

2

MODELING OF THE INTERACTION BETWEEN DENSIFICATION MECHANISMS IN POWDER COMPACTION

S. J. Subramanian and P. Sofronis

*Department of Theoretical and Applied Mechanics,
University of Illinois at Urbana-Champaign,
216 S. Wright Street, Urbana IL 61801, U.S.A.*

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ABSTRACT

Sinter-forging is a common means of forming ceramic or inter-metallic components for modern technological applications such as advanced structural materials, magnetic materials, thin ceramic layers on substrates and nanocomposites. Micrometer/nanometer-sized powders provide significant advantages in terms of reduced processing pressures/temperatures and increased control over the final microstructure. Although extensive research in the past twenty years has identified and analyzed the individual mechanisms acting at the microscale during densification, no comparable progress has been made towards understanding the full spectrum of interactions between the mechanisms. In this paper, a micromechanical model of powder consolidation is presented. It accounts for the elastic and power-law creep deformation of the bulk material along with stress-driven diffusion on the interparticle contacts and curvature-driven surface diffusion along the pore surfaces. The finite element method is used to obtain the time-dependent deformation of the powder aggregate from which macroscopic quantities such as relative density, densification rate, and strain rate as well as microscopic quantities such as volumetric flux and pore surface curvature are calculated. Comparisons are made with existing models and experimental data from the consolidation of micron-sized Al_2O_3 and TiAl.

1. INTRODUCTION

Powder processing is used in forming complex shaped components from ceramic or inter-metallic materials. Sinter-forging is an attractive means of producing dense fine-grained components and is especially suited to produce alloys and composites. It is convenient to speak of the consolidation process as occurring in two stages (Ashby, 1974; Helle *et al.*, 1985; Swinkels and Ashby, 1981). Stage I is characterized by connected porosity and existence of discrete necks between individual particles. When the relative density (density of compact/density of fully compact material) is greater than about 0.9, one enters stage II of the deformation, wherein the porosity is isolated with near equilibrium-shaped pores.

The closing of porosity typically occurs by various deformation mechanisms like linear elasticity, rate-independent plasticity, power-law creep and diffusion along the grain-boundaries and pore surfaces, and interparticle slip (Ashby, 1974; Swinkels and Ashby, 1981; Helle *et al.*, 1985). Arzt and coworkers (Arzt, 1982; Arzt *et al.*, 1983; Fischmeister and Arzt, 1983) developed a model for densification that took into account the evolving particle coordination number and contact size during densification. Subsequently, a variety of models have used their idea of an isotropically evolving contact area to predict the macroscopic behavior of a powder compact (Helle *et al.*, 1985; McMeeking and Kuhn, 1992).

Models for stage I densification have been proposed in which the single dominant mechanism is plasticity (Fleck *et al.*, 1992; Akisanya *et al.*, 1994; Akisanya and Cocks, 1995; Fleck, 1995) or power-law creep (Ashby, 1974; Swinkels and Ashby, 1981; Kuhn and McMeeking, 1992; Bouvard, 1993) or grain boundary diffusion (McMeeking and Kuhn, 1992) or interparticle slip (Casagrande and Sofronis, 1997). Cocks (1992) has

suggested a method to include the effect of an interface reaction stress (Ashby, 1969; Burton, 1972) on the diffusion process along the interparticle contacts. Work has also been done in the area of coupling interparticle diffusion to pore surface diffusion (Pan and Cocks, 1995; Zhang and Schneibel, 1995; Bouvard and McMeeking, 1996; Pan *et al.*, 1997) and linear elastic deformation to surface diffusion (Freund *et al.*, 1993; Suo and Wang, 1994; Wang and Suo, 1997; Xia *et al.*, 1997). Thouless (1993) has studied the effect of surface diffusion on the creep of sintered arrays of particles. Analogously, there are also models in the literature for stage II densification in which the dominant mechanism is one of those mentioned above, namely power-law creep (Cocks 1989; Duva and Crow, 1992; Sofronis and McMeeking, 1992), diffusional creep (Riedel *et al.*, 1994b) or interparticle diffusion and slip (Riedel *et al.*, 1994a). An overview of the analytical investigations into powder compaction is found in the work of Cocks (1994). In his paper, Cocks has laid out a methodology for formulating the constitutive laws for powder compacts. Furthermore, he has used appropriate kinematically and statically admissible fields to calculate bounds on macroscopic potentials that describe the densification process. Significantly, in this work, Cocks has taken into account the interaction between pairs of competing mechanisms in effecting the overall compaction of the particles.

Typically, in the work to date, one finds that not all the densifying mechanisms have been modeled in a synergistic framework. The interaction between the mechanisms could significantly affect the shapes of pores and the stress distribution along the interparticle contact areas, and hence macroscopic quantities like the relative density and the rate of densification. A systematic model that begins by considering the point contact between the particles and proceeds in time capturing and describing all the transient effects is still lacking.

There are also a number of assumptions that have been made in the past, the validity of which is an open question. A few of the models (McMeeking and Kuhn, 1992; Riedel *et al.*, 1994) have assumed the grains to be rigid and this implies that the normal velocity of separation between adjacent grains is a constant along their boundary. This simplifies the diffusion process on the interparticle contact areas and might give inaccurate results for the normal stresses and diffusional mass fluxes along the interparticle boundaries when other deformation processes are concurrently active. Treatment of the dihedral angle at the boundary of the interparticle contact area has also constituted a source of ambiguity. While it is certainly true that the well-known equilibrium expression for the dihedral angle in terms of the grain boundary and pore surface energies is valid for a static pore tip, the angle at the tip is not strictly equal to that dictated by the equilibrium relation when the interparticle area is growing. Still, it is not uncommon to find the strict enforcement of the dihedral angle at the tip during densification in several of the currently available models (Svoboda *et al.*, 1993; Bouvard and McMeeking, 1996; Pan and Cocks, 1995). Moreover, the equilibrium value of the angle is used to calculate the curvature at the tip of the pore. Thus, any untenable assumption regarding the angle could lead to a misleading description of the diffusion process at the tip. Lastly, even in the case when surface diffusion significantly affects

densification, one finds that the classical Laplace relationship connecting the curvature and surface energy of the pore to the normal stresses in the adjoining bulk material (Herring, 1951; Rice and Chuang, 1981; Freund *et al.*, 1993) is ignored. This relation is an important aspect of the underlying physics and its specification in the relevant initial boundary value problem cannot be omitted.

In the present work, the deformation of the powder aggregate is modeled by first determining the initial contact area due to elastic deformation that takes place instantaneously upon application of the external load. Subsequently, densification is considered to proceed under the combined action of linear elasticity and power-law creep in the bulk of the particles while diffusional mass transport occurs along the interparticle contact areas and pore surfaces. A finite element formulation is developed and implemented to solve the relevant initial boundary value problem under plane strain conditions. The pore surface energy appears in the formulation as it affects the pore surface diffusion and the magnitude of the stress acting normal to the pore surface. The surface energy of the interparticle contact area (grain boundary energy) is also accounted for in the formulation through the energy changes that occur when matter from the contact area is deposited onto the pore free surface. The proposed numerical scheme to solve the finite element equations is marched in time in order to capture the evolution of the pore shape and the closing of the porosity. The method provides an attractive means to keep track of the variation of the flux and curvature of the pore surface and the flux on the interparticle areas. The tip stress (i.e. the stress at the boundary of the interparticle area) and the corresponding curvature are calculated as part of the overall system equilibrium employed in the analysis and as a result, there is no need to make any restrictive assumptions with regard to their magnitude.

2. MATERIAL CONSTITUTIVE LAWS

The bulk of the particles is modeled as a homogeneous and isotropic material that deforms by linear elasticity and power-law creep. The total strain rate in the bulk is defined through the velocity as

$$\dot{\epsilon}_{ij} = (v_{i,j} + v_{j,i})/2, \quad (1)$$

where $(.)_{,j} = \partial(.)/\partial x_j$ and the superposed dot denotes differentiation with respect to time. The total strain is decomposed into elastic and creep strain components such that

$$\dot{\epsilon}_{ij} = \dot{\epsilon}_{ij}^e + \dot{\epsilon}_{ij}^c. \quad (2)$$

The elasticity of the particle is characterized by the standard Hooke's law

$$\sigma_{ij} = C_{ijkl} \epsilon_{kl}^e, \quad (3)$$

where C_{ijkl} are the components of the elastic modulus tensor, and the Einstein summation convention is implied over a repeated index. The creep strain rate $\dot{\epsilon}_{ij}^c$ which obeys incompressibility is derived from

$$\dot{\epsilon}_{ij}^c = \frac{\partial \phi^c}{\partial \sigma_{ij}}, \quad (4)$$

where

$$\phi^c = \frac{1}{1+n} \dot{\epsilon}_0 \sigma_0 \left(\frac{\sigma_e}{\sigma_0} \right)^{1+n} \quad (5)$$

is the power-law creep potential, $\sigma_e = \sqrt{3s_{ij}s_{ij}/2}$ is the effective stress, $s_{ij} = \sigma_{ij} - (\sigma_{kk}/3)\delta_{ij}$ is the deviatoric stress, n is the creep exponent, $\dot{\epsilon}_0$ and σ_0 are material constants, and δ_{ij} is the Kronecker delta. From Eqs. (4) and (5), one finds

$$\dot{\epsilon}_{ij}^c = 3C\sigma_e^{n-1}s_{ij}/2, \quad (6)$$

where $C \equiv \dot{\epsilon}_0/\sigma_0^n$ is the creep modulus.

Diffusion along the interparticle contact areas or the pore surface is motivated by chemical potential gradients such that

$$j_b = -\frac{D_b\delta_b}{kT} \frac{d\mu}{ds} \quad (7)$$

along the interparticle area, and

$$j_p = -\frac{D_p\delta_p}{kT} \frac{d\mu}{ds}, \quad (8)$$

along the pore surface. The parameter μ denotes the chemical potential of an atom, j_b and j_p are the volumetric fluxes per unit length along a direction s tangent to the interparticle area and pore surface respectively, D_b and D_p are the corresponding diffusion coefficients, δ_b and δ_p are the corresponding effective thicknesses through which matter diffuses, k is Boltzmann's constant and T is the absolute temperature.

Following Herring (1951), one can write

$$\mu = -\sigma_n \Omega \quad (9)$$

along the interparticle area, and

$$\mu = -\gamma_p(k_1 + k_2)\Omega \quad (10)$$

along the pore surface, where σ_n is the stress normal to the contact area, k_1 and k_2 are the principal curvatures of the pore surface, and Ω is the atomic volume of the diffusing species. Combining Eqs. (7) and (9), and (8) and (10), yields respectively

$$j_b = \mathcal{D}_b \frac{d\sigma_n}{ds} \quad (11)$$

and

$$j_p = \mathcal{D}_p \frac{d[\gamma_p(k_1 + k_2)]}{ds}, \quad (12)$$

where now $\mathcal{D}_b = D_b\delta_b\Omega/kT$ and $\mathcal{D}_p = D_p\delta_p\Omega/kT$ are correspondingly interparticle and pore surface diffusivities having dimensions of volume divided by stress per unit time.

Matter conservation along the interparticle contact area requires

$$\frac{dj_b}{ds} + \dot{h} = 0, \quad (13)$$

where \dot{h} is the normal velocity of interpenetration that would have resulted had the particles penetrated into one another under the action of the external loads. Denoting by

\dot{g} the rate of change of the gap g (see Fig. 1a) between two points on the boundaries of particles I and II which are coming into contact with each other with corresponding velocities v_i' and v_i'' , one can calculate the normal interpenetration velocity as

$$\dot{h} = (v_i'' - v_i') n_i + \dot{g}, \quad (14)$$

where n_i are the components of the unit normal to the contact area as shown in Fig. 1a. Notice that at any point along the contact area ($g = 0$, $\dot{g} = 0$), any change in the relative velocity $(v_i'' - v_i') n_i$ results directly in a change in the volumetric flux. Matter conservation for diffusion along the pore surface dictates that

$$\frac{dj_p}{ds} - v_n = 0, \quad (15)$$

where v_n is the velocity component normal to the pore surface with the unit normal vector pointing into the particle. A negative value of v_n implies that material is being deposited on the pore surface whereas a positive value implies erosion of the surface.

Considering the pore surface as a stretched membrane under the action of a constant surface tension and normal stresses σ_n from the adjoining bulk material, one can state the requirement for equilibrium at any arbitrary point by means of the standard Laplace equation (Gurtin and Murdoch, 1975; Rice and Chuang, 1981; Freund *et al.*, 1993)

$$\sigma_n = \gamma_p (k_1 + k_2). \quad (16)$$

The sign conventions for the curvature and the normal stress are shown in Fig. 1b. At the junction between the interparticle contact area and the pore surfaces, both volumetric flux and chemical potential are continuous. By Eqs. (10) and (11), chemical potential continuity requires that

$$(\sigma_n)_0 = [\gamma_p (k_1 + k_2)]_0, \quad (17)$$

where the subscript 0 is used to denote values at the junction.

3. THE UNIT CELL -- INITIAL BOUNDARY VALUE PROBLEM IN PLANE STRAIN

The problem of densification of a square array of cylinders under hydrostatic loading is considered (see Fig. 2). So the present analysis simulates hot isostatic pressing (HIPing) under plane strain deformation conditions. Pure HIPing was considered solely to reduce the amount of numerical complexities that would arise from interparticle slip if the applied macroscopic stress had a nonzero deviatoric component. It should be emphasized that the formulation developed in this work is equally valid for general loading states, though an appropriate unit cell has to be chosen to accommodate the applied stresses and appropriate modifications to the governing equations have to be made to account for interparticle slip (Needleman and Rice, 1980 and Casagrande and Sofronis, 1997). Due to the symmetry of the arrangement, the densification is studied by considering the deformation of just one quadrant of the cylinder as shown in Figs. 2 and 3. Clearly, this cell is only one of many that could have

been used. This particular choice has been made in view of the fact that the unit normals to the contact areas are known *a priori*. This offers a significant simplification with respect to the numerical evaluation of the contact areas and reduces the number of iterations and computation time considerably. As is evident from Fig. 3, there are two straight contact areas (line segments BC and DE) that develop upon loading, with the pore surface (arc CD) lying in between. Lastly, in view of the plane strain assumption, the only nonzero curvature is the in-plane one and it is denoted by k .

Extending the formulation of Needleman and Rice (1980) to include the effects of interface and pore surface energies, and pore surface diffusion, one can cast the governing equilibrium equations for the deformation of the cell in the following variational form

$$\begin{aligned} \int_{S_T} T_i \delta v_i ds = & \int_A \sigma_{ij} \delta \dot{\epsilon}_{ij} dA + \int_{S_b} \sigma_n \delta v_n ds + \int_{S_p} \sigma_n \delta v_n ds \\ & + \gamma_b \dot{a}_1 - \gamma_p (\delta v_{ip} \cos \phi)_C + \gamma_b \dot{a}_2 - \gamma_p (\delta v_{ip} \cos \phi)_D. \end{aligned} \quad (18)$$

Here, T_i is the specified traction on the boundary S_T over which tractions are prescribed, σ_{ij} is the stress tensor within the bulk region A occupied by the particles (Fig. 3), δv_i is an arbitrary virtual variation of the velocity that vanishes on the part where velocities are prescribed, $\delta \dot{\epsilon}_{ij}$ is the corresponding strain rate variation, σ_n is the normal stress either on the interparticle area S_b or the pore surface area S_p (Fig. 3), δv_n is the virtual variation of the normal velocity on S_b and S_p , s denotes arclength along the boundary BCDE measured from point B (Fig. 3), and ϕ is the dihedral angle at points C and D and is schematically shown in Fig. 1b. The parameters \dot{a}_1 and \dot{a}_2 are rates of change of the two contact areas and δv_{ip} is the velocity at the junctions measured positive if it causes an increase in the area of contact. The term on the left-hand side of the equation is the external virtual power delivered to the system and the terms on the right sum up to the total internal power. The integral evaluated on S_p represents the power expended in changing the pore surface area as it moves normal to itself during loading. The terms involving γ_b and γ_p represent the power expended/released as the densification proceeds and matter exchange occurs between interparticle boundaries and pore surface accompanied by corresponding changes in surface energy.

