2) where no flow has occurred in the $x$
Figure 23: Final composition versus distance for two cases of ingot mold solidification. (a) Steady state $v_y = 0$, within the liquid-solid zone except at $x' = L$. (b) Steady state $v_y = 0$ at $x_L < L$; $v_x = 0$ after $x_L = L$. 

$$\frac{v_x G}{\varepsilon} = \frac{\beta}{1-\beta} \quad \text{AT} \quad \frac{x}{L} > 0.2$$

$$v_x = 0 \quad \text{AT} \quad \frac{x}{L} = 0, \quad \frac{x}{L} = 1$$
direction. The shape of the curve in between these two points, however, is determined by the dependence of \( v_y \) on \( g_L \). Solution requires detailed consideration of fluid flow within the porous medium.

2. Contraction Strain.

In the foregoing, movement of solid during solidification has not been considered. Such movement might occur by (a) floatation or settling, (b) contraction or other strain. Contraction strains can result in internal fissuring and/or "filled hot tears" at any location, \( x, y, z \) in an ingot; one example of this is at the ingot centerline, considered below.

Consider the case examined in the preceding section of solidification with planar isotherms from two flat parallel faces meeting at a centerline as shown in Figure 19; except that "fissuring" will occur from thermal contraction of the already solidified metal, at the casting centerline, \( x' = L \). When the centerline just reaches the liquidus temperature, solid begins to form and is drawn away from the centerline at a velocity \( v \) from thermal contraction; hence, the "volume element" associated with this solid is also drawn away at velocity \( v' \). At subsequent temperatures \( T \) below the liquidus, other volume elements form at the centerline, in liquid of composition \( C_L^0 > C_0 \) (for \( k < 1 \)).

For the volume element which forms at the centerline at temperature \( T \) below the liquidus (when the liquid composition at the centerline is
raised to $C_L^0$), equation (20) is written:

$$\frac{g_L}{g_L} \frac{dg_L}{g_L} = - \int \frac{C_L}{C_L^0} (1 - \beta)(1 + \frac{v_G}{C_L}) \frac{dC_L}{C_L}$$  \hspace{1cm} (46)$$

The final position of the volume element at the end of solidification ($t_L + t_f$) is:

$$\Delta L = \int_{t_L}^{t_L + t_f} v'dt$$  \hspace{1cm} (47)$$

where: $t_L$ and $t_f$ are times for initiation and completion of solidification at x

$$\Delta L = \text{distance from casting centerline}$$

If the only strain is from thermal contraction and the strain is small:

$$x = x_E$$

$$v = \int Tdx'$$  \hspace{1cm} (48)$$

$$x = 0$$

where: $T = \text{temperature at x}$

$$\alpha = \text{linear coefficient of thermal contraction at T}$$

Qualitative conclusions from the foregoing are that thermal contraction can cause segregation at the centerline (or elsewhere) in the solidifying ingot. Segregation resulting from this contraction is additive to that resulting from solidification contraction.
Equations (46) and (48) can be used to evaluate thickness of segregated zone, and $g_L$ and $C_L$ at any location within the zone during solidification. Hence, final segregation at the centerline is readily calculated as above.

Consider, for example, the unidirectionally solidification aluminum-4.5 per cent copper alloy used as example above, constant $k$, $\beta$, $G$, $e$ linear distribution of solid in the liquid-solid zone (with $z$). Assume near the region in question fluid flow (to feed thermal contraction and shrinkage) occurs only by (a) flow down the ingot centerline, parallel to isotherms, and (b) flow perpendicular to isotherms at locations other than the centerline.

Hence, $\frac{vG}{\epsilon} = \frac{\beta}{1 - \beta}$ so equation (31) becomes:

$$g_L = \left( \frac{C^o_L}{C_L} \right) \frac{1}{1 - k}$$

Figure 24 plots $C_S$ versus $g_S$ for three values of $C^o_L$: (a) $C^o_L = C_0 = 4.5$ per cent copper, (b) $C^o_L = 20$ per cent copper, and (c) $C^o_L = 33$ per cent copper.

Assume now thermal contraction was such that each volume element moved .001 cm/per cent increase in solute at the centerline. Hence, a segregated centerline results of thickness $(.001)(33) = .033$ cm in which the average composition varies linearly from 4.5 per cent copper to 33 per cent, Figure 25.
Figure 24: Solid composition, $\bar{C}_L$, versus volume fraction solid, $g_s$, for three values of $C_L^*$: (a) $C_L^* = C_0 = 4.5$ per cent copper, (b) $C_L^* = 20$ per cent copper, (c) $C_L^* = 33$ per cent copper.
Figure 25: Centerline segregation resulting from thermal contraction of .001 cm/per cent increase in solute at the centerline.
C. Negative "Cone" of Segregation at Ingot Base

Consider solidification of the bottom portion of the ingot illustrated in Figure 19. Assume as limiting case that

(1) at the periods of solidification represented by Figure 19a, feed metal flows into the bottom "cone" (represented by the dendrites growing vertically) only from the fully liquid region, and

(2) after the dendrites growing from the sidewalls reach the centerline, flow into the "cone" is all through a small channel at the apex of the cone.

Now, assume an ingot of long rectangular cross section (long dimension perpendicular the planr of the paper). The "cone" then is a prism, and hereafter referred to as such. Moreover, fluid flow into this prism is exactly as in the prism ingot discussed earlier (Figure 13) and final composition within the prism is expected to be as given in Figure 14. Note that compositions particularly near the apex, are substantially below nominal, in qualitative agreement with experiments on large ingots which show what is commonly termed the "negative cone" of segregation.

D. Macrosegregation in Sand Cast Plates

In sand castings temperature gradients within the solidifying metal are generally small, because of the low heat diffusivity of the mold compared with that of the metal. Where these are negligible, and where feeding is complete, the term $\frac{vG}{\varepsilon}$ in equation (20) is zero.
For the plate shown in Figure 20 solidification is sketched in Figure 21. Significant macrosegregation is expected through the plate, approaching the maximum expected at a chill face in unidirectional solidification (as temperature gradient in the longitudinal direction goes to zero, assuming complete feeding). In the region near the riser, some thermal gradients are expected, and hence, some reduction in final composition in this area is also expected, Figure 26 illustrates this schematically for aluminum-4.5 per cent copper alloy.

In the end chilled plate, Figure 21b, steep longitudinal gradients are introduced by the chill, and although some heat is lost to the cold sand mold, isotherms are expected to be essentially planar, and perpendicular to the longitudinal (x') direction. Since the isotherms are assumed planar, relations developed in earlier sections for "unidirectional solidification" apply here as well. Movement of the isotherms, however, is expected to depend on plate geometry and heat diffusivity of the mold as well as on rate of heat removal through the chill. Figure 26 illustrates composition distribution expected assuming (1) movement of isotherms throughout most of the plate is as in unidirectional solidification from a metal chill, no interface resistance, but (2) isotherms approach each other near the riser end of the plate.
Figure 26: Schematic diagram of composition versus distance in a sand cast plate and a chill, sand cast plate.
E. **Under Riser Segregation Positive Segregation Near "Hot Top"**

For one example of under riser segregation consider the riservertical sand cast cube of Figure 22. Here, the riser chosen is inadequate to feed the cube and at any time during solidification the metal in the riser area will contain more solid than the metal in the casting, Figure 27.

Thermal gradients are assumed small in the horizontal (y and z) directions of casting and riser. The thermal gradient is also assumed small in the x direction except in the region between the casting and riser. To illustrate the problem simply, assume temperature difference between bulk riser and bulk casting are constant during solidification, and the temperature gradient in the vicinity of the casting and riser junction is constant.