Starting with Eq. (11) and using the divergence theorem, one can state the principle of virtual power on the interparticle contact areas as

$$\int_{S_b} \sigma_n \delta v_n ds = \int_{S_b} \frac{1}{\mathcal{D}_b} j_b \delta j_b ds + (\sigma_n \delta j_b)_D - (\sigma_n \delta j_b)_C, \quad (19)$$

where δj_b and δv_n are respectively virtual flux and normal velocity fields that satisfy the zero flux condition at points B and D and are related by $d\delta j_b/ds + \delta v_n = 0$. Substituting Eqs. (19) and (16) into Eq. (18), one obtains the final variational statement for the densification process as follows:

$$\begin{aligned}
\int_{S_r} T_i \delta v_i ds = & \int_A \sigma_{ij} \delta \varepsilon_{ij} dA + \int_{S_p} \gamma_p k \delta v_n ds + \int_{S_b} \frac{1}{\mathcal{D}_b} j_b \delta j_b ds + (\sigma_n \delta j_b)_D - (\sigma_n \delta j_b)_C \\
& + \gamma_b \dot{a}_1 - \gamma_p (\delta v_{tip} \cos \phi)_C + \gamma_b \dot{a}_2 - \gamma_p (\delta v_{tip} \cos \phi)_D.
\end{aligned} \tag{20}$$

Equation (20) is solved for displacements by the finite element method. Using appropriate interpolation matrices for the relevant quantities, the variational statement is converted into a set of non-linear equations for the nodal velocities, or equivalently, the nodal displacement increments. There are two sources of non-linearity in the resulting equations: the power-law creep nature of the material, and the development and evolution of the interparticle contact areas. The contact areas, which are domains for the interparticle diffusion process, are not known beforehand and are determined as part of the solution. To this end, an iterative scheme has been proposed to systematically evaluate the contact areas as well as the creep strains occurring in the particles.

With regard to the contact areas, the strategy is to assume sizes for the contact areas that are in excess of the expected, and solve the boundary value problem at a given time t_{n+1} for displacements and stresses when the solution is known at time t_n . The solution is obtained by calculating increments to the known displacements at time t_n such that the governing Eq. (20) is satisfied at time t_{n+1} . Then, the stresses along the contact areas are checked for consistency, i.e., whether the stresses are compressive and nodes assumed to be diffusive are indeed in contact. If either of these conditions is not met, the assumed contact areas are reduced by a node and the procedure is repeated till the solution obtained is consistent.

In setting up the finite element equations for the displacement increments, the creep strain increment portion (cf. Eq. (2)) of the total strain increment associated with a displacement increment is unknown. As a result, the procedure for obtaining the solution for each set of assumed contact areas BC and DE (Fig. 3) is also iterative in nature (see Appendix A). For an assumed displacement increment, a creep strain increment is calculated pointwise as shown in Appendix C by the Newton iteration method ("local" iterative scheme). The finite element equations are then set up and solved for a new displacement increment also by the Newton iteration method ("global" iterative scheme). The creep strain rates are then corrected and these coupled "local" and "global" iterative schemes continue till convergent increments of displacement and creep strain are obtained, of course, for the assumed contact areas BC and DE. The creep strain rate and displacement increments at time t_n were used to initiate the process. Thus, since the solution to the initial boundary value problem involves also the simultaneous evaluation of the contact areas, a three-level iterative scheme has been devised to perform the task.

On the pore surface, except at the tip nodes at C and D (Fig. 3), the curvature is calculated at nodal positions i using the local position coordinates and those of the two neighboring nodes, $i-1$ and $i+1$. Thus (see Appendix D),

$$k_i = k_i(x_{i-1}, y_{i-1}, x_i, y_i, x_{i+1}, y_{i+1}), \tag{21}$$

where x_i and y_i are nodal coordinates. For the curvature at the tip, a quadratic extrapolation based on the curvatures at the three neighboring pore surface nodes is used. It ought to be emphasized that this way of calculating curvature is a point of departure from most of the previous work wherein, the tip curvature was calculated from the value of the dihedral angle, which was assumed to be strictly valid during the entire densification process.

The variational statement (20) enforces equilibrium in the bulk of the particles and ensures the satisfaction of the diffusion equation Eq. (11) on the interparticle boundaries BC and DE. However, it does not enforce the pore surface diffusion relation given by Eq. (12) or the local stress-curvature relation given by Eq. (16). These equations are strictly enforced as side constraints at time t_{n+1} during the iteration process. Since the numerically obtained solution to the stress, flux and curvature at time t_{n+1} is a function of the displacement increment, strict enforcement of Equations (12) and (16) leads to multi-point constraints along the boundary S_p . Moreover, the flux vanishes at points B and E owing to the square arrangement of particles. At B, the flux is zero by construction (see Appendix A) and at E, the zero flux boundary condition is enforced through a multi-point constraint that involves all the degrees of freedom (nodal velocities) on S_b and S_p .

4. DIMENSIONLESS GROUPS

A dimensional analysis over the geometric, material and loading parameters of the system leads to the following functional representation for the dependent variables:

$$\mathbf{a} = \mathbf{f}(\sigma_a/E, \nu, n, \psi_b, \psi_p, \chi_b, \chi_p, t/t_d), \quad (22)$$

where \mathbf{a} is the non-dimensional version of a typical parameter such as displacement, strain-rate, stress etc., σ_a is the applied macroscopic stress on the aggregate (Fig. 3),

$$\psi_b = \frac{\gamma_b}{\sigma_a R}, \quad (23)$$

$$\psi_p = \frac{\gamma_p}{\sigma_a R}, \quad (24)$$

$$\chi_b = \frac{\mathcal{D}_b}{C \sigma_a^{n-1} R^3}, \quad (25)$$

$$\chi_p = \frac{\mathcal{D}_p \gamma_p}{C \sigma_a^n R^4}, \quad (26)$$

R is the particle radius, t_d is a time parameter that represents a characteristic time scale of the densification process, and isotropic elasticity was assumed for the particles with Young's modulus E and Poisson's ratio ν .

Clearly, ψ_b and ψ_p represent the strengths of the interface and pore surface tension in relation to the applied stress. The higher these values are, the larger is the energy required to be delivered by the applied load to change the respective areas of these surfaces. On the pore surface, a higher value of ψ_p implies higher normal

stresses in the adjoining bulk material. Next, it is shown that the dimensionless groups χ_b and χ_p characterize respectively the relative rates at which the interparticle and pore surface diffusion processes proceed in comparison to the power-law creep deformation of the bulk.

Groups χ_b and χ_p can be arrived at by considering the magnitudes of strain rates that are representative of the three densification processes, namely power-law creep deformation in the bulk, interparticle diffusion, and pore surface diffusion. A measure of the creep strain rate is readily expressed in terms of the creep constant and the applied stress through $\dot{\epsilon}^c = C\sigma_a^n$. The diffusion processes can be thought of as resulting in strain rates, which in plane strain can be described by

$$\dot{\epsilon}^b \approx \frac{\dot{A}_b}{A} \approx \frac{j_b}{R^2} = \frac{(\mathcal{D}_b \sigma_a / R)}{R^2} = \frac{\mathcal{D}_b \sigma_a}{R^3} \quad (27)$$

in the case of interparticle diffusion, and

$$\dot{\epsilon}^p \approx \frac{\dot{A}_p}{A} \approx \frac{j_p}{R^2} = \frac{(\mathcal{D}_p \gamma_p / R^2)}{R^2} = \frac{\mathcal{D}_p \gamma_p}{R^4} \quad (28)$$

in the case of pore surface diffusion, where \dot{A}_b and \dot{A}_p are measures of rates at which the interparticle and pore surface diffusion processes transport mass per unit length in the out-of-plane direction and A is a measure of the area of the particle.

Characteristic time scales for each of the three rate mechanisms can be defined as the time required for the rate process to yield a strain equal to a reference strain, which is chosen as the elastic strain corresponding to the applied stress σ_a . Thus, the following characteristic times can be defined:

$$t_c = \frac{\epsilon^e}{\dot{\epsilon}^c} = \frac{\sigma_a / E}{C\sigma_a^n} = \frac{1}{EC\sigma_a^{n-1}} \quad (29)$$

for power-law creep deformation,

$$t_b = \frac{\epsilon^e}{\dot{\epsilon}^b} = \frac{R^3}{\mathcal{D}_b E} \quad (30)$$

for interparticle diffusion, and

$$t_p = \frac{\epsilon^e}{\dot{\epsilon}^p} = \frac{R^4 \sigma_a}{\mathcal{D}_p E \gamma_p} \quad (31)$$

for pore surface diffusion. Therefore, by means of Eqs. (27), (29) and (30), definition (25) is rewritten as

$$\chi_b = \frac{\dot{\epsilon}_b}{\dot{\epsilon}_c} = \frac{t_c}{t_b} = \frac{\mathcal{D}_b}{C\sigma_a^{n-1}R^3}. \quad (32)$$

Similarly, by means of Eqs. (28), (29) and (31), definition (26) is rewritten as

$$\chi_p = \frac{\dot{\epsilon}_p}{\dot{\epsilon}_c} = \frac{t_c}{t_p} = \frac{\mathcal{D}_p \gamma_p}{C\sigma_a^n R^4}. \quad (33)$$

It is obvious from Eqs. (32) and (33) that, for instance, if $\chi_b = 10$ and $\chi_p = 100$, one can infer that the interparticle and pore surface diffusion processes are respectively one and

two orders of magnitude faster than the power-law creep process in the bulk of the particles. Since the speed of the densification process is controlled by the smallest time constant, $t_d = \min(t_c, t_b, t_p)$ in Eq. (22).

5. NUMERICAL RESULTS

A notable feature of the formulation is that if the interparticle diffusion coefficient is assumed to have a very small value, and the free surface diffusion coefficient is set equal to zero, and any surface energy effects are neglected, it is equivalent to a penalty method treatment of the initial elastic contact problem. This feature was employed to calculate the elastic solution at time $t = 0$, which serves as the starting point for the subsequent time-dependent deformation of the unit cell. The solution was obtained for successive time steps by the procedure outlined in Section 3 and macroscopic quantities such as density and densification rates were calculated. The undeformed particle radius was $R = 1$ and this corresponds to an initial area fraction of particles of 0.785. Eight-noded isoparametric elements were used with a 2×2 rule for integration of stiffness matrices. The incompressibility of the creep deformation was enforced by the method of Nagtegaal, Parks and Rice (1974). The mesh used in the calculation has 920 elements and 2481 nodes, out of which 321 were sequentially numbered and placed along the arc BCDE to discretize the contact areas and the pore surface. The elasticity of the particles was assumed to be isotropic, and unless stated otherwise, the finite element results were obtained for $\chi_b = 10$, $\chi_p = 100$, $\psi_p = \psi_b = 0.1$, $E/\sigma_a = 100$, $\nu = 0.33$, and creep exponent equal to 5. The applied macroscopic stress state was $\Sigma_{xx} = \Sigma_{yy} = -\sigma_a$ as shown in Fig. 3. With this choice of data, the smallest time constant is that for pore surface diffusion and hence $t_d = t_p$ (see Eq. (31)). The initial marching time step was $\Delta t = 10^{-6} t_p$, and as the powder aggregate relaxed and densified, the magnitude of the time step was increased. Calculations were carried out till the total elapsed time was equal to $t_p/2$. At this stage, nodal displacements were observed to be quite large and the present small-strain scheme with no nodal updating was deemed to be inadequate and hence no further computations were carried out. If t is the total elapsed time, then one can conclude from Eq. (29) that

$$\frac{\epsilon^e}{\epsilon^c} = \frac{t_c}{t} = \frac{\chi_p t_p}{t} \quad (34)$$

Since the largest value of t/t_p is 0.5 and $\chi_p = 100$, it is clear that during the entire range of deformation considered, the magnitude of the creep strains is small when compared to the elastic strains.

Figure 4 shows the flux profiles on the interparticle contacts and pore surface as a function of the arc length s at various values of the normalized time t/t_d . Note that $s = 0$ at point B and $s = s_{max}$ at point E (Fig. 3). The flux is normalized by a reference value $j_{ref} = \mathcal{D}_b \sigma_a / s_{max}$ and the stress by the magnitude of the applied stress σ_a . Peaks

in the magnitude of the volumetric flux coincide with the location of the boundaries of the interparticle areas (that is, points C and D), and the midpoint of the pore is a point of reflection for the flux profile. The anti-symmetry of the flux profiles about $s = s_{\max}/2$ derives from the hydrostatic compaction feature, that is, equal amounts of mass diffuse out of the two contact areas BC and DE toward the pore surface. The flux profiles flatten out as time elapses and this is again in accordance with diffusion-induced stress and pore curvature relaxation whereby the gradients in stress and curvature reduce with time, thus resulting in less amounts of volumetric diffusion.

A typical normal stress σ_n/σ_a profile is shown in Fig. 5 for $0 \leq s \leq s_{\max}$ along with the corresponding curve for flux at time $0.025t_d$. The normal stress becomes less compressive as one approaches the tips (points C and D in Fig. 4) and this translates to monotonically increasing values of volume flux. The tips are extremum points for the normal stress gradient and this is reflected in the maximum magnitude of the volume flux. As one moves past the tips and toward the pore surface, the normal stress variations become significantly less, but the higher value of pore surface diffusion parameter still leads to comparable flux magnitudes. The stress variation is symmetric with respect to the center of the pore surface and this agrees with the anti-symmetric nature of the flux profile. A snapshot of the flux and curvature variations on the pore surface at time $t = 0.025t_d$ is shown in Fig. 6. This plot illuminates the characteristics of surface diffusion over the pore. As one would expect from Eq. (12), the flux is negative when the curvature decreases, positive when it increases, and zero at maxima or minima. Figure 6 also indicates that the maximum absolute value of the non-dimensionalized curvature kR is less than 2. Therefore, for $\psi_p = 0.1$, Eq. (16) phrased in the form $\sigma_n/\sigma_a = kR\psi_p$, yields values for the non-dimensionalized stress σ_n/σ_a that are less than -0.2 . This result is in excellent agreement with the normal stress levels over the pore surface shown in Fig. 5.

In Fig. 7, the normalized curvature kR of the pore surface is shown plotted against normalized arclength s/s_{\max} at various instants of time. In accordance with the curvature sign convention, the curvature is negative throughout since the pore is convex. The expected symmetric variation about the mid-point is observed. As deformation proceeds, the entire curvature profile shifts upwards, i.e., the pore deforms so that its curvature gradients are neutralized and it becomes less curved.

Figure 8 shows the outline of the unit cell at different stages of deformation. One can readily observe the systematic increase in the sizes of the interparticle contact areas as well as the pattern of mass distribution. While the value of the surface diffusion parameter controls the pattern of mass distribution, thickening of the pore surface occurs due to deposition of mass moving out of the contact areas. For $\chi_p = 100$, while there seems to be a reasonably even distribution overall, more mass is deposited in the regions near the tip of the pore as compared to the interior.

In order to investigate the magnitude of the normal tip stress σ_0 (cf. Eq. (17)), additional calculations were carried out for $\chi_p = 10^4$, while the values of the other

parameters were held unchanged. Due to the nature of the loading, the stresses at the two tips are equal to each other. Figure 9 shows the variation of the tip stress plotted as a function of relative density. The deformation of the unit cell immediately following the initial elastic contact was characterized by sharp gradients in the stresses and rapid rates of increase in the contact sizes. As a result, the numerical scheme employed in this study was unable to capture the magnitude of the tip stresses with the desired accuracy. As mentioned in Appendix A, the scheme for calculation of the tip stress is an approximate one and this may be the reason for the lack of accuracy in the early stages. However, it should be emphasized that the solution for the displacement was always accurate, even during the early stages of deformation. The local tip stress values stabilized after additional densification and Fig. 10 shows the variation after the initial transients. It is seen that throughout the densification process, the stress at the tip is compressive. The initial, instantaneous elastic deformation leads to negligibly small compression at the beginning, but as soon as the diffusion processes set in, a dramatic drop in the stress occurs. Subsequently, as time elapses, there is a decrease in the magnitude of the tip stresses and this trend is milder in the case of $\chi_p = 10^4$ while the $\chi_p = 10^2$ case is characterized by a more pronounced variation. The figures also show the corresponding profiles of $(\gamma_p k)_0$ whose data points are indistinguishable from those of the tip stress, in accordance with Eq. (17).

Comparison with other models

The complete solution to the initial boundary value problem for the unit cell facilitates the calculation of the relative density D of the powder compact by determining the area fraction occupied by the particles. The densification rate \dot{D} is then computed through the change ΔD occurring during the time increment Δt . Expressions for \dot{D} are available in the literature from models based on different principal densifying mechanisms acting alone.

McMeeking and Kuhn (1992) considered the densification of an initially random close-packed aggregate of spherical particles deforming primarily by interparticle diffusion. They report the densification rate to be

$$\dot{D} = \frac{36(1 - D_0)^2}{D(D - D_0)^2} \left(\frac{D}{D_0} \right)^{\frac{2}{3}} \frac{\mathcal{D}_b p}{R^3}, \quad (35)$$

where D_0 is the initial relative density which for random close-packing is 0.64 and p is the applied pressure. The numerical results of the present model are compared with those of Eq. (35) at the same macroscopic pressure prevailing during the densification process. This macroscopic pressure is essentially elastic, in view of the negligible creep response of the cell, and therefore it is taken to be $p = 2(1 + \nu)\sigma_a/3$ under plane strain. Then, recalling definition (25) of χ_b , one can recast Eq. (35) into

$$\dot{D} = 24(1 + \nu) \frac{(1 - D_0)^2}{D(D - D_0)^2} \left(\frac{D}{D_0} \right)^{\frac{2}{3}} \chi_b C \sigma_a^n. \quad (36)$$

A slightly different result was obtained by Helle *et al.* (1985) for densification under pure interparticle diffusion. Making similar substitutions, one can rewrite their formula for the densification rate as

$$\dot{D} = \frac{86}{3}(1 + \nu) \frac{(1 - D_0)^2}{(D - D_0)^2} \chi_b C \sigma_a^n. \quad (37)$$

For the case of compaction by power-law creep deformation alone, Kuhn and McMeeking (1992) calculate that the densification rate for purely hydrostatic compression is given by

$$\dot{D} = \frac{27\pi}{16\sqrt{3}} \left[\frac{2(1 + \nu)}{9} \right]^n D \left(\frac{D - D_0}{1 - D_0} \right)^{\frac{1}{2}} \left[\frac{(1 - D_0)}{D^2(D - D_0)} \right]^n C \sigma_a^n. \quad (38)$$

Under the same conditions, Helle *et al.* (1985) calculate the densification rate to be

$$\dot{D} = \frac{5.3}{\sqrt{3}} \left[\frac{2(1 + \nu)}{9} \right]^n (D^2 D_0)^{\frac{1}{3}} \left(\frac{D - D_0}{1 - D_0} \right)^{\frac{1}{2}} \left[\frac{(1 - D_0)}{D^2(D - D_0)} \right]^n C \sigma_a^n. \quad (39)$$

Figure 10 shows the comparison of the densification rate obtained using the finite element results of the present model at $\chi_b = 10$ and two different values of χ_p , namely 10^2 and 10^4 , with the corresponding results that Eqs. (36)-(39) yield. The interparticle diffusion models of Eqs. (36) and (37) agree very well with each other and so do the two power-law creep models of Eqs. (38) and (39). The densification rates furnished by the interparticle diffusion models are several orders of magnitude greater than those from the power-law creep models, as expected, since densification by interparticle diffusion is much faster than by power-law creep in the bulk. The present investigation predicts initial densification rates an order of magnitude higher than those of the grain boundary diffusion models, but these decrease rapidly and fall between the values calculated from the models employing interparticle diffusion alone and those using power-law creep alone. Obviously, the very high initial densification rates during the early stages (at low relative density) result from the rapid diffusion processes associated with the initial, severe elastic stress gradients. As densification proceeds and the diffusional relaxation mechanisms alleviate and neutralize these gradients (see Fig. 4), the densification rates drop to lower levels. The parameter χ_p seems to have a moderate but definite effect on the densification rate. It should be borne in mind that the diffusional creep models of Eqs. (36) and (37) are predicated on pore surface diffusion that is very fast in comparison to the interparticle diffusion. Therefore, the finite element results of Fig. 10 denoted by the lower curve ($\chi_p = 10^4$) should be considered in comparing the models.

Comparison with experimental data

Casagrande *et al.* (1996) performed a series of constant strain rate densification experiments using cylindrical compact specimens of porous alumina specimens under uniaxial compression over a wide variety of temperatures. The imposed strain rates varied between 3×10^{-6} and $3 \times 10^{-5}/s$ and the temperatures between $1050^\circ C$ and $1200^\circ C$. Analyzing their results, Casagrande *et al.* observed that when the densification rates were cast in the non-dimensional form

$$\alpha(t) = \frac{\dot{E} k T \mu d^2}{D_b \delta_b b^2} \left(\frac{1}{\Sigma(t)} \right)^2, \quad (40)$$

where \dot{E} is the applied compressive strain rate, Σ the measured macroscopic stress on the specimen, μ the shear modulus, d the particle diameter, and b the Burgers vector, the calculated values of α for all the experiments ranged from 1 to 10, and collapsed onto a narrow band that decreases in magnitude with increasing relative density.

Model calculations were carried out using data for alumina (Tables 1 and 2) at $1100^\circ C$ for three of the stress levels Casagrande *et al.* measured in their experiments, namely 100 MPa, 110 MPa and 125 MPa. The calculated macroscopic dilatation rate $\dot{E}_{kk} = -2\dot{E}_a$ (Fig. 3) was then used to derive a one-dimensional macroscopic compressive strain rate \dot{E} such that $3\dot{E} = -\dot{E}_{kk}$. Similarly, the macroscopic pressure $p = 2(1 + \nu)\sigma_a/3$ was set equal to the pressure $\Sigma/3$ in simple uniaxial compaction under compressive stress Σ . Next, Eq. (40) was used to calculate the values of the non-dimensionalized strain rate α associated with the finite element calculation results. These values are illustrated in Fig. 11. At a given stress Σ , the strain rate \dot{E} decreases with time and since $\alpha(t)$ is proportional to \dot{E} (Eq. (40)), so does α as shown in the figure. The values of α computed from strain rates obtained from the model calculations are shown by hollow symbols connected by dotted lines while the values of α corresponding to the experimental strain rates, calculated using Eq. (40), are shown by solid symbols connected by solid lines. Due to the limitations imposed by the small-strain assumption employed in the current scheme, calculations were performed over a small range of densification. The lowest strain rates corresponding to the finite element data shown in Fig. 11 are of the order of $10^{-4}/s$ which is still 1-2 orders of magnitude greater than those used in the experiments of Casagrande *et al.* Thus, the corresponding lowest values of α obtained from the present numerical calculations are of the order of around 500 which is notably also 1-2 orders of magnitude greater than the range of 1-10 obtained by Casagrande *et al.*. Keeping this observation in mind and looking at the monotonically decreasing trend exhibited by α , one may reasonably expect the values of α one could obtain numerically for the strain rates employed in the experiments of Casagrande *et al.* to agree well with those obtained experimentally.

Using cylindrical specimens, Kim (1997) carried out sinter-forging experiments on Ti-48Al powder with an average grain size of 10 microns. Compaction took place

under uniaxial compressive stress at values varying from 200–500 MPa and held constant during densification at a temperature of 625°C. With the use of material data for γ -TiAl (Tables 3 and 4) model calculations were performed under macroscopic stresses of $\sigma_a = 300, 400$ and 500 MPa. As with the definition of the macroscopic strain rate \dot{E} , a uniaxial compressive macroscopic strain E was defined through $E = 2(E^{el} + \int_0^t \dot{\epsilon}_a dt')/3$, where E^{el} is the strain in the unit cell at the end of the initial

elastic contact. Figure 12 shows this numerically calculated strain E as a function of time. As observed by Kim, the strains increase monotonically with time, and not surprisingly, higher stresses lead to higher strains. Experiments were performed by Kim for time durations as long as 5500 seconds, whereas due to the limitations of the current numerical scheme mentioned in the preceding paragraph, computations only spanned much shorter periods of time. In spite of this, quantitative agreement with Kim's experiments seems to be quite good. For instance, under a stress of 500 MPa, a strain of 0.8% was calculated at $t = 40s$, while the corresponding Fig. 44 of Kim (1997) shows a strain of around 1%.

6 DISCUSSION

From Fig. 4 it is seen that at time $t = 0.005t_d$, the flux reverses sign at two pairs of symmetrically placed points on the pore surface and by Eq. (12), the curvature exhibits stationary values at these points as shown in Fig. 6. However at time $t = 0.03t_d$, there is only one pair of points at which this happens and for the times $t = 0.125t_d$ and $t = 0.300t_d$, the flux decreases monotonically from tip to tip. Thus, the surface diffusion process is very effective in neutralizing the local curvature gradients and setting up a linearly varying flux pattern.

The calculations shown in Fig. 5 indicate that the interparticle boundaries and the pore surface are all in a state of compression. The compressive stress on the interparticle areas decreases as one moves towards the tip, thus setting up stress gradients that force the mass out through the boundaries and onto the pore surface. The chemical potential is higher on the pore surface at regions of higher convexity and atoms there tend to move toward regions of less convexity or concavity. Interparticle diffusion is the primary source of rearrangement of matter and hence, χ_b has a central role to play in determining the magnitude of mass flux at the tip of the interparticle contacts. However, it is also essential that the pore surface be capable of efficiently absorbing the mass emanating from the contact areas and therefore, the rate of mass transfer also depends directly on the magnitude of the surface diffusion parameter χ_p . Consequently, so does the pattern of redistribution of the flux coming onto the pore from the interparticle areas. Slow surface diffusion leads to an accumulation of mass near the tips and this in turn retards the flux at the tip. As a result, there is a decrease in the densification rate. On the other hand, fast surface diffusion enables the pore

surface to absorb all the mass flowing out of the contact areas and this leads to higher densification rates.

The above argument is clearly supported by Fig. 10 wherein the densification rates obtained for the two different values of χ_p are noticeably different. One can also see from the figure that modeling the processes accurately is very crucial in determining the macroscopic behavior. Depending on which mechanism is assumed to dominate, rates of densification that are several orders of magnitude apart can be obtained. The results shown in Fig. 10 also suggest that densification that occurs solely by grain boundary diffusion or power-law creep should be treated as a limiting case. At the outset, the densification rates are closer to those predicted by models of densification dominated by grain boundary diffusion, but as time elapses and diffusion slows down due to stress and curvature gradient relaxation, the bulk deformation in the particles tends to affect the process. It seems that at very high relative densities, the densification rates tend to approach those predicted by the power-law creep models.

As shown in Fig. 11, the present finite element model predicts dimensionless densification rates for Al_2O_3 for three different stress levels that are compatible with the experimental results of Casagrande *et al.* (1996). Also, there is good agreement between the predicted and experimentally obtained strain in the case of TiAl compacts (Kim, 1997). However, it is important to note that both sets of experiments were uniaxial compression tests while the present model simulates plane strain densification. Therefore, the comparisons that have been made using the derived one-dimensional stress and strains as explained in Section 5 involve a certain amount of approximation. Casagrande *et al.* concluded that powder compaction in their experiments was controlled by the interparticle diffusion mechanism. This is in agreement with our numerical results that indicate a synergistic dominance of interparticle and pore surface diffusion, at least during the early stages of deformation.

An important result of the present finite element calculations regards the prediction of the evolving shape of the pore surface during densification (see Fig. 8). Clearly, the curvature of the pore remains negative (convex) during the early stages of the densification. Thus, the phenomenon of curvature sign reversal along the pore surface, as postulated and employed in the analysis of some two-particle densification models, was not observed in the calculations. However, simulations spanning longer times are required in order to assess whether pore convexity continues to be maintained during the entire stage I densification process or, if that is not the case, to identify the time at which the pore surface begins to develop concave segments as the process advances toward stage II.

7 CLOSURE

Densification of a powder aggregate is a complex physical phenomenon involving multiple mechanisms characterized by a large number of material and geometric parameters. In the present study, important dimensionless groups have been

identified and this leads to a dramatic reduction in the number of the case calculations required to analyze the process.

A finite element methodology has been developed to study the densification process of a particle aggregate based on the interaction between the mechanisms acting at the microscale. In view of the complexity of the governing equations, one may consider the devised numerical techniques as fairly robust in predicting the values of both microscopic and macroscopic parameters when pore and particle shape changes are small. For predictions pertaining to times closer to the end of stage I densification, the present model needs to be modified to account for geometric nonlinearities.

The finite element calculations show that the overall densification rate is sensitive to the strength of the pore surface diffusion. Available models based on a single operating mechanism have been shown to either overestimate (diffusional creep) or underestimate (power-law creep) the densification rates due to the neglect of the coupling between the densification mechanisms. Lastly, during the early stages of the densification process, the pore surface curvature maintains the same sign all along the pore surface.

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TABLE 1: Material properties for Al_2O_3 (Helle *et al.*, 1985; Ashby, 1990).

Surface energy	$= 0.90 \text{ J/m}^2$
Atomic Volume Ω	$= 1.42 \times 10^{-29} \text{ m}^3$
Melting point T_m	$= 2320 \text{ K}$
Burgers vector b	$= 4.76 \times 10^{-10} \text{ m}$
Young's modulus E	$= 395 \text{ GPa}$ at room temperature
Shear modulus	
$\mu = \mu_0 \left[1 + \frac{(T - 300)}{T_m} \left(\frac{T_m d\mu}{\mu_0 dT} \right) \right]$	$\mu_0 = 155 \text{ GPa}$ at 300K
$\left(\frac{T_m d\mu}{\mu_0 dT} \right)$	$= -0.35$
<u>Power-law creep:</u>	$\dot{\epsilon} = D_c \left(\frac{\sigma}{\sigma_{ref}} \right)^n, D_c = 10^{-6} \exp\left[\frac{-Q_c}{RT_m} \left(\frac{T_m}{T} - 2 \right)\right]$
Activation energy Q_c	$= 477.0 \text{ kJ/mol}$
Power-law creep exponent n	$= 3$
Reference Stress σ_{ref}	$= 1250 \text{ MPa}$
<u>Volume diffusion:</u>	$D_v = D_{0v} \exp\left(-\frac{Q_v}{RT}\right)$
Activation energy Q_v	$= 477.0 \text{ kJ/mol}$
Pre-exponential D_{0v}	$= 1.2 \times 10^{-6} \text{ m}^2/\text{s}$
<u>Grain boundary diffusion:</u>	$D_b \delta_{ib} = D_{0b} \delta_{0b} \exp\left(-\frac{Q_b}{RT}\right)$
Activation energy Q_b	$= 419.0 \text{ kJ/mol}$
Pre-exponential $D_{0b} \delta_{0b}$	$= 8.6 \times 10^{-10} \text{ m}^3/\text{s}$
<u>Surface diffusion:</u>	$D_p \delta_p = D_{0p} \delta_{0p} \exp\left(-\frac{Q_p}{RT}\right)$
Activation energy Q_p	$= 500.0 \text{ kJ/mol}$
Pre-exponential $D_{0p} \delta_{0p}$	$= 8.6 \times 10^{-6} \text{ m}^3/\text{s}$

TABLE 2: Values of the dimensionless groups for alumina at 1100°C at different stresses.

Σ (MPa)	χ_b	χ_p	$\psi_b = \psi_p$	E/σ_a	$t_d (= t_p)$ (s)
100	43	81	0.228	8381	21.7
110	35	61	0.208	7619	23.9
125	27	42	0.183	6704	27.1

TABLE 3: Material properties for γ -TiAl (Ashby, 1990).

Surface energy $\gamma = 1.76 \text{ J/m}^2$
Atomic Volume $\Omega = 3.11 \times 10^{-29} \text{ m}^3$
Melting point $T_m = 1730 \text{ K}$
Young's modulus $E = 168 \text{ GPa}$ at room temperature
Power-law creep: $\dot{\epsilon} = D_c \left(\frac{\sigma}{\sigma_{ref}} \right)^n$, $D_c = 10^{-6} \exp\left[\frac{-Q_c}{RT_m} \left(\frac{T_m}{T} - 2 \right)\right]$
Activation energy $Q_c = 320.0 \text{ kJ/mol}$
Power-law creep exponent $n = 3$
Reference Stress $\sigma_{ref} = 1800 \text{ MPa}$
Grain boundary diffusion: $D_b \delta_{ib} = D_{0b} \delta_{0b} \exp\left(-\frac{Q_b}{RT}\right)$
Activation energy $Q_b = 180.0 \text{ kJ/mol}$
Pre-exponential $D_{0b} \delta_{0b} = 7.6 \times 10^{-13} \text{ m}^3/\text{s}$
Surface diffusion: $D_p \delta_p = D_{0p} \delta_{0p} \exp\left(-\frac{Q_p}{RT}\right)$
Activation energy $Q_p = 150.0 \text{ kJ/mol}$
Pre-exponential $D_{0p} \delta_{0p} = 7.6 \times 10^{-10} \text{ m}^3/\text{s}$

TABLE 4: Values of the dimensionless groups for γ - TiAl at 625°C at different stresses.

Σ (MPa)	χ_b	χ_p	$\psi_b = \psi_p$	E/σ_a	$t_d (= t_p)$ (s)
300	40	6485	0.0029	1269	78.1
400	22	2735	0.0022	952	104.1
500	14	1400	0.0018	761	130.1

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APPENDIX A: FINITE ELEMENT FORMULATION AND SOLUTION OF THE INITIAL BOUNDARY VALUE PROBLEM

A1. Formulation

Under plane strain conditions, the interparticle contact areas are assumed to be the two plane segments BC and DE of Fig. 3. Arclength s is measured starting from point B and increases toward point E. As time elapses, the contact areas develop and

typically different nodes define the end of the contact areas at different times. These nodes are not known at the start of the calculation and have to be determined as part of the solution. One contact segment is defined by the line $y = -R$ and the other by $x = R$. The normal n is defined on the contact areas as positive when it points inwards as shown in Fig. 1a. A non-negative scalar gap function $g(s)$ at any point of the particle surface is defined to be its normal distance from the contact plane. If a node is already in contact, the value of the gap function is set to 0. From the initial position of the nodes and their subsequent deformation, $g(s)$ is easily calculated since the normal and the position of the contact surface do not change with time. Once the gap function is established, for $v'' = 0$ and $v' = v$, Eq. (14) yields

$$\dot{h} = \dot{v}_n + \dot{g} = v_\alpha n_\alpha + \dot{g}. \quad (A1)$$

Here $\dot{g} = g/\Delta t$ and measures the rate at which a node that was separated by an amount g from the contact plane bridges the gap between them. With the above definition, Eq. (13) is modified as

$$\frac{dj_b}{ds} + v_\alpha n_\alpha + \dot{g} = 0, \quad (A2)$$

where $j_b(s)$ is the flux on the contact areas measured as a function of s and is positive if the mass flow is along the direction of increasing s . In Fig. 3, $j_b > 0$ if the flow is from B to C and from D to E.