Consider first solidification in the bulk riser, well away from the casting riser junction; here, $G = 0$. Since $v_x$ and $v$ are finite $\frac{v_x G}{\varepsilon} = 0$ and maximum segregation, as at a chill face, is expected (provided no pores form), except in the region at the top of the riser where negative segregation occurs as a result of liquid depletion.

Similarly, while feeding occurs, solute redistribution in the bulk casting must be as at a chill face (at locations well away from the casting-riser junction). However, at some critical fraction solid, $g^c_s$, which must occur not later than when the riser reaches $T_E$, its non-equilibrium solidus, all feeding is expected to cease. Final composition in the bulk casting is therefore expected to be less than
Figure 27: Sketch of solidification for cube casting of Figure 22.
the chill face maximum, by an amount determined by the time during solidification when feeding ceases. This composition will however, be at least slightly greater than $C_0$.

In the region at, or just below, the riser-casting junction, we expect composition to be significantly higher than in the bulk casting. This is because, as long as flow occurs in this region, it is flowing up a temperature gradient; hence $\frac{vG}{\epsilon}$ is positive and compositions in excess of the chill face maximum are possible. Compositions in excess of the chill face maximum must result as long as there is a net flow of solute up a temperature gradient, and this flow continues to the last stages of solidification.
III. EXPERIMENTAL RESULTS AND DISCUSSION

In this section, results are presented from experiments designed to illustrate and substantiate certain aspects of the general macrosegregation expression previously developed. In particular, consideration is given to the parameter, \( \frac{v_x G}{\epsilon} \) in equation (20).

\[
\frac{d g_L}{g_L} = - \left( \frac{1 - \beta}{1 - k} \right) \left( 1 + \frac{v_x G}{\epsilon} \right) \frac{d C_L}{C_L}
\]  \hspace{1cm} (20)

This solute redistribution expression is for the case of unidirectional heat flow, but where fluid flow is not necessarily unidirectional.

As previously discussed for a particular alloy system and composition, values can be assigned which effectively average \( k \) and \( \beta \) over the complete solidification range, even though they are usually functions of \( C_L \). Using the values \( \beta = .055 \) and \( k = .172 \) as the constants for this particular alloy composition, consideration can then be given to the effect of solidification variables on the parameter \( \frac{v_x G}{\epsilon} \). The following experimental results are presented to show how variations in this parameter can cause compositional changes throughout a casting or ingot.

A. Unidirectional Heat and Fluid Flow

1. Constant Cross Section.

Consideration is first given to experimental results on four ingots of constant cross section where both heat and fluid flow are
unidirectional. The ingots were made using the experimental set up described in Appendix E. Ingots 1 – 3 are of square cross section 7.5 cm per side, 20 cm high. Ingot number 4 is circular 11.5 cm in diameter and 20 cm high.

(a) Ingot Number 1.

Ingot number 1 was cast and solidified on a water cooled chill. Furnace temperature was maintained high enough to obtain a completely columnar casting structure. Cooling curves for the ingot are shown in Figure 28. From these curves Figure 29 was constructed, showing liquidus and eutectic isotherm movement as a function of time. It is interesting to note that even though a water cooled chill was used, after 720 seconds the mushy zone extends the entire length of the casting. That is, at the same time the metal at the chill first becomes completely solid, initial solid is forming at the top. This means that at least the initial stages of solidification, at all locations, is under the initial transient conditions previously considered (i.e., \( x < x_c \)).

Calculation of composition up the length of the ingot using only thermal data was done as described in Appendix G. Agreement between this and the experimentally determined curve (using fluorescent analysis for chemical analysis, see Appendix F) is shown in Figure 30.

Calculated agreement becomes poor at distances greater than 10 centimeters from the chill. It can be seen from the thermal data of Figure 28 that at this location gradients become shallow (typically
Figure 28: Cooling curves for Ingot No. 1 at locations 1, 2, 3, 4, 5 and 7 inches from the chill.
Figure 30: Experimental and calculated (from thermal data) solid composition versus distance from chill in Ingot No. 1.
2 - 5°C/cm), a condition which can lead to porosity and therefore lower composition.

(b) Ingot Number 2.

Ingot number 2 was also cast and solidified against a water cooled chill. Furnace temperature was lower than for the preceding ingot, and resulted in an equiaxed ingot structure, even though heat flow was unidirectional. Insufficient thermal data is available for complete analysis, but liquidus and solidus isotherm movements were recorded. An initial transient of about 5 centimeters (Figure 31) exists which is seen from the increased composition at locations less than 5 cm, Figure 32.

In the region after the initial transient, \( n_q \) (equation 34), which reflects the relationship between isotherms, changes very little. At locations greater than 5 cm, \( n_q \) varies from .8 to 1.0. Figure 10 shows that within this range of \( n_q \) values composition is not expected to vary more than .03 per cent and within experimental limits should be relatively constant. This is the case in the region from 5 to about 15 cm from the chill where composition is about 4.68 per cent copper. It is also expected (Figure 10) that within this region (i.e., 5 - 15 cm, \( n_q = .8 - 1.0 \)) average composition will be slightly greater than \( C_0 \). For this ingot \( C_0 \) is 4.60 per cent copper, less than the composition in the 5 - 15 cm range (4.68 per cent). Toward the top of casting (greater than 15 cm) composition begins to drop off since solute must be conserved.
Figure 31: Liquidus and eutectic isotherm movement for InRot No. 2.
Figure 32: Solid composition versus distance from the chill in Ingot No. 2.
(c) Ingot Number 3.

Ingot number 3 was cast and solidified against a chill which had both air and water for a coolant. Figure 33 presents the experimental and calculated (from thermal data) composition curves. At locations greater than 15 centimeters where gradients are shallow and porosity forms, agreement between experimental and calculated curves breaks down. The general method used for calculating the curve is outlined in Appendix G.

(d) Ingot Number 4.

Ingot number 4 was cast against a chill and air cooled to start solidification. After the liquidus had extended 2.2 - 2.5 cm into the casting, before the liquid at the chill reached the eutectic temperature, the air was replaced by a water coolant. The resulting thermal curves for some typical isotherms are shown in Figure 34. Resulting composition near the chill face is shown in Figure 35.

When comparison is made between the liquidus and eutectic isotherms the initial transient \((x < x_c)\) extends almost 4 cm into the casting. This precludes quantitative comparison in this region of the liquidus and eutectic isotherm slopes to obtain \(n_q\) because use of equation (34) requires \(x > x_c\). Also, complete analysis as in ingots 1 and 3 was not possible due to insufficient thermal data in the critical region. However a qualitative comparison can be made between the liquidus isotherm and an isotherm less than 100 per cent solid (eutectic). For example consider the liquidus and 30 per cent solid isotherms in
Figure 31: Experimental and calculated (from thermal data) solid composition versus distance from chill in Ingot No. 3.
Figure 34: Thermal curves for some typical isotherms in Ingot No. 4.
Figure 35: Solid composition versus distance from the chill in Ingot No. 4.
Figure 34. We can define \( n_{qx} \) as:

\[
n_{qx} = \frac{n_L - n_x}{n_x}
\]

where \( n_x \) = slope of an isotherm movement line where \( 0 < x < 100 \) per cent

Now \( n_{qx} \) can be calculated which will give a qualitative indication of whether compositions should be greater or less than \( C_0 \) even though quantitative results are not available. In the region where \( x \) is less than 2.5 cm comparison between the liquidus and the 30 per cent solid isotherm slopes gives \( n_{qx} \) as -.9. The maximum negative deviation possible between two isotherms is -1, therefore significant negative departure from \( C_0 \) is expected in the region up to 2.5 cm. After \( x = 2.5 \) cm the relationship between the liquidus and all other slopes becomes about the same; that is, \( n_{qx} \) becomes constant which predicts a constant composition (Figure 35).