Integrating the above equation with $j_b(s_B) = 0$, one can express the flux on BC as

$$j_b(s) = -\int_0^s v_\alpha(s') n_\alpha ds' - G_1(s), \quad (A3)$$

where

$$G_1(s) = \int_0^s \frac{g(s')}{\Delta t} ds' \quad (A4)$$

and the subscript on G indicates that the integration is carried out over S_{b1} .

Since the flux is continuous at the junction of the contact areas and the pore surface, the flux on the pore surface is easily expressed as

$$j_p(s) = j_b(s_C) - \int_{s_C}^s v_\alpha(s') n_\alpha ds'. \quad (A5)$$

Similarly, the flux on S_{b2} is obtained as

$$j_b(s) = j_p(s_D) - \int_{s_D}^s v_\alpha(s') n_\alpha ds - G_2(s). \quad (A6)$$

Following Needleman and Rice (1980), one can interpolate the velocity on the contact area and pore surface using piecewise linear functions as

$$v_\alpha = \sum_i v_\alpha^i \phi_i(s), \quad (A7)$$

wherein i denotes the node number and the summation is carried out over all the nodes on $(S_{b1} + S_p + S_{b2})$, $\alpha = 1, 2$ and ϕ_i has the familiar form

$$\phi_i(s) = \left\{ \begin{array}{ll} \frac{s - s_{i-1}}{s_i - s_{i-1}}, & s_{i-1} \leq s < s_i \\ \frac{s_{i+1} - s}{s_{i+1} - s_i}, & s_i \leq s < s_{i+1} \\ 0, & s < s_{i-1} \text{ or } s > s_{i+1} \end{array} \right\}. \quad (\text{A8})$$

The flux on the interfaces and the pore surface can now be calculated analytically in terms of the nodal velocities by introducing these velocity interpolation functions, (A8) into the flux integrals, Eqs. (A3), (A5), and (A6), and carrying out the integrations. If one defines m_α^i as

$$m_\alpha^i(s) = \int_0^s \phi_i(s') n_\alpha ds', \quad (\text{A9})$$

then the flux on the interface and the pore surface can be generally expressed as

$$j(s) = - \sum_i m_\alpha^i(s) v_\alpha^i - G(s) \quad (\text{A10})$$

and the function $G(s)$ is evaluated according to the relation

$$G(s) = \left\{ \begin{array}{ll} G_1(s), & s \leq s_C \\ G_1(s_C), & s_C < s \leq s_D \\ G_1(s_C) + G_2(s), & s > s_D \end{array} \right\} \quad (\text{A11})$$

In matrix notation,

$$j(s) = \{M(s)\}^T \{v^N\} - G(s), \quad (\text{A12})$$

where $\{M(s)\}$ is a $2(|N_E - N_B| + 1) \times 1$ array whose entries are the terms $m_\alpha^i(s)$, $\{v^N\}$ is the array of nodal velocities on the surface ($S_{b1} + S_p + S_{b2}$) and N_E and N_B are the node numbers at the points E and B respectively.

If Δs is an infinitesimal segment of S_p , it can be readily shown by geometric argument that

$$k \delta u_n \Delta s = \delta(\Delta s) \quad (\text{A13})$$

and thus,

$$\int_{S_p} \gamma_p k \delta v_n ds = \int_{S_p} \gamma_p \delta ds = \sum_{i=1}^{N_{el}^p} \gamma_p \delta l_i, \quad (\text{A14})$$

where N_{el}^p is the number of element sides on the pore surface and l_i is the length of the i th side. Here, it is to be noted that the lengths of the elements change solely due to the motion of the pore surface normal to itself.

In order to cast the above relation in terms of the virtual nodal velocities and displacement increments, consider the i th pore surface edge l^i defined by the nodes i and $i+1$. Let the angles that the tangents to the pore surface make with the x-axis at these nodes be θ^i and θ^{i+1} respectively at time t_n . Virtual velocities δv_x^i and δv_y^i act at

node i . Then, the virtual velocity along the direction normal to the pore surface pointing into the particle is

$$\delta v_n^i = -\delta v_x^i \sin \theta^i + \delta v_y^i \cos \theta^i. \quad (\text{A15})$$

If one denotes the position at time t_{n+1} of node i by the vector X^i , then the new virtual position x^i due to its virtual motion in the normal direction is given by

$$\begin{aligned} x^i &= X^i + \delta v_n^i \Delta t n \\ &= X^i + \Delta t (-\delta v_x^i \sin \theta^i + \delta v_y^i \cos \theta^i) (-\sin \theta^i e_x + \cos \theta^i e_y) \\ &= X^i + \Delta t [(\delta v_x^i \sin^2 \theta^i - \delta v_y^i \sin \theta^i \cos \theta^i) e_x + (-\delta v_x^i \sin \theta^i \cos \theta^i + \delta v_y^i \cos^2 \theta^i) e_y] \end{aligned} \quad (\text{A16})$$

where e_x and e_y are the Cartesian unit vectors.

Use of the shorthand $S^i = \sin \theta^i$ and $C^i = \cos \theta^i$ permits one to write

$$x^i = X^i + \Delta t [(\delta v_x^i (S^i)^2 - \delta v_y^i S^i C^i) e_x + (-\delta v_x^i S^i C^i + \delta v_y^i (C^i)^2) e_y] \quad (\text{A17})$$

Similarly, for the node $i+1$ one obtains

$$x^{i+1} = X^{i+1} + \Delta t [(\delta v_x^{i+1} (S^{i+1})^2 - \delta v_y^{i+1} S^{i+1} C^{i+1}) e_x + (-\delta v_x^{i+1} S^{i+1} C^{i+1} + \delta v_y^{i+1} (C^{i+1})^2) e_y] \quad (\text{A18})$$

The new virtual length of the edge is the virtual distance between the two nodes after application of the virtual velocities, i.e.,

$$l_{new}^i = |(x^{i+1} - x^i)| \quad (\text{A19})$$

Expressing the position vectors of the nodes at time t_{n+1} as $X^i = X^i e_x + Y^i e_y$ and $X^{i+1} = X^{i+1} e_x + Y^{i+1} e_y$ and neglecting terms that are $O(\Delta t^2)$, the new length of the edge can be shown to be

$$l_{new}^i = l_{old}^i \left[1 + \frac{\Delta t}{(l_{old}^i)^2} \left((X^{i+1} - X^i)(-\delta v_x^i (S^i)^2 + \delta v_y^i S^i C^i + \delta v_x^{i+1} (S^{i+1})^2 - \delta v_y^{i+1} S^{i+1} C^{i+1}) \right. \right. \\ \left. \left. - (Y^{i+1} - Y^i)(-\delta v_x^i S^i C^i + \delta v_y^i (C^i)^2 + \delta v_x^{i+1} S^{i+1} C^{i+1} - \delta v_y^{i+1} (C^{i+1})^2) \right) \right] \quad (\text{A20})$$

where $l_{old}^i = |(X^{i+1} - X^i)|$ is the length of the edge before application of the virtual velocities. With the new length thus defined, the virtual rate of change of edge length can be evaluated straightforwardly as $\delta \dot{l}_i = (l_{new}^i - l_{old}^i) / \Delta t$, which in conjunction with Eq. (A20) above can be written symbolically as

$$\delta \dot{l}_i = \{\delta v^N\}^T \{P\}, \quad (\text{A21})$$

where $\{P\}$ and $\{\delta v^N\}$ are given by

$$\{P\} = \begin{Bmatrix} -(X^{i+1} - X^i)(S^i)^2 + (Y^{i+1} - Y^i)S^i C^i \\ (X^{i+1} - X^i)S^i C^i - (Y^{i+1} - Y^i)(C^i)^2 \\ (X^{i+1} - X^i)(S^{i+1})^2 - (Y^{i+1} - Y^i)S^{i+1} C^{i+1} \\ -(X^{i+1} - X^i)S^{i+1} C^{i+1} + (Y^{i+1} - Y^i)(C^{i+1})^2 \end{Bmatrix} \quad \text{and} \quad \{\delta v^N\} = \begin{Bmatrix} \delta v_x^i \\ \delta v_y^i \\ \delta v_x^{i+1} \\ \delta v_y^{i+1} \end{Bmatrix}. \quad (\text{A22})$$

All the position coordinates in Eq. (A22) are those at time t_{n+1} and these can be related to the known position coordinates at time t_n through the displacement increments. If the corresponding coordinates at time t_n are denoted with the subscript n , then the following holds:

$$X^r = X_n^r + \Delta u_x^r, \quad \text{and} \quad Y^r = Y_n^r + \Delta u_y^r, \quad \text{for } r = i, i+1 \quad (\text{A23})$$

Then, $\{P\}$ can be explicitly written as function of the displacement increments as

$$\{P\} = \begin{Bmatrix} -(X_n^{i+1} - X_n^i + \Delta u_x^{i+1} - \Delta u_x^i)(S^i)^2 + (Y_n^{i+1} - Y_n^i + \Delta u_y^{i+1} - \Delta u_y^i)S^i C^i \\ (X_n^{i+1} - X_n^i + \Delta u_x^{i+1} - \Delta u_x^i)S^i C^i - (Y_n^{i+1} - Y_n^i + \Delta u_y^{i+1} - \Delta u_y^i)(C^i)^2 \\ (X_n^{i+1} - X_n^i + \Delta u_x^{i+1} - \Delta u_x^i)(S^{i+1})^2 - (Y_n^{i+1} - Y_n^i + \Delta u_y^{i+1} - \Delta u_y^i)S^{i+1} C^{i+1} \\ -(X_n^{i+1} - X_n^i + \Delta u_x^{i+1} - \Delta u_x^i)S^{i+1} C^{i+1} + (Y_n^{i+1} - Y_n^i + \Delta u_y^{i+1} - \Delta u_y^i)(C^{i+1})^2 \end{Bmatrix}. \quad (\text{A24})$$

Terms associated with changes in the areas of contact resulting from matter transport can be rewritten in matrix form following a similar procedure. Clearly, $(\delta v_{ip})_C = \delta v_x^C$ and the rate of change of the length of contact area BC is quite simply $(\delta v_x^C - \delta v_x^B)$. Since we are considering only one of the two contacting particles, $\dot{a}_1 = (\delta v_x^C - \delta v_x^B)/2$. Similarly, for the contact area DE, we get $(\delta v_{ip})_D = -\delta v_y^D$ and $\dot{a}_1 = -(\delta v_y^D - \delta v_y^E)/2$. Thus,

$$\begin{aligned} & \gamma_b \dot{a}_1 - \gamma_p (\delta v_{ip} \cos \phi)_C + \gamma_b \dot{a}_2 - \gamma_p (\delta v_{ip} \cos \phi)_D \\ &= \frac{1}{2} \gamma_b (\delta v_x^C - \delta v_x^B - \delta v_y^D + \delta v_y^E) - \gamma_p (\delta v_x^C \cos \phi_C - \delta v_y^D \cos \phi_D) \\ &= \{\delta v^N\}^T \{F_\gamma\}, \end{aligned} \quad (\text{A25})$$

where

$$\{F_\gamma\} = \begin{Bmatrix} -\frac{1}{2} \gamma_b \\ \frac{1}{2} \gamma_b - \gamma_p \cos \phi_C \\ -\frac{1}{2} \gamma_b + \gamma_p \cos \phi_D \\ \frac{1}{2} \gamma_b \end{Bmatrix}. \quad (\text{A26})$$

Setting $\{v^N\} = \{\Delta u^N\}/\Delta t$, and employing the standard interpolation matrices $[A]$ for the velocity and $[B]$ for the strain-rate inside the particle A , substitution of Eqs. (A12), (A22), and (A26) into the governing Eq. (20), yields the non-linear finite element equations

$$\int_A [B]^T \{\sigma\} dA + [K^d] \{\Delta u^N\} = \{F\}, \quad (\text{A27})$$

where $[K^d]$ is the diffusion matrix given by

$$[K^d] = \frac{1}{\mathcal{D}_b \Delta t} \int_{S_b} \{M(s)\} \{M(s)\}^T ds \quad (\text{A28})$$

and $\{F\}$ is the force vector defined as

$$\{F\} = \frac{1}{D_b} \int_{s_b} G(s) \{M(s)\}^T ds + \int_{s_r} [A]^T \{T\} ds - \sum_{i=1}^{N_d^p} \gamma_p \{\dot{P}\} - \{F_\gamma\} - \sigma_n(s_D) \{M(s_D)\} + \sigma_n(s_C) \{M(s_C)\}. \quad (\text{A29})$$

All entities occurring in Eqs. (A27) are referred to time t_{n+1} and thus, Eqs. (A27) are also satisfied at this time.

A2. Solution

Upon setting

$$\{\mathcal{Q}\} = \int_A [B]^T \{\sigma\} dA + [K^d] \{\Delta u^N\} - \{F\}, \quad (\text{A30})$$

the finite element equations are stated as

$$\{\mathcal{Q}\} = \{0\}. \quad (\text{A31})$$

The stresses $\{\sigma\}$ at time t_{n+1} are calculated as $\{\sigma(t_{n+1})\} = \{\sigma(t_n)\} + \{\Delta\sigma\}$, where $\{\Delta\sigma\}$ depends on both the displacement and creep strain increments (see Appendix B). Thus, the set of equations (A31) has to be solved for both $\{\Delta u^N\}$ and $\{\Delta\epsilon^c\}$ and since the creep strain increments are not known *a priori*, the solution procedure is iterative in nature.

Newton's iteration method is employed to solve Eqs. (A31). Each "global" iteration involves the solution of the following linearized equation for the correction $d\{\Delta u^N\}$ to the displacement increment $\{\Delta u^N\}$:

$$\{\mathcal{Q}\} + \left[\frac{\partial \mathcal{Q}}{\partial \Delta u^N} \right] d\{\Delta u^N\} = \{0\}, \quad (\text{A32})$$

From the definition of $\{\mathcal{Q}\}$, one obtains

$$\left[\frac{\partial \mathcal{Q}}{\partial \Delta u^N} \right] = \int_A [B]^T \left[\frac{\partial(\Delta\sigma)}{\partial(\Delta\epsilon)} \right] [B] dA + [K^d] + \sum_{i=1}^{N_d^p} \gamma_p \left[\frac{\partial P}{\partial(\Delta u^N)} \right] + \left[\frac{\partial F_\gamma}{\partial(\Delta u^N)} \right] + \{M(s_D)\} \left\{ \frac{\partial \sigma_n(s_D)}{\partial(\Delta u^N)} \right\}^T - \{M(s_C)\} \left\{ \frac{\partial \sigma_n(s_C)}{\partial(\Delta u^N)} \right\}^T, \quad (\text{A33})$$

where $[\partial(\Delta\sigma)/\partial(\Delta\epsilon)]$ depends on the values of stress $\{\sigma\}$ at time t_{n+1} and is calculated in Appendix B. During the iterations, only the force terms arising from the gap function and the external loads are independent of the deformation. All other terms present in $\{\mathcal{Q}\}$ are functions of the current value of the nodal displacements at t_{n+1} and hence, give rise to derivatives with respect to $\{\Delta u^N\}$.

As seen from Eq. (A24), the components of $\{P\}$ are linear in the displacement increments and so, the matrix $[\partial P/\partial(\Delta u^N)]$ is calculated easily from that equation. On the other hand, the components of $\{F_\gamma\}$ depend on the displacements indirectly through $\cos\phi_C$ and $\cos\phi_D$ and therefore the computation of $[\partial F_\gamma/\partial(\Delta u^N)]$ entails that of the derivatives of $\cos\phi_C$ and $\cos\phi_D$. Similarly, the vectors $\{\partial \sigma_n(s_C)/\partial(\Delta u^N)\}$ and

$\{\partial \sigma_n(s_D)/\partial(\Delta u^N)\}$ are dependent on the displacements at time t_{n+1} and the method of calculating these are detailed later in the section.

In the following, the procedure for calculating $\cos \phi_C$ and $\cos \phi_D$ and their derivatives with respect to the displacement increments is discussed. Let $\mathbf{X}^C = X^C \mathbf{e}_x + Y^C \mathbf{e}_y$ be the position of the node occupying point C and $\mathbf{X}^{C+1} = X^{C+1} \mathbf{e}_x + Y^{C+1} \mathbf{e}_y$ that of the subsequent node on the pore at time t_{n+1} . Further, denote by ΔX the difference $X^{C+1} - X^C$ and by ΔY the difference $Y^{C+1} - Y^C$. Then, the length of the edge l_C at time t_{n+1} is $\sqrt{(\Delta X)^2 + (\Delta Y)^2}$ and $\cos \phi_C = \Delta X / l_C$. Suppose displacement increments $\Delta \mathbf{u}^C = \Delta u_x^C \mathbf{e}_x + \Delta u_y^C \mathbf{e}_y$ and $\Delta \mathbf{u}^{C+1} = \Delta u_x^{C+1} \mathbf{e}_x + \Delta u_y^{C+1} \mathbf{e}_y$ are applied to the two nodes respectively. Adopting a procedure similar to the one for calculating the vector $\{P\}$, we get value of ϕ_C at C:

$$\begin{aligned} \cos(\phi + \Delta \phi)_C &= \Delta X / l_C + \frac{1}{(l_C)^3} \begin{bmatrix} -(\Delta Y)^2 \\ \Delta X \Delta Y \\ (\Delta Y)^2 \\ -\Delta X \Delta Y \end{bmatrix}^T \begin{bmatrix} \Delta u_x^C \\ \Delta u_y^C \\ \Delta u_x^{C+1} \\ \Delta u_y^{C+1} \end{bmatrix} \\ &= \cos \phi_C + \frac{1}{(l_C)^3} \begin{bmatrix} -(\Delta Y)^2 \\ \Delta X \Delta Y \\ (\Delta Y)^2 \\ -\Delta X \Delta Y \end{bmatrix}^T \begin{bmatrix} \Delta u_x^C \\ \Delta u_y^C \\ \Delta u_x^{C+1} \\ \Delta u_y^{C+1} \end{bmatrix} \end{aligned} \quad (\text{A34})$$

Similarly, if we denote the pore surface node nearest to the node occupying point D by the superscript $D-1$, we obtain the following relation for the dihedral angle at D:

$$\cos(\phi + \Delta \phi)_D = \cos \phi_D + \frac{1}{(l_D)^3} \begin{bmatrix} -\Delta X \Delta Y \\ (\Delta X)^2 \\ \Delta X \Delta Y \\ -(\Delta X)^2 \end{bmatrix}^T \begin{bmatrix} \Delta u_x^D \\ \Delta u_y^D \\ \Delta u_x^{D-1} \\ \Delta u_y^{D-1} \end{bmatrix} \quad (\text{A35})$$

Evidently, the dependence of $\cos \phi_C$ and $\cos \phi_D$ on the displacement increments is linear and the derivatives $\{d(\cos \phi_C)/d(\Delta u^N)\}$ and $\{d(\cos \phi_D)/d(\Delta u^N)\}$ can be calculated by inspection.

The normal stresses at the tip are calculated numerically by averaging the values at the stations of integration of the element(s) surrounding it. If the tip node is the mid-side node of an element edge, then the relevant integration stations are confined within that element itself, whereas if the tip node is shared by two elements, each element contributes one integration station to the calculation of the normal stress. In the former case, the tip stress would depend on the 16 nodal displacement increments associated with the single element while in the latter all 26 nodal displacement increments from

the two elements would be involved in the computation. This procedure is extended to evaluate the stress derivatives in Eq. (A33) for nodes at C and D. A test value $u_i = 10^{-3}$ is assigned to one of the displacement increments Δu^i on which the tip stress depends, while all other nodal displacement increments are set to zero. The creep strains are retained at the levels dictated by the solution at time t_n and the resulting tip stress σ_n is calculated by the averaging procedure indicated above. The ratio (σ_n/u_i) of this computed tip stress to the test value is the required derivative of the tip stress with respect to the non-zero nodal displacement increment $(\partial\sigma_n/\partial\Delta u_i)$. This procedure is repeated for all the relevant degrees of freedom to give the entire matrix of tip stress derivatives.

Thus, all the quantities that are needed to calculate $[\partial\mathcal{G}/\partial\Delta u^N]$ and \mathcal{G} are completely determined and Eq. (A32) is solved for $d\{\Delta u^N\}$. The displacement increment $\{\Delta u^N\}$ is updated by adding this correction to the existing value and a new value of the total strain increment $\{\Delta\epsilon\} = [B]\{\Delta u^N\}$ is obtained from this. The creep constitutive equations are then used to isolate the creep strain portion of this total strain increment. Calculation of this creep strain portion requires the solution of a system of 4 nonlinear equations at each integration station and the details of this "local" iterative scheme are presented in Appendix C.

A3. Constraint equations

As mentioned before, the finite element equations are to be solved under constraints imposed for the satisfaction of the flux-curvature and stress-curvature relations on the pore surface. Moreover, one also needs to satisfy the condition of zero flux at E. From the preceding formulation, the flux at point E has the form

$$j(s_E) = -\sum m_\alpha^i(s_E)v_\alpha^i - G_1(s_C) - G_2(s_E) \quad (\text{A36})$$

Since the velocities are approximated as the displacement increments divided by the time step, setting the flux at E equal to zero is easily achieved by imposing a constraint on the nodal displacement increments. This constraint involves all the nodes on the contact areas and the pore surface.

The curvature at any node on the pore surface at time t_{n+1} is calculated from nodal coordinates and is clearly dependent on the unknown displacement increments (see Appendix D). The curvature gradient at the midpoint of each two-noded element edge is calculated by a central difference scheme and the satisfaction of the flux-curvature relation at this point generates one constraint in terms of the displacement increments.

Finally, the stress-curvature relation on the pore surface is satisfied by generating and satisfying one constraint at each pore surface node (except the first and the last) such that at time t_{n+1} , the relation $\sigma_n = \gamma_p k$ is satisfied. It is seen that the satisfaction of a similar constraint at the first and last nodes leads to a deterioration of the quality of the numerical scheme and greater difficulty in achieving convergence. However, the

curvature values at the tip obtained from extrapolation agree very well with those obtained from the stress, as discussed in sections 5 and 6.

APPENDIX B: CALCULATION OF $[\partial(\Delta\sigma)/\partial(\Delta\epsilon)]$

The finite element forms of constitutive Eqs. (3) and (6) are written respectively as

$$\{\sigma\} = [C^e]\{\epsilon^e\} \quad (B1)$$

and

$$\{\dot{\epsilon}^c\} = \{f(\{\sigma\})\} \quad (B2)$$

where $\{\sigma\}^T = [\sigma_{11}, \sigma_{22}, \sigma_{33}, \sigma_{12}]$, $\{\epsilon^e\}^T = [\epsilon_{11}^e, \epsilon_{22}^e, \epsilon_{33}^e, \epsilon_{12}^e]$, and $\{\dot{\epsilon}^c\}^T = [\dot{\epsilon}_{11}^c, \dot{\epsilon}_{22}^c, \dot{\epsilon}_{33}^c, \dot{\epsilon}_{12}^c]$.

From equation (6), it is evident that for the isotropic case under plane strain conditions, $\{f\}$ takes the form

$$\{f\} = \begin{Bmatrix} 3C\sigma_e^n s_1/2 \\ 3C\sigma_e^n s_2/2 \\ 3C\sigma_e^n s_3/2 \\ 3C\sigma_e^n s_4 \end{Bmatrix} \quad (B3)$$

where the s_i ($i = 1, 4$) are the deviatoric stresses defined as

$$\begin{aligned} s_1 &= \sigma_{11} - \sigma_{kk}/3, \\ s_2 &= \sigma_{22} - \sigma_{kk}/3, \\ s_3 &= \sigma_{33} - \sigma_{kk}/3, \\ s_4 &= \sigma_{12} \end{aligned} \quad (B4)$$

and $\sigma_e = \sqrt{3[s_1^2 + s_2^2 + s_3^2 + 2s_4^2]}/2$ is the equivalent stress.

Suppose the complete solution to the problem at time t_n is available and the solution at time t_{n+1} is required. Then,

$$\begin{aligned} \{\sigma(t_{n+1})\} &= \{\sigma(t_n)\} + \{\Delta\sigma\} \\ &= \{\sigma(t_n)\} + [C^e]\{\Delta\epsilon^e\} \\ &= \{\sigma(t_n)\} + [C^e]\{\Delta\epsilon\} - [C^e]\{\Delta\epsilon^c\}. \end{aligned} \quad (B5)$$

Given $\{\Delta\epsilon\}$, $\{\sigma(t)\} + [C^e]\{\Delta\epsilon\}$ is completely determined and used as the elastic predictor $\{\sigma^e(t_{n+1})\}$. Thus, once $\{\Delta\epsilon^c\}$ is known, stress $\{\sigma(t_{n+1})\}$ is calculated through

$$\{\sigma(t_{n+1})\} = \{\sigma^e(t_{n+1})\} - [C^e]\{\Delta\epsilon^c\} \quad (B6)$$

For the creep strain increment $\{\Delta\epsilon^c\}$, one obtains the approximation

$$\{\Delta\epsilon^c\} = \{\dot{\epsilon}^c\}\Delta t \quad (B7)$$

and the trapezoidal rule is used to calculate the strain rate

$$\{\dot{\epsilon}^c\} = \frac{1}{2}[\{f(\{\sigma(t_n)\})\} + \{f(\{\sigma(t_{n+1})\})\}] \quad (B8)$$

Combining Eqs. (B7) and (B8), one gets the following implicit equation for the creep strain increments in terms of the time step Δt and the assumed total strain increment $\{\Delta\epsilon\}$:

$$\{\Delta\epsilon^c\} - \frac{1}{2}\Delta t \left[\{f(\{\sigma(t_n)\})\} + \{f(\{\sigma(t_{n+1})\})\} \right] = \{0\}. \quad (\text{B9})$$

From Eqs. (B5), it follows that

$$\left[\frac{\partial(\Delta\sigma)}{\partial(\Delta\epsilon)} \right] = [C^e] - [C^e] \left[\frac{\partial(\Delta\epsilon^c)}{\partial(\Delta\epsilon)} \right]. \quad (\text{B10})$$

In order to evaluate the matrix of creep strain derivatives, one first obtains from Eqs. (B9),

$$\left[\frac{\partial(\Delta\epsilon^c)}{\partial(\Delta\epsilon)} \right] = \frac{1}{2}\Delta t \left[\frac{\partial f}{\partial \sigma} \right] \left[\frac{\partial(\Delta\sigma)}{\partial(\Delta\epsilon)} \right], \quad (\text{B11})$$

where the matrix of derivatives $[\partial f / \partial \sigma]$ are evaluated at time t_{n+1} . Thus, the unknown matrix of stress derivatives reappears on the right-hand side above and combining Eqs. (B10) and (B11) gives

$$\left[\frac{\partial(\Delta\sigma)}{\partial(\Delta\epsilon)} \right] = [C^e] - \frac{1}{2}\Delta t [C^e] \left[\frac{\partial f}{\partial \sigma} \right] \left[\frac{\partial(\Delta\sigma)}{\partial(\Delta\epsilon)} \right] \quad (\text{B12})$$

which can be rearranged to yield

$$\left[\frac{\partial(\Delta\sigma)}{\partial(\Delta\epsilon)} \right] = \left[[C^e]^{-1} + \frac{1}{2}\Delta t \left[\frac{\partial f}{\partial \sigma} \right] \right]^{-1} \quad (\text{B13})$$

Carrying out a term by term differentiation of the vector $\{f\}$ in Eq. (B3), one gets the following expressions for the components of $[\partial f / \partial \sigma]$:

$$\frac{\partial f_i}{\partial \sigma_j} = \left\{ \begin{array}{ll} C\sigma_e^{n-1} \left[1 + \frac{9}{4}(n-1) \left(\frac{s_i s_j}{\sigma_e^2} \right) \right], & i = j = 1, 2, 3 \\ 3C\sigma_e^{n-1} \left[1 + 3(n-1) \left(\frac{s_i s_j}{\sigma_e^2} \right) \right], & i = j = 4 \\ C\sigma_e^{n-1} \left[-\frac{1}{2} + \frac{9}{4}(n-1) \left(\frac{s_i s_j}{\sigma_e^2} \right) \right], & i \neq j \text{ \& } i, j = 1, 2, 3 \\ \frac{9}{2} C\sigma_e^{n-1} (n-1) \left(\frac{s_i s_j}{\sigma_e^2} \right), & i = 1, 2, 3 \text{ \& } j = 4 \end{array} \right\}. \quad (\text{B14})$$

Notice that

$$\frac{\partial f_i}{\partial \sigma_j} = \frac{\partial f_j}{\partial \sigma_i}. \quad (\text{B15})$$

Thus, given the stress state from Eq. (B6), once $\{\Delta\epsilon^c\}$ is determined as shown in Appendix C, Eqs. (B13) and (B14) in combination yield the matrix of derivatives $[\partial(\Delta\sigma) / \partial(\Delta\epsilon)]$.

APPENDIX C: "LOCAL" ITERATIVE LOOP FOR THE CALCULATION OF $\{\Delta\epsilon^c\}$ FROM $\{\Delta\epsilon\}$

Eq. (B9) of Appendix B forms the basis for the calculation of the creep strain portion of the total strain increment $\{\Delta\epsilon\} = [B]\{\Delta u^N\}$. Defining

$$\{g(\Delta\epsilon^c)\} = \{\Delta\epsilon^c\} - \frac{1}{2}\Delta t \left[\{f(\{\sigma(t_n)\})\} + \{f(\{\sigma^e(t_{n+1})\}) - [C^e]\{\Delta\epsilon^c\}\} \right], \quad (C1)$$

one can cast Eq. (B9) as $\{g\} = \{0\}$, and this system of non-linear equations is solved for $\{\Delta\epsilon^c\}$. The Newton iteration method is once again implemented at each station of integration in the bulk of the particles and this results in the solution of the linear system of equations:

$$\{g(\Delta\epsilon^c)\} + \left[\frac{\partial g(\Delta\epsilon^c)}{\partial(\Delta\epsilon^c)} \right] d\{\Delta\epsilon^c\} = \{0\} \quad (C2)$$

Upon differentiation of Eq. (C1), one obtains

$$\frac{\partial g_i(\Delta\epsilon^c)}{\partial(\Delta\epsilon_j^c)} = \frac{\partial(\Delta\epsilon_j^c)}{\partial(\Delta\epsilon_j^c)} - \frac{1}{2}\Delta t \frac{\partial f_i}{\partial\sigma_k} \frac{\partial(C_{kl}^e\Delta\epsilon_l^c)}{\partial(\Delta\epsilon_j^c)}. \quad (C3)$$

Once again, the derivatives $\partial f_i/\partial\sigma_k$ are evaluated at time t_{n+1} .

The three creep strains $\Delta\epsilon_1^c$, $\Delta\epsilon_2^c$ and $\Delta\epsilon_3^c$ are not independent of each other; they are related through the incompressibility of creep deformation, i.e., $\Delta\epsilon_1^c + \Delta\epsilon_2^c + \Delta\epsilon_3^c = 0$. Therefore, one needs to eliminate one of them from the above equations; for instance, one could set $\Delta\epsilon_3^c = -(\Delta\epsilon_1^c + \Delta\epsilon_2^c)$ and eliminate $\Delta\epsilon_3^c$. Then, for the plane strain case, with $(k=1,4)$,

$$\frac{\partial(C_{kl}^e\Delta\epsilon_l^c)}{\partial(\Delta\epsilon_j^c)} = \begin{cases} C_{kj}^e & j = 4 \\ C_{kj}^e - C_{k3}^e & j = 1,2 \end{cases} \quad (C4)$$

or, more generally,

$$\frac{\partial(C_{kl}^e\Delta\epsilon_l^c)}{\partial(\Delta\epsilon_j^c)} = C_{kj}^e - C_{k3}^e\delta_{1j} - C_{k3}^e\delta_{2j} \quad j = 1,2,4 \quad (C5)$$

Thus, the tangent stiffness matrix for the local loop becomes

$$\frac{\partial g_i(\Delta\epsilon^c)}{\partial(\Delta\epsilon_j^c)} = \delta_{ij} + \frac{1}{2}\Delta t \frac{\partial f_i}{\partial\sigma_k} (C_{kj}^e - C_{k3}^e\delta_{1j} - C_{k3}^e\delta_{2j}) \quad (C6)$$

Here, the matrix on the left is a 3×3 matrix and the indices i and j take on values 1, 2 and 4 for stresses $\sigma_{11}, \sigma_{22}, \sigma_{12}$ and strains $\epsilon_{11}, \epsilon_{22}, \epsilon_{12}$, whereas the index k runs from 1 through 4. With the tangent stiffness completely determined, Eqs. (C2) can now be solved to yield the correction $d\{\Delta\epsilon^c\}$. The iterative procedure continues until the measure

$$e = \sqrt{\frac{d(\Delta\epsilon_{ij}^c)d(\Delta\epsilon_{ij}^c)}{\Delta\epsilon_{ij}^c\Delta\epsilon_{ij}^c}} \quad (C7)$$

falls below 2.5×10^{-4} .