2. Reduced Cross Section.

The next group of castings were reduced in cross section as shown in the schematic diagram, Figure 11. While the castings will be considered again in a later section, attention in this portion is directed only at the effect of the reduction on composition through the central portion of the ingot. That is consideration is given to a section 3/4 inch by 3/4 inch extending from the chill surface to the top of the casting through the geometric center.
Figure 36 shows the composition variations up the length of an ingot in the center section where area reduction is 9/1. As expected there is a sharp minimum at the intersection of the large and small sections (see theoretical section, Figure 12, variable ingot cross section). To the left of the intersection there is no change from \( \Delta \) to zero slope as in the theoretical case (Figure 12). This is because the theoretical model assumes constant \( v_x \) at any location \( y \) from the chill. In the actual ingot, just below the reduction, \( v_x \) will be greatest at the center of the ingot (i.e. absolute value) and decrease toward the edges. This will be considered further in a later ingot, where extensive thermal data was available.

An experiment was next conducted similar to the previous, but with the area change 2/1; results are presented in Figure 37. For this ingot, \( P \) equals 2 (Figure 3) with a predicted minimum composition at the intersection about 4.1 per cent copper. Experimental value is 4.16 per cent. The general shape of the composition curve is similar to that of the previous figure, except for a secondary minimum about 12 cm. This may be related to a sudden velocity increase at this location when the final liquid solidifies just below the intersection. This, however, would require verification by thermal analysis and has not been done.

The final ingot presented in this section is a repeat of the 9/1 reduction ingot made with thermocouples cast in place so that experimental and theoretical curves could be compared with curves calculated from the thermal data. Figure 38 shows the measured
Figure 36: Solid composition versus distance from the chill in an ingot of 9 to 1 reduction in area (no thermal data).
Figure 37: Solid composition versus distance from the chill in an ingot of 2 to 1 reduction in area.
composition using fluorescent analysis. Figure 39 shows the theoretical curve for a 4.5 per cent alloy solidifying with a mushy zone 0.2L at steady state. This curve had previously been presented in Figure 12. The curve as presented in Figure 39b was calculated using thermal data from the actual ingot, as described in Appendix G. The sharp breaks at the reduction in area of the theoretical and calculated curves are due to the assumption that \( v_x \) is constant at a given time over the entire \( y-z \) plane at a given distance \( x \) from the chill. This requires discontinuous change in \( v_x \) at the reduction. In the actual case this is not to be expected since it would require, just below the reduction, that feed metal flow in a thin layer across the top of the large section and then down, so that \( v_y \) and \( v_z \) equal zero at all locations except precisely at the junction. Experimental data from this ingot, presented in Figure 40, show that final solid composition just below the junction is much lower near the center of the casting than it is toward the edges. This clearly results from finite \( v_y \) and \( v_z \), and is discussed in a later section.

As a final example of the effect of section reduction on composition, partial results of an ingot with metal plates to divert fluid flow are presented in Figure 41. The plates are spaced so they extend half the width into the casting (total width 7.5 cm). The side of the casting with the plates will be considered in a following section. On the side with no plates, the minimum composition occurs at approximately 4 cm from the chill. The 2/1 reduction in area from the inserted plates occurs at 5 cm. That the expected minimum occurs below the reduction seems possible since from fluid flow considerations
Figure 38: Solid composition versus distance from the chill in an ingot of 9 to 1 reduction in area (thermal data, Appendix G).
Figure 39: Composition versus distance from chill for reduced section ingot. (a) Theoretical - $C_0 = 4.5$ per cent copper, 10/1 reduction. (b) Calculation from experimental data $C_0 = 4.68$ per cent copper, 9/1 reduction.
Figure 40: Solid composition versus distance from the chill at two locations in the bottom portion of an ingot of 9 to 1 reduction in area.
Figure 41: Solid composition versus distance from the chill in an ingot with horizontal plates (as sketched) extending half the width into casting. Composition determined on side without plates.
the plates may act similar to an orifice with the maximum in velocity occurring slightly downstream from the reduction.

The 2/1 reduction is the same as in one of the reduced section ingots previously discussed, Figure 37. In fact the shapes of the two composition curves are very similar.

If comparison is made of Figures 36, 37, and 41 it can be seen, as expected from Figure 3, that the minimum composition is a function of the area reduction.

B. Unidirectional and Bi-Directional Heat Flow; Non-Unidirectional Fluid Flow

In the theoretical section \( v_x \) was discussed as a part of the \( \frac{v_x G}{\Delta} \) parameter, \( \frac{v_x G}{\Delta} \). When \( v_x = 0 \) the general macrosegregation equation (20) reduces to that for solidification at a chill face. This is the maximum composition obtainable in unidirectional solidification (in the absence of melting and solidification) and for an aluminum-4.5 per cent copper alloy is 4.92 per cent copper. The minimum composition is \( kC_0 = .77 \) per cent copper for 4.5 per cent = \( C_0 \) which will be approached as \( v_x \) approaches minus infinity. For solidification where both heat and fluid flow are non-directional \( v_x \) can become positive and local compositions greater than 4.92 per cent copper are possible. In general, for solidification where heat flow is unidirectional and no melting and resolidification occurs, as \( v_x \) increases negatively from zero, local average composition will decrease.

For unidirectional solidification \( v_x \) refers to that component of fluid flow which is perpendicular to isotherms (parallel to heat flow). In this section three castings will be considered in which \( v_x \) values and the resulting compositions varied considerably.

The first experimental ingot considered is shown in Figure 42. The "ingot" comprises the bottom portion of the mold cavity (3.8 cm, by 15 cm by 10 cm high). The hot top is as shown, feeding through a neck at one end of the ingot. Ingot, neck, and hot top are "perfectly" insulated so all heat flow is downwards, out through the water-cooled base plate.

The ingot is designed to simulate heat and fluid flow in solidification of a slice of a large ingot. Specifically, the base of the ingot is intended to represent the mold wall, the "top" of the ingot represents the centerline, and flow occurs left to right from the neck, as it would downwards from the hot top of a large ingot.

Solidification in the portion of the ingot near the base is as in a simple, constant cross section, unidirectionally solidified ingot. This is confirmed by the final segregation seen in this region (at locations 4.5, 6.1 and 7.6 cm from the chill, Figure 43). At any given location along the casting, composition is essentially the same at all three distances from the chill.
Figure 42: Simulated centerline ingot.
Figure 43: Solid composition along the simulated centerline ingot at locations 4.5, 6.1 and 7.6 cm from the chill.
As the dendrite tips approach the ingot top some differences in flow behavior are to be expected:

(1) fluid flow can no longer be perpendicular to isotherms throughout the ingot if solidification shrinkage is to be completely fed, (2) downward flow in the vicinity of the neck must be much greater than that in a classical unidirectional ingot at a similar distance from the chill. In addition, it may be expected that isotherms in the vicinity of the neck may become crowded relative to the rest of the ingot, and alter $G/\varepsilon$; this, however, did not significantly affect solute redistribution in the similar ingot of Figure 40, in which any change in movement of isotherms did not appreciably affect calculate segregation.

Figure 44 shows the composition along the ingot at locations 7.9 and 9.8 cm from the chill (i.e., 2.1 and 0.2 cm from the casting top). Remarkably strong macrosegregation results, where compositions near the ingot top vary systematically by as much as .83 per cent which is nearly 20 per cent of the base alloy composition (aluminum-4.7 per cent copper alloy).

Most notable feature of the data is the low solute region below the neck. This would be predicted from (2) above. In addition the composition of the ingot top at locations removed from the neck appears to be above $C_0$; this would be expected (in the case of complete feeding) since the flow here cannot be perpendicular to isotherms. That is, $v_x$ is approaching zero and lies between zero and the steady
Figure 44: Solid composition along the simulated centerline ingot at locations 7.9 and 9.8 cm from the chill.
state value. This will be discussed further in the centerline segregation portion.