APPENDIX D: CALCULATION THE PORE SURFACE CURVATURE

For all nodes on the pore surface except those at the tips, the curvature is determined by their position and those of their two closest neighbours. Denote by (x_{i-1}, y_{i-1}) , (x_i, y_i) and (x_{i+1}, y_{i+1}) the position of nodes $(i-1)$, i and $(i+1)$ respectively. One can uniquely determine a circle of radius r centered at (a, b) passing through these three points, i.e., the three variables r , a and b can be obtained by solving the following system of equations:

$$\begin{aligned} (x_{i-1} - a)^2 + (y_{i-1} - b)^2 &= r^2 \\ (x_i - a)^2 + (y_i - b)^2 &= r^2 \\ (x_{i+1} - a)^2 + (y_{i+1} - b)^2 &= r^2 \end{aligned} \quad (D1)$$

The exact expressions for r , a and b have been obtained analytically in terms of the position coordinates; however, they are too cumbersome to report and one can state in symbolic terms

$$\begin{aligned} r &= r(x_{i-1}, y_{i-1}, x_i, y_i, x_{i+1}, y_{i+1}) \\ a &= a(x_{i-1}, y_{i-1}, x_i, y_i, x_{i+1}, y_{i+1}) \\ b &= b(x_{i-1}, y_{i-1}, x_i, y_i, x_{i+1}, y_{i+1}). \end{aligned} \quad (D2)$$

The reciprocal of the radius of the circle gives the curvature at node i , though due care should be given to the sign of the curvature. If the angle made with the x -axis by the chord joining nodes i and $(i-1)$ is less than that made by the chord joining nodes i and $(i+1)$, then the circle constructed as above is deemed convex with respect to the interior of the particle and the curvature is signed negative and vice versa.

$$k_i = k_i(x_{i-1}, y_{i-1}, x_i, y_i, x_{i+1}, y_{i+1}) = (\text{sign}) \frac{1}{r(x_{i-1}, y_{i-1}, x_i, y_i, x_{i+1}, y_{i+1})} \quad (D3)$$

The curvature at time t_{n+1} is calculated as the sum of the curvature at time t_n and an increment obtained from changes in nodal positions due to the addition of displacement increments. From Eq. (D3), this increment is

$$\Delta k_i = \sum_{j=i-1}^{i+1} \left(\frac{\partial k_i}{\partial x_j} \Delta u_x^j + \frac{\partial k_i}{\partial y_j} \Delta u_y^j \right). \quad (D4)$$

The required derivatives of k_i with respect to each of the six coordinates involved are calculated analytically from Eq. (D3). The curvature of the nodes at the tips of the pore is calculated by a quadratic extrapolation of the curvatures at the three adjacent pore surface nodes. Assigning the subscript 0 to the values at the tip and 1, 2 and 3 to those at the three neighboring nodes, one finds that the curvature at the tip takes the form

$$k_0 = As_0^2 + Bs_0 + C \quad (D5)$$

where the constants A , B and C satisfy the system of equations

$$\begin{bmatrix} s_1^2 & s_1 & 1 \\ s_2^2 & s_2 & 1 \\ s_3^2 & s_3 & 1 \end{bmatrix} \begin{Bmatrix} A \\ B \\ C \end{Bmatrix} = \begin{Bmatrix} k_1 \\ k_2 \\ k_3 \end{Bmatrix}. \quad (D6)$$

The curvature increments Δk_1 , Δk_2 and Δk_3 are obtained as explained in the preceding paragraph and these increments in turn are used to calculate the increment to the tip curvature, i.e.,

$$\Delta k_0 = \sum_{j=1}^3 \left\{ s_0^2 \left(\frac{\partial A}{\partial k_j} \right) \Delta k_j + s_0 \left(\frac{\partial B}{\partial k_j} \right) \Delta k_j + \left(\frac{\partial C}{\partial k_j} \right) \Delta k_j \right\} \quad (D7)$$

Once all curvature increments are calculated as functions of the displacement increments, the constraints discussed in Appendix A (Section A3) are generated and applied.

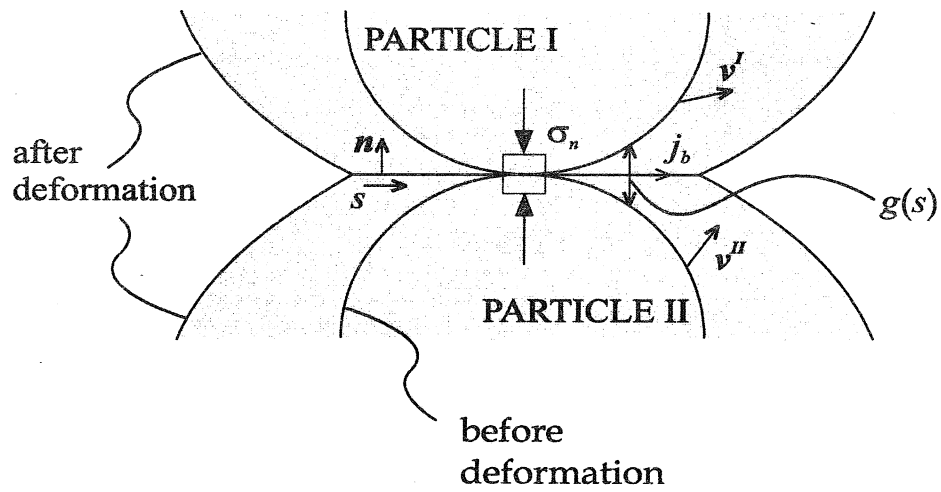


Fig. 1a. The interparticle contact area: s is the arclength, \mathbf{n} is the unit normal vector, σ_n is the normal stress, j_b is the volumetric flux and $g(s)$ is the non-negative gap function.

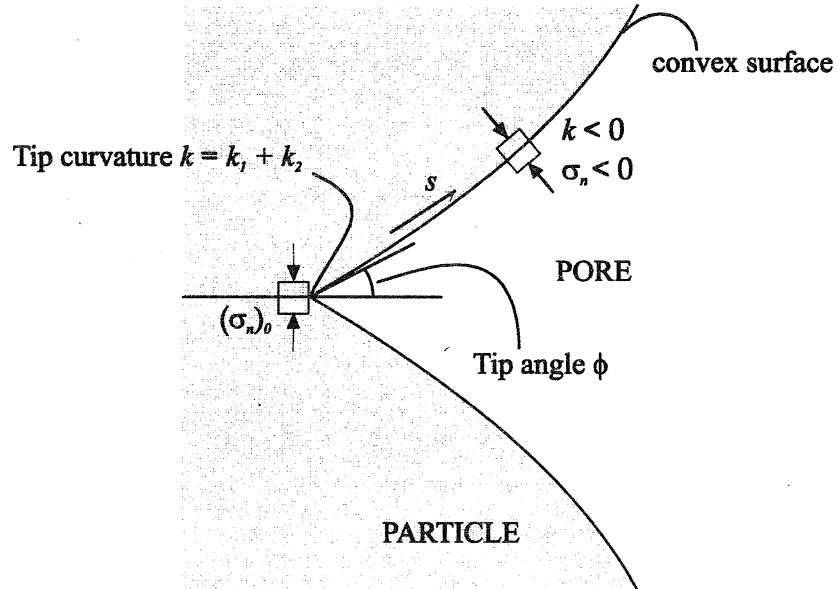


Fig. 1b. Sign convention for curvature on the pore surface: k is negative if the center of curvature is in the region occupied by the particle and correspondingly the stress is compressive.

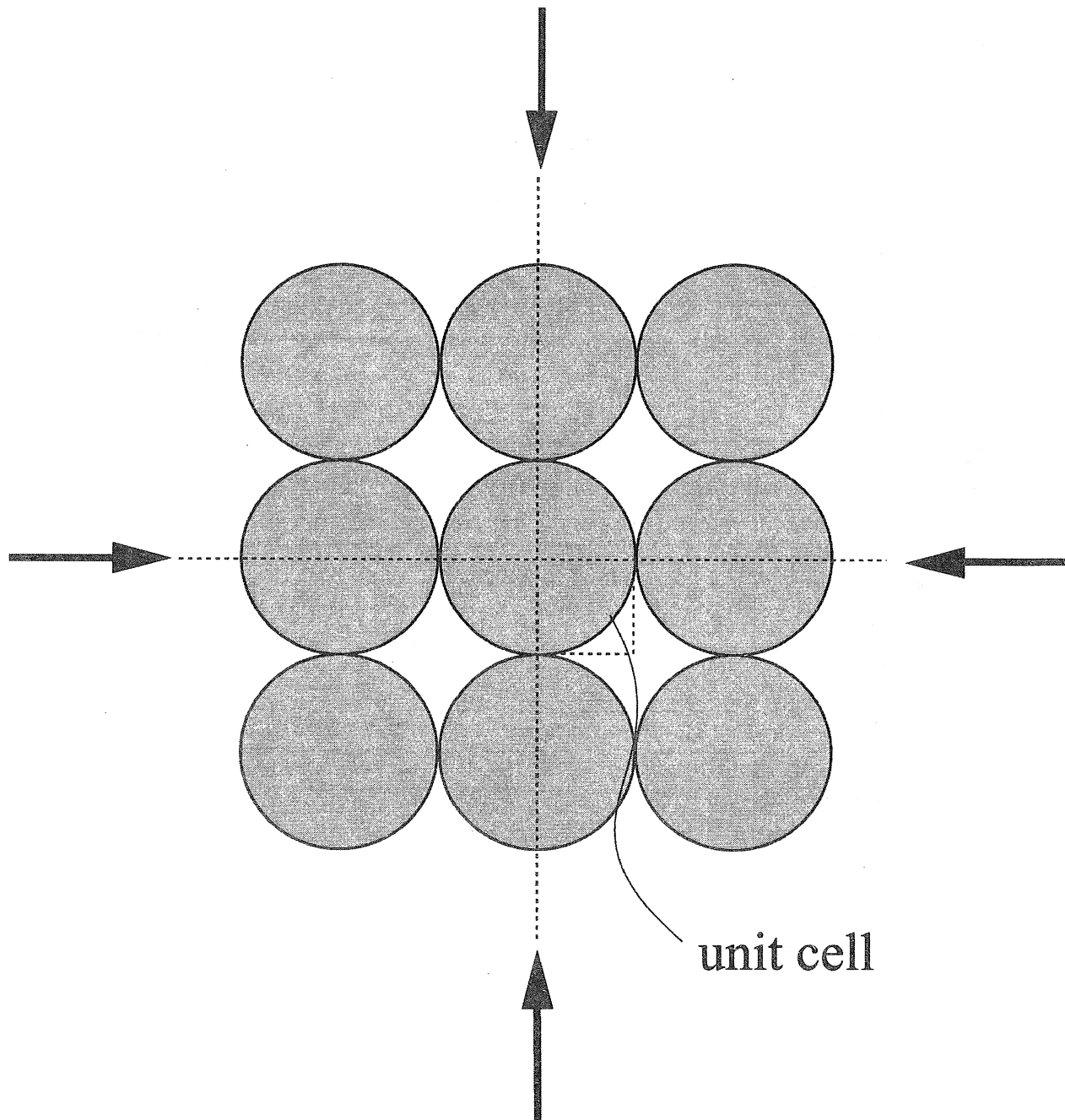


Fig. 2. Square arrangement of the powder particles under plane strain conditions. The square formed by the dashed lines is the unit cell chosen.

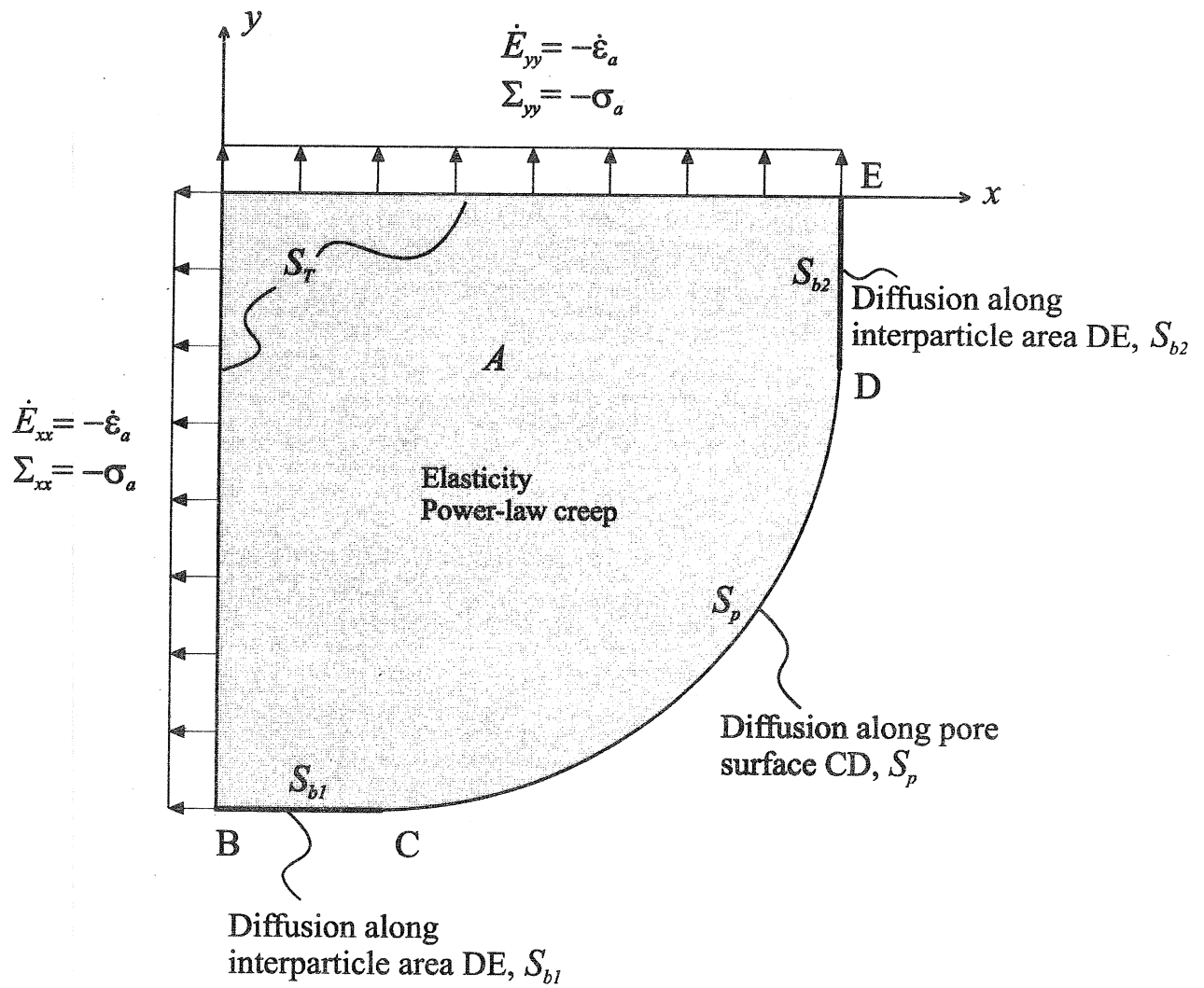


Fig. 3. Definition of the domains and boundary conditions on the unit cell.

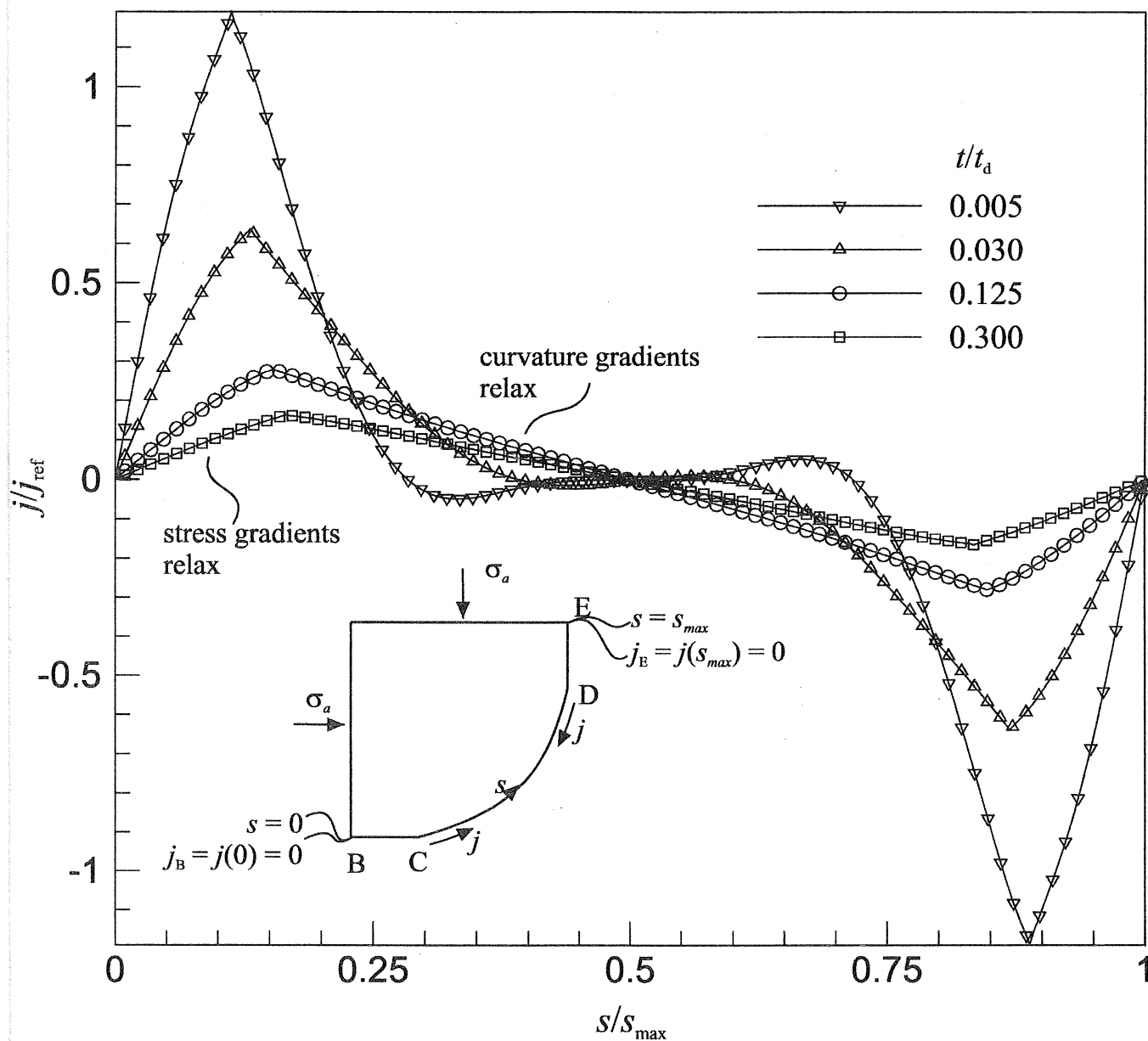


Fig. 4. Flux on the interparticle contact areas and pore surface for $\chi_b = 10$, $\chi_p = 100$, $\psi_b = \psi_p = 0.1$, $E/\sigma_a = 100$. Notice that $j = j_b$ on S_b and $j = j_p$ on S_p ; $j_{ref} = \mathcal{D}_b \sigma_a / s_{max}$.