The second example considered in this section is the portion of the previously described ingot that had stainless steel inserts.

Figure 45 shows composition as a function of distance from the chill for that portion of the ingot shown which had the plates to restrict fluid flow. The decrease in composition near the chill, with increasing distance from the chill, is as expected in usual unidirectional solidification, the minimum at 2.5 cm, however, is lower than expected; it is not clear why this is so.

It is interesting to note, by comparing Figures 41 and 45, that the minimum in the section with the plates occurs at a different distance from the chill than minimum in the other portion of the ingot. At the location where the right portion reaches a minimum (at 4.40 per cent copper) the left portion is about the same composition. The left side (without plates), however, continues to decrease to a minimum of 4.16 per cent copper. At this location the right portion (with plates) has a local average composition greater than 5.0 per cent copper. This means that a composition difference of more than .8 per cent copper exists over a horizontal distance of less than 2.5 cm.

Near the plates where \( v_x \) is expected to be 0 the composition should be the same as the chill face composition (i.e., approximately 5.00 per cent for the 4.56 per cent = \( C_0 \) alloy). That fact that the composition
Figure 45: Solid composition versus distance from the chill in an ingot with horizontal plates (as sketched) extending half the width into casting. Composition determined on side with plates.
Just under and between the plates is higher could be due to some "exudation" occurring as a result of solid contraction.

Just above the plate, solidification also is expected to occur with $v_x = 0$ and composition here is the same as at the chill face. Further above this plate, a minimum is observed which may be related to another set of plates (not shown) 15.25 cm from the chill.

The third example considered is the previous ingot (9/1 reduction) at one side of the heavy section (Figure 11). As discussed previously when theoretical calculations were made for this ingot, velocity $v_x$ at a location $y$ (for a given distance from the chill) was assumed constant except exactly at the riser-casting junction. In the actual case, however, this is not expected to be true. As feed metal enters the larger section the absolute value of $v_x$ must be a maximum at the center and, just below the section reduction is expected to be greater than the flow velocity in unidirectional solidification, constant cross section; hence, final composition is here less than $C_0$, Figure 40.

To feed the heavy section out toward the edges, $v_x$ is expected to approach zero for $x = L_R$ where $L_R$ is distance from the chill to section reduction. Hence, composition at the top of the heavy section (away from the reduced section) should be that at a chill where $v_x$ is also zero (5.15 per cent copper for a 4.68 per cent = $C_0$ alloy). This is seen in Figure 40.
2. Centerline Segregation.

While centerline segregation is considered as a separate problem, it can be treated as a limiting special case of non-unidirectional heat and fluid flow. As discussed in the theoretical section, Part III, the increased solute content at the centerline of an ingot can be related to the magnitude and component of the fluid flow velocity $v_x$. As an example, consider the simulated centerline ingot previously discussed, Figure 42. In Figure 46 the results of chemical analysis away from chill are presented at one location along a direction parallel to the heat flow direction. Composition remains relatively constant until the top is approached. At locations close to the chill, fluid flow to feed solidification is perpendicular to isotherms and $v_x$ approaches a steady state value \( \frac{\beta}{1 - \beta} \frac{c}{G} \). As the dendrite tips touch the top, fluid flow can no longer be perpendicular to isotherms and $v_x$ approaches zero. As $v_x$ approaches zero, local average composition increases. The maximum occurring when $v_x = 0$ and composition approaches the same as at a chill face. That this maximum was not reached in this ingot at the simulated centerline was presumably due to incomplete feeding along the very top of the ingot.

It can now be visualized how centerline segregation can occur in a large ingot where solidification fronts from the ingot walls meet at a central region. Before the dendrite tips touch, solidification is as above, when $v_x$ is perpendicular to isotherms. Once the tips touch, however, flow is no longer perpendicular to isotherms but can even become parallel. When flow is parallel, $v_x = 0$ and local average
Figure 46: Solid composition versus distance from the chill in the simulated centerline ingot at the location shown.
composition will approach chill face composition (for perfect feeding). The extent that local average composition approaches chill face composition depends on:

1. How much solidification in the volume element occurred with \( v_x = 0 \).
2. How closely \( v_x \) approaches zero (which depends on details of fluid flow as discussed in the previous section).

As example of this, consider the ingot solidifying toward a common center from two flat chill walls, using the experimental set-up described in part I-B, Appendix E. Figure 47 presents the results of chemical analysis across this ingot at two locations. Both show increases in composition at the center even though casting width was only 3.75 cm and total solidification time was 140 second. The qualitative similarity of the segregation curves in Figure 47 to those measured in large ingots is remarkable; see, for example, Reference 17.

Finally, consideration can again be given to both the reduced cross section ingot and the ingot with metal plates inserted. In these ingots, \( v_x \) is expected to approach zero at the following locations, as well as at the chill face:

1. In the reduced section ingot at the top of the heavy section (see Figure 40b).
2. In the ingot with inserted plates; on either side of the plates and inside the plates (see Figure 41).
Figure 47: Solid composition versus distance between two chills in a bi-directionally solidified casting at two locations. (T) near top of casting, (M) middle portion of casting.
Local average composition at these locations increases in a manner similar to the top of the simulated centerline ingot and the actual centerline of the experimental ingot.
IV. CONCLUSIONS

1. Flow of solute-rich liquid to feed solidification and thermal contractions, is the basic mechanism responsible for many apparently different types of segregation. Among these are:
   (a) inverse segregation
   (b) centerline segregation
   (c) banding
   (d) negative cone of segregation
   (e) localized solute rich regions
      (1) exudation
      (2) "hot tearing" type segregation
      (3) solute enrichment under risers and hot tops.

2. The final composition at any point x, y, z in a casting or ingot (and hence macrosegregation) can be calculated from knowledge of (a) phase diagram and other physical constants of the alloy,
   (b) thermal history at x, y, z during solidification, and (c) amount and direction of fluid flow at x, y, z to feed shrinkage of parts of the casting at lower temperatures.

3. Laboratory experiments on aluminum-4.5 per cent copper alloy qualitatively and quantitatively confirm analyses presented. These experiments comprise measurement of macrosegregation in small ingots of the following mold designs: (a) simple unidirectional, (b) unidirectional with restricted fluid flow,
   (c) unidirectional with variable heat extraction rate to
illustrate banding, (d) unidirectional with reduced cross section, (e) edge-risered unidirectional to illustrate centerline segregation, and (f) ingot with bi-directional heat flow to illustrate centerline segregation.
V. SUGGESTIONS FOR FUTURE WORK

While much experimental work has been done in the area of unidirectional heat and fluid flow, further experimental work seems desirable on non-directional solidification. Specifically, work should be done on "hot tear" type segregation, segregation in sand castings, and under-riser segregation.

In conjunction with this work consideration should be given to the effect of porosity on macrosegregation results. While a theoretical and qualitative presentation is made, quantitative results should prove helpful.

A new area of investigation would be the effect of magnetic fields on macrosegregation. A sufficiently strong magnetic field should influence segregation by altering fluid flow during solidification.

Another area needing study is that concerning the details of fluid flow in non-unidirectional solidification. Limiting cases have been considered herein, but treatment of intermediate areas (which more closely approximate real conditions) requires detailed fluid flow analysis.

A more applied area for future work would be application of the ideas presented herein to actual ingot solidification processes (static, continuous, consumable electrode). Such application would require numerical analysis and clearly would require use of high
speed digital computers. However, the application should be feasible, and certainly of great engineering usefulness.
VI. REFERENCES


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APPENDIX A

List of Symbols

\(a_L, a_E\) \hspace{1cm} \text{intercepts of liquidus and eutectic isotherm movement equations}

\(a_T\) \hspace{1cm} \(a_L - a_E\)

\(A, A', A_E, A'_0\) \hspace{1cm} \text{cross sectional areas at } x, x', x_E, \text{ and chill face, respectively}

\(\bar{C}\) \hspace{1cm} \text{local average composition of liquid and solid (weight fraction, weight per cent solute)}

\(C_a\) \hspace{1cm} \text{average composition of the entire amount of primary } a \text{ phase in the final solid ingot (weight fraction, weight per cent solute)}

\(C_E\) \hspace{1cm} \text{composition of eutectic (weight fraction, weight per cent solute)}

\(C_L\) \hspace{1cm} \text{liquid composition (weight fraction, weight per cent solute) within volume element}

\(C_L^C\) \hspace{1cm} \text{liquid composition (weight fraction, weight per cent solute) at } x' \text{ when } x_E = 0

\(C_L^v\) \hspace{1cm} \text{liquid composition (weight fraction, weight per cent solute) when feeding stops}

\(C_O\) \hspace{1cm} \text{overall composition (weight fraction, weight per cent solute) in a binary alloy}

\(\bar{C}_S\) \hspace{1cm} \text{average solid composition (weight fraction, weight per cent solute) after solidification}

\(C_S^g\) \hspace{1cm} \text{solid composition (weight fraction, weight per cent solute) at the liquid-solid interface}

\(\Delta C\) \hspace{1cm} \text{macrosegregation (weight fraction, weight per cent solute)}

\(f_L\) \hspace{1cm} \text{weight fraction liquid}

\(f_S\) \hspace{1cm} \text{weight fraction solid}

\(g_E\) \hspace{1cm} \text{volume fraction eutectic}

\(g_L\) \hspace{1cm} \text{volume fraction liquid}
\( s^c_L \)  
volume fraction liquid at \( x' \) when \( x_E = 0 \)

\( s^v_L \)  
volume fraction liquid when feeding stops completely

\( s_S \)  
volume fraction solid

\( G \)  
temperature gradient

\( k \)  
equilibrium partition ratio

\( L \)  
total solidification length from a chill or wall (i.e., unidirectional solidification \( L = \) total length of ingot, ingot solidification from two walls \( L = \) width/2 etc)

\( n_L, n_E \)  
slope of liquidus and eutectic isotherm movement equation

\( n_q \)  
\( (n_L - n_E)/n_E \)

\( p \)  
dimensionless group of variables  
\( \frac{v_x G}{c} \frac{(1 - \beta)}{\beta} \)

\( t_v \)  
time when feeding stops completely

\( t_{v^o} \)  
time when dendrite tips reach \( L \)

\( T_E \)  
non-equilibrium solidus

\( T_L \)  
liquidus temperature

\( U_E \)  
velocity of eutectic isotherm \( T_E \)

\( v \)  
local velocity of liquid relative to solid (resulting from solidification contraction), cm/sec

\( v_x \)  
local velocity of liquid relative to solid in the \( x \) direction (perpendicular to isotherms) resulting from solidification contraction

\( x, y, z \)  
coordinate axles for volume element under consideration

\( x', y', z' \)  
general coordinate axles

\( x_c \)  
location of the liquidus isotherm when the eutectic isotherm is at \( x_E = 0 \)
\( x_L, x_E \)  
location of the liquidus and eutectic isotherms (distance from the chill)

\( x_v \)  
location of "volume element" when feeding stops

\( \beta_E \)  
solution shrinkage \((\rho_S - \rho_L)/\rho_S\)

\( \rho_E \)  
solid solution shrinkage of eutectic

\( \epsilon \)  
rate of temperature change, °C/sec

\( \bar{\rho} \)  
local average density (liquid and solid), \( \text{gm/cm}^3 \)

\( \rho_a \)  
density of a phase \( \text{gm/cm}^3 \)

\( \rho_L \)  
liquid density \( \text{gm/cm}^3 \)

\( \rho_S \)  
solid density \( \text{gm/cm}^3 \)

\( \rho_{SE}, \rho_{LE} \)  
densities of solid and liquid eutectic respectively, \( \text{gm/cm}^3 \)
APPENDIX B

Calculations of Solute Redistribution, Unidirectional Solidification Constant k, \( \beta \)

1. \( x > x_c \)

Assume first the distribution of liquid in the mushy zone is as sketched in Figure 5b. Solidification conditions are described by:

\[
\begin{align*}
x_L &= n_L t^b + a_L \quad \text{(B1)} \\
x_E &= n_E t^b + a_E \quad \text{(B2)}
\end{align*}
\]

\[
\frac{g_L'}{x' - x_E} \frac{1 - g_E}{x_L - x_E} = \quad \text{(B3)}
\]

Differentiate (B3) with respect to time at constant \( x' \):

\[
\frac{dg_L'}{dt} = - \frac{1 - g_E}{x_L - x_E} \frac{dx_E}{dt} + \frac{x' - x_E}{x_L - x_E} \frac{dx'}{dt} (x_L - x_E) \quad \text{(B4)}
\]

Rewrite (B3) for \( x' = x \), \( g_L' = g_L \); differentiate with respect to time at constant \( g_L \). Set equal to \( -\ell/G \) for linear \( g_L \) and \( T \) versus \( x \):

\[
\left( \frac{\partial x}{\partial t} \right)_{g_L} = (\frac{g_L - g_E}{1 - g_E})(\frac{\partial}{\partial t})(x_L - x_E) - \frac{\partial}{\partial t} x_E = -\frac{\ell}{G} \quad \text{(B5)}
\]

Now, for constant \( \rho_L \), equation (22) is written:

\[
g_L \frac{x}{x_E} = \frac{\beta}{1 - \beta} \left[ \int \frac{dg_L'}{dt} \left( x' - g_E \right) \frac{dx_E}{dt} \right] \quad \text{(B6)}
\]
Substitute (B4) in (B6) and integrate:

\[ g_L \frac{dx}{dt} = - \left( \frac{\beta}{1-\beta} \right) \left[ (1-g_E) \frac{x-x_E}{x_L-x_E} \frac{dx}{dt} + \frac{1}{2} \left( \frac{x-x_E}{x_L-x_E} \right) \frac{d}{dt} (x_L-x_E) \right] + \frac{dx_E}{dt} \]

\[ \ldots \ldots \text{(B7)} \]

Combine (B5) and (B7):

\[ \frac{v_x G}{\varepsilon} = \frac{\beta}{1-\beta} \frac{1}{g_L} \left[ (1-g_E) \frac{x-x_E}{x_L-x_E} \frac{dx}{dt} + \frac{1}{2} \left( \frac{x-x_E}{x_L-x_E} \right) \frac{d}{dt} (x_L-x_E) \right] + \frac{dx_E}{dt} \]

\[ \ldots \ldots \text{(B8)} \]

Substitute in

\[ \frac{x-x_E}{x_L-x_E} = \frac{g_L-g_E}{1-g_E} \]

\[ \ldots \ldots \text{(B9)} \]

\[ \frac{dx}{dt} = \frac{3x_E}{3t} = bn_E t^{b-1} \]

\[ \ldots \ldots \text{(B10)} \]

\[ \frac{d}{dt} (x_L-x_E) = \frac{3}{3t} (x_L-x_E) = b (n_L-n_E) t^{b-1} \]

\[ \ldots \ldots \text{(B11)} \]

\[ n_q = \frac{n_L-n_E}{n_E} \]

\[ \ldots \ldots \text{(B12)} \]

and (B8) becomes

\[ \frac{v_x G}{\varepsilon} = \frac{1}{2} \left( \frac{\beta}{1-\beta} \right) \left[ \frac{g_L (g_L n_q + 2b) + n_q s_E^2}{g_L (g_L n_q + b)} \right] \]

\[ \ldots \ldots \text{(B13)} \]

where \( b = 1 - g_E - n_q g_E \)

Solute redistribution, and from this, macrosegregation can now be determined analytically by substituting (B13) in equation (20) and integrating from \( g_L = 1 \) at \( C_L = C_0 \) to \( g_L, C_L \). The analytic expression so obtained is cumbersome. For small \( g_E \), however, (B13) reduces to:
\[
\frac{v_i G}{\varepsilon} = \frac{1}{2} \left( \frac{\beta}{1 - \beta} \right) \left[ \frac{\kappa}{\gamma L q} \right] + 2 \frac{\kappa}{\gamma L q} + 1
\]  
(B14)

which is also the expression obtained by assuming the distribution of liquid in the mushy zone as given in Figure 5c. Substitution of typical values for unidirectionally solidified aluminum-4.5 per cent copper shows that results obtained from (B14) are at most a few per cent different from those obtained using (B13), and (B14) is therefore used for subsequent analysis.