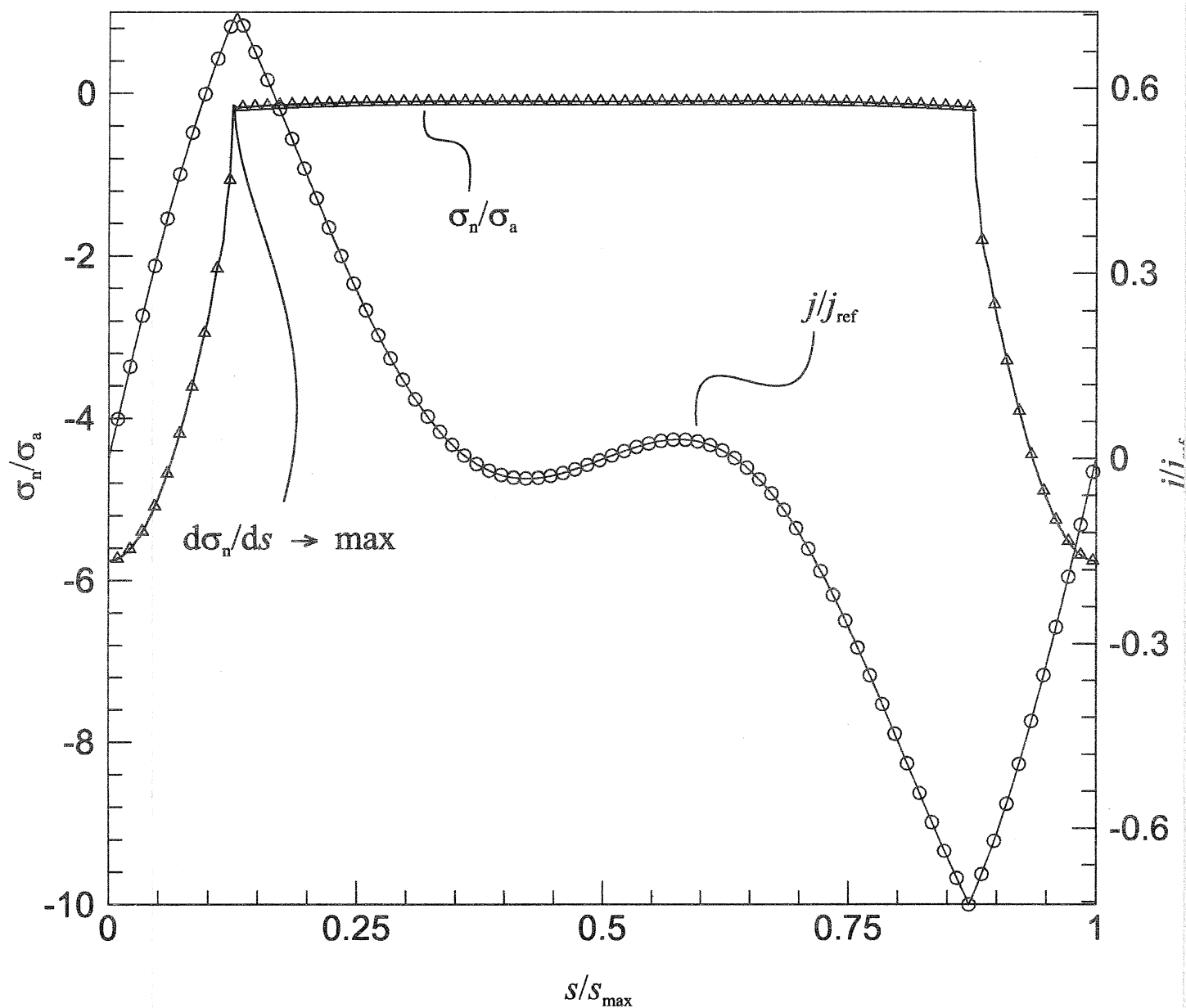


Fig. 5. Flux and normal stress on the interparticle contact areas and pore surface for $\chi_b = 10$, $\chi_p = 100$, $\psi_b = \psi_p = 0.1$, $E/\sigma_a = 100$ at time $t/t_d = 0.025$. Normalizing flux j_{ref} is equal to $\mathcal{D}_b \sigma_a / s_{\max}$.

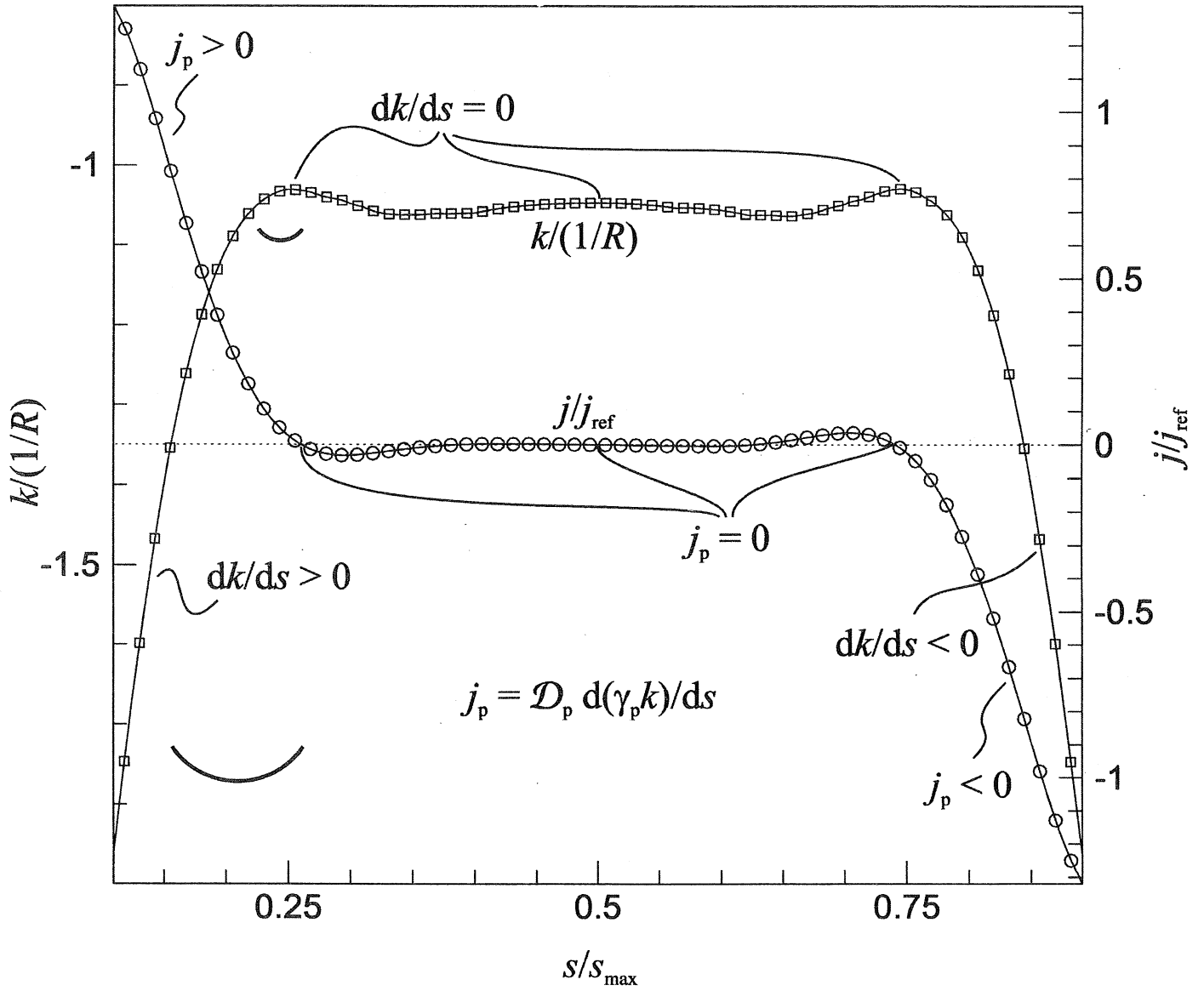


Fig. 6. Curvature and flux on the pore surface: $\chi_b = 10$, $\chi_p = 100$, $\psi_b = \psi_p = 0.1$, $E/\sigma_a = 100$ at time $t/t_d = 0.025$. Normalizing flux j_{ref} is equal to $\mathcal{D}_b \sigma_a / s_{\text{max}}$.

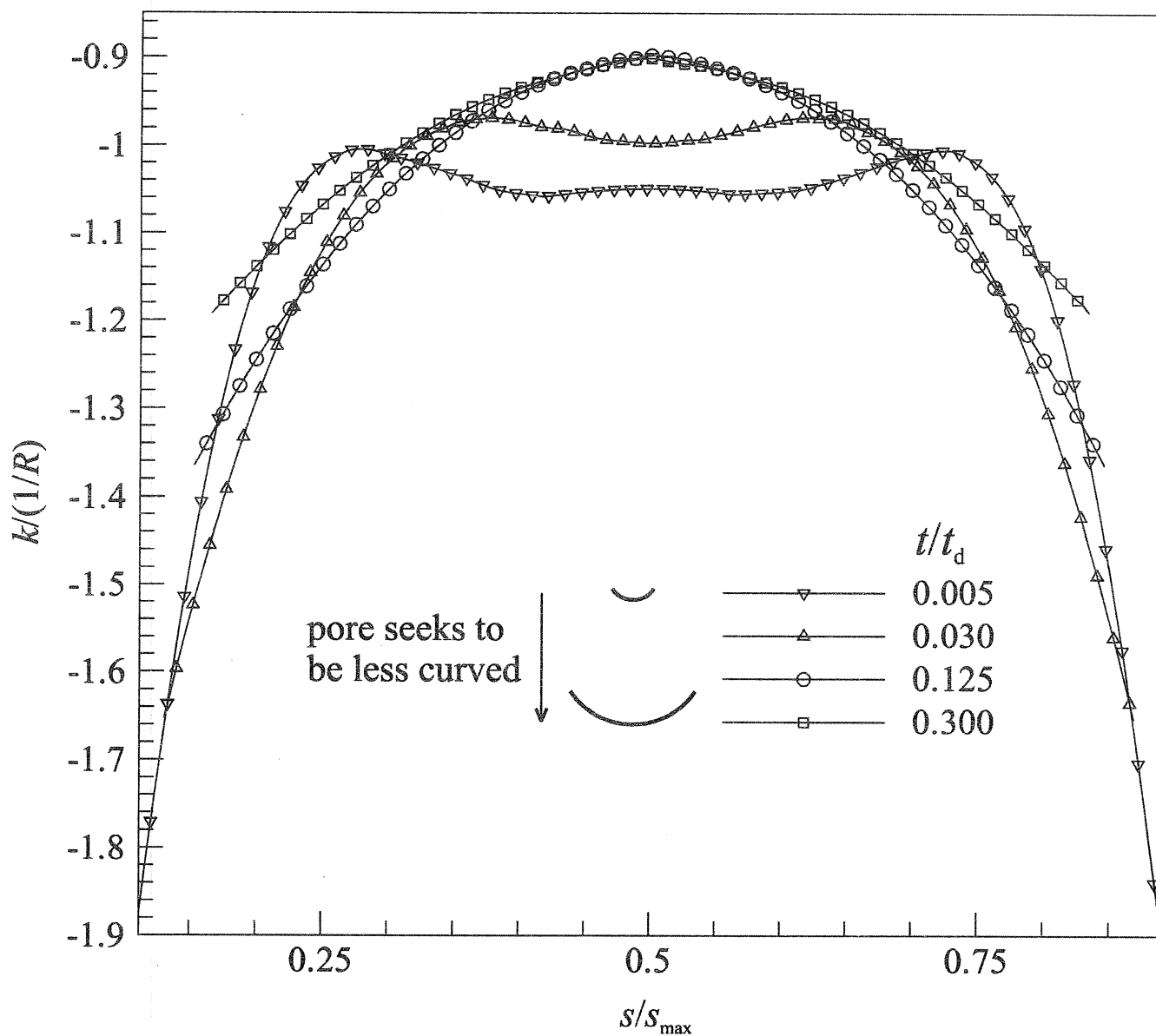


Fig. 7. Variation of pore surface curvature for $\chi_b = 10$, $\chi_p = 100$, $\psi_b = \psi_p = 0.1$, $E/\sigma_a = 100$.

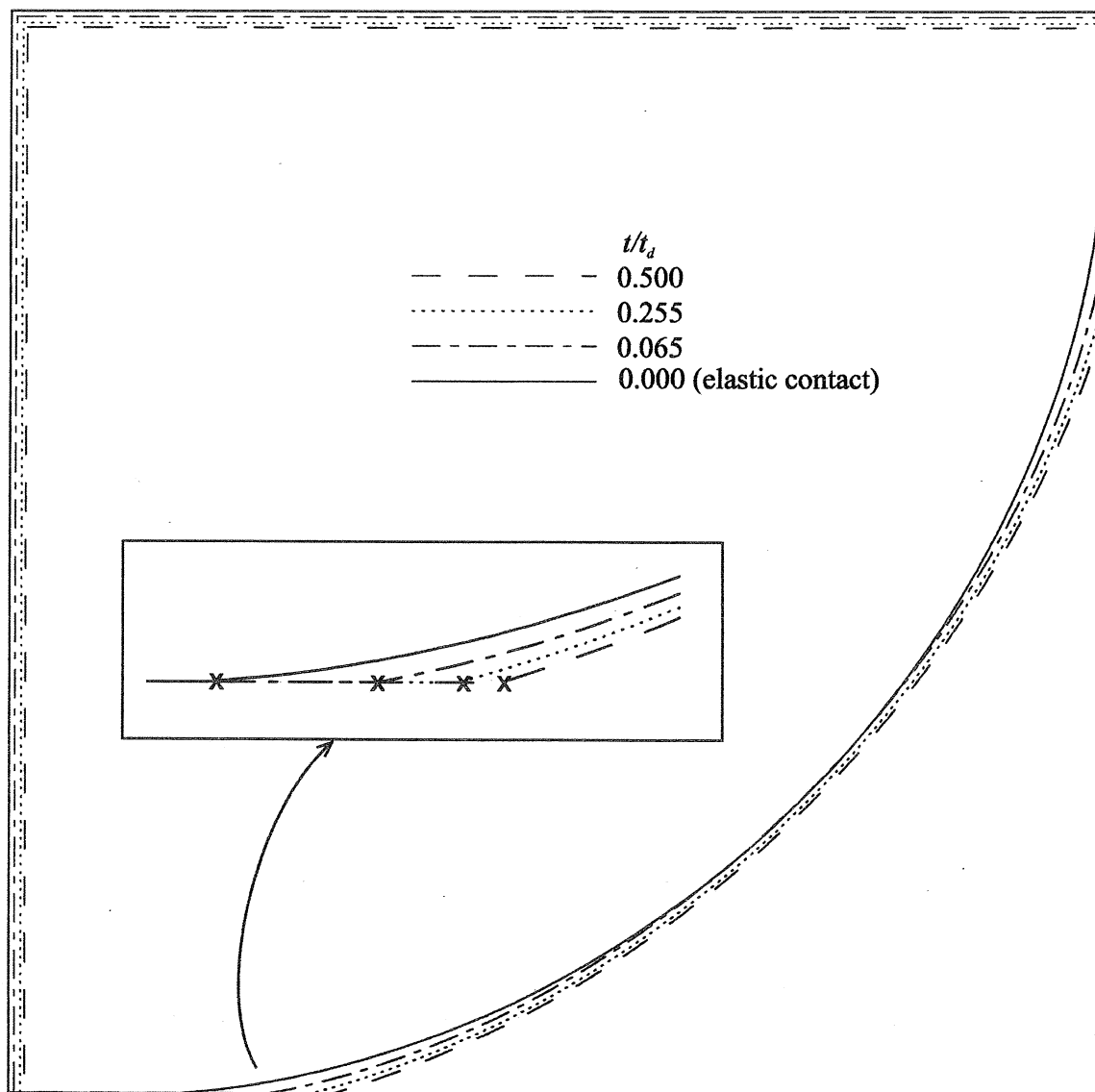


Fig. 8. Deformed mesh outlines at various times for $\chi_b = 10$, $\chi_p = 100$, $\psi_b = \psi_p = 0.1$, $E/\sigma_a = 100$. Sequential tip positions are shown in inset.