As example of the difference, consider the case of \( k = 0.172, \beta = 0.055, n = 0.52 \). Calculated values of \( \frac{v_i G}{\varepsilon} \) obtained from the two equations are given in Table B1:

<table>
<thead>
<tr>
<th>( \kappa )</th>
<th>( \frac{v_i G}{\varepsilon} ) (Eq. B14)</th>
<th>( \frac{v_i G}{\varepsilon} ) (Eq. B13)</th>
<th>Per Cent Difference</th>
</tr>
</thead>
<tbody>
<tr>
<td>1.0</td>
<td>-</td>
<td>-</td>
<td>-</td>
</tr>
<tr>
<td>.9</td>
<td>.0489</td>
<td>0.0480</td>
<td>- 1.6</td>
</tr>
<tr>
<td>.5</td>
<td>.0522</td>
<td>.0517</td>
<td>- 1.0</td>
</tr>
<tr>
<td>.1</td>
<td>.0567</td>
<td>.0578</td>
<td>+ 1.9</td>
</tr>
<tr>
<td>.0902</td>
<td>.0569</td>
<td>.0582</td>
<td>+ 2.3</td>
</tr>
</tbody>
</table>

(eutectic)
Now, substituting equation (B14) in equation (20), and integrating from the limits $g_L = 1$ at $c_L = C_0$ to $g_L$, $C_L$, the final solute redistribution expression is:

$$C_L = C_0 g_L \left[ \frac{g_L n q + 1}{n a + 1} \right]$$

where $a = 1 - \frac{\beta}{2}$

2. $x < x_c$

For calculations of solute redistribution at locations close to the chill face a procedure similar to the foregoing is followed except that for $x < x_c$ and $g_L > g_L^c$, $g_L v_x$ (see equation B6 above) is calculated by integration of equation (22) from the limits 0 to $x$, not $x_E$ to $x$.

The resulting expression for $g_L = 0$ is:

$$\frac{v_x G}{\varepsilon} = \frac{x}{2} \left( \frac{\beta}{1 - \beta} \right) \frac{2(x - x_E) + g_L n q (x - 2x_E)}{(x - x_E)^2(g_L n q + 1)}$$

where

$$x_E = \frac{x + g_L a (n_q - 1) + \alpha E}{g_L n q + 1}$$

Substitute (B17) in (B16)

$$\frac{v_x G}{\varepsilon} = \frac{x}{2} \left( \frac{\beta}{1 - \beta} \right) \frac{2P + (g_L 0 - x) n q}{g_L P^2}$$
where

\[
P = n_q x - a_{E n} + a_E - a_L \quad \text{(B19)}
\]

\[
Q = n_q x - 2a_{E n} q + 2a_E - 2a_L \quad \text{(B20)}
\]

Substitute (B16) in equation (20)

\[
- \frac{1}{1 - k} \frac{dC_L}{C_L} = \frac{dg_L}{N + Mg_L} \quad \text{(B21)}
\]

where

\[
N = \frac{6x(2P - xn_q)}{2P^2} \quad \text{(B22)}
\]

\[
M = \frac{xkQ q + 2(1 - \beta)P^2}{2P^2} \quad \text{(B23)}
\]

Integrating between the limits \( g_L = 1 \) at \( C_L = C_0 \) and \( g_L, C_L \)

\[
C_L = C_0 \left( \frac{N + Mg_L}{N + M} \right)^{1 - k} \quad \text{(B24)}
\]
APPENDIX C

Calculation of Solute Redistribution in Final Transient Region During Unidirectional Solidification

From Figure (C-1)

\[
(1 - x_v) = - \frac{1}{v} \int_{t_v}^{t^*} v_x \, dt
\]

for \( n_q = 0, x_L - x_E = \text{constant} \)

\[
v_x = \frac{\beta}{1 - \beta} \frac{dx_E}{dt}
\]

\[
g_L^v = \frac{x_v - x_E}{x_L - x_E}
\]

where \( g_L^v = \text{fraction liquid at } x_v \)

Differentiating (C3) and rearranging

\[
dx_E = dx_v - (x_L - x_E) \, dg_L^v
\]

Substituting in (C1)

\[
1 - x_v = \int_{t_v}^{t^*} \frac{1}{v} - \frac{\beta}{1 - \beta} [dx_v - (x_L - x_E) \, dg_L^v] \]

at time \( t_v, x_v = x_v^*, g_L^v = g_L^v \)

at time \( t^*, x_v = L, g_L^v = 1 \)
Substituting these limits in (C5)

\[
L - x_v = \frac{\beta}{1 - \beta} \int_1^{x_v} \frac{1}{1 - \beta} \frac{\delta L}{x_v} \int (x_L - x_L L - x_E) \delta L
\]  

(C6)

Integrating between limits

\[
1 - x_v = \frac{\beta}{1 - \beta} [(x_v - L) - (x_L - x_E)(g_L^v - 1)]
\]  

(C7)

Solving for \( g_L^v \) yields

\[
g_L^v = 1 - \frac{(1 - x_v)}{(x_L - x_E) \beta}
\]  

(C8)
Figure C-1: Transient zone, end of solidification at ingot top: (a) time, $t^*_v$, when dendrite tips first reach top of ingot, (b) time, $t_v$, at a later stage of solidification (before completely solid).
APPENDIX D

Calculation of Solute Redistribution: Unidirectional Solidification; Constant $k$, $\beta$; Pyramid Shape

1. $x > x_c$

From equation (23) when $g_E = 0$

$$ A g_L \frac{dx}{dt} = -\frac{\beta}{1 - \beta} \int x \frac{dg_L'}{dt} x' A' $$

(D1)

where

$$ A' = (1 - \frac{x'}{x_L}) A_0 $$

(D2)

from (B4) with $\eta = 0$, $x_L = x_E$ = constant

$$ \frac{dg_L}{dt} = \left( \frac{1}{x_L - x_E} \right) \left[ \frac{dx_E}{dt} \right] $$

(D3)

Substitution of (D2) and (D3) in (D1)

$$ (1 - \frac{x}{x_L}) g_L \frac{dx}{dt} = -\frac{\beta}{1 - \beta} \int x \frac{1}{x_L - x_E} \frac{dx_E}{dt} \frac{1}{x_L - x_E} (1 - \frac{x'}{x_L})^2 x' $$

(D4)

integration yields

$$ v_x = -\frac{\beta}{1 - \beta} \left( \frac{1}{x_L - x} \right) \frac{dx_E}{dt} \left[ \frac{(x - x_E)}{x_L - x_E} \right] - \frac{x^2 - x_E^2}{2(x_L - x_E)} $$

(D5)

from (B9) when $g_E = 0$

$$ \frac{x - x_E}{x_L - x_E} = g_L $$

(D6)
From (B17) when \( n_q = 0 \)

\[ x_E = x + a_T g_L \]  \hspace{1cm} (D7)

where \( a_T = a_L - a_E \) \hspace{1cm} (D8)