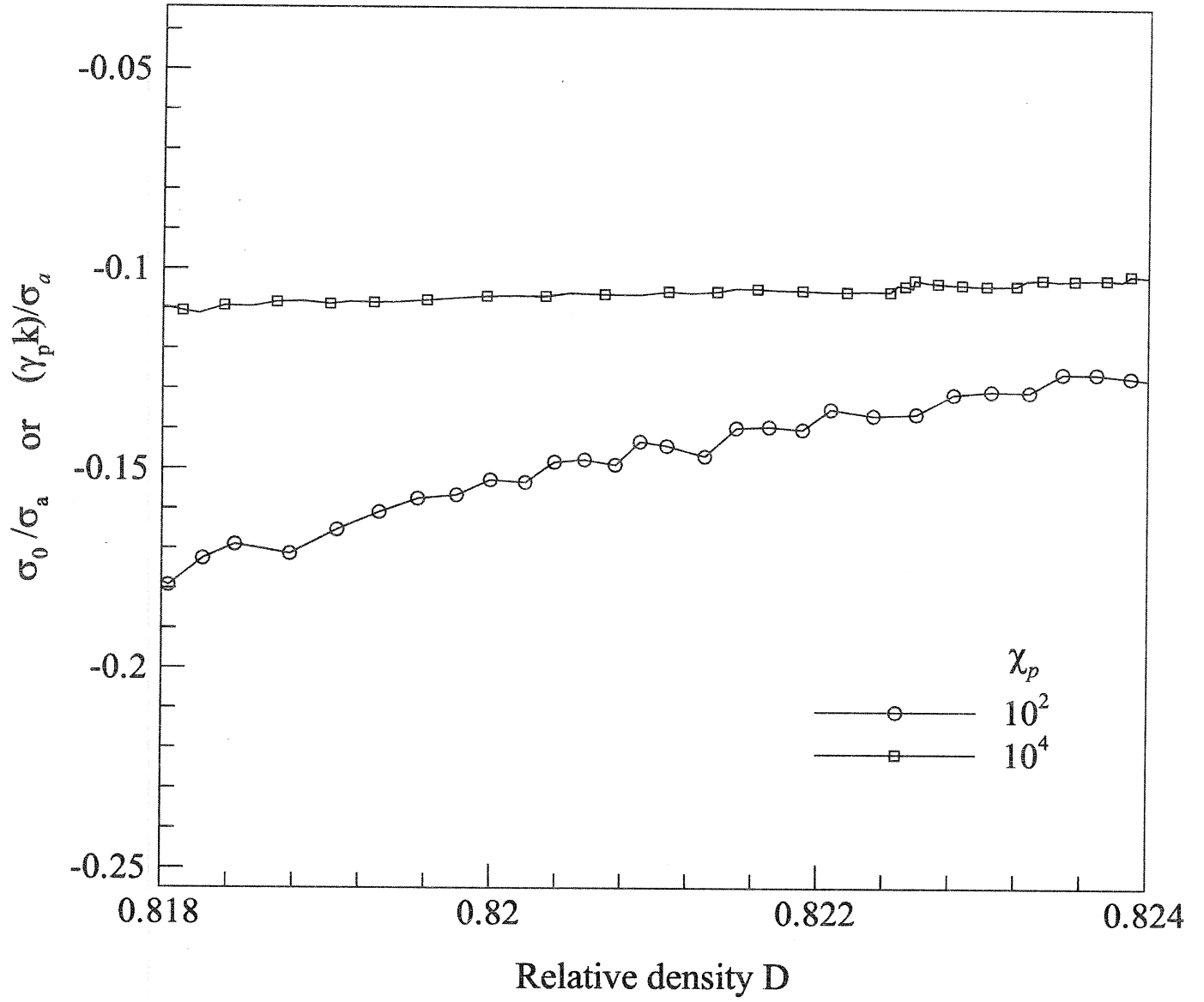


Fig. 9. Comparison of tip stresses for different surface diffusion parameters for $\chi_b = 10$, $\psi_b = \psi_p = 0.1$, $E/\sigma_a = 100$.

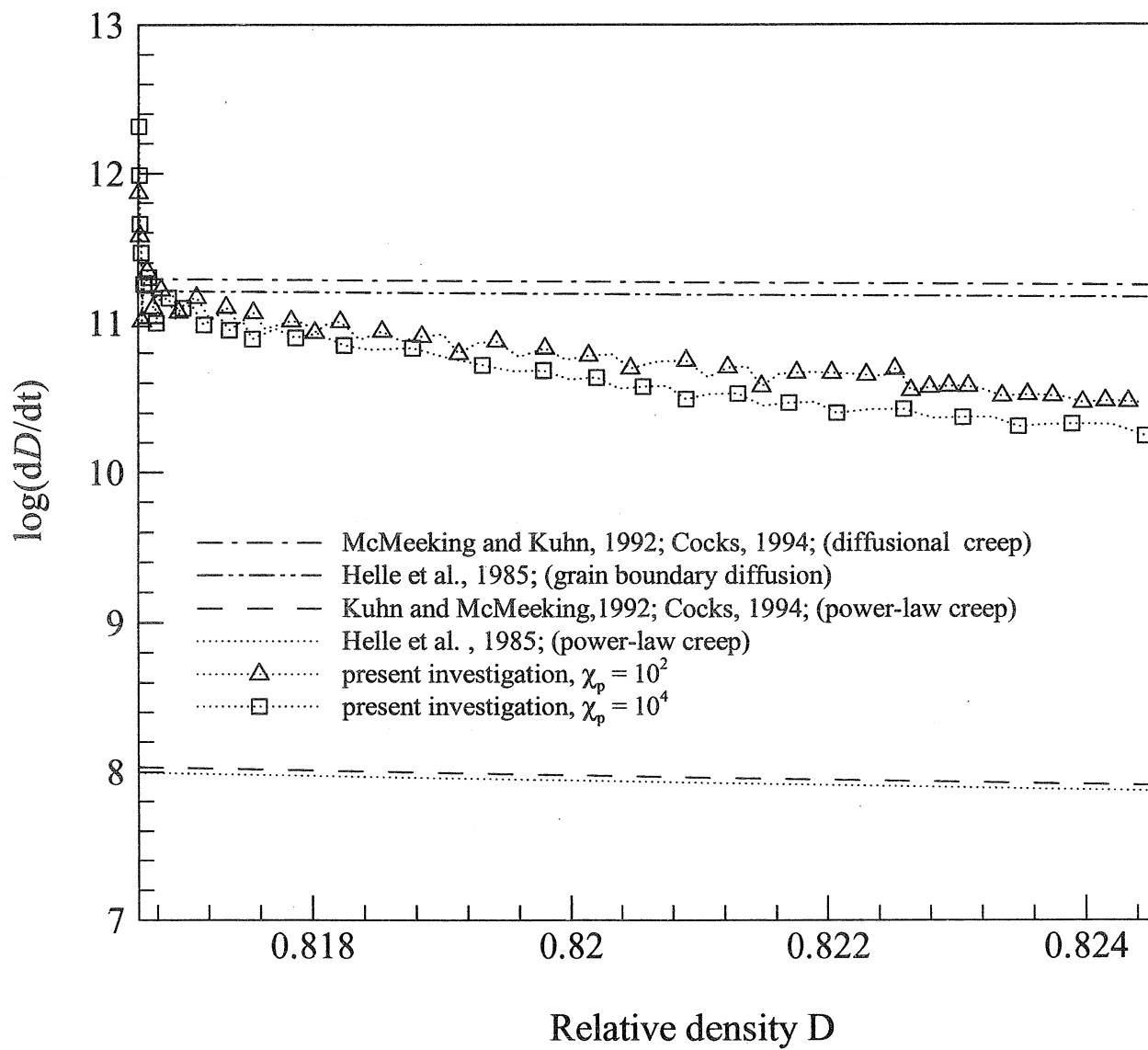


Fig. 10. Comparison of calculated densification rates with other analytical models.

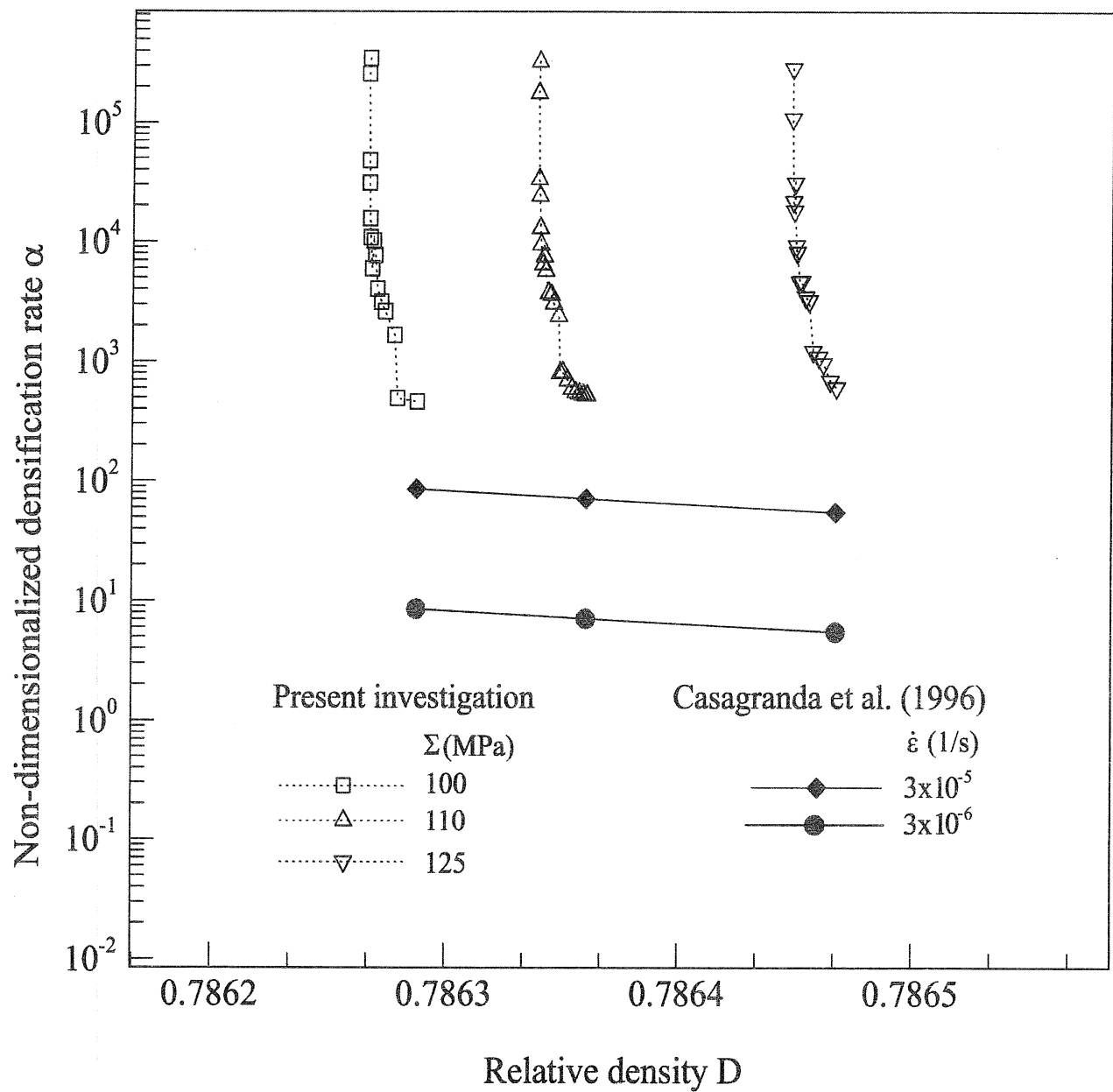


Fig. 11. Variation of consolidation rate with macroscopic stress (numerical calculations) and with strain rate (experimental results) for Al_2O_3 .

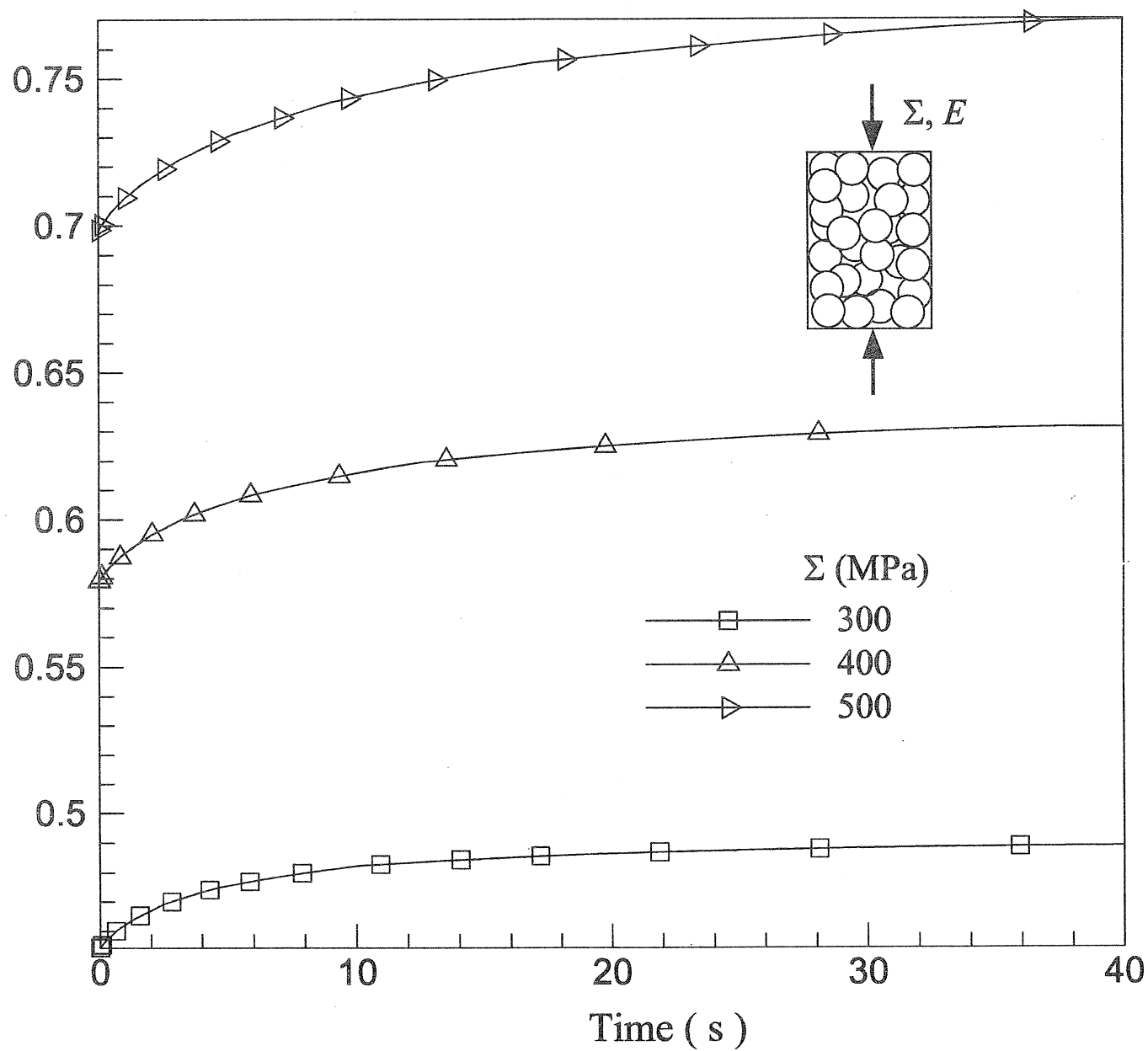


Fig. 12. Finite element calculation of uniaxial strain as a function of time for TiAl.

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