From (B5)

\[ \frac{\varepsilon}{G} = - \frac{dx_E}{dt} \] \hspace{1cm} (D9)

Substitution of (D7) and (D9) in (D5)

\[ \frac{v_x G}{\varepsilon} = \frac{\beta}{1 - \beta} \left( \frac{1}{L - x} \right) \left( \frac{2L(x_L - x_E) + 2xa_T + a^2_T g_L}{2(x_L - x_E)} \right) \]

2. \( x < x_c \)

Integration limits on (D1) change

\[ A g_L v_x = - \frac{\beta}{1 - \beta} \int_0^x \frac{d\theta}{dt} \theta x' A' \] \hspace{1cm} (D10)

Substitution of (D2) and (D3), integration and substitution of (D9) yields

\[ \frac{v_x G}{\varepsilon} = - \frac{\beta}{1 - \beta} \left( \frac{1}{L - x} \right) \left[ \frac{x(2L - x)}{2g_L} \right] \] \hspace{1cm} (D11)
APPENDIX E

Experimental Casting Procedure

A. Introduction


The typical experimental set up shown in Figure E-1 was used, except for one series of centerline segregation experiments, for all castings and ingots made as part of the current investigation. Since unidirectional solidification is desired, essentially all heat must be extracted from one surface of the ingot. This was accomplished by doing all casting and solidification in an electrically heated furnace.

After placing the entire assembly in the furnace with coolant pipes extending outside, the unit was heated to 600-700°C. When the plaster reached the desired temperature (usually 1-2 hours after the furnace temperature) the mold was filled through a funnel extending from the outside of the furnace into the cavity. The assembly, with liquid was held for several minutes to allow convection to subside. The coolant, either air or water (depending on the desired solidification rate) was turned on and the ingot solidified. With the four sides and top insulated, heat removal for solidification was through the bottom surface, resulting in fully columnar structures.

Casting and ingot designs were varied, as discussed in Experimental Results and Discussion section, to produce desired macrosegregation effects. However, for casting solidified with unidirectional heat flow
the top was always open to the furnace atmosphere and temperature, the bottom surface rested on a chill through which the heat was extracted, and all other surfaces were insulated with plaster.

2. Centerline Segregation Experiment.

The centerline segregation experiment was designed to produce two unidirectionally solidifying fronts meeting at the ingot center, Figure E-2. Casting procedure was the same as in the straight unidirectional ingot with water or air used as coolant. (Piping is not shown in schematic drawing, however, continuous flow was supplied to both chills.)

B. Mold Preparation

In many ingots, thermocouples were used to obtain continuous records of solidification. In all cases these were chromel-alumel couples with output continuously monitored on a L & N Speedomatic W, Multipoint (12) recorder.

When it was necessary to locate many thermocouples accurately, the pattern for the ingot was first shaped from foamed polystyrene. The thermocouples (alumina sheathed, tips coated with boron nitride), were placed in the styrofoam pattern at the desired locations. The assembly was placed inside a stainless steel flask, the flask was then filled with liquid plaster which was allowed to set. The result was that the thermocouples remained accurately located since they were held in place by the plaster.
The mold was next fastened to the chill and placed in the furnace with the foamed polystyrene still in place. Upon subsequent heating the polystyrene vaporized and left a deposit-free cavity.

Then castings were made without thermocouples, wood patterns were used rather than styrofoam. The greased wood pattern was placed inside a flask, liquid plaster poured around the pattern and allowed to set. After setting, the pattern was removed and the mold fastened to the chill. The assembly could then be placed in the furnace.

C. Metal Melting Procedure

All melts were prepared from aluminum-4.5 per cent master heats. These had been previously cast into 5 to 7 pound ingots from high purity aluminum (99.99 per cent) and 50-50 aluminum-copper OFHC master alloy. All melting was done in "Fiberfrax" coated graphite crucibles using a gas-fired furnace. After preheating the crucible for an hour or more, the charge was melted and superheated to a temperature of 690-720°C. After stirring the melt, a chlorine degas was carried on for 10 minutes. After stirring again for one minute a gas sample, under reduced pressure, was taken. When the sample was gas-free, the crucible was removed from the furnace, a chemical analysis specimen taken up into a 7 mm vycor tubing, and skimmed. Pouring temperature into the mold was about 680 to 700°C. After pouring, another chemical analysis specimen was taken from the remaining liquid metal.
Figure E-1: (a) Schematic diagram of unidirectional solidification, mold-chill assembly.  
(b) Cross section of assembly.
Figure E-2: (a) Schematic diagram of centerline segregation, mold-chill assembly.
(b) Cross section of assembly.
A. Introduction

When any material is bombarded with electrons of X-rays of high enough energy a characteristic line spectrum is emitted. This can be used as a basis for either qualitative or quantitative chemical analysis. For qualitative analysis the various elements in a sample are made to emit their "characteristic lines", these elements may then be identified by analyzing the emitted radiation and showing that the specific wavelengths characteristic of the element are present. By comparing the intensity of these lines with the intensities of lines from suitable standards, a method of quantitative analysis is available. For multicomponent systems, comparison with standards is complicated by interactions between elements. When dealing with binary systems however, it is a rapid, accurate method for obtaining reproducible chemical analysis.

In the present work a General Electric XRD-5, X-ray Spectrometer was used with a tungsten target tube. The primary white radiation from the tube was used to fluoresce aluminum-copper samples on an area 1.25 cm by 1.8 cm. This sample size insured that the analysis was representative and no microsegregation effects are recorded. That is, we are measuring local composition by integrating over many dendrites, which are typically 10-1000µ center to center spacing.
The secondary radiation from the aluminum-copper alloy is then diffracted from a single crystal of known d spacing (lithium fluoride, d spacing 2.014).

The Bragg law predicts that for each angular setting of the crystal, radiation of only a single wavelength is reflected. The spectrometer was set at 45.08° which for the crystal and spacing used reflects the copper, \( K_\alpha \) wavelength. The intensity of this "characteristic line" was measured and compared to a standard intensity curve.

B. Standards Preparation

Standards were prepared with the same apparatus used by Bower in a study of solidification in thin plate castings. (Shown schematically in Figure F-1). By drawing a vacuum and placing the molds in the melt molten metal is forced between the two copper chills, spaced 1/8 inch apart, and allowed to solidify. Compositional variations were obtained by successively adding high copper master alloy to low alloy heats.

A 10 pound heat of aluminum-1 per cent copper was melted in a gas furnace using a graphite crucible. After degasing with chlorine, samples were sucked into the chill mold at 1500°F. Master alloy of 50 per cent copper-aluminum composition was added to increase alloy content of the melt to about 2.5 per cent copper. Again the melt was degassed and samples sucked into the chill mold. By progressively adding alloy, a series of standards were prepared. After cooling to room temperature the samples were sanded on coarse paper to remove any exuded layer. They were then sanded down to 300 grit emery, a finish smooth enough
for use in fluorescent analysis. Table F-1 lists the compositions obtained.

\[
\begin{array}{ll}
\text{Table F-1} \\
\hline
\text{Standard} & \text{Composition - Z Copper} \\
A & 1.12 \\
B & 2.69 \\
C & 4.35 \\
D & 5.83 \\
E & 8.48 \\
\hline
\end{array}
\]

After sanding the cast surfaces on the edge, two pieces weighing about one gram each were removed from opposite sides and used for wet analysis. The average of these compositions was taken to be the composition of the plate.

C. Standard Curve Determination

Fluorescence of the five standards was carried out as previously described. The intensity from Standard E (maximum copper) was measured first with the power adjusted to give approximately 100,000 counts per second. This approaches the limits of the scaler used for counting. At greater than 100,000 counts/sec the unit can be flooded and inaccuracies result. After the intensity from Standard E was determined, Standards A through D were measured. Since the intensity of copper, K\text{a}, is being used for comparison, decreasing copper content means decreasing intensity. After all five samples are measured Standard E is normalized...
to $I/I_o = 1$. The intensity of standards is A to D are divided by $I_o$ (the intensity of Standard E) to give normalized intensities. (These values are plotted in Figure F-2).

**D. Composition Determination**

When the intensity from a series of unknowns is measured, Standard E is also measured. The intensity of the unknown divided by the intensity of Standard E ($I_o$) gives a fraction less than one (if the composition of the unknown is less than 8.48 per cent copper). This fraction can be directly translated to a per cent copper from Figure F-2.

Standard practice was to check one of the Standards A to D each time a series of unknowns was determined to be sure reproducibility of results was obtained. As a further check, periodically samples which had been analyzed by fluorescence were also checked by wet chemical methods. Agreement was found generally to be within ± 0.05 copper. Another method of insuring accuracy was the redetermination of standard curves. Three different times during the investigation the standard curves were redetermined. Each time, no significant difference was found between curves.
Figure F-1: Schematic diagram of apparatus used to prepare standard chemical analysis specimens.
Figure F-2: Standard chemical analysis curve.
The following is an outline of the method used to calculate chemical composition at various locations in unidirectionally solidified ingots, using thermal data and relationships developed in the theoretical section. As example, the calculations are given for a point 7.9 cm from the chill, in the reduced section ingot described in the Experimental Results and Discussion section.

A. Thermal Data

Temperature versus time curves at the different thermocouple locations are made directly from the output recorded on the L & N Speedomatic W Recorder (Figure G-1).

B. Temperature Versus Fraction Liquid

A temperature versus fraction liquid curve is drawn assuming $g_E = 0$ and fraction liquid distribution in the mushy zone is linear (Figure G-2a). For comparison, a curve of actual fraction liquid in the mushy zone is shown, Figure G-2b (see Macrosegregation Versus Distance from Chill in Theoretical Section).

C. Calculation of $v_x$

By using the data from Figures G-1 and G-2a, Figure G-3 is constructed to be used for graphical integration to find $v_x$ as follows:
(1) At a constant \( x \), \( \Delta g_L \Delta x \) is determined by adding up the number of squares from \( g_L = 0 \) to \( g_L = 1 \) between \( t_n \) and \( t_{n+1} \) and multiplying by the area of each square. Area to the left of the ingot reduction is weighted 9 times that to the right (ratio of area reduction). That is in equation form.

(2) Equation (23) is solved for the case of no eutectic:

\[
\frac{A_2 g_L x}{1 - \beta} = \frac{1}{A_1} \int_{t_n}^{t_{n+1}} \frac{\Delta g_L' x}{t_{n+1} - t_n} \, dx' + \frac{1}{A_2} \int_{t_n}^{t_{n+1}} \frac{\Delta g_L' x}{t_{n+1} - t_n} \, dx' 
\]

\[\text{............... (G1)}\]

Equation (G-1) is written for the reduced cross section ingot, for \( x > x_c \) and for \( x_E > 0 \) where:

\( x_c \) = distance from chill to reduction in cross section

\( x \) = distance from chill to location where composition is being calculated

\( \Delta g_L \) = change in fraction liquid at \( x \) from \( t_n \) to \( t_{n+1} \)

\( A_1 \) = area of lower (large) section

\( A_2 \) = area of upper section

For the general case of constant cross section at locations in the ingot unaffected by the reduction, equation (G-1) becomes:

\[
g_L' x = \frac{\beta}{1 - \beta} \left( \frac{\Delta g_L \Delta x}{t_{n+1} - t_n} \right)
\]

\[\text{(G2)}\]
D. $\frac{v}{x}$ Versus $g_L$

Figure G-4 is obtained by taking $v_x$ as being linear and averaged over the interval $\Delta g_L$. That is, the determined value is plotted at:

$$g_L = \frac{(g_L)_n + (g_L)_{n+1}}{2}$$  \hspace{1cm} (G3)

where $g_L =$ fraction liquid at $x$ at time $t = \frac{t_n + t_{n+1}}{2}$.

This is plotted as Figure G-4.

E. Calculation of $G/\epsilon$

$G/\epsilon$ is calculated by first replotting the data of Figure G-3 as in Figure G-5 and then the reciprocal of these slopes are $G/\epsilon$ at a given $x$ for various $g_L$. In Figure G-5 sample curves for two values of $g_L$ are shown. In practice a whole family of curves is necessary to calculate $G/\epsilon$ as a function of $g_L$.

F. $G/\epsilon$ Versus $g_L$

Figure G-6 shows $G/\epsilon$ as a function of $g_L$. Using this and Figure 4, $\frac{v_x G}{\epsilon}$ as a function of $g_L$ can be calculated.

G. Calculation of Local Average Composition

From the steady state solute redistribution expression $\frac{v_x G}{\epsilon}$ is found as a function of $C_L$. $C_L$ being calculated from the values of $g_L$ assuming steady state values. The final plot is $\frac{v_x G}{\epsilon}$ versus $\ln C_L$ which is necessary to solve the solute redistribution (equation 20). This is shown in Figure G-7.
Average composition is found as follows:

Solve by taking two cases of macrosegregation equation:

\[
(1 - k) \frac{dg_L}{g_L} = - (1 - \beta)(1 + \frac{v_x G}{\epsilon}) \frac{dC_L}{C_L} \quad (G4)
\]

also take case where \( v_x = 0 \), at chill face where maximum segregation
can be calculated

\[
(1 - k) \frac{dg_L}{g_L^*} = - (1 - \beta) \frac{dC_L}{C_L} \quad (G5)
\]

Subtracting (1a) from (1)

\[
(1 - k) \ln \frac{g_L}{g_L^*} = (1 - \beta) \int \frac{v_x G}{\epsilon} \frac{dC_L}{C_L} \quad (G6)
\]

A plot of \( \frac{v_x G}{\epsilon} \) versus \( \ln C_L \), as in Figure G-7 allows graphical
integration to solve the general macrosegregation equation.

The area under the curve will give the term inside the integral of
equation (3):

\[
\text{area} = - \frac{(1 - k)}{(1 - \beta)} \ln \frac{g_L}{g_L^*} \quad (G7a)
\]

to find \( g_E \)

\[
\frac{- \text{area} (1 - \beta)}{(1 - k)} = \ln \frac{g_E}{g_E^*} \quad (G7b)
\]

\( g_E \) (33% Copper in eutectic) = \( C_E \)

\( C_a = 1.51 - 1.54 \)

\( C_S = C_a + C_E \)
Figure G-3: Fraction liquid versus distance from the chill for the 9 to 1 reduced section ingot.
Figure G-4: Calculated $v_x$ versus $g_L$ for the 9 to 1 reduced section ingot at $x = 7.9$ cm from the chill.
(cm from chill). Numbers are location of thermocouple from chill presented in Figure 30.) Reduced section of Fig. 8 (composition versus distance)

Figure G-1: Thermal data for 9 to 1 reduced section from Pouring (sec) after cooling.

Time After Pouring (sec)
Figure C-2: Temperature versus fraction liquid-Iiquid-solid region.

(a) Steady state solution

(b) Assumption linear distribution

center copper alloy

\( C_0 = 4.68 \) per
Figure G-5: Constant fraction liquid curves plotted as a function of time versus distance from the chill.
Figure G-6: $G/c$ versus fraction liquid ($g_L$) at 7.9 cm from chill in the 9 to 1 reduced section ingot.
Figure G-7: Calculated data plotted in a form suitable for graphical integration using equations (G-4) - (G-7b).